THESIS

THE DEVELOPMENT OF LIGANDS FOR C–H FUNCTIONALIZATION

UTILIZING AMINO ACID DERIVED DIRECTING GROUPS

Submitted by

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ABSTRACT

THE DEVELOPMENT OF LIGANDS FOR C–H FUNCTIONALIZATION

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The functionalization of unreactive bonds has become a focus of new reaction methodology. The foremost difficulty lies within achieving high levels of chemo-, regio-, and stereoselectivity without the need for molecular complexity. The aim of my project was to develop a catalyst system that could direct a C–H functionalization by forming a transient covalent attachment to a simple substrate and release the substrate after the transformation. I have developed two asymmetric catalysts for the C–H acetoxylation of sp^3 and sp^2 bonds. The sp^3 C–H acetoxylation occurs with high levels of diastereoselectivity, demonstrating the compatibility of this idea for enantioselective C–H functionalization.

In the course of my asymmetric ligand development, I engineered a stereoretentive synthesis for the formation of quaternary asymmetric amino amides. The straightforward synthesis involves facile, high yielding conversion of L-proline into a broad scope of differentially substituted amides with excellent enantioselectivity.

Furthermore, my exploration into amino acid derived ligands has uncovered a new method for C–C bond formation. The coupling of aryl substrates, originating from aryl sulfonamides, with olefins in a Heck-type transition metal catalyzed process has been discovered from examining serine scaffolds. This transformation may become a useful addition to the arsenal of C–C bond forming reactions.

Acknowledgements

I would first like to thank Eric for being a fantastic mentor during my time at Colorado State University. It's been very exciting to be a part of a brand new start up lab, and I want to thank him for letting me be a part of it. I also want to thank him for being so understanding in recent months with the tough decisions I've had to make.

I would next like to thank my group members who certainly made things interesting. I'd like to thank Doug and Paul who were part of the group's first class as we all worked together to figure out how a lab is supposed to work. I'd also like to thank them for teaching me more about food than I could ever care to know; I'd guess over half of our conversations concerned food as a major topic. I'd like to thank Eric "Little Man" for helping me edit my thesis, and also for playing some of the worst "music" I've ever heard, which really puts things in perspective. I'd like to thank Curtis for contributing to my real life know-how, and that's all I should really say about that. Lastly, I'd like to thank Brian "Oily Pete", among others, for taking so much guff from all of us upperclassman and just keeping things interesting. Overall, the guys have been great, super supportive and I will really miss all the talk about food.

My family has been so understanding of my decision to move to Fort Collins (about 1600 miles from Wisconsin) and attend graduate school. It has not been easy to be so far apart, but we have made the best of the situation at hand, and I have always felt very connected even though I've been so far away. I'd like to thank my dad for keeping me updated on Wisconsin sports, as Colorado could never steal away my love for the Packers. I'd like to thank my sisters and their families for staying so connected with me and visiting me when they had a chance. It makes me a little less homesick. I'd also like to thank my mom for supporting all of my tough decisions,

while helping me make better choices for my future. I could not have made it this far without her constant affection and understanding.

And I suppose I'll have to wrap this up by thanking Todd. I've been joking for a long time that I wouldn't acknowledge him at all, mostly because it's a lot of fun for me. But that just wouldn't be very fair or honest. Having met Todd by coming to graduate school, my entire experience has been influenced by him. He's been here for me through good times and really really bad times. He's dealt with all my craziness and insecurities about my abilities to complete this degree and has always pushed me to be my very best. I can honestly say that I would not have completed this work without him. I do not have enough time or space to express all of my gratitude.

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LIST OF ABBREVIATIONS

Ac   Acetyl

Acac   acetylacetonate

Ac_2O acetic anhydride

Bu   butyl

Bn   benzyl

Boc   *t*-butoxy carbonyl

Boc_2O di-*t*-butyl dicarbonate

BQ   benzoquinone

CAN   cerric ammonium nitrate

Cp   cyclopentaldienyl anion

CSA   camphorsulfonic acid

Cy   cyclohexyl

d   day

Dba   dibenzylideneacetone

DCC   N,N'-Dicyclohexylcarbodiimide

DCE   1,2-dichloroethane

DIPEA   diisopropylethylamine

DMF   N,N-dimethylformamide

DMP   Dess Martin periodinane

DMSO   dimethylsulfoxide

Dppe   1,2-bisdiphenylphosphinoethane

DMAP   4-dimethylaminopyridine

EDC   1-ethyl-3-(3-dimethylaminopropyl)carbodiimide

EtOAc   ethyl acetate

HOBt   hydroxybenzotriazole

h   hour

IPA   isopropyl alcohol

KHMDS   Potassium bis(trimethylsilyl)amide

LAH   Lithium aluminum hydride

LDA   lithium diisopropylamide

LHMDS   lithium bis(trimethylsilyl)amide

m-CPBA meta-chloroperbenzoic acid

Mes   mesityl, (2,4,6-trimethyl phenyl)

min   minute

Ms   methane sulfonyl

MTPA   α-methoxy,α-trifluoromethyl phenyl acetic acid

Ns   4-nitrophenyl sulfonyl

PGME   phenyl glycine methyl ester

Ph   phenyl

PTSA   *p*-toluene sulfonic acid

TBS   tert-butyldimethylsilyl

TEMPO   2,2,6,6-tetramethylpiperidine-1-oxy radical

Tf   trifluoromethanesulfonyl

TFA   trifluoroacetic acid

THF   tetrahydrofuran

Ts   *p*-toluene sulfonyl

Chapter 1

C–H Functionalization

Carbon-hydrogen bond functionalization, particularly in the formation of new carbon-carbon and carbon-heteroatom bonds, is a primary focus of modern methodological studies.[1] The development of new reactions relying on C–H functionalization is widely pursued in an effort to construct complex molecules from those with a low degree of functionality. Our research will exploit a transient directing group to accomplish the C–H functionalization of simple substrates without the necessity for a high degree of initial functionality within the molecule.

1.1 Research Objectives

Recent demonstrations of C–H bond functionalization involve an internalized ligating group to achieve high levels of chemo- and regioselectivity.[1] Our methodology will focus on the development of a ligand (**A**) that contains a heteroatom ligating group that can transiently attach to a substrate with a low degree of functionality (Scheme 1.1.1). We endeavor to attach the ligand to an aldehyde substrate through an acetal linkage. The acetalization component of the ligand will condense onto an aldehyde to afford a substrate-ligand adduct (**B**). The internal ligating group can then coordinate a metal (**C**) and place it in position to perform a C–H functionalization (**D**). Upon hydrolysis, the functionalized product will be released and the ligand regenerated. We will demonstrate the utility of this methodology with respect to the current examples of C–H functionalization.

Scheme 1.1.1
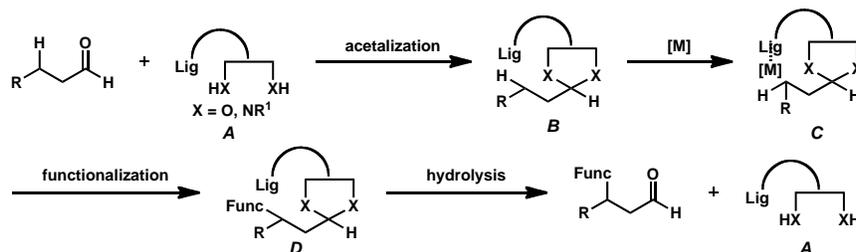
1.2   Recent Examples of C–H Functionalization

Directed C–H bond activation was first accomplished in the 1960s, when both Dubeck

and Cope discovered the cyclometallation product of azobenzene (Scheme 1.2.1).[2,3] Since this

discovery, remarkable work has advanced the application of C–H activation to catalysis and

functionalization. In the course of these advances, much insight into the mechanism of C–H

activation has been achieved leading to the progression of reaction development.

Scheme 1.2.1
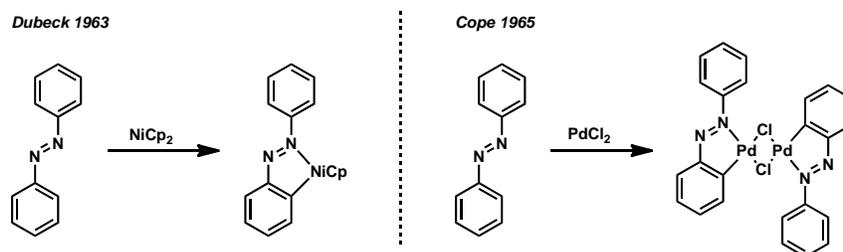
Directed C–H bond functionalization via Pd^{II}/Pd^{IV} catalysis has been widely explored in

recent years by numerous groups.[1] Carbon-oxygen bond formation has been one of the most

widely explored areas of C–H functionalization. In 2004, Sanford and coworkers showed that

7,8-benzoquinoline would undergo acetoxylation in the presence of catalytic palladium and

stoichiometric iodobenzene diacetate (Scheme 1.2.2).[4] A variety of directing groups have

mediated C–H functionalization in these aromatic systems. Mechanistically, a Pd^{II}/Pd^{IV} catalytic

cycle was envisioned featuring initial C–H activation followed by oxidation and reductive

elimination (*vide supra*).   Additionally, Sanford and coworkers discovered the same

transformation can be induced in the functionalization of *sp³* C–H bonds (Scheme 1.2.2).[5]   In

rigid systems, 2° C–H bonds undergo acetoxylation diastereoselectively, suggesting that either

C–H activation or reductive elimination proceeds stereoselectively.
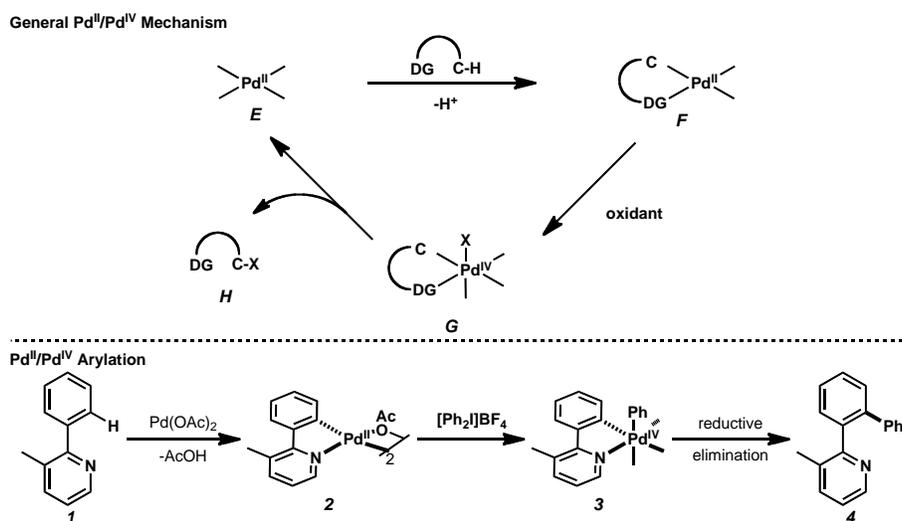
Scheme 1.2.2
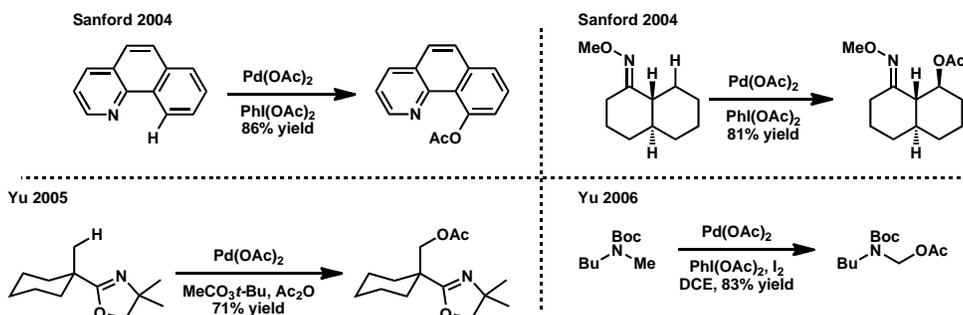
Similarly, Yu and coworkers have demonstrated oxazolines to be competent directing

groups in these transformations.[6]   In the presence of catalytic palladium (II) and *tert*-butyl

peracetate in acetic anhydride, the acetoxylated product was afforded in 71% yield (Scheme

1.2.2). The mechanism is presumed to follow that outlined by Sanford and coworkers. Further

investigations revealed a dual role for Ac_2O, both assisting in the oxidation to palladium (IV) and

regeneration of the $Pd(OAc)_2$ catalyst. Changing the solvent to propionic or isobutyric anhydride

afforded the corresponding carboxylate product. Similarly, they have observed IOAc to be a

competent oxidant in these transformations, generated *in situ* from I_2 and $PhI(OAc)_2$.[7] Boc-

protected methyl amines were acetoxylated at the 1° C–H bond in the presence of catalytic

$Pd(OAc)_2$ and IOAc in 83% yield.

In the first proposed mechanism, it is envisioned that an internal ligating group directs the

palladium to undergo C–H activation to form the palladacycle (**F**) (Scheme 1.2.3). Subsequent

oxidation to Pd^IV (**G**) followed by reductive elimination affords the acetoxylated product (**H**) in

86% yield.[4] These Pd^IV complexes have been isolated and well-studied in comparable reductive

eliminations.[8] Dissociation to the products is thought to occur through a pre-dissociation of a

ligand, or a direct reductive elimination. This mechanism was demonstrated in a C-C bond forming arylation reaction executed by Sanford and coworkers.[9] Cyclometallation of **1** *via* Pd(OAc)$_2$ afforded palladacycle **2**, which upon treatment with a diaryliodonium salt oxidized Pd^{II} to Pd^{IV} (**3**). Reductive elimination from the Pd^{IV} species formed the new carbon-carbon bond of product **4** and regenerated the palladium catalyst. This mechanism is thought to be applicable to all C–H oxidation reactions.

Scheme 1.2.3

Recently, new mechanistic evidence has suggested that a bimetallic Pd^{III}/Pd^{III} or Pd^{II}/Pd^{IV} species may be the active catalyst.[10] The mechanistic hypothesis is shown in Scheme 1.2.4. Upon cyclometallation, two palladacycles (**5**) dimerize to form intermediate **6**. This species can undergo a bimetallic oxidative addition to form a Pd^{III}-Pd^{III} bond (**7**). A bimetallic reductive elimination can occur to release the functionalized product (**8**) and regenerate the catalyst. This reductive elimination is thought to occur via a 1,1-concerted reductive elimination. Ritter and coworkers have compiled extensive kinetic and mechanistic work to support this hypothesis. The rate-determining step (RDS) of this transformation is highly dependent on the strength of the oxidant. In the case of PhI(OAc)$_2$, the RDS of acetoxylation is C–H activation, negating the

significance of a Pd^{III}-Pd^{III} dimer versus the conventional Pd^{II}/Pd^{IV} catalytic cycle. In the case of

N-chlorosuccinimide, however, the RDS of chlorination is the oxidative step, where the

mechanistic support for a Pd^{III}-Pd^{III} dimer materialized. If it is assumed that two Pd atoms are

required to come together to form a dimeric species in the RDS, the reaction should be second

order in palladium. If the catalyst resting state is dimeric and the RDS is dimeric in palladium

the reaction should be first order in palladium. Because the catalyst resting state is in

equilibrium between monomeric and dimeric species, the kinetic data revealed that the reaction

is 1.5 order in palladium. These bimetallic high oxidation state species have been isolated and

characterized and have shown kinetic competency in C–H functionalization reactions.

Scheme 1.2.4
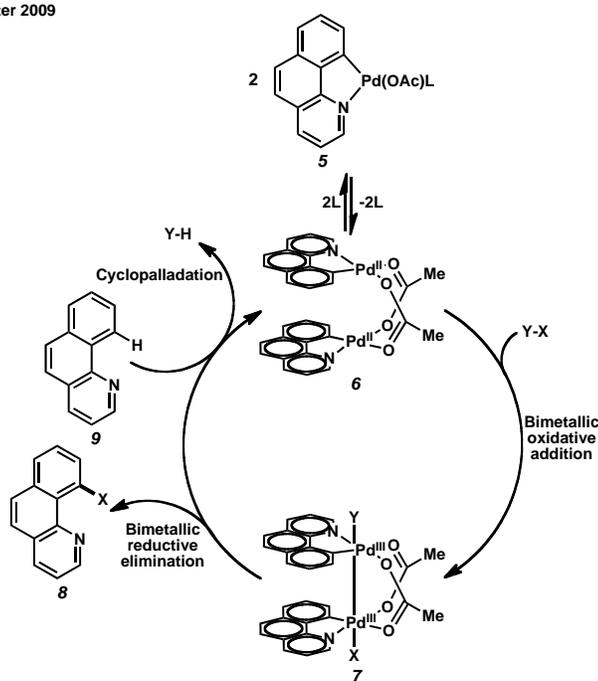
While the mechanism of oxidation in these transformations has been well-studied, the

mechanism of the cyclometallation step requires additional explanation. In the context of

directed C–H activation, three possible pathways have been postulated: oxidative addition, σ-

bond metathesis and electrophilic activation.[11-13] C–H activation via palladium is thought to

occur by either an electrophilic aromatic substitution mechanism or through an agostic interaction leading to a concerted metallation-deprotonation (CMD) event (Scheme 1.2.5).[11] In both cases, directing group coordination places the metal in position to undergo C–H activation. The electrophilic aromatic substitution reaction involves the formation of the arenium intermediate **J**, with subsequent re-aromatization via base-assisted deprotonation. While this mechanism cannot be unambiguously discredited, recent computational studies by Davies and coworkers suggest that an agostic C–H bond interaction for the initiation of activation is most probably followed by a CMD event.[13] Investigations with **10** and $Pd(OAc)_2$ suggest that initiation occurs via displacement of one η^2 acetate arm with the C–H bond to afford the 6-membered agostic intermediate **K**.

Fagnou and coworkers have also investigated the mechanism of acetate-assisted C–H activation.[14] It was found that acetate derivatives were essential for the realization of the reaction, which supports the agostic interaction-CMD mechanism. Additionally, they observed large kinetic isotope effects consistent with hydrogen abstraction during the rate-determining step. Lastly, they observed that electron poor aromatics outcompete electron rich systems in direct competition, which is contradictory with an electrophilic aromatic substitution mechanism. All of these observations suggest that the mechanism of C–H activation more closely aligns with a CMD event, rather than electrophillic aromatic substitution.

Scheme 1.2.5
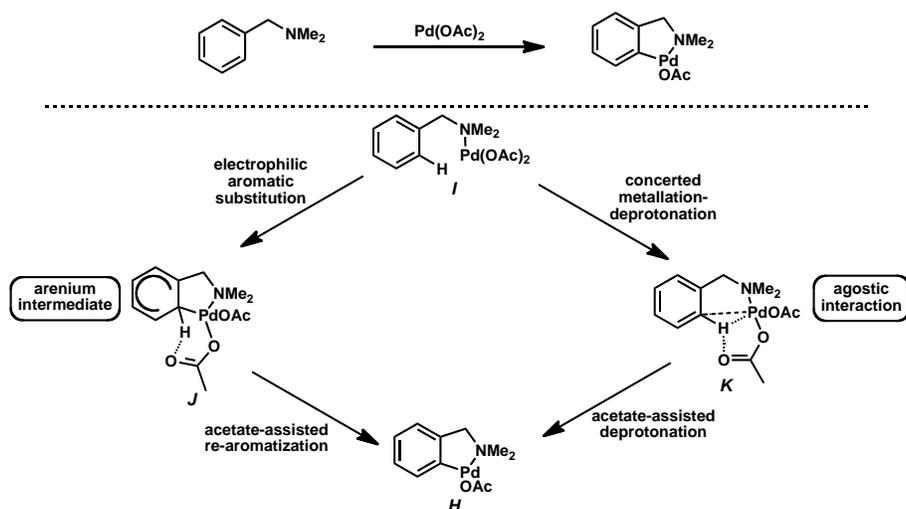
The directing groups in these transformations can play an important role in the rate of functionalization. Sanford and coworkers have further investigated nitrogen based directing group ability.[15] Utilizing benzylpyridine derivatives and optimized reaction conditions, competition studies were conducted and revealed that directing groups with electron donating substituents reacted preferentially (Scheme 1.2.6, **11a** and **12a**). Subsequently, individual kinetic studies were conducted on the substrates, and revealed that electron-withdrawing substituents on the pyridine ring enhanced the reaction rate for acetoxylation. Presumably, a more basic pyridine will bind palladium (II) preferentially over a less basic pyridine. C–H activation, however, occurs more rapidly with a less basic pyridine as a directing group, reducing the amount of electron density around the palladium species. In addition to substituent effects, competition studies were conducted between different directing groups to ascertain the reactivity and revealed that more basic directing groups reacted preferentially (Scheme 1.2.6). Based on the data, a relative reactivity trend in $Ac_2O/AcOH$ was compiled and ranked directing groups in terms of reactivity with the pyridine **11a** being most reactive and amide **18a** being least reactive.

Individual studies were conducted on the substrates, revealing that all of the directing groups (**11a-18a**) were competent for directing C–H functionalization.
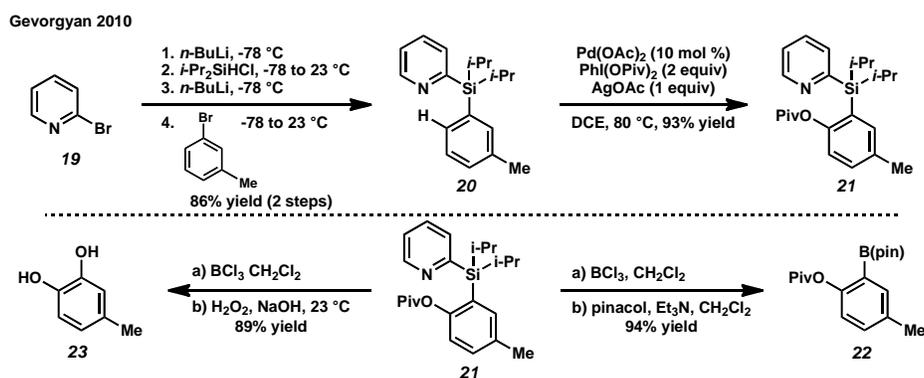
Scheme 1.2.6
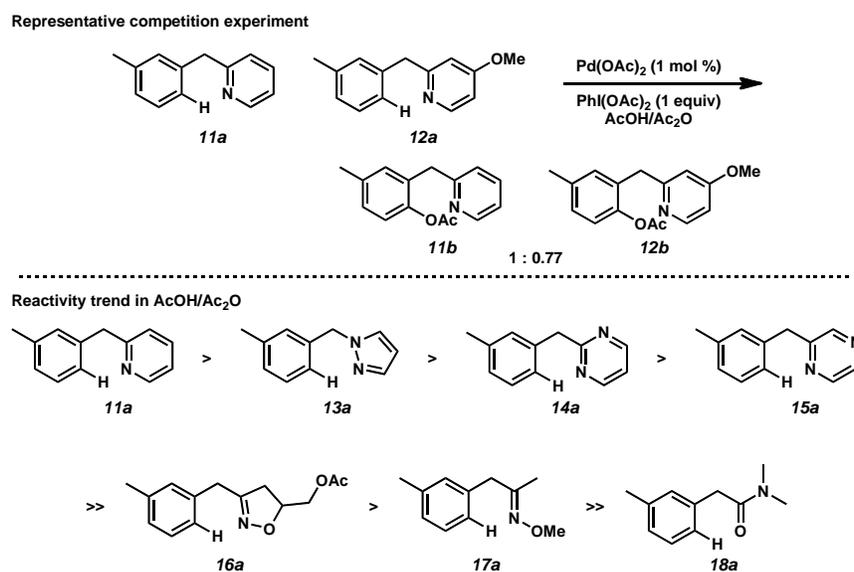
1.3   Convertible Directing Groups

A limitation inherent in traditional directed C–H functionalization is the difficulty of removal of the required directing group. There have been recent examples of these processes occurring in high yield. Gevorgyan and coworkers used silicon-tethered pryidine directing groups to promote C–H pivaloxylation and acetoxylation via palladium catalysis with oxidation via a hypervalent iodine species (Scheme 1.3.1).[16] By treating 2-bromopyridine (**19**) with *n*-butyllithium and diisopropylsilyl chloride, followed by *n*-butyllithium and 3-bromotoluene they attached the pyridinediisopropylsilyl directing group in two steps with good yield. Treatment of **20** with pivaloxylation conditions afforded **21** in excellent yield, while treatment with PhI(OAc)$_2$ afforded the corresponding acetoxylated product in 88% yield. The addition of AgOAc in the functionalization reaction provided higher yields, presumably by increasing the concentration of Pd(OAc)$_2$ in solution.[17] From the functionalized product, the silicon-tethered directing group

was removed and the functional group transformed in one pot to reveal a variety of different compounds. Treatment with borontrichloride followed by pinacol afforded the pinacolborane derivative (**22**) in excellent yield, which can be further derivatized in Suzuki couplings. Alternatively, treatment with borontrichloride followed by oxidizing conditions afford the catechol derivative (**23**). Furthermore, they were able to replace the directing group with H by treatment with AgF in MeOH in 93% yield. This method provides a selective high yielding four-step C–H oxidation via a transformable directing group.
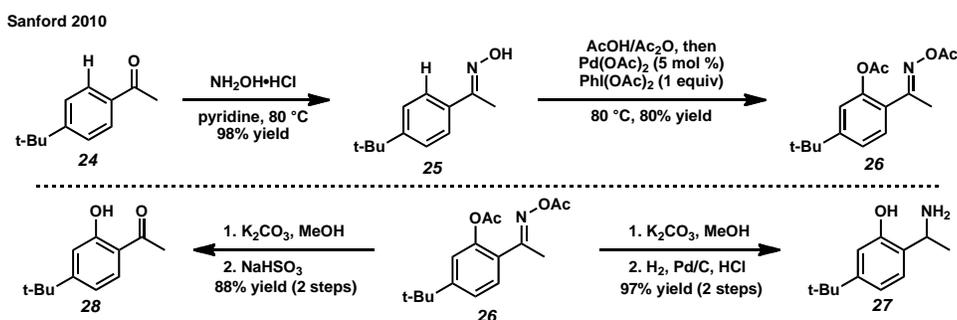
Scheme 1.3.1

Sanford and coworkers developed an alternative method using another convertible directing group.[18] They found *O*-acetyl oximes to be competent directing groups in the acetoxylation of aryl sp^2 C–H bonds as well as sp^3 C–H bonds (Scheme 1.3.2). In one step in high yield, ketone **24** was converted to the corresponding oxime (**25**). Treatment with an AcOH/Ac$_2$O mixture, presumably to make the *O*-acetyl oxime, followed by treatment with Pd(OAc)$_2$ and PhI(OAc)$_2$ afforded the desired acetoxylated product (**26**) in good yield. *O*-acetyl oximes are easily transformable directing groups, and after removal of the acetate, different chemical modifications afforded a diverse number of products. Treatment with hydrogenation conditions afforded the corresponding primary amine (**27**), while treatment with sodium bisulfite

provided the original ketone (**28**). This method provides a highly selective four-step procedure to generate acetoxylated sp^2 and sp^3 C–H bonds in good yield.[19]

Scheme 1.3.2

Gevorgyan and coworkers also developed a C–H alkenylation reaction with a directing group that can be easily removed (Scheme 1.3.3).[20] Using an ortho silanol directing group, they were able to effect sp^2 alkenylation, which upon treatment with TBAF revealed the functionalized product devoid of the necessary directing group. In three steps, which can be performed semi-one-pot, phenol **29** was converted to silanol **30** in good yield. The C–H alkenylation reaction was effected under palladium catalysis with an amino acid derived ligand developed by Yu and coworkers.[21] Upon completion of the functionalization, treatment with TBAF in a semi-one-pot reaction afforded product **31** in excellent yield. This method provided a five-step process in two pots to provide C–H alkenylation products with excellent selectivity.[22]
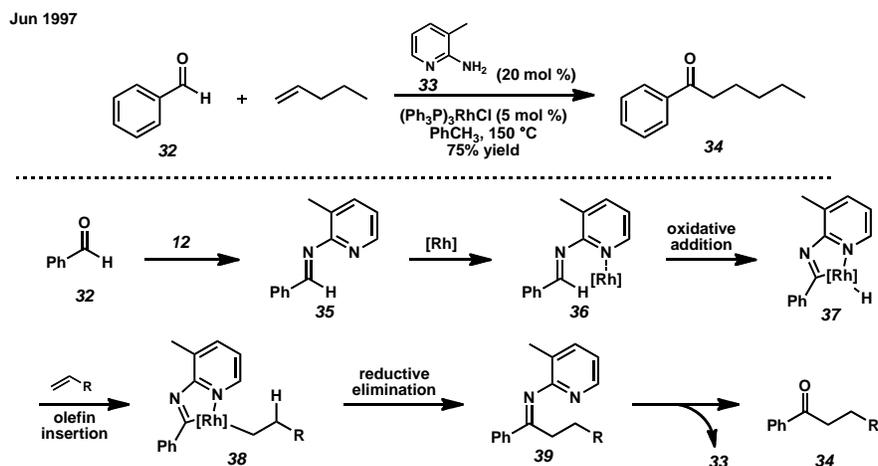
Scheme 1.3.3
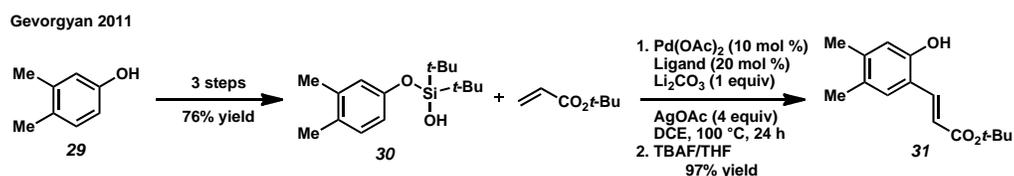
1.4 Transient Directing Group Scaffolds

While there have been a few examples of easily convertible directing groups, these methods still require several steps to both install and remove the necessary functionality for high

levels of selectivity in each substrate. The goal of our research is to accomplish directed C–H functionalization with an *in situ* removable directing group. In a few demonstrated chelation-assisted transformations a directing group is attached to the desired substrate to form a transient reactive intermediate, which can then undergo some kind of functionalization. Jun and coworkers reported a chelation-assisted hydroacylation of benzaldehyde with a picoline catalyst (**33**) (Scheme 1.4.1).[23] In this transformation, an internal directing group is prepared *in situ* to enable C–H activation. Treatment of benzaldehyde with 1-pentene in the presence of catalytic rhodium and 2-amino-3-picoline (**33**) afforded the hydroacylated product in 75% yield. Jun and coworkers proposed that **33** condenses onto benzaldehyde to give imine **35** (Scheme 1.4.1). Coordination to imine **35** by rhodium gives amino-rhodium species **36**, which is converted to 5-membered rhodacycle **37** via oxidative C–H activation. Olefin insertion to give rhodacycle **38** followed by reductive elimination affords hydroacylated intermediate **39**. Hydrolysis then provides desired hydroacylated product **34** and regenerates the directing ligand. Although the same transformation can be effected without the picoline additive, its use prevents an undesired decarbonylation reaction.

Scheme 1.4.1

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Breit and coworkers employed a bifunctional strategy towards the hydroacylation of aryl aldehydes (Scheme 1.4.2).[24] In this method, a picoline catalyst containing a pendant phosphine (**42**) was utilized to direct hydroacylations. This design allows bidentate-coordination between the catalyst and the metal. As a result, a lower catalyst loading was achieved while maintaining high reactivity by generating a higher effective concentration of activated substrate **44**. Using this catalyst, a hydroacylation of benzaldehyde with 1-octene was achieved in 83% yield. An intramolecular hydroacylation was also promoted with this catalyst system, converting **45** to cyclized **46** in excellent yield. In general, intramolecular hydroacylation suffers from complications with polymerization and decarbonylation. This transformation was effected with low catalyst loadings and high yield with the bifunctional catalyst.

Scheme 1.4.2
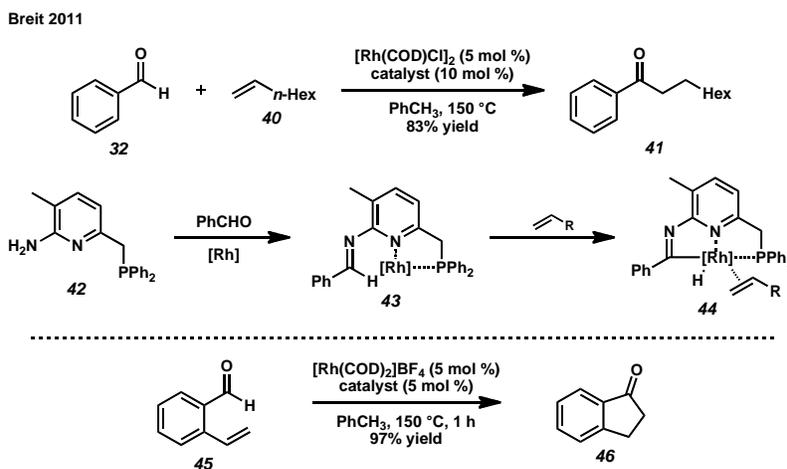
Bedford and coworkers demonstrated a C–H arylation using a transient directing group.[25] In this method, a rhodium-catalyzed ortho-arylation of phenols was described using a phosphinite directing group (Scheme 1.4.3). Mechanistically, it was proposed that the rhodium (I) species undergoes an oxidative addition of an aryl bromide to generate **53**. When coordinated to the rhodium (III) complex, **49** can then undergo an orthometallation to give rhodacycle **50**. Reductive elimination provides the arylated phosphinite product (**52**). Upon a transesterifcation

process, a new substrate exchanges with the arylated phosphinite to release the cross-coupled product (**48**). The use of the phosphinite catalyst provided a single arylated product, ortho to the phenol. One limitation of this chemistry is the need for a bulky *t*-butyl group at the other ortho position to achieve high yields. Decreasing the steric bulk of this position provided lower yields, and with no ortho substitution, no product was isolated.
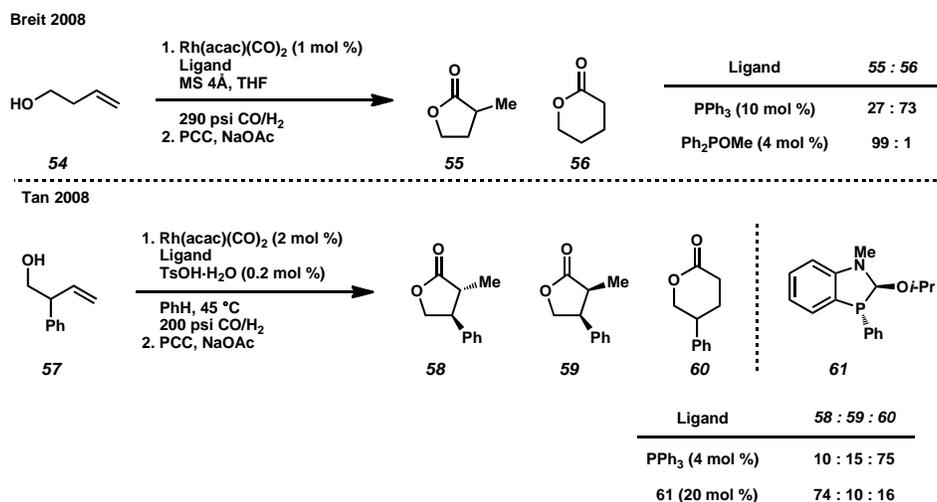
Scheme 1.4.3
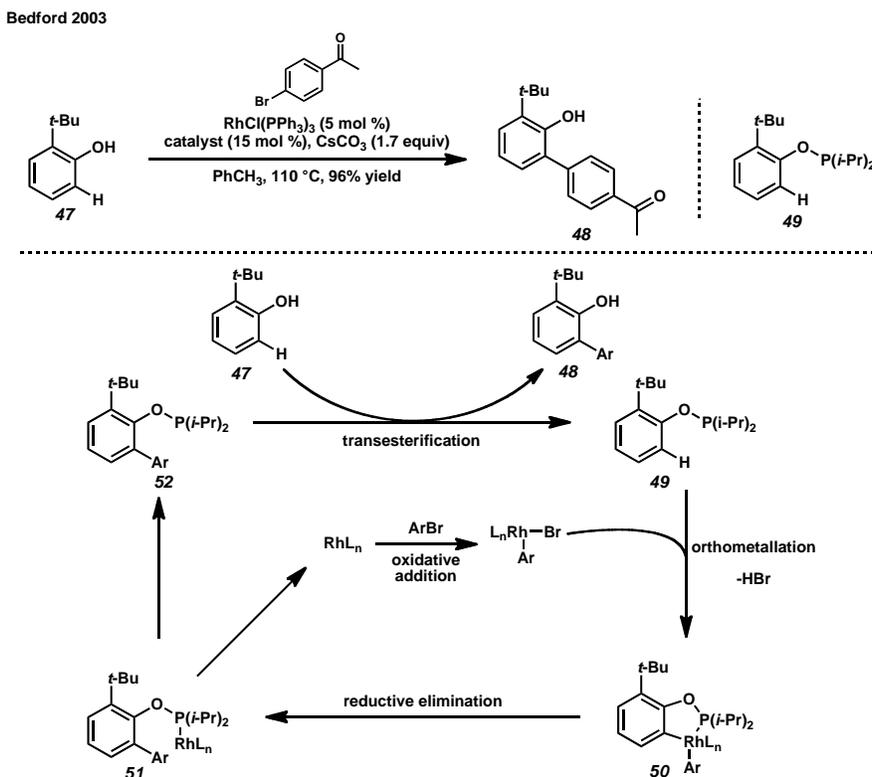
Breit and coworkers have also used this concept in hydroformylation reactions, where they employed a phosphinite catalyst to promote desired reactivity.[26] With triphenylphosphine as an external directing group, hydroformylation of **54** was observed under rhodium catalysis to give lactols, which afforded lactones **55** and **56** upon oxidation (Scheme 1.4.4). Good selectivity for the linear hydroformylation product (**56**) was observed under these conditions. When triphenylphosphine was exchanged for the phosphinite ligand, however, the hydroformylation

reaction proceeded with excellent selectivity for the branched product (**55**). Breit and coworkers

proposed that the alcohol substrate exchanges with methoxy group, as a ligand on phosphorous,

generating an internal phosphine-directing group.   This intermediate directed the

hydroformylation of **54** with enhanced reactivity to give the branched product (**55**) in excellent

selectivity.   Tan and coworkers employed a similar model to achieve high regio- and

diastereoselectivity in hydroformylation reactions (Scheme 1.4.4).[27]   In their system,

hydroformylation occured in the presence of an external phosphine catalyst to give linear product

60 in good selectivity. Introduction of a specialized phosphine catalyst (**61**), which can undergo

ligand exchange with the alcohol substrate (**57**), allowed a now internal phosphine catalyst to

direct hydroformylation to give the branched product (**58**) in good regio- and diastereoselectivity

(Scheme 1.4.4).

Scheme 1.4.4

Breit 2008

Ligand	55 : 56
PPh$_3$ (10 mol %)	27 : 73
Ph$_2$POMe (4 mol %)	99 : 1

Tan 2008

Ligand	58 : 59 : 60
PPh$_3$ (4 mol %)	10 : 15 : 75
61 (20 mol %)	74 : 10 : 16

Tan and coworkers have recently developed a new enantioselective hydroformylation

using a transient directing group scaffold.[28]   In comparison to their original catalyst, the new

scaffold contains an additional non-epimerizable stereocenter (Scheme 1.4.5). In catalyst **61**,

both stereocenters are epimerizable under the reaction conditions, preventing any high degree of

stereoinduction. In catalyst **64**, however, the stereocenter with the isopropyl group is permanently set, preserving the stereochemical integrity of the other stereocenters, as they maintain an *anti, anti* configuration in order to prevent *syn*-pentane interactions. Using this catalyst, Tan and coworkers obtained the hydroformylated product (**63**) in good yield and high enantioselectivity. Rational catalyst design of this nature to incorporate stereoselectivity may be very important in the development of enantioselective C–H functionalization.

Scheme 1.4.5
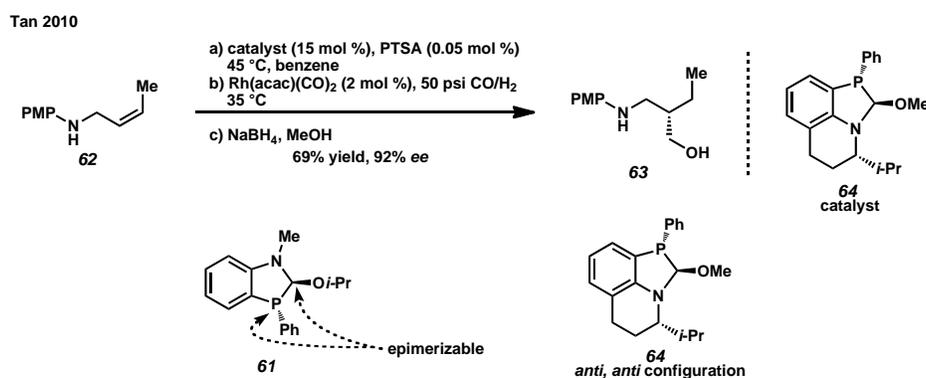
Yu and coworkers have recently developed a carboxylic acid directed C–H olefination reaction.[29] Subjecting achiral **65** to catalytic Pd(OAc)$_2$, BQ, KHCO$_3$ and an amino acid ligand in the presence of O$_2$ afforded the desymmeterized olefinated product (**66**) in 97% *ee* (Scheme 1.4.6). The amino acid, Boc-L-isoleucine, acts as a ligand on palladium to perform a diastereoselective C–H activation to generate **67**. Migratory insertion into the olefin and β-hydride elimination afforded olefinated product **66** and Pd^0. Reoxidation via BQ and O$_2$ afforded the active Pd^{II} catalyst. This example demonstrates that rational ligand design and reaction development can generate enantiopure products from simple achiral starting materials without the need for molecular complexity.

Scheme 1.4.6
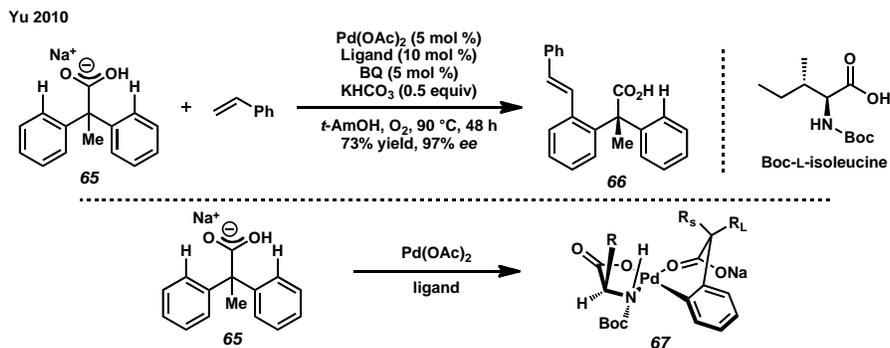
1.5 Research Objectives Revisited

In an effort to unify the power of directed C–H activation with chelate-assisted functionalization, we aim to develop C–H activation methodology mediated by a transient directing group. We envision designing a ligand that has both an acetalization component and an attached ligating group (Scheme 1.5.1). The acetalization component will condense onto a carbonyl substrate, forming a covalently bound intermediate with an internal directing group. The tethered ligating group will then direct metal-catalyzed functionalization to a remote carbon-hydrogen or carbon-carbon bond. Hydrolysis of the ligand will reveal the functionalized product and regenerate the free ligand. Proposed to be catalytic in both ligand and metal, this functionalization will generate molecular complexity in rapid fashion without the need for additional directing group removal steps. With proper design, the scaffold will be able to direct a C–H functionalization with high regio- and stereoselectivity.

Scheme 1.5.1
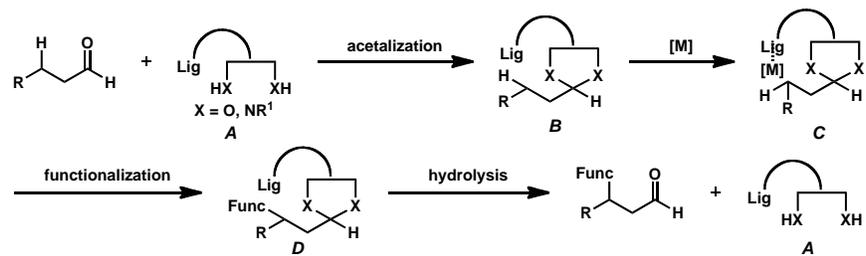
1.6   References and Notes

[1] Lyons, T. W.; Sanford, M. S. *Chem. Rev.* **2010**, *110*, 1147-1169.

[2] Kleimann, J. P.; Dubeck M. *J. Am. Chem. Soc.* **1963**, *85*, 1544-1545.

[3] Sieckman, R. W.; Cope, C. A. *J. Am. Chem. Soc.* **1965**, *87,* 3272-3273.

[4] Dick, A. R.; Hull, K. L.; Sanford, M. S. *J. Am. Chem. Soc.* **2004**, *126*, 2300-2301.

[5] Desai, L. V.; Hull, K. L.; Sanford, M. S. *J. Am. Chem. Soc.* **2004**, *126*, 9542-9543.

[6] Giri, R.; Liang, J.; Lei, J. –G.; Li, J. –J.; Wang, D. –H.; Chen, X.; Naggar, I. C.; Guo, C.; Foxman, B. M.; Yu, J. –Q. *Angew. Chem. Int. Ed.* **2005**, *44*, 7420-7424.

[7] Wang, D. –H.; Wu, D. –F.; Yu, J. –Q. *Org. Lett.* **2006**, *8*, 3387-3390.

[8] Dick, A. R.; Kampf, J. W.; Sanford, M. S. *J. Am. Chem. Soc.* **2005**, *127*, 12790-12791.

[9] Kalyani, D.; Deprez, N. R.; Desai, L. V.; Sanford, M. S. *J. Am. Chem. Soc.* **2005**, *127*, 7330-7331.

[10] (a) Powers, D. C.; Ritter T. *Nat. Chem.* **2009**, *1*, 302-309. (b) Powers, D. C.; Giebel, M. A. L.; Klein, J. E. M. N.; Ritter, T. *J. Am. Chem. Soc.* **2009**, *131*, 17050-17051. (c) Deprez, N. R.; Sanford, M. S. *J. Am. Chem. Soc.* **2009**, *131*, 11234-11241.

[11] Albrecht, M. *Chem. Rev.* **2010**, *110*, 576-623.

[12] Colby, D. A.; Bergman, R. G.; Ellman, J. A. *Chem. Rev.* **2010**, *110*, 624-655. Oxidative addition and σ-bond metathesis are more common with rhodium and iridium and are generally not applicable in the C–H activation with palladium.

[13] Davies, D. L.; Donald, S. M. A.; Macgregor, S. A. *J. Am. Chem. Soc.* **2005**, *127*, 13754-13755.

[14] Lapointe, D.; Fagnou, K. *Chem. Lett.* **2010**, *39*, 1118-1126.

[15] Desai, L. V.; Stowers, K. J.; Sanford, M. S. *J. Am. Chem. Soc.* **2008**, *130*, 13285-13293.

[16] Chernyak, N.; Dudnik, A. S.; Huang, C.; Gevorgyan, V. *J. Am. Chem. Soc.* **2010**, *132*, 8270-8272.

[17] AgOAc provides a source of acetate in the reaction to regenerate Pd(OAc)₂. It's complete role in the reaction is not well understood.

[18] Neufeldt, S. R.; Sanford, M. S. *Org. Lett.* **2010**, *12*, 532-535.

[19] Aldoximes do not function well under these conditions, as they have a propensity to eliminate to form the nitrile.

[20] Huang, C.; Chattopadhyay, B.; Gevorgyan, V. *J. Am. Chem. Soc.* **2011**, *133,* 12406-12409.

[21] Conditions using (+)menthyl(O₂C)-Leu-OH as a catalyst were developed by Yu and coworkers for the alcohol directed C–H alkenylation of sp^2 C–H bonds. Lu, Y.; Wang, D. –H.; Engle, K. M.; Yu, J. –Q. *J. Am. Chem Soc.* **2010**, *132*, 5916-5921.

(+)menthyl(O₂C)-Leu-OH
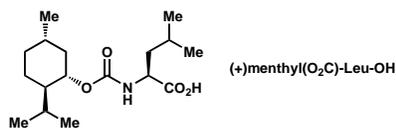
[22] These conditions work well for very electron rich aromatic systems, but the yields decrease significantly when employing electron-deficient aromatics. Furthermore, only activated olefins are utilized in these transformations.

[23] (a) Lee, H.; Hong, J. –B.; Jun, C. –H. *J. Org. Chem.* **1997**, *62,* 1200-1201. (b) Hong, J. –B.; Lee, D. –Y.; Jun, C. –H. *Synlett*, **1999***,* 1-12.

[24] Vautravers, N. R.; Regent, D. D.; Breit, B. *Chem. Commun.* **2011***, 47*, 6635-6637.

[25] Bedford, R. B.; Coles, S. J.; Hursthouse, M. B.; Limmert, M. E. *Angew. Chem. Int. Ed.* **2003***, 42*, 112-114.

[26] (a) Grünanger, C. U.; Breit, B. *Angew. Chem., Int. Ed.* **2008**, *47*, 7346-7349. (b) Grünanger, C. U.; Breit, B. *Angew. Chem. Int. Ed.* **2010**, *49*, 967-970.

[27] Lightburn, T. E.; Dombrowski, M. T.; Tan, K. L. *J. Am. Chem. Soc.* **2008**, *130*, 9210.

[28] Worthy, A. D.; Joe, C. L.; Lightburn, T. E.; Tan, K. L. *J. Am. Chem. Soc.* **2010**, *132*, 14757-14759.

[29] (a) Shi, B. -F.; Zhang, Y. -H.; Lam, J. K.; Wang, D. –H.; Yu, J. –Q. *J. Am. Chem. Soc.* **2010**, *132*, 460-461. (b) Wang, D. –H.; Engle, K. M.; Shi, B. –F.; Yu, J. –Q. *Science* **2010***, 327*, 315.

Chapter 2

Acetoxylation of *sp²* and *sp³* C–H Bonds

In an effort to unify the power of directed C–H functionalization with chelate-assisted functionalization, we aimed to develop a C−H acetoxylation of *sp²* and *sp³* C–H bonds mediated by a transient directing group (Scheme 2.0.1). We envisioned a ligand that has both an acetalization component and a ligating group. Based on work by Sanford[1] and Yu,[2] we imagined using an oxazoline or pyridyl ligating group to direct a metal-catalyzed functionalization of a remote C−H bond. The acetalization component needs to condense onto an aldehyde, therefore we envisioned using a combination of an alcohol or acid and amine or amide (**M**), which would afford substrate-ligand adduct **N**. Treatment with acetoxylation conditions would afford functionalized intermediate **P**. Subsequent hydrolysis of the ligand would reveal the functionalized product (**Q**), and regenerate the free ligand. Proposed to be catalytic in both ligand and metal, this method would require synchronization of initial acetalization, functionalization, and transacetalization.

Scheme 2.0.1. General Concept
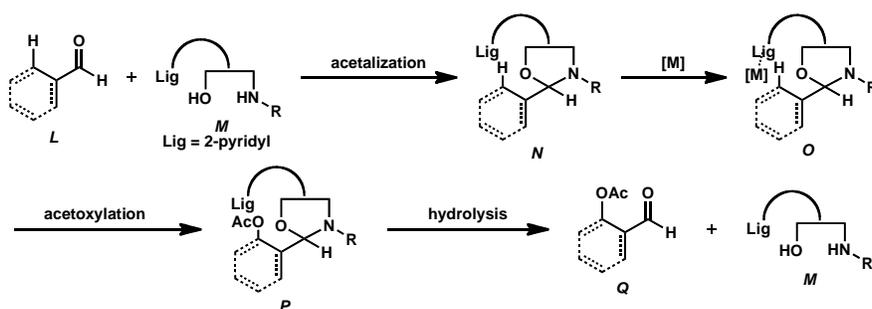
2.1 Ligand Synthesis

Our initial studies began with ligand design and synthesis. We imagined an amino alcohol could serve as the acetalization component, which should readily condense onto an

aldehyde forming an *N,O*-acetal. For a ligating group, an oxazoline or 2-pyridyl moiety would be straightforward to install, and these have already been demonstrated as competent directing groups in C–H acetoxylation.[1-2] What remained to be determined was the structural alignment of the ligand. Based on our hypothesis, the formation of our substrate ligand adduct **N**, can occur with a *cis* or *trans* relationship between the substrate to be functionalized and the ligating group (Scheme 2.1.1). If the **O-*cis*** isomer were obtained, the relationship between the ligating group and substrate would allow for C–H functionalization. If the **O-*trans*** isomer were obtained, however, the ligating group and substrate would not be in proximity to promote a directed functionalization. Literature precedent revealed that [3.3.0] systems can provide high levels of the desired stereoselectivity. Seebach has demonstrated self-reproduction of chirality using amino acids to build structurally complex unnatural amino acids (Scheme 2.1.1).[3] L-Proline (**68**) was condensed onto pivaldehyde to afford **69** as a single diastereomer. Alkylation with benzyl bromide afforded **70**, again as a single diastereomer, with a *cis* relationship between the α-substituent and the alkyl group of the acetal. We believed we could exploit this method to achieve a *syn* relationship between our ligating group and substrate. In the event that trace amounts of the *anti* isomer is generated, it should be unreactive to functionalization conditions, and isomerization would presumably yield the *syn* diastereomer.

Scheme 2.1.1. Formulation of Our Approach
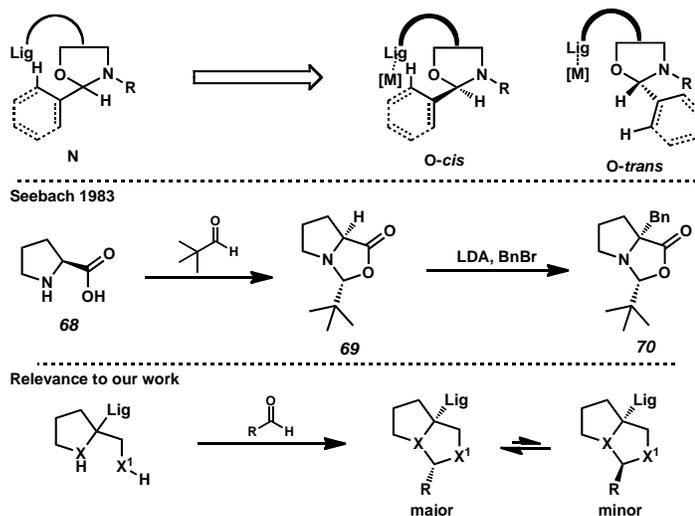
We imagined this [3.3.0] system could be used, yielding high levels of diastereoselectivity[3] and providing a rigid ligand scaffold. With these concepts in hand, we designed our ligand retrosynthesis from (*S*)-proline (Scheme 2.1.2). Installation of the directing group could be accomplished upon protection of the amino acid; deprotection and reduction of the acid would afford the desired ligand.

Scheme 2.1.2. Ligand Retrosynthesis
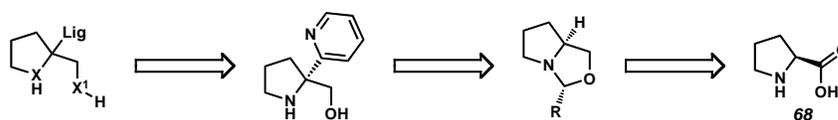
Employing Seebach's acetalization/alkylation chemistry, the *N,O*-acetal (**69**) was formed as a single diastereomer (Scheme 2.1.3).[3] Acetal **69** was unstable under air, and rapidly decomposed to the starting materials. An attempted arylation using Hartwig's enolate coupling chemistry[4] was unsuccessful at generating the desired pyridine ligand.

Scheme 2.1.3. Initial Ligand Synthesis
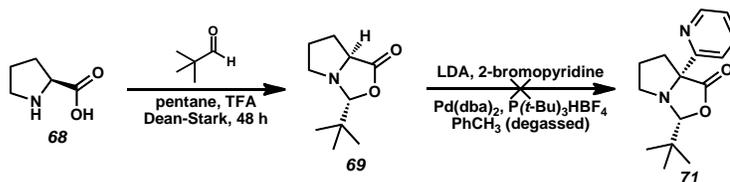
Realizing that compound **69** would be difficult to alkylate, we looked to find different condensation conditions that would render a more stable *N,O*-acetal. Following a revised Seebach procedure,[3] chloral hydrate could be condensed with proline (Scheme 2.1.4). Using these conditions we obtained **72** as a single diastereomer in 60% yield. Treatment with arylation conditions, however, proved unsuccessful to generate the pyridine ligand (**73**).

Scheme 2.1.4. A Revised Route
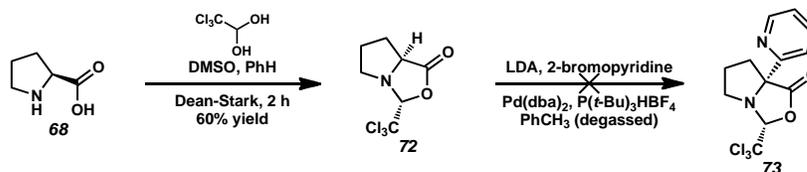
We expected using an oxazoline ligating group, rather than a pyridine, would better facilitate ligand synthesis. Alkylation with ethyl formate afforded the desired aldehyde as a stable isolable compound (Scheme 2.1.5). Several conditions were tried to oxidize the aldehyde to the acid,[5] with KMnO₄ being the most successful oxidant.[6] Treating **75** with oxalyl chloride afforded the corresponding acid chloride, which upon treatment with an amino alcohol afforded the desired amide. We investigated several procedures to transform the amide into the corresponding oxazoline.[7] Although we were able to generate the alkyl chloride in good yield, the base induced cyclization to afford the oxazoline was not reproducible.

Scheme 2.1.5. Synthesis of an Oxazoline Ligating Group
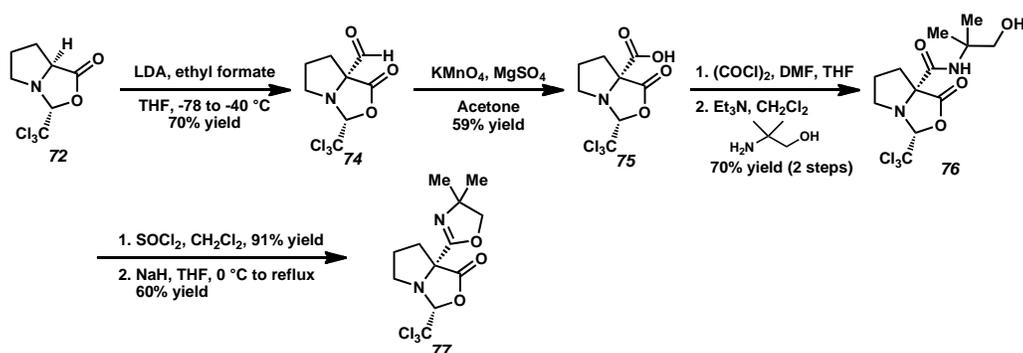
Ultimately, we employed an *in situ* oxidation/cyclization of the aldehyde to produce the corresponding oxazoline.[8] Treatment of aldehyde **74** with 2,2-dimethyl-1-aminoethanol, base and iodine afforded oxazoline **77** in one step in good yield (Scheme 2.1.6). To generate the amino alcohol ligand directly, we treated oxazoline **77** with a variety of different reducing agents, but were unable to isolate any of the desired product. Additionally, we tried to generate the methyl ester by treatment with anhydrous HCl in methanol, but were unsuccessful.

Scheme 2.1.6. Attempted Cleavage of the *N,O*-acetal
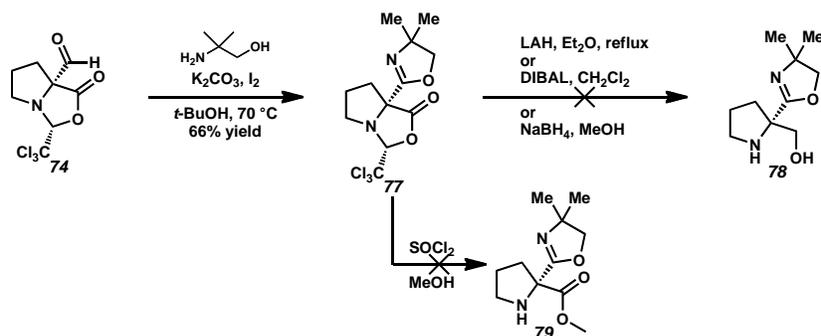
Believing the oxazoline to be too sensitive to *N,O*-acetal cleavage conditions, we thought a pryidine ligating group might be more robust and provide us with a ligand. Eager to install this ligating group to test our original hypothesis, we tried a standard alkylation procedure using 2-bromomethyl pyridine hydrobromide (Scheme 2.1.7). The alkylation was successful in installing

the pyridine directing group to generate **80** in modest yield. Several different bases were

examined in an effort to improve the yield of the alkylation, but were unsuccessful.

Scheme 2.1.7. Installation of a Pyridyl Ligating Group

Base	yield
LDA	19%
KHMDS	0%
LiHMDS	0%
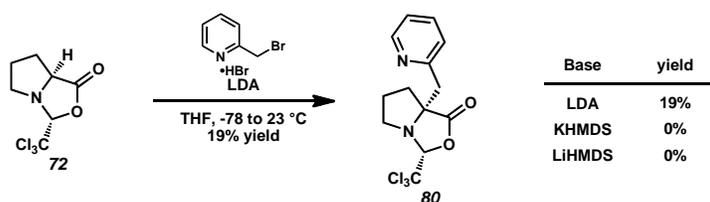
Treatment with typical acid hydrolysis conditions, however, did not afford the desired

amino acid (**81**) (Scheme 2.1.8). Believing the existence of pyridine and acid moieties to be

problematic, we tried to bypass the intermediate by generating the methyl ester (**82**) using

anhydrous HCl and methanol, but again were unsuccessful. In an attempt to generate amino

alcohol **83** directly we treated **80** with LiAlH$_4$ to reduce the acetal, but only observed

decomposition of the starting material. Furthermore, treating **72** with a reducing agent failed to

afford amino alcohol **84**, indicating that the *N,O*-acetal could not be cleaved under reductive
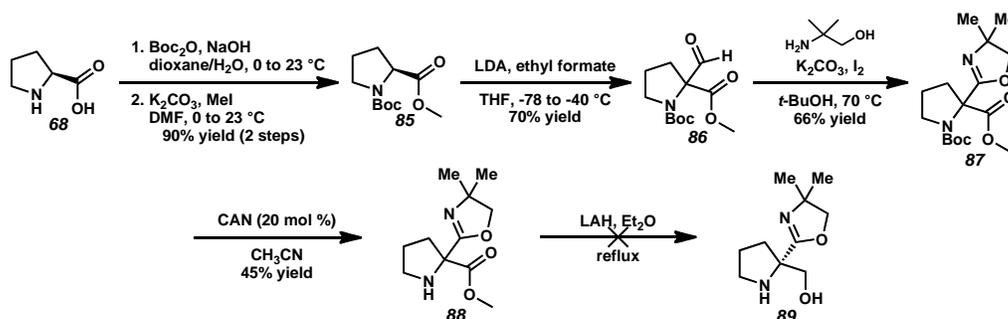
conditions.

Scheme 2.1.8. Hydrolysis of the *N,O*-acetal
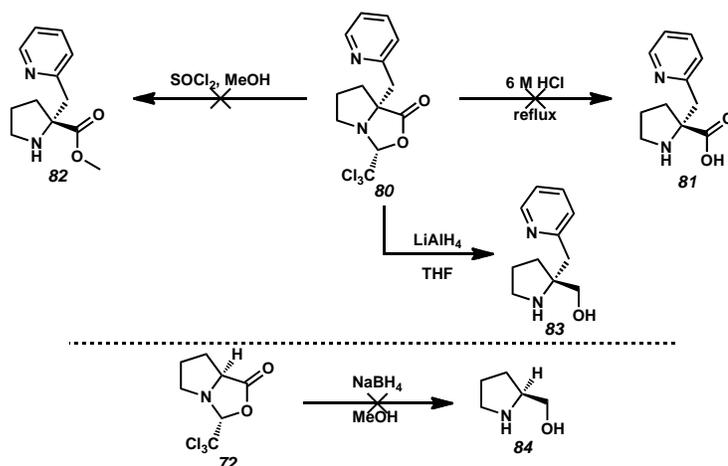
As the ligand synthesis via the *N,O*-acetal was proving unsuccessful, we considered that

protecting the amine and carboxylic acid independently would provide a more successful

approach. Formation of the *t*-butyl carbamate with subsequent esterification afforded (*S*)-Boc-proline methyl ester **85** (Scheme 2.1.9). To install the oxazoline, **85** was formylated with ethyl formate. Treatment with oxidative conditions and the amino alcohol afforded the desired oxazoline (**87**) in good yield. After trying several deprotection conditions[9] the Boc group was removed with catalytic CAN in CH_3CN.[10] Treatment with LAH to reduce the methyl ester, however, failed to provide any amino alcohol ligand, again demonstrating the high reactivity of the oxazoline moiety in our system.

**Scheme 2.1.9. A Revised Synthesis of an Oxazoline Ligand**

Believing a pyridine to be much more robust to the deprotection steps, we again sought to install this ligating moiety. Treatment with the arylation conditions was still unsuccessful, most likely due to the steric bulk of the protecting groups. Installing the methyl pyridine to generate **91** was low-yielding under several different conditions (Scheme 2.1.10). We attributed this yield to the low solubility of the electrophile salt and the super stoichiometric amount of LDA needed to neutralize the electrophile, which resulted in the competitive addition of LDA into the bromomethyl pyridine. We developed a procedure of neutralizing the electrophile outside of the main reaction with NaH (washed) in DMF at 0 °C. The addition of this suspension enolate improved the yield of **91**.

**Scheme 2.1.10. Revised synthesis of a pyridyl ligand**
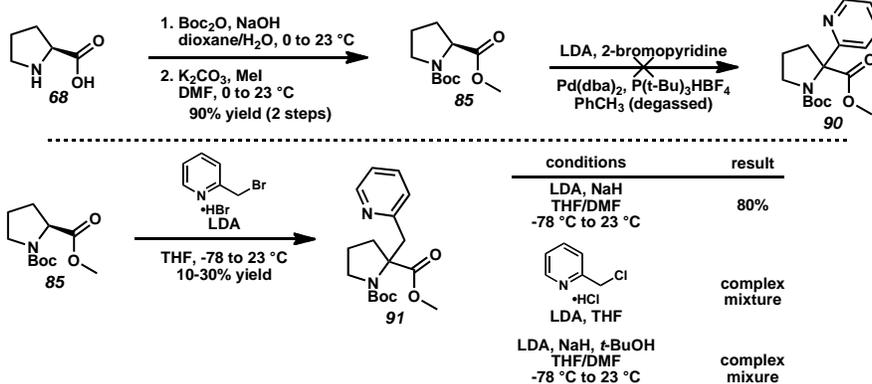
The removal of the Boc group was straightforward and occurred in high yield. Initial reduction of the methyl ester with DIBAL was unsuccessful (Scheme 2.1.11). Treatment of **91** with LAH, however, afforded the desired ligand (**92**) in good yield. The stability of **92** to air and water was not high, and led to its decomposition to a red oil. Limited handling in open atmosphere was necessary to use the ligand in further experiments. With the ligand in hand, functionalization was attempted, although initially unsuccessful. Treating cyclohexanecarboxaldehyde with a catalytic amount of ligand and acetoxylation conditions was ineffective at generating acetoxylated product **95** (Scheme 2.1.11). Rather than trying to functionalize a 2° sp^3 C–H bond, which has proven difficult,[11] we synthesized aldehyde **94** which contains a 1° sp^3 C–H bond. Treatment of this aldehyde with catalytic amounts of ligand and acetoxylation conditions was also unsuccessful.

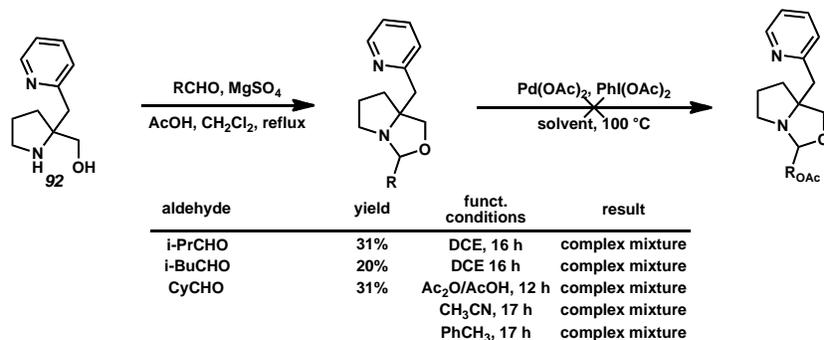**Scheme 2.1.11. Initial attempts at catalytic functionalization**
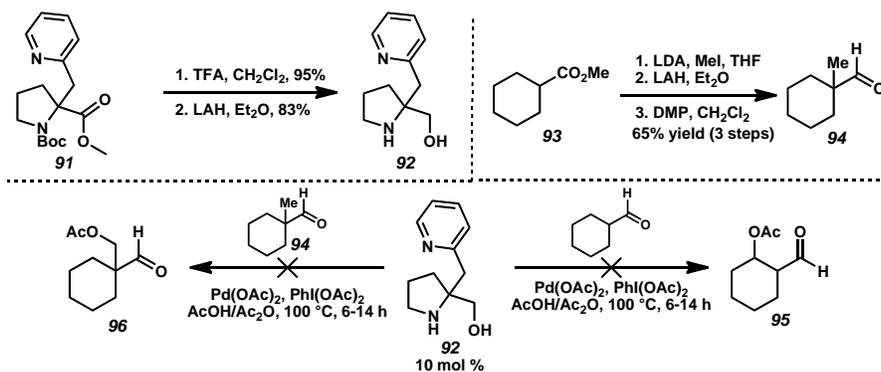
Realizing that the synchronization of acetalization, functionalization, and transacetalization was extremely challenging, we sought to study the system in a stoichiometric sense. This approach afforded the opportunity to study the nature of the *N,O*-acetal and the competency of our ligand for C–H functionalization. We first tried a one-pot, two-step procedure treating the aldehyde (**94**) and ligand **92** with AcOH in CH_2Cl_2, followed by treatment with acetoxylation conditions, but no functionalized ligand adduct was isolated (Scheme 2.1.12). Condensation onto octanal resulted in a highly labile *N,O*-acetal (**99**). When subjected to typical functionalization conditions, no desired acetoxylation product **100** was observed. Interestingly, when **99** was exposed to functionalization conditions at 100 °C, we observed acetoxylation of the α-position (to the acetal). Upon treating the substrate with stoichiometric oxidant in the absence of palladium we still observed the acetoxylated product; the product therefore likely arises from an enamine-catalyzed acetoxylation.[12] Treatment of octanal with acetoxylation conditions afforded no α-acetoxylation product.

**Scheme 2.1.12. Attempts at stoichiometric functionalization**
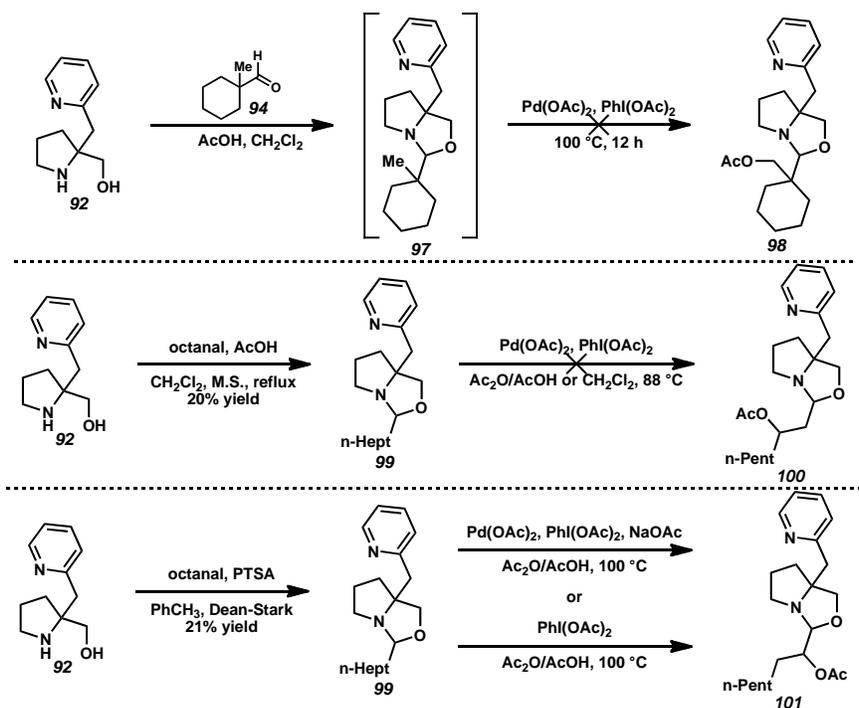
We believed the enamine induced processes were occurring as a result of *N,O*-acetal lability. To slow down these processes with aliphatic aldehydes we employed α-disubstituted aldehydes (Table 2.1.1). After condensation of the ligand with an aldehyde, each substrate-ligand adduct was treated to acetoxylation conditions. All of the substrates independent of solvent resulted in a complex mixture of products that could not be deconvoluted.

Table 2.1.1. Examination of α-substituted aldehydes

aldehyde	yield	funct. conditions	result
i-PrCHO	31%	DCE, 16 h	complex mixture
i-BuCHO	20%	DCE 16 h	complex mixture
CyCHO	31%	Ac₂O/AcOH, 12 h	complex mixture
		CH₃CN, 17 h	complex mixture
		PhCH₃, 17 h	complex mixture

Lastly, due to our unsuccessful acetoxylation attempts with aliphatic aldehydes, we turned to functionalize aromatic aldehydes (Scheme 2.1.13). Condensation of the ligand (**92**) onto benzaldehyde afforded **105** in moderate yield. Treatment with acetoxylation conditions, however, was unsuccessful at providing functionalized products.

**Scheme 2.1.13. Attempted functionalization of benzaldehyde**
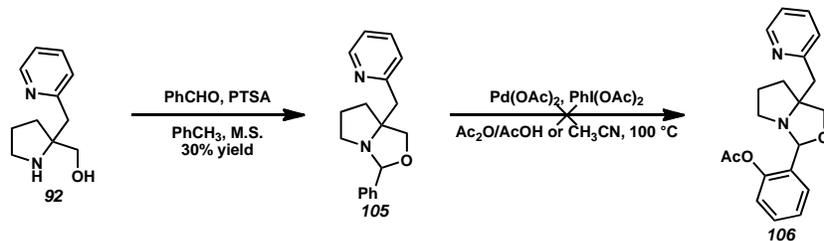
While we have been able to synthesize the desired ligand scaffolds and condense them onto aldehydes, the products of these condensations have shown to be rather unstable. Seemingly the instability of the acetal may lend satisfactorily to the transacetalization we hope to achieve. If the linkage is too labile however, oxidative side products may dominate the reactions, and prevent the desired C–H functionalization.

2.2 Acetoxylation of Isobutyraldehyde

We attributed our difficulty in effecting functionalization to the lability of the *N,O*-acetal and attempted to find a more stable acetalization component. Literature accounts show that when amino amides were condensed onto aromatic aldehydes, the resulting *N,N*-aminal was a stable, isolable product.[13] We sought to convert protected amino ester **91** to the amide via the carboxylic acid; however, saponification was unsuccessful (Scheme 2.2.1). Direct conversion to the methyl amide (**108**) via methylamine was ineffective. Exposing the methyl ester to aniline in toluene at 100 °C did not afford any of the phenyl amide product (**109**). Likewise, heating to 200 °C in xylene, or treating with KCN in toluene or THF were ineffective at generating the

desired amide. We also tried to generate the aluminum amide, but this was also ineffective at displacing the methyl ester.

Scheme 2.2.1. Conversion to an amino amide
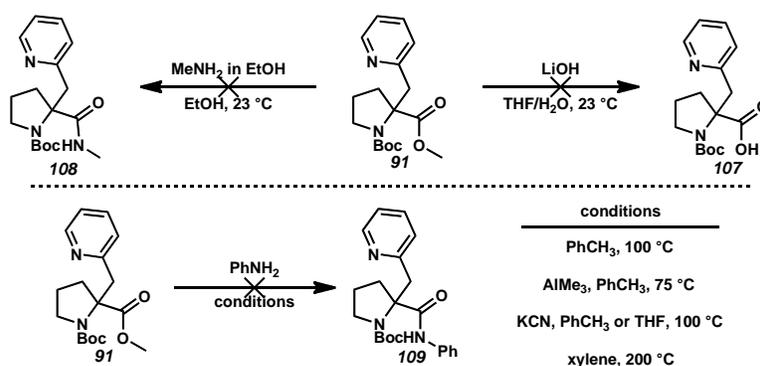
The installation of an amide from the methyl ester appeared to be an impractical route to an amino amide. Rather than employing a saponification of the methyl ester, we imagined using a hydrogenation of a benzyl ester to generate the desired acid. Formation of the benzyl ester occurred in high yield, and alkylation with 2-bromomethyl pyridine installed the ligating group (Scheme 2.2.2). Hydrogenation with Pd/C and H_2 appeared to afford the desired carboxylic acid. Any attempt to convert the acid into the amide, however, was unsuccessful.

Scheme 2.2.2. An alternative approach to an amino amide
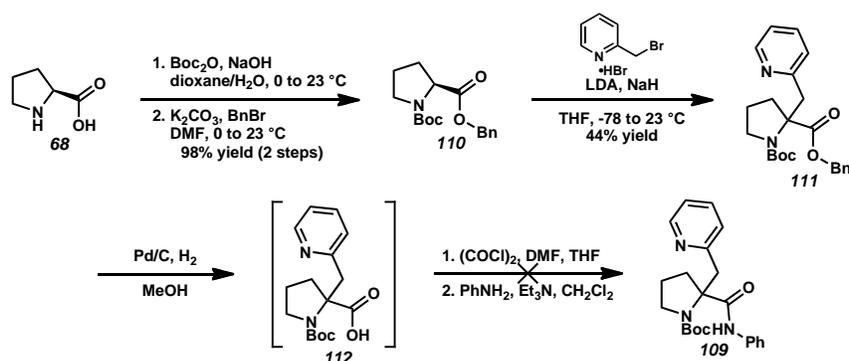
Troubled by the difficulty of generating the amino amide ligand with the methyl pyridine ligating group, we began to examine a new ligating group. We imagined that a pyridine directly attached to the [3.3.0] system may be successful. By changing the conformational flexibility of

the system, differential activity may be imparted, and may enable us to convert the methyl ester into the desired amide. Nucleophilic aromatic substitution with **85** afforded the arylated product (**113**) (Scheme 2.2.3). Treatment with trimethylaluminum and aniline directly afforded the desired amide.[14] Deprotection with TFA proceeded with high yield to provide amino amide ligand **114**. Isolation of this compound proceeded smoothly and decomposition was not observed under ambient conditions. Condensation onto isobutyraldehyde afforded **115** in good yield as a stable *N,N*-aminal.
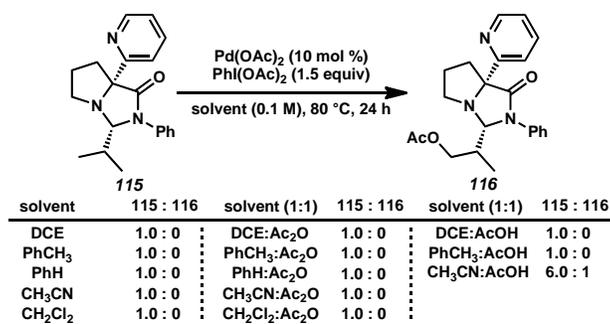
Scheme 2.2.3. Generation of an amino amide
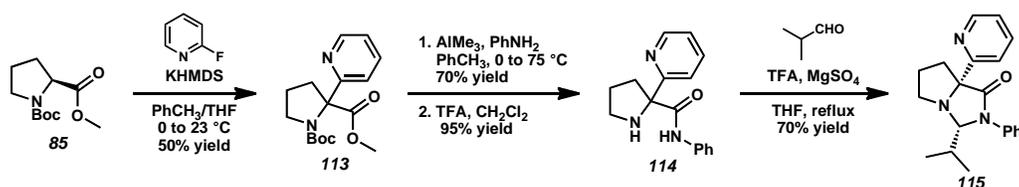
Initial attempts at acetoxylation were unsuccessful, using either $PhI(OAc)_2$ or IOAc as the stoichiometric oxidant at 100 °C in DCE (Scheme 2.2.4). When reacted under Sanford's original conditions with catalytic $Pd(OAc)_2$ and stoichiometric $PhI(OAc)_2$ in $AcOH/Ac_2O$ at 85 °C, monoacetoxylated product **116** was observed in 10% isolated yield, observed as a single diastereomer. In the absence of palladium no reaction was observed, suggesting that the transformation was a palladium-catalyzed C–H acetoxylation. Initial optimization attempts revealed that as temperature increased, an increase in the ratio of product to starting material was observed, with the highest ratio occurring at 85 °C, but still only providing 10% yield. Above 100 °C, only iminium-catalyzed products and decomposition products were observed.

Scheme 2.2.4. Initial acetoxylation of isobutyraldehyde

temp °C	115 : 116
110	1.0 : 0
70	5.9 : 1
75	6.0 : 1
80	2.8 : 1
85	2.7 : 1 (10% yield)
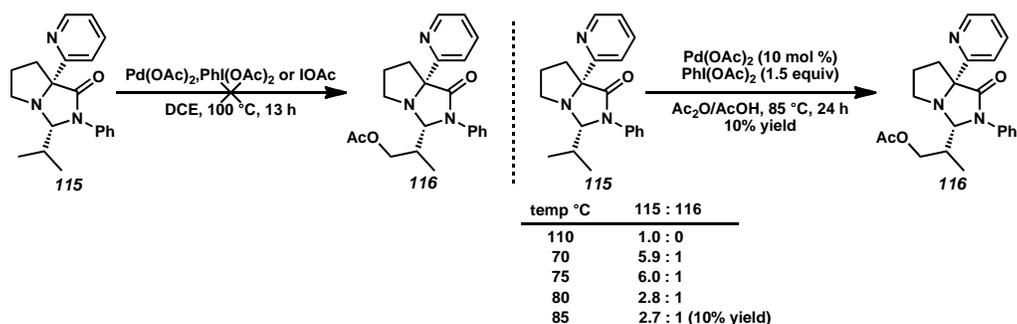
We began the optimization of the acetoxylation reaction by a solvent screen. As seen in Table 2.2.1, when the reaction was run in any solvent but AcOH/Ac$_2$O, no product was observed. Yu and coworkers found that Ac$_2$O was essential for regenerating the active palladium catalyst.[2] Attempts with various mixed solvent systems with Ac$_2$O again provided no product. Mixed solvent systems with AcOH yielded no product with DCE or PhCH$_3$, but did provide a small amount of product when using CH$_3$CN. Believing AcOH and Ac$_2$O to be essential to the reaction, we tried adding multiple equiv of each to reactions run in DCE, PhCH$_3$ and CH$_3$CN, but no product was observed.

Table 2.2.1. Solvent screen

solvent	115 : 116	solvent (1:1)	115 : 116	solvent (1:1)	115 : 116
DCE	1.0 : 0	DCE:Ac$_2$O	1.0 : 0	DCE:AcOH	1.0 : 0
PhCH$_3$	1.0 : 0	PhCH$_3$:Ac$_2$O	1.0 : 0	PhCH$_3$:AcOH	1.0 : 0
PhH	1.0 : 0	PhH:Ac$_2$O	1.0 : 0	CH$_3$CN:AcOH	6.0 : 1
CH$_3$CN	1.0 : 0	CH$_3$CN:Ac$_2$O	1.0 : 0		
CH$_2$Cl$_2$	1.0 : 0	CH$_2$Cl$_2$:Ac$_2$O	1.0 : 0		

In an attempt to isolate a putative C–H activated palladacycle, we treated the isobutyraldehyde substrate (**115**) with stoichiometric palladium in the absence of oxidant (Scheme 2.2.5). Subsequent treatment with diphenylphosphinoethane (dppe) released the

starting material. Based on this result, we believe that the palladium coordinated to the substrate in a bi-dentate fashion to afford **117**, but based on ^{1}H NMR, no C–H activation was observed.

**Scheme 2.2.5. Generation of a palladacycle**
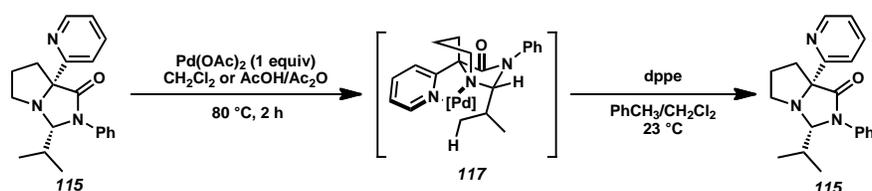
The low levels of conversion in these reactions were curious to us. Treatment with stoichiometric palladium and the addition of oxidant gave near complete consumption of starting material, affording the major product as the monoacetoxylated compound (**116**) and the minor as the diacetoxylated product (**118**) (Scheme 2.2.6). Due to these findings and low catalyst loading providing only a small amount of desired product (<10%), we examined the catalytic turnover of the palladium catalyst. Increasing the catalyst loading of palladium to 25 mol % followed by treatment with dppe afforded an increase in the ratio of product to starting material, suggesting less than two catalytic turnovers. Indeed, when catalyst loading was increased to 50 mol %, over 75% conversion of the starting material was observed. We postulated that the low turnover may be due to the formation of PdI$_2$ which is inactive in acetoxylation reactions.[15] Treatment of **115** with two equiv of AgOAc in the presence of acetoxylation conditions, however, failed to improve the turnover number. We attribute the low conversion to the incredibly strong binding of the ligand to palladium. It is interesting to note, however, despite the high catalyst loadings, the product was isolated with high diastereoselectivity, speaking to the high selectivity of the C–H functionalization. Further optimization found 50 mol % Pd(OAc)$_2$ with 1.8 equiv of oxidant for 24 h to yield 66% of **116** in 10 to1 dr and 13% of the diacetoxylated product (**118**).

Scheme 2.2.6. Catalyst turnover

|	115 : 116 : 118
25 mol %	10.4 : 5.7 : 1
50 mol %	1.9 : 6.0 : 1
100 mol %	1 : 4.8 : 1.1
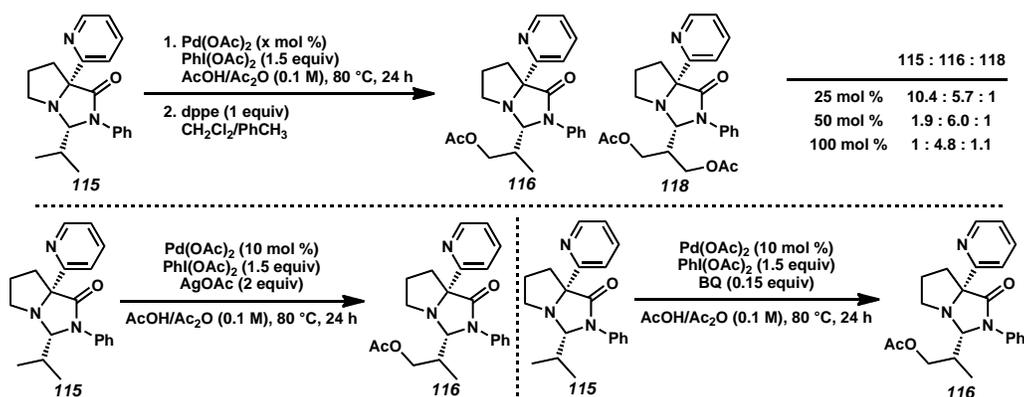
In order to learn more about the directing capability of our scaffold, we tried to establish that the pyridine-ligating group was essential for functionalization. In our system it is possible that the pyrrolidine nitrogen, phenyl amide or pyridine could be directing the functionalization. The role of the pyridyl group could be determined by removing it and observing the outcome. Formation of the phenyl amide via mixed anhydride chemistry and subsequent deprotection allowed us to generate amino amide **122** (Scheme 2.2.7). Condensation onto isobutyraldehyde afforded the desired *N,N*-aminal. With no pyridyl ligating group we expected that functionalization could not occur. Rather than no reaction, however, we observed oxidation of the pyrrolidine ring to the corresponding pyrrole (**124**), necessitating the presence of a α-substituent. We installed a benzyl group at the α-position to mimic the size of the ligating group, without the nitrogen. When subjected to the functionalization conditions, no acetoxylation was observed, suggesting that the pyridine-ligating group is essential for the C–H acetoxylation in our system.
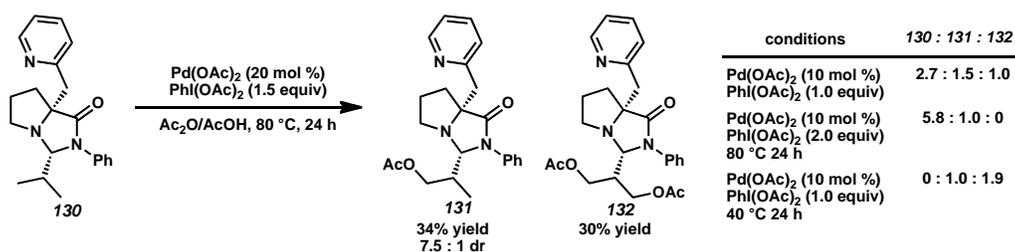
Scheme 2.2.7.
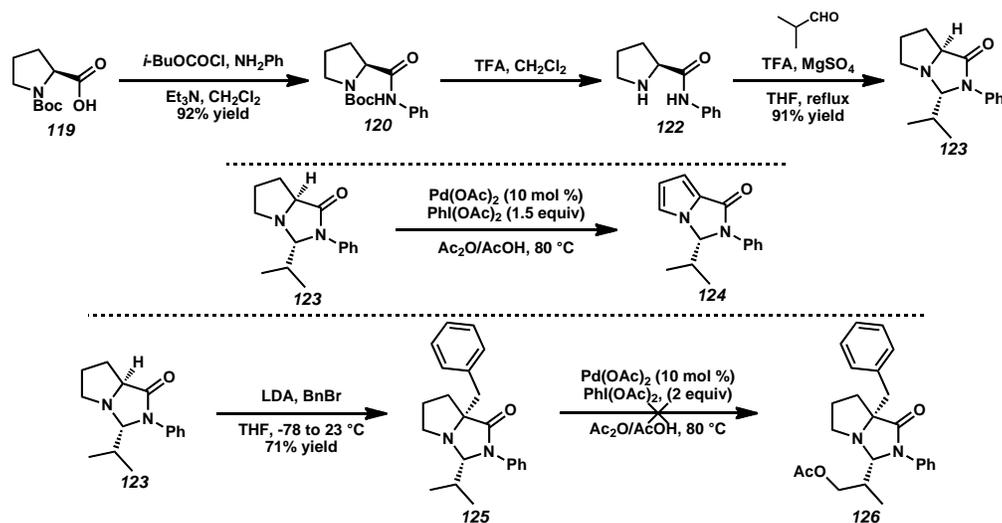
Encouraged by our ligand design providing stoichiometric C–H acetoxylation with high selectivity, we imagined changing the connectivity of the pyridine ligating group to our acetalization component by adding a methylene linker. This new scaffold would likely allow for greater conformational flexibility, and thereby likely impart differential reactivity. Despite all of our unsuccessful attempts to convert methyl ester **91** to phenyl amide **109**, we eventually treated the ester with phenyl amide to afford the desired product in modest yield. Subsequent deprotection afforded the amino amide ligand, which could be condensed onto isobutyraldehyde to afford test substrate **128**. Ultimately, we realized we could use our synthesis of **125** (*vide supra*) to synthesize the amino amide scaffolds stereoselectively (Scheme 2.2.8). Subjecting **123** to the appropriate base and electrophile afforded both **129** and **130** in good yield with complete diastereoselectivity.

Scheme 2.2.8. Asymmetric ligand synthesis
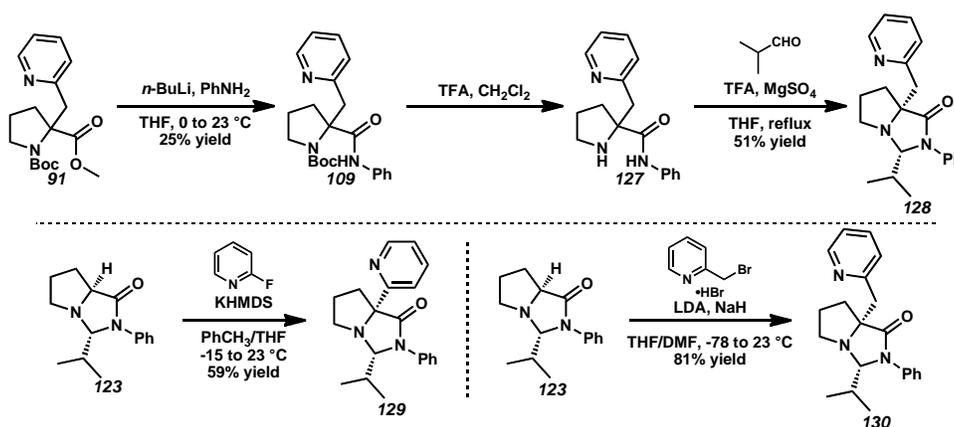
Treating substrate **130** with acetoxylation conditions afforded a nearly 1 : 1 mixture of mono- and diacetoxylated products (Scheme 2.2.9). The addition of a methylene linker to the pryidine ligating group dramatically increased the reactivity of the functionalization event. Decreasing the catalyst loading and equiv of oxidant resulted in incomplete conversion to product. Increasing the number of equiv of oxidant resulted in higher amount of diacetoxylated product (**132**). When the acetoxylation was run at lower temperatures, limited conversion to monoacetoxylated product **131** was observed.

Scheme 2.2.9. Acetoxylation of isobutyraldehyde

| conditions	130 : 131 : 132
Pd(OAc)$_2$ (10 mol %) PhI(OAc)$_2$ (1.0 equiv)	2.7 : 1.5 : 1.0
Pd(OAc)$_2$ (10 mol %) PhI(OAc)$_2$ (2.0 equiv) 80 °C 24 h	5.8 : 1.0 : 0
Pd(OAc)$_2$ (10 mol %) PhI(OAc)$_2$ (1.0 equiv) 40 °C 24 h	0 : 1.0 : 1.9

Analysis of the length of reaction was done to determine the extent of diacetoxylation as a function of time, as well as decomposition. At 8 h, good conversion to monoacetoxylated product was observed (Scheme 2.2.10). With longer reaction times, the conversion appeared to decrease, suggesting that decomposition of the products may be occurring.

Scheme 2.2.10. Time screen for acetoxylation

time (h)	*130 : 131 : 132*
8	1 : 4.2 : 2.8
10	1 : 3.4 : 2.5
16	1 : 3.4 : 4.0
18	1 : 2.3 : 2.3
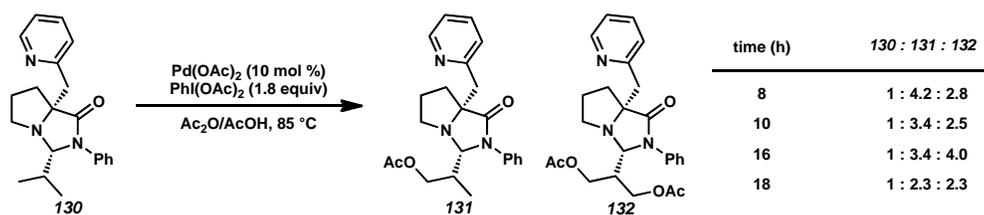
Upon optimization we found that changing the temperature of the reaction affected both the amount of diacetoxylated product and the diastereoselectivity of the monoacetoxylated product (Scheme 2.2.11). At 55 °C after 24 h 41% of the monoacetoxylated product was observed with 10 : 1 dr, with only 13% diacetoxylated product. Upon increasing the temperature to 70 °C, after 13 h 48% of the monoacetxoylated product was observed with 5.9 : 1 dr, and 19% diacetoxylated product was isolated.

Scheme 2.2.11

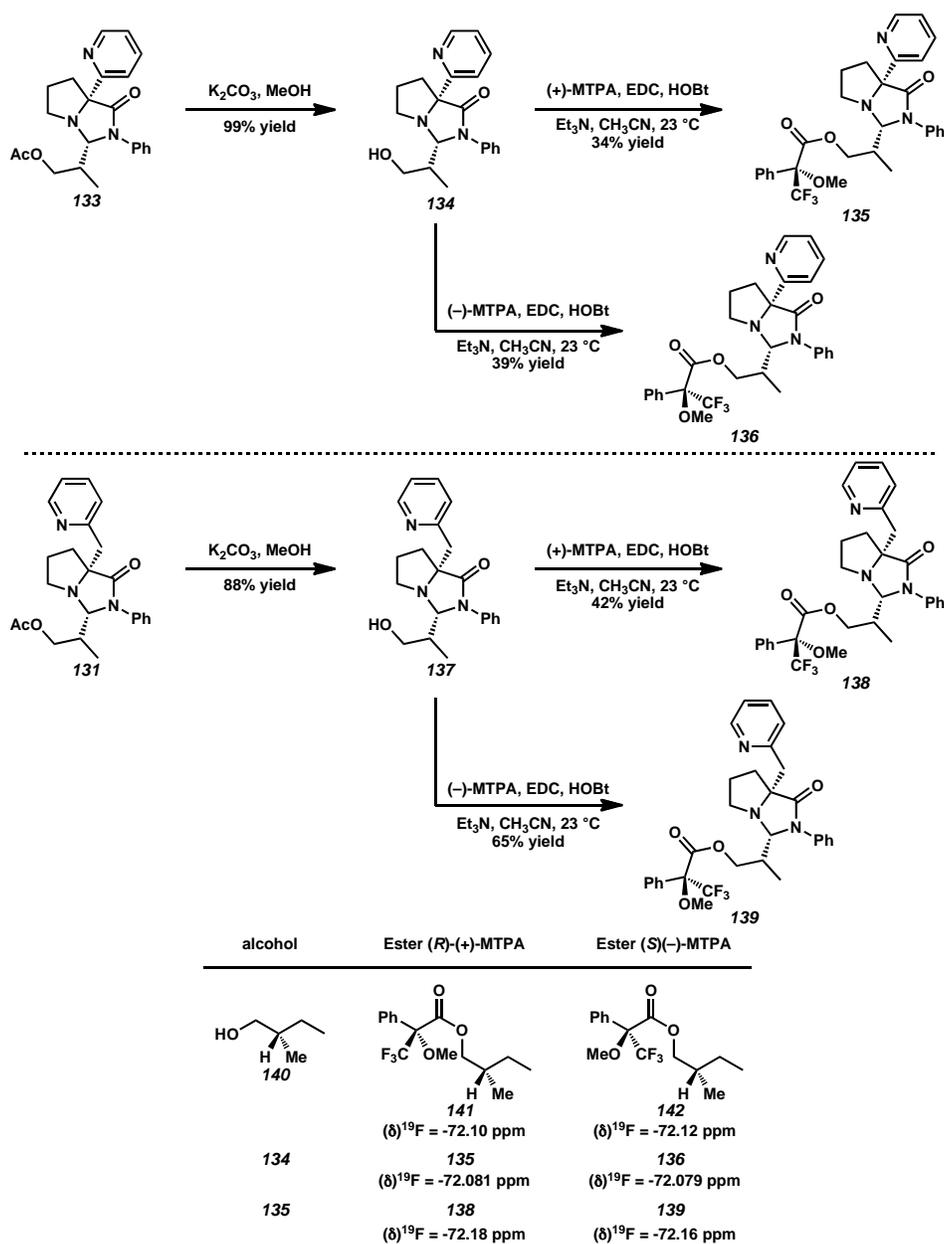| time (h)	temp (°C)	*131 : 132 (%)*	dr
24	55	41 : 13	10 : 1
13	70	48 : 19	5.9 : 1
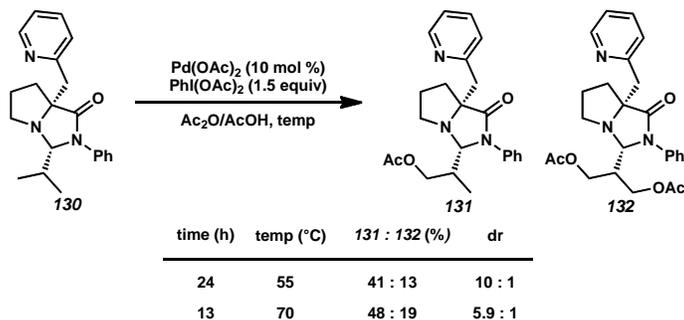
2.3 Determination of Absolute Configuration

Our observation of high levels of diastereoselectivity in the monoacetoxylation of substrates **116** and **131** is a promising entrance into the possibility of asymmetric C–H functionalization. In order to understand the mode of selectivity, however, we needed to determine which methyl of the isopropyl group was being functionalized. Due to our asymmetric ligand synthesis and highly diastereoselective functionalization reaction, we should be generating a single enantiomeric product. We imagined a chiral shift reagent may be able to

tell us the absolute configuration of the new stereocenter generated through functionalization.[16] Seebach and coworkers have used Mosher's acid as a chiral shift reagent to determine the absolute configuration of primary alcohols.[17] We imagined coupling the functionalized products with both enantiomers and comparing the ^{19}F NMR shifts to literature data.

Treatment of asymmetric **133** with K_2CO_3 in MeOH afforded the primary alcohol in excellent yield (Scheme 2.3.1). The coupling reaction with both (+)- and (–)-MTPA mediated by EDC and HOBt afforded esters **135** and **136** as single diastereomers, reinforcing the highly stereoselective synthesis of our substrates. Treatment of **131** with the same sequence of conditions afforded esters **138** and **139** as single diastereomers. The low conversion to the esters was attributed to the low reactivity of the substrates. When increasing the catalyst loading of HOBt to almost one equiv, the reaction proceeded in greater than 50% yield. The optical rotation of the remaining starting material was compared to that of the original starting mixture and found to be the same, indicating that a kinetic resolution process was not occurring. The literature precedence by Seebach suggested that when the (*S*)-alcohol is coupled with (*S*)-MTPA, the ^{19}F NMR peak shifts further upfield than that of the (*S*)-alcohol when coupled with the (*R*)-MTPA acid (Scheme 2.3.1). In ester **136**, the (*S*)-MTPA adduct had a downfield shift relative to the (*R*)-MTPA adduct. Similarly, ester **139** from the coupling with (*S*)-MTPA had a downfield shift relative to the (*R*) adduct. This data suggests that the pro-*S* methyl is selectively functionalized under acetoxylation conditions.

Scheme 2.3.1. Determination of absolute configuration

alcohol	Ester (R)-(+)-MTPA	Ester (S)(–)-MTPA
HO, H Me **140**	Ph, F₃C OMe, H Me **141** (δ)^{19}F = -72.10 ppm	Ph, MeO CF₃, H Me **142** (δ)^{19}F = -72.12 ppm
134	**135** (δ)^{19}F = -72.081 ppm	**136** (δ)^{19}F = -72.079 ppm
135	**138** (δ)^{19}F = -72.18 ppm	**139** (δ)^{19}F = -72.16 ppm

Eager to have additional evidence to support our findings of the absolute configuration, we imagined converting the primary alcohol to an acid, in order to have a α-stereocenter which may provide corroboration to our previous findings. Oxidation of the primary alcohol to the corresponding acid occurred via a mild bleach procedure (Scheme 2.3.2).[18] Peptide coupling with (R)-PGME and (S)-PGME afforded the corresponding amides. If an (S),(S) relationship is

established between the PGME and new stereocenter, the aminal proton should be better shielded by the phenyl ring of the PGME.[19] If an (R),(R) relationship is established, the aminal proton should be relatively less shielded, and appear further downfield than in the (S)-PGME amide. A shift of 4.73 ppm was observed for amide **144**, while a shift of 4.49 ppm was observed for amide **145**, again supporting that the pro-S methyl is selectively functionalized.

Scheme 2.3.2. Another analysis of absolute configuration
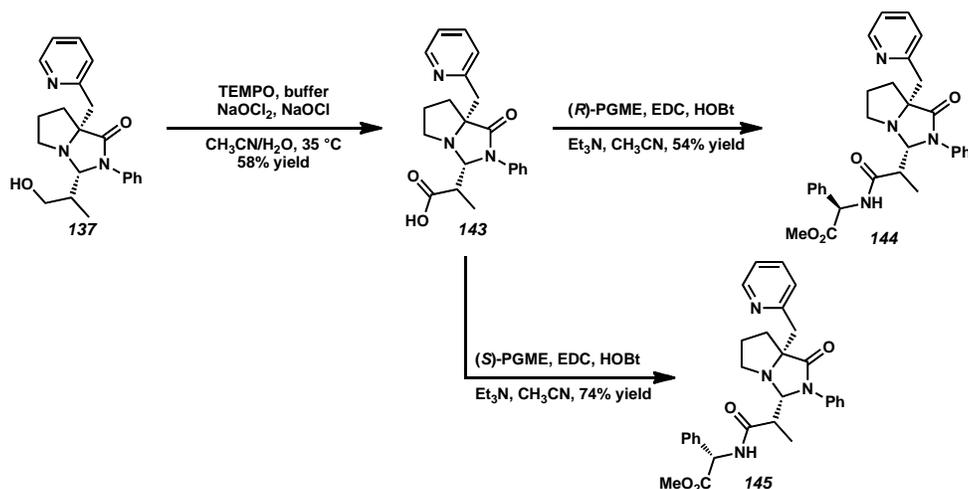
Ultimately, we sought out an X-ray crystal structure to confirm the absolute stereochemistry of the new stereocenter. While an X-ray structure cannot provide the absolute configuration on its own, the use of a chiral starting material as a template provides a known stereocenter from which to set the remaining stereocenters. Coupling with p-nitrobenzoic acid afforded ester **145** (Scheme 2.3.3). The compound was crystallized and afforded the X-ray structure. From the structure it is clear to see that the pro-S methyl is selectively functionalized under the acetoxylation conditions, which is consistent with the evidence provided by the esters and amides.

Scheme 2.3.3. X-ray crystal analysis
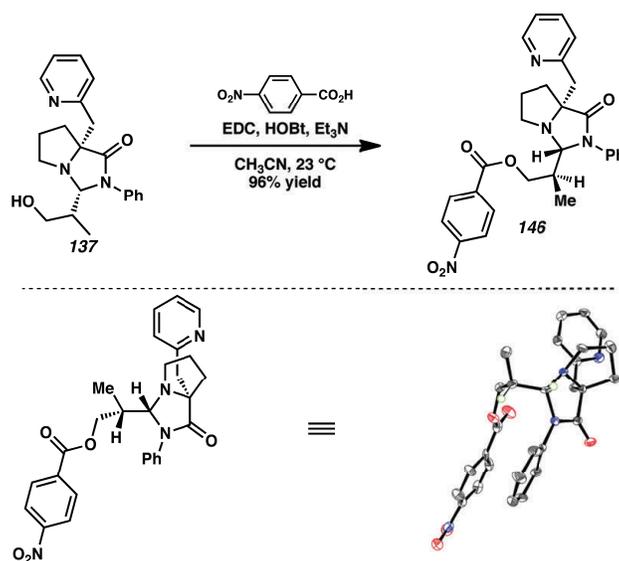
The acetoxylation of sp^3 C–H bonds in both **115** and **130** occurs with high levels of diastereoselectivity. While we have not been able to achieve the functionalization of 2° C–H bonds, the high levels of selectivity observed thus far may translate into an asymmetric C–H functionalization reaction, if the appropriate conditions are realized. The ligand synthesis from a chiral starting material is straightforward and is conducted with what we believe to be high levels of stereoretention.

2.4 Other Attempts at sp^3 C–H Functionalization

In addition to the functionalization of isobutyraldehyde, we have attempted the acetoxylation of several other aliphatic aldehydes. Condensation of ligand **114** onto cyclohexanecarboxaldehyde afforded *N,N*-aminal **147** as a single diastereomer (Scheme 2.4.1). Treatment of this substrate with acetoxylation conditions, however, did not afford any of the desired functionalized product. We attribute this lack of reactivity to the difficulty of activating 2° C–H bonds, and that the appropriate geometry for functionalization may not be possible in our system. Knowing that *t*-Bu groups in general are easy to activate, we imagined installing

pivaldehyde as the aldehyde to be activated.[20] Condensation onto pivaldehyde afforded the

aminal as a single diastereomer. Alkylation with 2-fluoropyridine afforded test substrate **150** in

modest yield. Treatment of the more sterically hindered substrate with functionalization

conditions, however, resulted in no acetoxylated product. We imagine that the steric interaction

between a *t*-butyl methyl and the phenyl amide prevents functionalization. We also synthesized

the hydrocinnamaldehyde derivative **153**. After condensation with amino amide **122**, alkylation

provided a single diastereomer of **152**. Treatment with acetoxylation conditions, however, did

not afford any of the benzylic acetate. The conformationally flexible nature of the

hydrocinnamaldehyde likely contributes to the lack of functionalization.

Scheme 2.4.1. Other substrates for acetoxylation
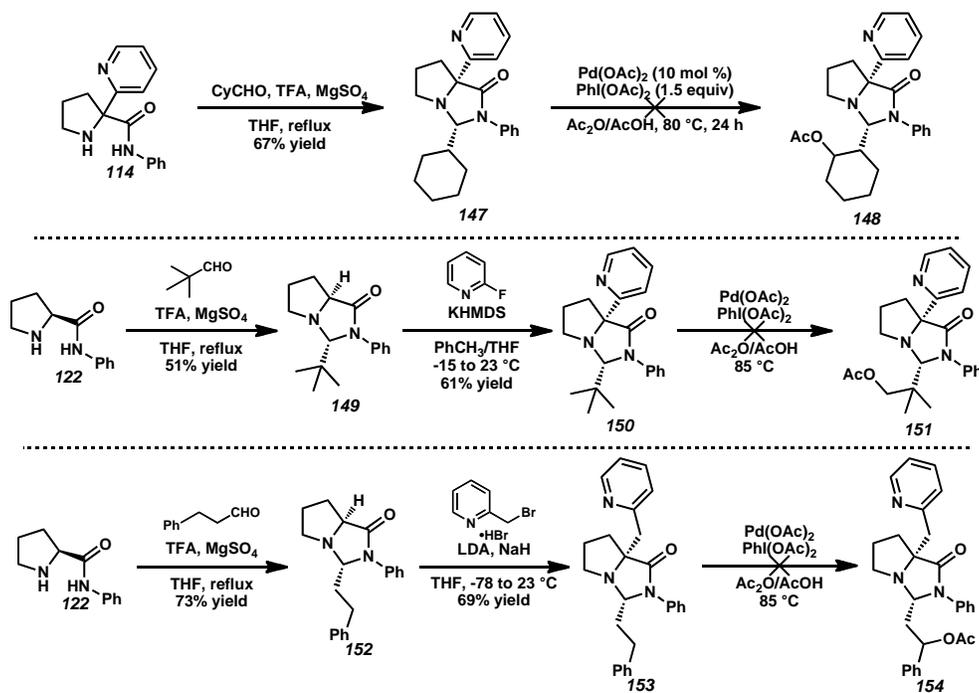
Corey and coworkers have previously reported a diastereoselective sp^3 C–H

acetoxylation reaction of 2° C–H bonds.[21] In order to achieve this reactivity, as 2° C–H bonds

are very unreactive to these electrophillic activation conditions, they added Mn(OAc)$_2$ as an

activator for the C–H functionalization reaction. Presumably manganese acts as a Lewis acid to

activate the palladium for C–H activation. We imagined this might allow acetoxylation of our unreactive substrates. We generated pyridine **155** from amino amide **114** and ethylbutyraldehyde in good yield and as a single diastereomer (Scheme 2.4.2). Subjecting pyridine **155** to Corey's acetoxylation conditions, however, afforded none of the desired product, and only starting material was recovered. We also generated aminal **158** from a condensation with amino amide **122** and cyclohexanecarboxaldehyde, followed by an alkylation with 2-bromomethyl pyridine as a single diastereomer. Treatment with the acetoxylation conditions, however, afforded none of the acetoxylated product.

Scheme 2.4.2. Acetoxylation of 2 ° C–H bonds
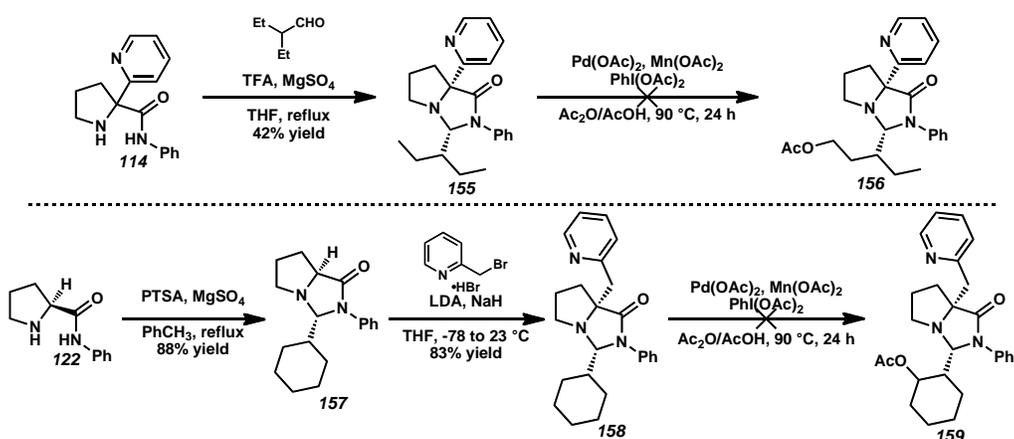
Besides acetoxylation we tried a number of different carbon-heteroatom and carbon-carbon bond formations utilizing substrates **129** and **130**. There have been numerous examples of C–Cl bond formation using electrophillic chlorine sources, such as NCS.[22] Subjecting pyridine **129** or pyridine **130** to NCS under palladium catalysis, however, afforded none of the carbon-chlorine bond formed products and only led to decomposition (Scheme 2.4.3). We believe that NCS may not be a strong enough oxidant to reach a higher valent Pd^{III} or Pd^{IV} species, and therefore only decomposes the substrate through other oxidative processes.

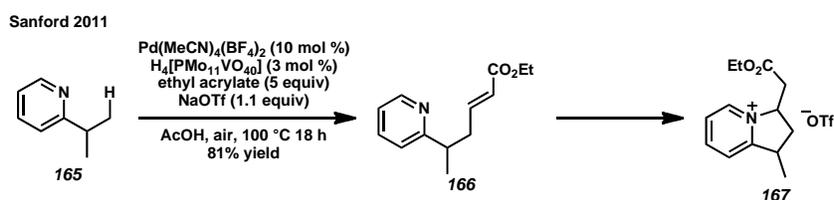**Scheme 2.4.3.** *sp³* **C–H chlorination**
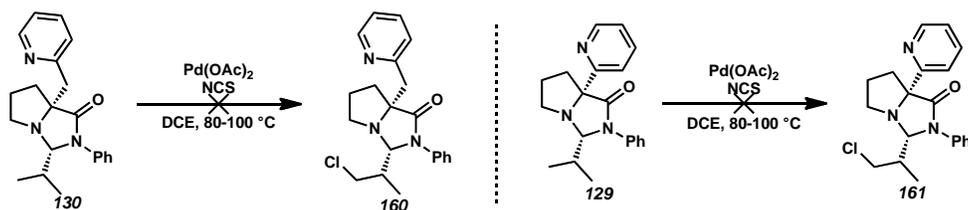
We also tried several carbon-carbon bond forming functionalization reactions under palladium catalysis. Yu and coworkers[23] demonstrated that Suzuki-type coupling can occur with catalytic palladium and PhB(OH)$_2$ with Cu(OAc)$_2$ and air acting to reoxidize Pd^0 to Pd^{II} (Scheme 2.4.4). When these conditions were employed with pyridines **129** and **130**, no arylated product was observed from the reaction. Additionally, we tried to perform a Heck-type coupling under palladium catalysis, but again no olefination product was isolated. Daugulis and coworkers have developed arylation conditions utilizing PhI under palladium catalysis.[24] Mechanistically, they envisioned an oxidation from Pd^{II} to Pd^{IV} via PhI, followed by reductive elimination to give the new carbon-carbon bond. Subjecting pyridine **130** to these conditions, however, generated none of the desired arylated product. Likewise, utilizing Sanford's diaryliodonium chemistry,[25] arylated pyridine **164** was not isolated from the reaction.

**Scheme 2.4.4. Other *sp³* C–H functionalizations**

46
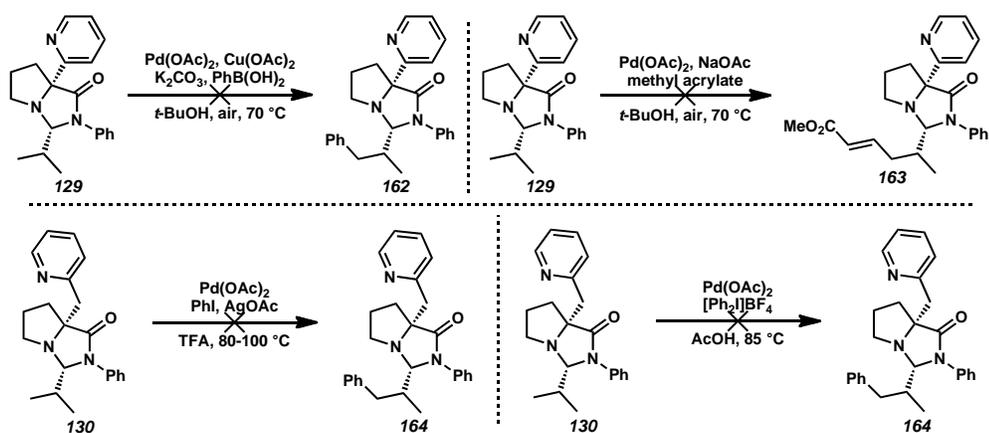
Sanford and coworkers have recently developed a new olefination reaction using a polyoxometalate co-catalyst (Scheme 2.4.5).[26] Subjecting 2-ethylpyridine to $Pd(MeCN)_4(BF_4)_2$ and $H_4[PMo_{11}VO_{40}]$ in the presence of ethyl acrylate and air in AcOH affords the olefinated intermediate (**165**). Pyridine can then act as a nucelophile and add into the Michael acceptor to form pyridinidum salt **166**. While the role of the polyoxometalate is not entirely clear, it is believed that it is involved in the reoxidation of Pd^0 to Pd^{II} to regenerate the active catalyst. It may also be involved in the C–H activation step, activating the palladium salt for cyclometallation.

Scheme 2.4.5. Olefination of *sp³* C–H bonds

Sanford 2011

Pd(MeCN)$_4$(BF$_4$)$_2$ (10 mol %)
H$_4$[PMo$_{11}$VO$_{40}$] (3 mol %)
ethyl acrylate (5 equiv)
NaOTf (1.1 equiv)

AcOH, air, 100 °C 18 h
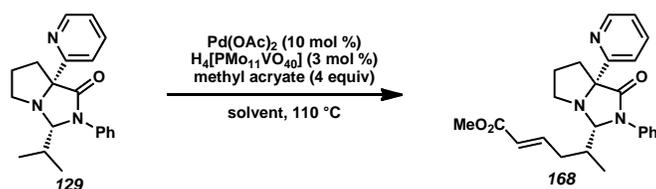81% yield

165

166

167

We imagined we may be able to utilize this chemistry for our *sp³* C–H functionalization reactions. Subjecting pyridine **129** to similar conditions in AcOH at 100 °C afforded none of the olefinated product (**168**) (Scheme 2.4.6). Treatment with the same conditions in DCE at 110 °C, however, afforded almost 20% conversion to olefinated product **168** as confirmed by LCMS. Interestingly, when we subjected pyridines **130**, **169**, and **158** to the same conditions, no reaction was observed. While prydidines **158** and **169** have not been reactive to any functionalization conditions, pyridine **130** normally shows enhanced reactivity relative to pyridine **129**. Clearly each oxidation system is influenced by subtle structural effects.

Scheme 2.4.6. Initial olefination of isobutyraldehyde
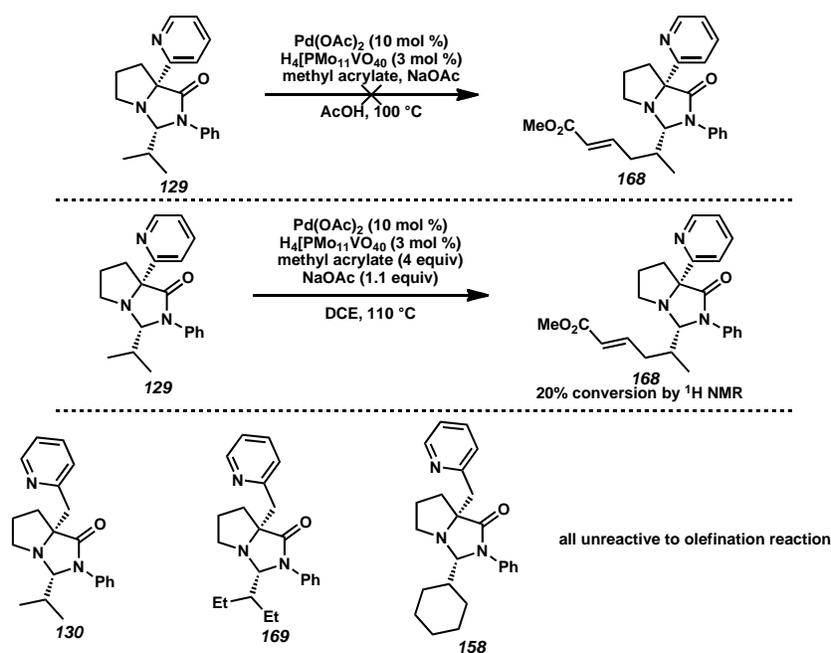
With the initial reactivity realized, we sought to begin optimization of the reaction (Table 2.4.1). Increasing the equiv of NaOAc decreased the percent conversion of the transformation. Running the same reaction in *t*-AmOH afforded none of the olefinated product, and only starting material was recovered. In CH$_3$CN nearly identical conversion was observed compared to that in DCE. Changing the base to Na$_2$CO$_3$ and adding NaOPiv only resulted in a trace amount of product formation. Changing the base to K$_2$CO$_3$ with no NaOPiv additive provided none of the desired product. When the reaction was run in CF$_3$CH$_2$OH, 23% conversion to olefinated product was observed. When the reaction was run in DMF, however, no olefinated product **168** was isolated. In Sanford's work, it was noted that Pd(OAc)$_2$ provided higher conversion to product when acac was used as a ligand. These conditions, however, were unsuccessful at providing any of the functionalized product with our substrate. Furthermore, we hypthesized adding just 1 equiv of AcOH might promote the reaction further, but again this was unsuccessful at improving the conversion.
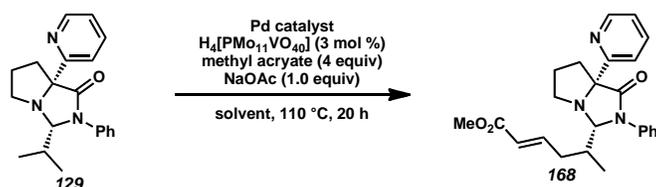
Table 2.4.1. Optimization of solvent and base

entry	solvent	base	additive	% conversion
1	DCE	NaOAc (2 equiv)	none	18
2	t-AmOH	NaOAc (2 equiv)	none	0
3	CH$_3$CN	NaOAc (2 equiv)	none	17
4	DCE	Na$_2$CO$_3$ (1 equiv)	NaOPiv (0.2 equiv)	trace
5	DCE	K$_2$CO$_3$ (1 equiv)	none	0
6		NaOAc (1.1 equiv)	none	23
7	DMF	NaOAc (1.1 equiv)	none	0
8	DCE	NaOAc (1.0 equiv)	acac (10 mol %)	0
9	CF$_3$CH$_2$OH	NaOAc (1.0 equiv)	AcOH (1.0 equiv)	trace

We next examined our palladium catalyst, eager to see if an alternative catalyst would enhance the conversion. Utilizing the very electrophillic catalyst that Sanford and coworkers used, we saw no improvement in conversion to **168** in either CF$_3$CH$_2$OH or DCE (Table 2.4.2). With Pd(OAc)$_2$ nearly 30% conversion was observed to product **168** using 1 equiv of NaOAc. Employing PdCl$_2$ afforded none of the product, and Pd(TFA)$_2$ only converted 14% of the starting material to the olefinated product. Rather than adding acac to the reaction, we used the F$_6$ version of Pd(acac)$_2$ and observed only 11% conversion to the product in CF$_3$CH$_2$OH. It may follow that the H$_6$ version being more basic may enhance the reaction relative to the F$_6$ variant.

Table 2.4.2. Optimization of palladium catalyst

entry	catalyst	solvent	% conversion
1	$Pd(CH_3CN)_4(BF_4)_2$	CF_3CH_2OH	20
2	$Pd(CH_3CN)_4(BF_4)_2$	DCE	18
3	$Pd(OAc)_2$	DCE	28
4	$PdCl_2$	DCE	0
5	$Pd(TFA)_2$	DCE	14
6	$Pd(acac-F_6)_2$	CF_3CH_2OH	11

Lastly we sought to examine the reoxidation step of the transformation, to ascertain whether or not this was hindering the conversion. Presumably, the polyoxometalate is partially responsible for the reoxidation of Pd^0 to Pd^{II} with the assistance of oxygen. We thought that our reaction conditions (run in a sealed vial under ambient air) may not be oxygen rich enough to regenerate the active catalyst. We found that if we flush the vial with pure oxygen before sealing, we see an increase in the conversion to product **168** (Scheme 2.4.7). In CF_3CH_2OH we see greater than 30% conversion to product, although in DCE we see less than 20% conversion to product **168**.

Scheme 2.4.7. Addition of O_2 for conversion
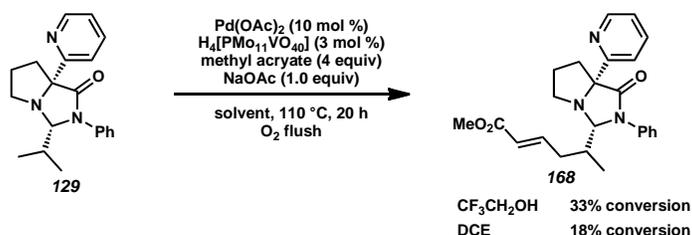
The olefination reaction is the only functionalization other than acetoxylation that has been successful with our model substrates. The reaction has not been fully optimized, with 33% conversion being the best observed. Part of the low conversion may be due to exceptional

binding of the substrate to the palladium catalyst, preventing turnover, as longer reaction times do not improve the conversion. It is apparent that CF_3CH_2OH gives the best conversion of all the solvents employed in this transformation. Further optimization of this reaction is required before synchronization of functionalization and acetalization can be attempted, although the mild reaction conditions may lend satisfactory results.

2.5 Acetoxylation of sp^2 C–H Bonds

Due to our success in the stoichiometric functionalization of the isobutyraldehyde substrates we decided to examine the functionalization of aromatic aldehydes. Condensing racemic amino amide **127** onto benzaldehyde afforded the desired *N,N*-aminal (**170**, Scheme 2.5.1). Alternatively, we synthesized the test substrate via our asymmetric ligand synthesis, which afforded a mixture of diastereomers, the minor of which corresponded to condensation product **170**. An NOE analysis of the diastereomers revealed the major product of the alkylation to be the syn diastereomer. Conversely, the condensation reaction provided only the anti diastereomer, showing an NOE between the acetal proton and a benzylic proton.

Scheme 2.5.1. Synthesis of a benzaldehyde ligand adduct
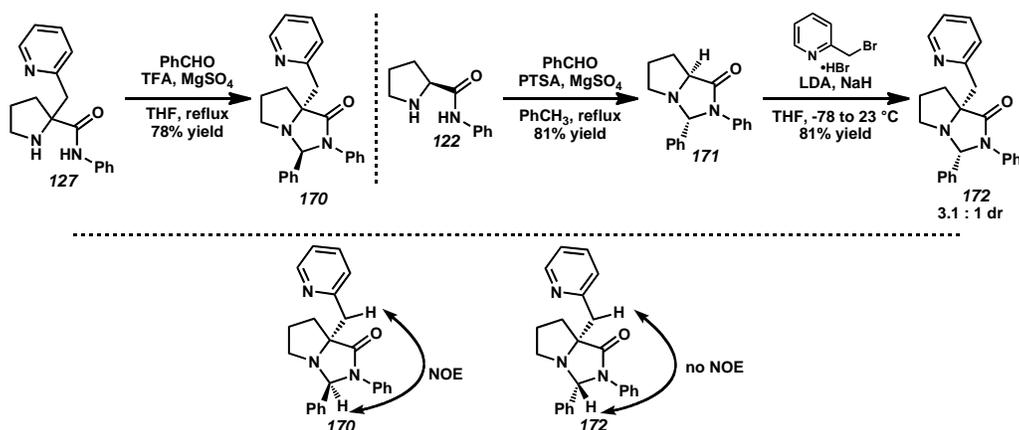
With the desired substrate in hand, we subjected **172** to acetoxylation conditions and were delighted to isolate 31% of the apparent functionalized product (Scheme 2.5.2). Treatment

with aqueous acid followed by Claisen's alkali afforded the phenol product in good yield, confirming the identity of an aryl acetate. A brief solvent screen was conducted with $PhCH_3$, CH_2Cl_2, DCE and CH_3CN, all affording the desired acetoxylated product in less than 10% yield. In alcohol solvent, such as MeOH, IPA or t-AmOH, no product was observed from the reaction. These results speak to the necessity of the $AcOH/Ac_2O$ solvent combination.

Scheme 2.5.2. Initial acetoxylation of benzaldehyde
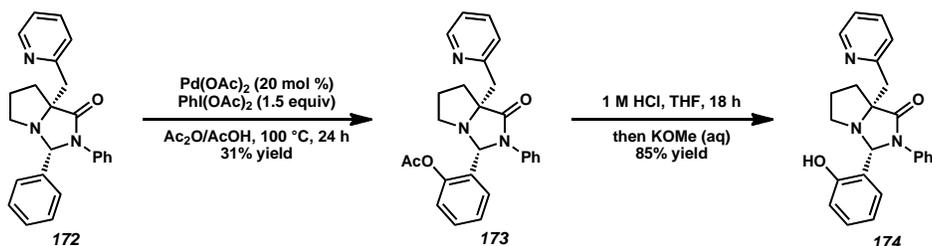
A combined solvent and temperature screen began our optimization of the acetoxylation reaction. At 100 °C, either AcOH or a combined $AcOH/CH_2Cl_2$ solvent mixture provided less than 10% of the desired product (Table 2.5.1). Ac_2O alone provided no acetoxylated product. At 23 °C there was no reaction in the mixed solvent system. Warming to 80 °C provided 50% of the desired product in a mixed solvent system, although heating to 90 °C caused a decrease in yield, presumably due to increased decomposition. Treatment with functionalization conditions in AcOH provided 25% yield but again lost effectiveness at higher temperatures.

Table 2.5.1. Solvent screen

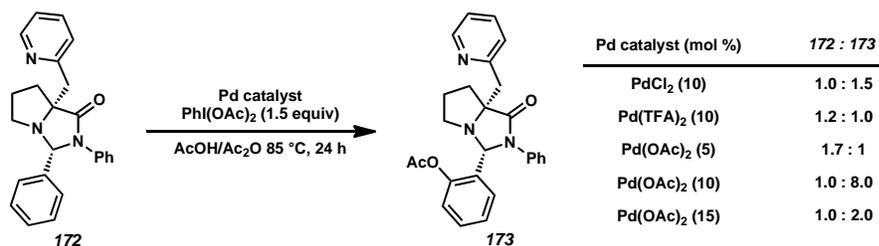| solvent	temp (°C)	yield	solvent	temp (°C)	yield
AcOH	100	<10%	AcOH:Ac₂O	80	50
Ac₂O	100	0%	AcOH	80	25
AcOH:CH₂Cl₂	100	<10%	AcOH:Ac₂O	90	36
AcOH:Ac₂O	23	0%	AcOH	90	18

Optimization continued with further probing the solvent composition and concentration. AcOH/Ac$_2$O as a solvent provided much higher levels of conversion and yield as compared to AcOH alone (Table 2.5.2). Moving from 0.05 M to 0.1 M provided a dramatic increase in yield, but increasing the concentration even more showed a decrease in yield. Presumably higher concentrations lead to higher levels of decomposition. While conversion noticeably improved, yield did not.

Table 2.5.2. Solvent concentration screen

solvent	concentration (M)	*172 : 173*	yield (%, crude)
Ac$_2$O/AcOH	0.05	1.0 : 2.0	40
Ac$_2$O/AcOH	0.1	1.0 : 3.0	62
Ac$_2$O/AcOH	0.2	1.0 : 8.3	50
AcOH	0.05	2.3 : 1.0	12
AcOH	0.1	1.3 : 1.0	18
AcOH	0.2	1 : 1	27

An examination of catalyst type and loading next directed our efforts towards optimization. Utilizing PdCl$_2$ afforded a lower conversion than Pd(OAc)$_2$ at the same catalyst loading (Scheme 2.5.3). The use of Pd(TFA)$_2$ provided even less product, and the formation of an oxidized side product was becoming more obvious (*vide infra*). As Pd(OAc)$_2$ appeared to be the most effective catalyst, different loadings were examined. From our results, it is evident that the reaction is sensitive to catalyst loading, with 10 mol % being the most effective concentration.

Scheme 2.5.3. Effect of palladium catalyst

Pd catalyst (mol %)	*172 : 173*
PdCl$_2$ (10)	1.0 : 1.5
Pd(TFA)$_2$ (10)	1.2 : 1.0
Pd(OAc)$_2$ (5)	1.7 : 1
Pd(OAc)$_2$ (10)	1.0 : 8.0
Pd(OAc)$_2$ (15)	1.0 : 2.0

We sought to examine the effect of the oxidant on yield and the formation of the oxidized side product. Increasing the equiv of PhI(OAc)$_2$ to 2 afforded higher conversion to the acetoxylated product (Scheme 2.5.4). Increasing to 3 equiv, however, began to lead to decomposition pathways and an increased amount of formation of an oxidized side product. Changing the oxidant to a perester[2a] provided only a small amount of product **173**. Using a stronger oxidant like Oxone only resulted in decomposition of the starting material and no conversion to functionalized product.

Scheme 2.5.4. Effect of oxidant on acetoxylation

entry	oxidant	yield (%)
1	PhI(OAc)$_2$ (1.5 equiv)	62
2	PhI(OAc)$_2$ (2 equiv)	65
3	PhI(OAc)$_2$ (3 equiv)	41
4		<10
5	oxone	decomp
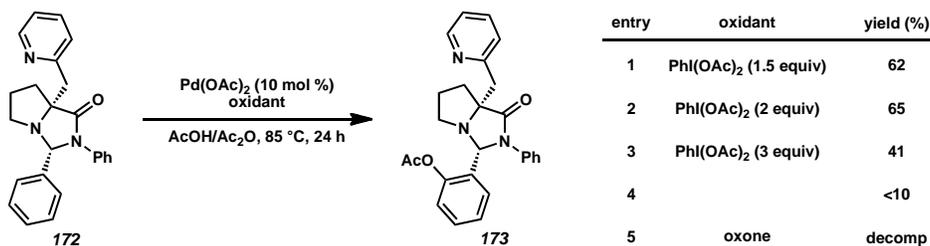
We eventually identified the oxidized side product as amide **174** (Table 2.5.3). We screened a range of conditions to ascertain the mechanism of amide formation. Under our typical acetoxylation conditions, a 4.6 : 1 ratio of product **173** and amide **174** was observed. Increasing the concentration to 0.4 M increased the formation of amide **174**, where a <2 : 1 ratio of **173** to **174** was observed. Functionalization in AcOH at 0.4 M afforded only the acetoxylated product, and none of the oxidized amide was observed. Treatment with acetoxylation conditions in only Ac$_2$O now afforded amide **174** in a 1.1 : 1 ratio to functionalized product, albeit with low

conversion. Under atypical functionalization conditions with the addition of PivOH as the acetate source and K_2CO_3 as the corresponding base in CH_2Cl_2, only the oxidized amide product was afforded. Furthermore, in highly acidic TFA, only the formation of the amide was observed; no acetoxylated product was isolated.

Table 2.5.3. Formation of an oxidized byproduct
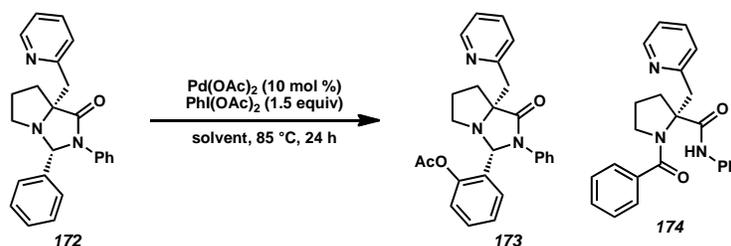
entry	conditions	173 : 174
1	Ac$_2$O/AcOH, 0.1 M	4.6 : 1
2	Ac$_2$O/AcOH, 0.4 M	1.7 : 1
3	AcOH, 0.4 M	1 : 0
4	K$_2$CO$_3$, PivOH, CH$_2$Cl$_2$, 0.1 M	0 : 1
5	Ac$_2$O, 0.1 M	1 : 1.1
6	TFA, 0.1 M	0 : 1

Based on these results, we believed the rate of formation of the amide byproduct was directly dependent on the acidity of the functionalization conditions. Under acidic conditions, the pyrrolidine nitrogen is protonated and the pyridyl directs the acetoxylation to the phenyl ring of the aminal (Scheme 2.5.5). Conversely, if the conditions are neutral, or even basic, the pyrrolidine nitrogen is not protonated and can attack PhI(OAc)$_2$ to generate ammonium intermediate **176**. Deprotonation to eliminate PhI and acetate anion occurs to form iminium **177**. Hydrolysis of the iminium reveals the oxidized amide. Under our conditions in Table 2.5.3, the pattern appears to follow our proposed mechanism. In entry 6, however, treatment with a much more acidic TFA should result in the formation of acetoxylated product if the pyrrolidine nitrogen is protonated. We attribute this contradiction to the formation of a much stronger oxidant PhI(TFA)$_2$, which can perform the oxidation to the amide in a much more facile manner.

**Scheme 2.5.5. Possible mechanism of aminal oxidation**
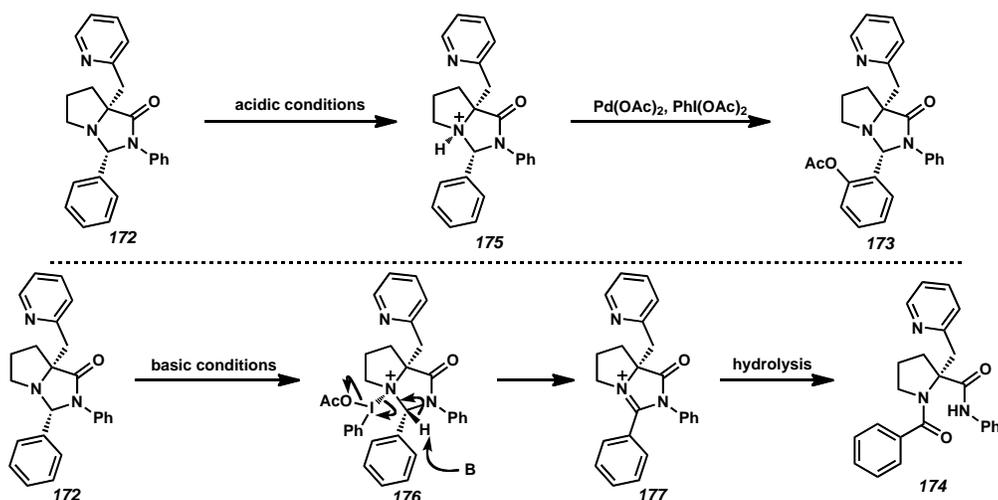
Interested in learning more about this oxidative transformation, we sought to electronically modify the amide component of our ligand to see its effect on the rate of acetoxylation and aminal oxidation. We imagined using both an electron rich aromatic amide and an electron poor aromatic amide. Starting from L-Boc proline we generated the amide bond and deprotected the pyrrolidine nitrogen for both **178**-**OMe** and **178-CF₃** in good yield (Scheme 2.5.6). Condensation onto benzaldehyde afforded the corresponding *N,N*-aminals in good yield. Alkylation with 2-bromopyridine afforded **180-OMe** and **180b-OMe** in a 2.4 : 1 mixture of diastereomers. Likewise alkylation of **179-CF₃** afforded a 2.1 : 1 mixture of diasteromers in excellent yield.

Scheme 2.5.6. Synthesis of electronically different amides

178

X = OMe   78% yield (2 steps)
X = CF₃   85% yield (2 steps)

179

X = OMe   75% yield
X = CF₃   82% yield

180a   *180b*

X = OMe   85% yield   2.4 : 1 mixture
X = CF₃   98% yield   2.1 : 1 mixture
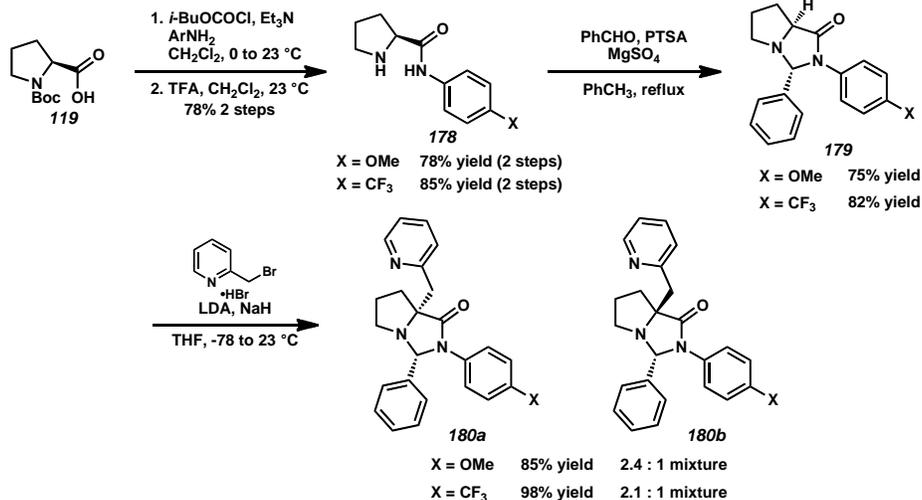
We subjected both differentially substituted amides to our standard acetoxylation conditions. The monoacetoxylation of **180-OMe** occurred in 32% yield with a large degree of aminal oxidation and decomposition (Scheme 2.5.7). The very electron rich nature of the amide makes the aminal center much more prone to oxidation. Monoacetoxylation of **180-CF₃** occurred much more smoothly in 64% yield. The occurrence of aminal oxidation was much less prevalent, likely due to the now more electron deficient nature of the aminal center.

Scheme 2.5.7. Acetoxylation of electronically modified amides

180   *181*

X = OMe   32% yield
X = CF₃   64% yield
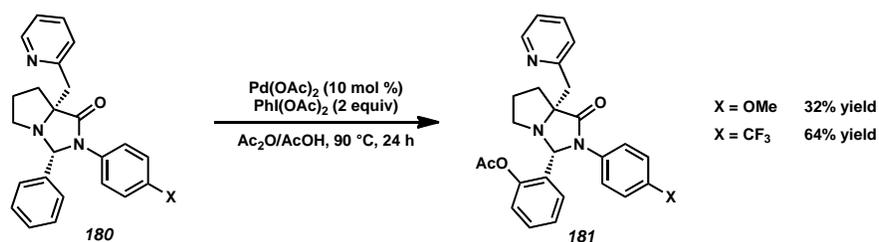
Expanding the substrate scope, we condensed amino amide **122** onto *p*-tolualdehyde to afford aminal **182** in good yield (Scheme 2.5.8). Alkylation with 2-bromomethyl pyridine afforded a 3.2 : 1 mixture of syn to anti diastereomers. Treatment of the syn diastereomer with acetoxylation conditions afforded acetate **185** in modest yield.

Scheme 2.5.8. Acetoxylation of *p*-tolualdehyde
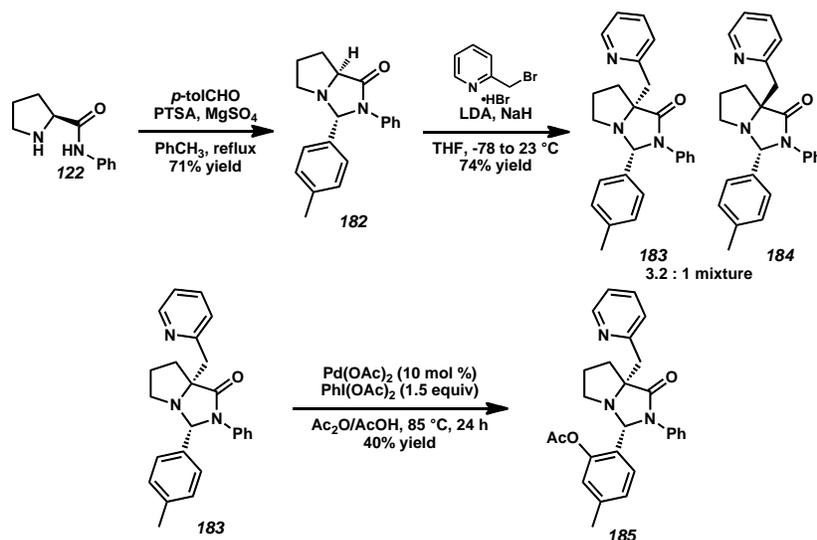
We imagined by utilizing the *p*-methoxyphenyl amide and *p*-tolualdehyde as the substrate to functionalize that the aminal center would be very sensitive to oxidation, even under our typical conditions. Condensing amino amides **178-OMe** and **178-CF₃** onto *p*-tolualdehyde afforded aminals **186-OMe** and **186-CF₃** in good yield (Scheme 2.5.9). Alkylation with 2-bromomethyl pryidine afforded pyridines **187** and **188** in a 2.1 : 1 mixture of diastereomers for both *p*-OMe and *p*-CF₃. Acetoxylation of the syn diastereomer (**187-OMe**) with our typical functionalization conditions afforded the desired product in 21% yield, and the amide byproduct in 19% yield. Conversely, treatment of **187-CF₃** with acetoxylation conditions afforded the desired product in 50% yield, with only 14% of the amide product isolated.

Scheme 2.5.9. Further extension of substrate scope

178

186
X = OMe   60% yield
X = CF₃   92% yield

187   188
X = OMe   88% yield   2.1 : 1 mixture
X = CF₃   85% yield   2.1 : 1 mixture

189
X = OMe   21% yield
X = CF₃   50% yield
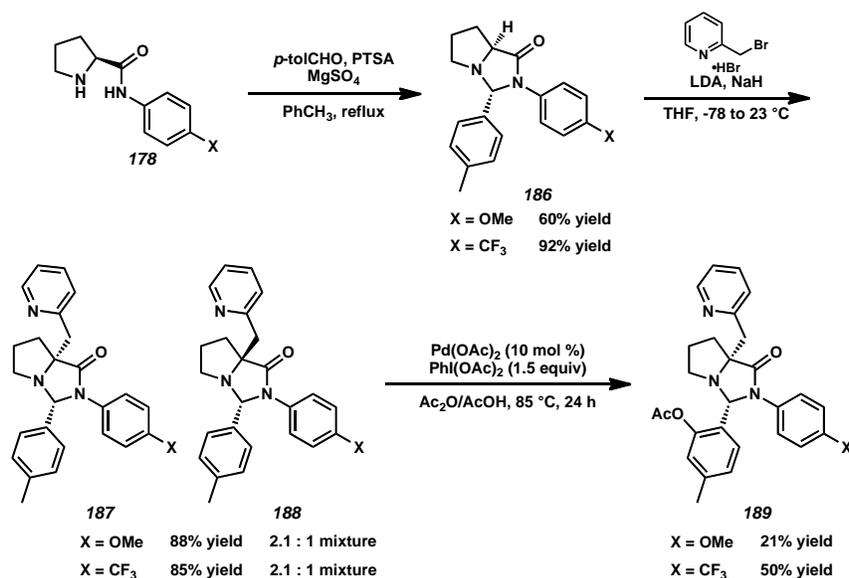
Based on the degree of aminal oxidation we imagined further decreasing the electron rich nature of the aminal center by using a more electron deficient amide. Coupling with 2,6-difluoroaniline and deprotection afforded amino amide **190** (Scheme 2.5.10). Condensation with benzaldehyde afforded aminal **191** in good yield. Alkylation with 2-bromomethyl pyridine afforded **192** as a single diastereomer. This enhancement of selectivity could be attributed to the electron deficient nature of the amide. Interestingly, treating pyridine **192** with acetoxylation conditions afforded none of the desired functionalized product. This may be due to activation of one of the C-F bonds or coordination to the palladium.

Scheme 2.5.10. Synthesis of the 2,6-difluorophenyl amide
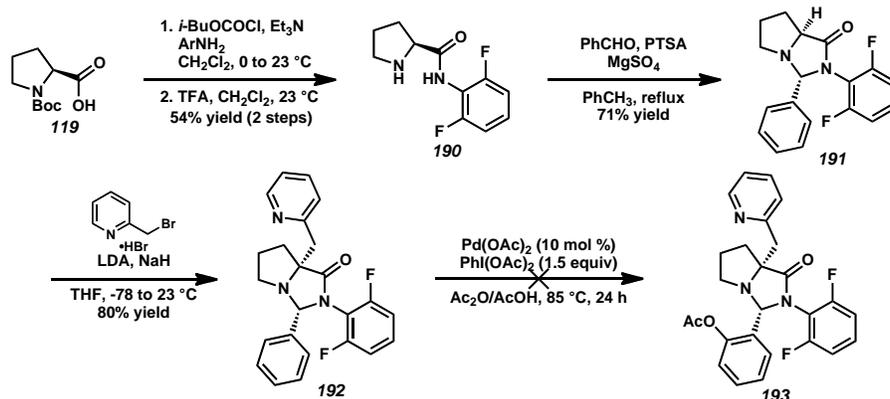
Seeking to establish the validity of the pyridyl ligating group directing the functionalization reaction, we wanted to remove the pyridyl group and observe if any functionalization could occur in its absence. From substrate **171** we were able to install a benzyl group via alkylation to generate a mixture of diastereomers (Scheme 2.5.11). We believed the benzyl group would have the same steric component as the pyridyl, without the nitrogen ligating component. We subjected the mixture of diastereomers to standard acetoxylation conditions. No acetoxylated product was observed from the reaction, and only aminal oxidized byproducts were observed other than starting material. This result establishes the necessity of the pyridyl ligating group for acetoxylation.

Scheme 2.5.11. Installation of a benzyl group
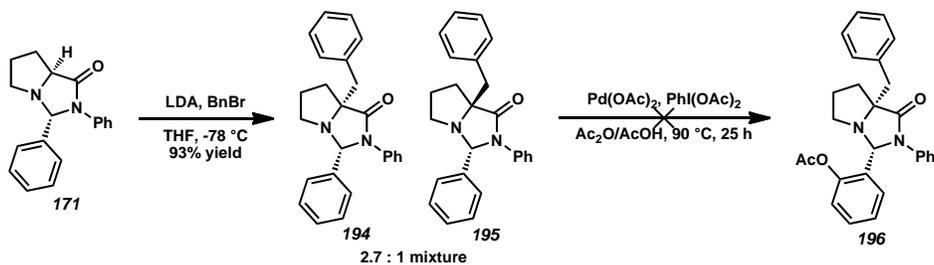
Lastly we were interested in examining the potential for isolation of a palladacycle. Treating substrate **172** with stoichiometric palladium in CH_2Cl_2, CH_2Cl_2/AcOH or just AcOH at 40 °C overnight afforded incomplete conversion to a complex that by ^1H NMR appeared to be

the palladacycle (Scheme 2.5.12). Treating substrate **172** with palladium in AcOH at 80 °C for 2 h afforded the same structure with complete conversion. We were able to crystallize the complex and obtain an X-ray crystal structure. From the structure it is obvious that the pyridyl and pyrrolidine nitrogen are coordinated to palladium. A palladium-carbon bond is also observed at the ortho position of the desired aryl ring. The syn orientation between the pyridyl group and aminal group allows for the desired C–H activation.

Scheme 2.5.12. Generation of a palladacycle
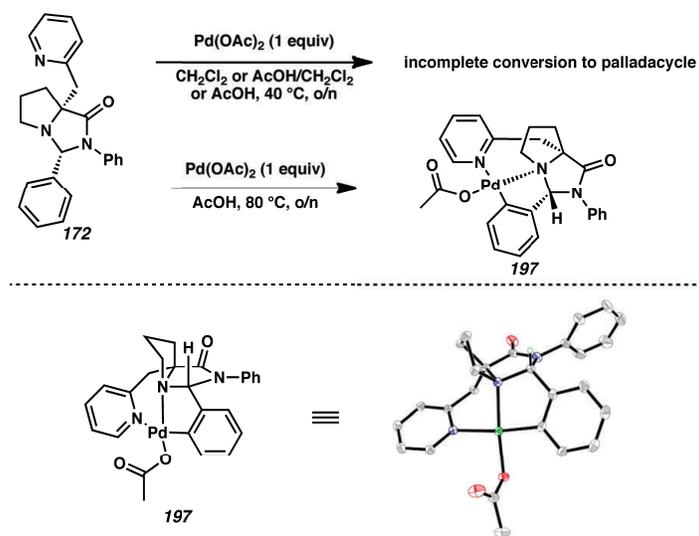
While we had isolated and characterized palladacycle **197**, we were not sure of its kinetic competency in these acetoxylation reactions. Subjecting the palladacycle to stoichiometric oxidant in AcOH/Ac$_2$O at 85 °C, followed by treatment with dppe, afforded none of the desired acetoxylated product **273** (Scheme 2.5.13). If the oxidation was proceeding via a Pd^{III}-Pd^{III} dimer, the size of the ligand may prevent this metal-metal bond formation, thereby inhibiting product formation. To circumvent this, we imagined adding an additional 5 mol % palladium that may be able to then form the dimeric metal species upon oxidation. Indeed, upon workup with dppe, we observed a small amount of the functionalized product (**173**). This low conversion may indicate that the isolated palladacycle is not an active intermediate under our

reaction conditions. Much more exploration into these complexes is necessary to understand more about the mechanism of functionalization.

Scheme 2.5.13. Examination of the kinetic competency of the palladacycle
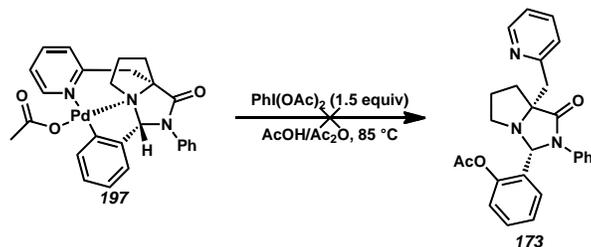
Based on the success of our acetoxylation of sp^2 C–H bonds using substrate ligand adduct **172**, we imagined we may be able to impart similar reactivity with ligand **114**. Racemic **114** was condensed with benzaldehyde in THF to afford *N,N*-aminal **198** of unknown stereochemistry (Scheme 2.5.14). While we were easily able to determine the relative stereochemistry of substrate **172** with NOE data, this substrate did not allow for an analogous analysis. We therefore subjected **198** to the functionalization conditions, believing that based on all of our data, the anti diastereomer would not functionalize under our conditions. Treatment with catalytic palladium and stoichiometric oxidant in AcOH/Ac$_2$O afforded a new product resembling the desired monoacetoxylated product (**199**). The reaction also afforded a new compound that contained no acetate, but appeared similar in structure to the starting aminal (**198**). We compared product **199** to the condensation product (**201**) of acetoxylated benzaldehyde and ligand **114**, which did not afford the same product by [^1]H NMR.

Scheme 2.5.14. Functionalization of benzaldehyde with a new ligand
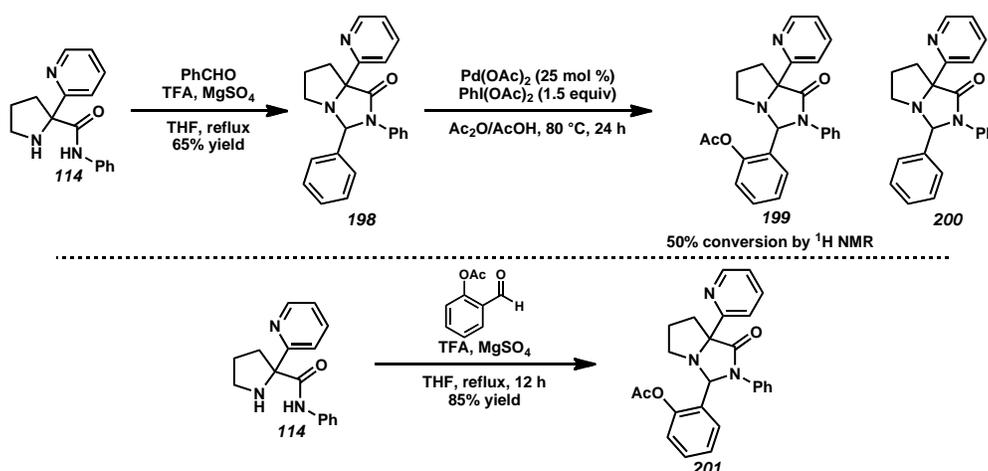
We conducted a ^{1}H NMR analysis of the aminal protons of each substrate in order to determine the stereochemistry of each. We found that the aminal shifts of both condensation products **198** and **201** were similar and downfield at 6.5 ppm (Scheme 2.5.15). In contrast, the shift of the aminal proton of acetate product **199** was much further upfield at 5.94 ppm. The unknown product of the functionalization reaction appeared to be the opposite diastereomer of **198** with a ^{1}H NMR shift of 5.64 ppm. Based on this large difference in shift, we believed that the condensation reactions were providing the anti diastereomer of the *N,N*-aminal. Under the functionalization conditions however, we believed that the anti diastereomer was isomerizing and then functionalization was occurring, affording **199**. The appearance of the syn diastereomer (**200**) confirmed the isomerization event under acetoxylation conditions.

Scheme 2.5.15. ^{1}H NMR analysis of aminal protons

63
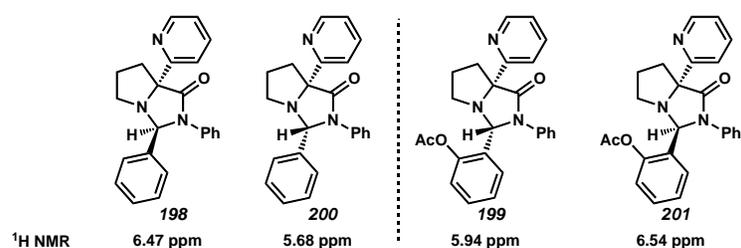
Perplexed by the formation of the anti diastereomer via condensation, we imagined we could synthesize the desired substrate in a similar manner to the above ligand syntheses. Subjecting aminal **171** to alkylation conditions with 2-fluoropryidine, however, did not afford a single diastereomer, but rather a 1 : 1 mixture of syn (**202**) to anti (**203**) asymmetric pryidines (Scheme 2.5.16). We again turned to the condensation reaction to ascertain whether we could obtain the syn diastereomer selectively. In our original conditions and with PTSA in PhCH$_3$, we only observed the anti diastereomer. Based on the functionalization conditions providing isomerization, we imagined that the addition of AcOH might lead to the syn diastereomer preferentially. Indeed, when AcOH was used as the acid catalyst in the condensation of asymmetric **204** and benzaldehyde, we observed greater than a 4 : 1 mixture of syn to anti. Increasing the ratio of AcOH : PhCH$_3$ (1 : 5) led to a 6.3 : 1 mixture of syn to anti, and using a 1 : 1 ratio of PhCH$_3$ to AcOH afforded a 6 to 1 ratio. When run on scale, we observed a 4 : 1 ratio of syn to anti in 90% yield.

Scheme 2.5.16. Obtaining the syn diastereomer selectively

acid	solvent	temp (°C), time (h)	a : b
TFA (10 mol %)	THF	72, 18 h	0 : 1.0
PTSA (5 mol %)	PhCH$_3$	110, 24 h	0 : 1.0
AcOH (1 M)	PhCH$_3$	110 °C, 24 h	4.3 : 1.0
AcOH (0.5 M)	PhCH$_3$	110 °C, 24 h	6.3 : 1.0
AcOH (0.1 M)	PhCH$_3$	110 °C, 24 h	6.0 : 1.0
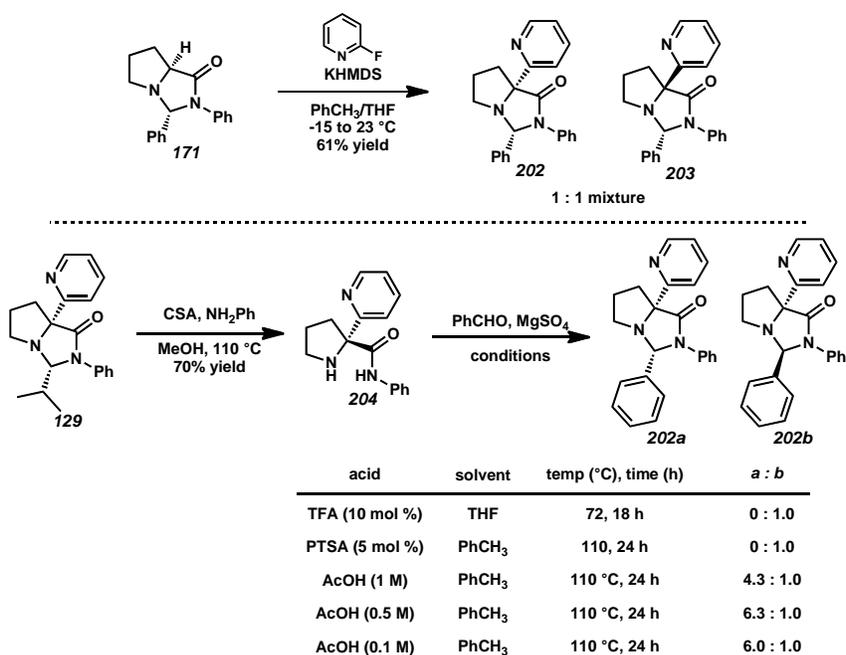
We therefore synthesized the corresponding syn diasteromers selectively using AcOH in the condensation reactions (Table 2.5.4). Utilizing electron neutral benzaldehyde, the syn diastereomer was obtained in a 4 : 1 ratio in good yield. More electron rich *p*-tolualdehyde afforded aminal **205** as a 6 : 1 ratio of syn to anti diastereomers. The *m*-tolualdehyde aminal was synthesized with slightly less selectivity in the ratio of syn to anti. Interestingly, the *o*-tolualdehyde derivative when condensed with the chiral ligand still afforded the anti diastereomer (**207b**) in a 1.5 : 1 ratio in good yield. We attribute this reversal in selectivity to the more sterically hindered nature of the aldehyde, inhibiting the syn condensation. Electron deficient *p*-Cl benzaldehyde was much less selective for the isomerization of the anti to syn diastereomer, providing only a 3.8 : 1 ratio for syn. Conversely, the 2-naphthyl derivative provided the highest level of syn to anti selectivity, with greater than 7 : 1 ratio.

Table 2.5.4. Expanding aminal substrate scope

ArCHO	product	dr (*a : b*)	yield (%)	ArCHO	product	dr (*a : b*)	yield (%)
Ph	202	4.0 : 1	89	*p*-OMe	208	5.8 : 1	78
p-Me	205	6.0 : 1	86	*p*-Cl	209	3.8 : 1	67
m-Me	206	5.8 : 1	76		210	7.3 : 1	80
o-Me	207	1 : 1.5	87				

Subjecting the syn diastereomers to the functionalization conditions required lower temperatures and less equiv of oxidant for completion of the reaction as compared to ligand substrate adduct **172**. In most cases, the starting material was completely consumed (Table 2.5.5). When a 4 : 1 ratio of **202** was treated with the functionalization conditions, a 55% yield of monoacetoxylated product was isolated. An additional 6% of diacetoxylated product was also observed. Likewise, subjecting the *p*-tolualdehyde derivative to the functionalization conditions

resulted in moderate yield of monoacetoxylated product and 16% yield of diacetoxylated product. The *m*-tolualdehyde derivative was completely selective for the less hindered ortho position in good yield. Furthermore, the *o*-tolualdehyde derivative was completely selective for sp^2 C–H functionalization in excellent yield. Subjecting electron rich *p*-anisaldehyde derivative **208** to milder acetoxylation conditions provided the functionalized product (**215**) in good yield, with less than 10% diacetoxylated product. Lastly, the 2-naphthyl derivative was completely selective for the less hindered position, and afforded acetoxylated arene **217** in 58% yield.

Table 2.5.5. Acetoxylation of syn diastereomers

ArCHO	product	yield (%) Ar$_{ox}$		ArCHO	product	yield (%) Ar$_{ox}$
Ph	211	55[a]		*p*-OMe	215	37[c]
p-Me	212	47[b]		*p*-Cl	216	39
m-Me	213	72			217	58
o-Me	214	85				

[a] 6% yield diacetoxylated product. [b] 16% diacetoxylated product.
[c] 9% diacetoxylated product

Interested in examining the difference in yield between utilizing the syn or anti diastereomer in acetoxylation reactions, we sought to synthesize a few anti diastereomers (Table 2.5.6). Condensation with benzaldehyde using PTSA and toluene afforded the anti diastereomer as the only product in good yield. Using the more electron rich *p*-tolualdehyde afforded a mixture of diastereomers in a 1 : 3 ratio. Condensation with *m*-tolualdehyde was less selective, affording a 1 : 5 mixture of **206** as syn to anti diastereromers. Electron rich *p*-anisaldehyde afforded a 4 : 1 mixture of anti to syn diastereomers in moderate yield. Finally, condensation with 1 and 2-naphthyl aldehydes provided only the anti diastereomers (**218** and **210**) in good yield.

Table 2.5.6. Synthesis of the anti diastereomers

ArCHO	product	dr (*a : b*)	yield (%)	ArCHO	product	dr (*a : b*)	yield (%)
Ph	202	0 : 1.0	87				
p-Me	205	1 : 3.0	83	1-naphthyl (1-naphthaldehyde)	218	0 : 1.0	85
m-Me	206	1 : 5.0	82				
p-OMe	208	1 : 3.9	69	2-naphthyl (2-naphthaldehyde)	210	0 : 1.0	72
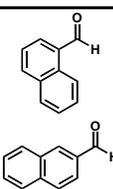
We subjected the diastereomeric mixtures to standard acetoxylation conditions utilizing two equiv of oxidant for 24 h. Functionalization of the electron neutral benzaldehyde derivative provided 42% yield of the monoacetoxylated product (**211**) and less than 5% of the diacetoxylated product (Table 2.5.7). Using the syn diastereomer provided a 13% higher yield than the corresponding anti diastereomer. The remaining starting material was isolated as a 2 : 1 mixture of syn and anti diastereomers. Utilizing derivative **205** provided nearly a 1 : 1 mixture of mono- to diacetoxylated products, displaying much higher reactivity. *m*-Tolualdehyde derivative **206** was selectively functionalized at the less hindered ortho position, with no recovered starting material. Employing the more electron rich *p*-anisaldehyde derivative **208** provided only a moderate amount of product, with the mass balance consisting of decomposition. This is most likely due to the very electron rich nature of the starting aminal. Conversely, 1-naphthyl derivative **218** was completely unreactive under the acetoxylation conditions and afforded no product. The 2-naphthyl derivative, however, was cleanly functionalized at the less sterically hindered position to give acetoxylated product **217** in good yield.

Table 2.5.7. Acetoxylation of anti diastereomers

ArCHO	product	yield (%) Ar$_{ox}$	ArCHO	product	yield (%) Ar$_{ox}$
Ph	211	42			
p-Me	212	31 (28)[a]	(1-naphthaldehyde)	219	no rxn
m-Me	213	51			
p-OMe	215	23[b]	(2-naphthaldehyde)	217	44

[a] yield in parentheses corresponds to observed diacetoxylated product.
[b] Run at 80 °C for 20 h.

We attributed the low mass recovery of some of the above transformations to the isomerization event necessary for the syn diastereomer to form in order for functionalization to occur. During the isomerization event, the substrate may be more prone to oxidative decomposition, resulting in loss of starting material and/or product. Utilizing the syn diastereomers rather than the anti isomers prevented the unnecessary isomerization event and decreased the amount of oxidative decomposition, thereby giving higher yields of the corresponding acetoxylated products.

We believed that in the functionalization reactions utilizing anti diastereomers, that the anti isomer was first isomerizing to the syn diastereomer and then undergoing functionalization. Subjecting anti **202b** to acetoxylation conditions in PhCH$_3$, we observed no acetoxylated product, and only saw the formation of aminal oxidized byproducts (Scheme 2.5.17). In PhCH$_3$ the anti diastereomer cannot isomerize to syn **202a**, therefore functionalization cannot occur. When syn **202a** was subjected to the same conditions, however, no acetoxylated product was observed. For this particular transformation it seems essential that AcOH/Ac$_2$O be present for functionalization to occur; therefore, it is very difficult to find conditions in which the anti diastereomer cannot isomerize to the syn. Additionally, in all functionalization reactions

utilizing the anti diastereomer, syn isomeric starting material was observed. Treating a 4 : 1 mixture of syn **202** to PTSA in refluxing PhCH$_3$ (conditions that typically provide the anti diastereomer) for 5 days afforded complete isomerization of anti **202b** to syn, and not the reverse. Based on this result, it is unlikely that the syn diastereomer could isomerize to the anti and then undergo functionalization. Furthermore, we have not observed any of anti acetoxylated **220** in any of the functionalization reactions.

Scheme 2.5.17.
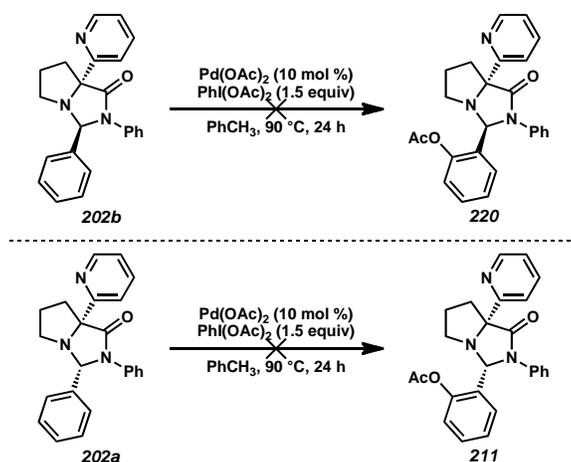
The acetoxylation of sp^2 C–H bonds has been relatively straightforward. The substrate scope for this transformation has been much more extensive than that of sp^3 C–H bond functionalization in our system. For that reason we anticipated that the synchronization of functionalization and transacetalization may be most feasible with aromatic aldehydes.

2.6   Other Attempts at Functionalization of sp^2 C–H Bonds

We have tried additional functionalization reactions with substrate **172** (Table 2.6.1). In an attempt to form a carbon-carbon bond, we employed catalytic palladium and silver acetate in the presence of iodobenzene, but observed none of the desired arylated product.[24] Furthermore, we treated **172** with Sanford's diaryliodonium conditions, but were unable to isolate any of the functionalized product. We also tried to employ an electrophillic rhodium catalyst in the

presence of an acetate source and diphenyl acetylene, but we did not observe any of the olefinated product.[28]

Table 2.6.1. Other *sp²* C–H functionalization reactions

entry	conditions	X
1	Pd(OAc)$_2$, PhI AgOAc, AcOH 130 °C, 48 h	Ph
2	Pd(OAc)$_2$, [Ph$_2$I]BF$_4$ AcOH, 85 °C, 24 h	Ph
3	RhCp*(MeCN)$_3$[SbF$_6$]$_2$ PivOH, PhCCPh PhCH$_3$, 24 h	(styryl)
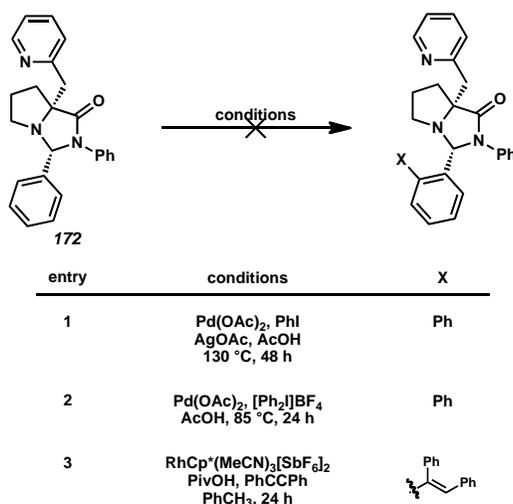
While we have not observed any other transformation other than C–H acetoxylation, there are numerous other transformations and conditions that should be examined. Specifically, the olefination chemistry developed by Sanford and coworkers, which has been successful in our *sp³* C–H olefination reaction, has not been examined in the context of *sp²* C–H bond functionalization in our system. A non-directed version has been developed by Ishii and coworkers, which may prove successful with our ligand-substrate adduct.[29]

2.7 Electronic Modification of Ligating Groups

We imagined electronic and steric differentiation of the pyridyl ligating group may render differential reactivity. Synthesis of the *p*-CN pyridyl was straightforward, with hydroxymethylation[30] of cyanopyridine followed by chlorination to afford the pyridyl chloride (**223**) (Scheme 2.7.1). Installation of the ligating group via alkylation afforded substrate **224**. Treatment with acetoxylation conditions afforded a mixture of 41% diacetoxylated to 36 % monoacetoxylated product. We attributed this increase in reactivity to the electron deficient

nature of the pyridyl ligating group, which should increase the rate of C–H activation, presumably the rate-determining step.

Scheme 2.7.1. Installation of an electron deficient pyridine
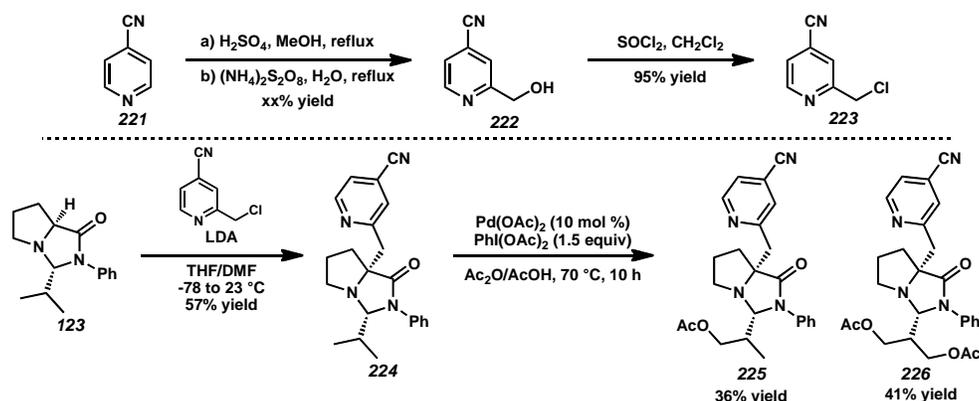
Conversely, we sought to make the pyridyl ligating group more electron rich to observe the effect on the product distribution of functionalization. Construction of the ligating group began with the conversion of 2-methylpyridine into the *p*-NO$_2$-pyridine *N*-oxide via an *m*-CPBA oxidation followed by nitration to afford **228** (Scheme 2.7.1). Conversion of the *p*-NO$_2$ group into the *p*-OMe pyridine via nucleophilic aromatic substitution afforded **229** in excellent yield. Rearrangement in Ac$_2$O followed by cleavage of the resultant acetate, and conversion of the benzylic alcohol into the corresponding chloride afforded **230** in good yield over 3 steps.[31] Alkylation of **123** with the benzylic chloride afforded the substrate in moderate yield. Upon treatment with functionalization conditions afforded 27% monoacetoxylated and 22% yield diacetoxylated product, with 16% of starting material recovered. With a much more electron rich directing group, the C–H activation step is presumably slower, due to the much more electron richness of the coordinated palladium.

Scheme 2.7.1. Installation of an electron rich pyridine ligating group
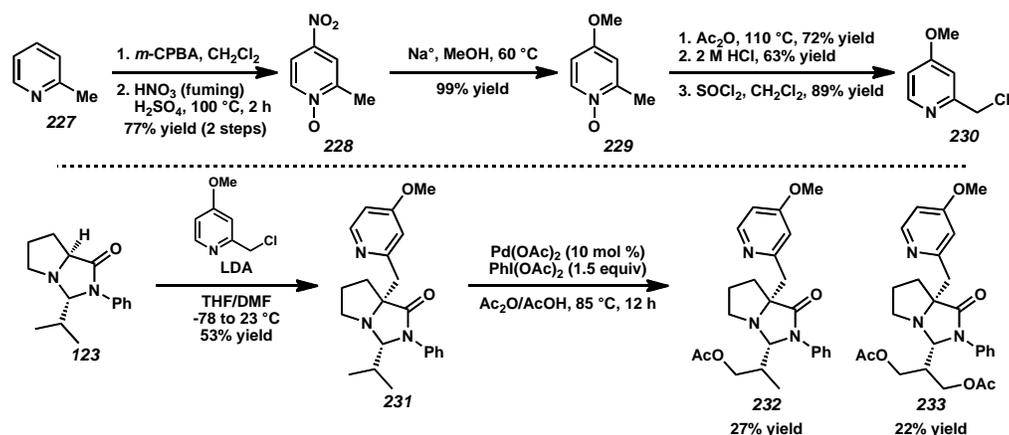
With the addition of the methylene spacer between the pyridyl ligating group and the [3.3.0] system we had observed increased diacetoxylation. We attributed this increased reactivity to the high conformational flexibility imparted by this structure. In an effort to tune this reactivity, we imagined generating a more sterically hindered pyridine which might provide more controlled levels of selectivity. Synthesis of the 6-methylpyridyl ligand began from 2,6-dimethylpyridine, which was converted into the *N*-oxide and then rearranged in Ac_2O to afford the benzylic acetate (**235**) (Scheme 2.7.3). Removal of the acetate and conversion of the resultant alcohol into the chloride afforded pyridine **236**. Alkylation with **123** afforded pyridine **237** in good yield. Treatment with acetoxylation conditions, however, afforded no functionalized product, even after prolonged reaction times at 120 °C. Presumably the pyridyl ligating site is now too sterically hindered to coordinate the palladium effectively to promote acetoxylation.

Scheme 2.7.3. Installation of sterically hindered pyridine
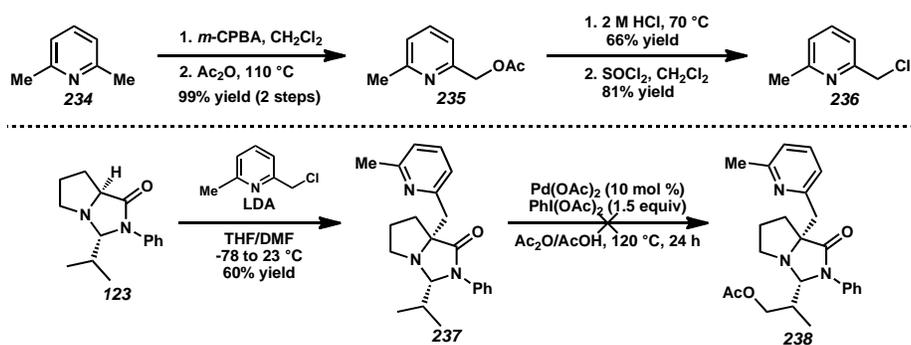
We also imagined restricting the flexibility of the directing group by substituting the benzyl position of the pyridine ligand. We envisaged this could provide higher levels of diastereoselectivity as well as prohibiting the amount of diacetoxylated product. We also believed we could tune the electronic nature of the pyridyl with the group we were to install at the benzylic position. We installed an electron withdrawing carbonyl at the benzylic position by alkylating **123** with 2-ethylpicolinate in good yield (Scheme 2.7.4). Treatment of pyridine **239** with acetoxylation conditions, however, afforded none of the desired product. We attributed this inactivity to the likely chelation of palladium between the pyridyl nitrogen and the oxygen of the newly installed carbonyl.

Scheme 2.7.4. Installation of conformationally restricted pyridine
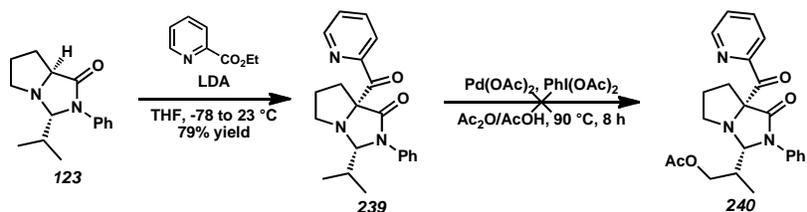
Additionally, we tried modifying the position of the ligating group on our scaffold as well as its electronics. From **119** we were able to synthesize amino amide **241** with a pyridyl ligating group on the amide (Scheme 2.7.5). Condensation onto isobutyraldehyde afforded aminal **242** in good yield. We chose to alkylate the α-position with methyl iodide to avoid any oxidation of the

pyrrolidine ring. Treating substrate **243** with acetoxylation conditions, however, was unsuccessful at generating any of the functionalized product. The palladium catalyst may be bound tightly between the pyridyl nitrogen, pyrrolidine nitrogen and even the amide, which could prevent C–H functionalization.

Scheme 2.7.5. Repositioning the ligating group
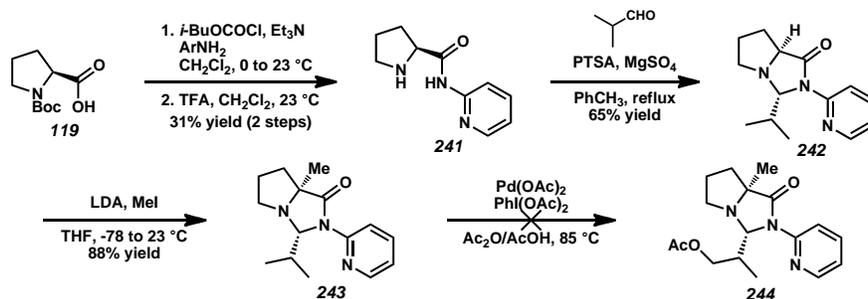
Along with examining electronically modified pyridines in sp^3 C–H acetoxylation, we tested the functionalization of sp^2 C–H bonds using the same ligating groups. Alkylation with **230** afforded a 2.3 : 1 mixture of syn to anti diastereomers (Scheme 2.7.6). Treatment of the mixture with functionalization conditions provided only 40% conversion to the monoacetoxylated product (**247**), with the mass balance containing mostly starting material.

Scheme 2.7.6. Acetoxylation of sp^2 C–H bonds
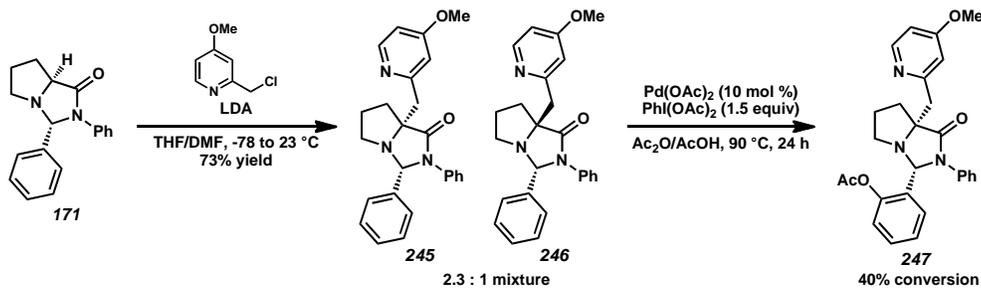
Based on the results of electronically modifying the pyridyl ligating group, it appears that using a slightly more electron deficient ligating group, such as *p*-CN pyridyl may promote C–H functionalization reactions more readily than the electron rich *p*-OMe pyridyl. Very little

investigation into the effects of these different pyridyl ligating groups has been done, and may eventually expand the types of C–H functionalization possible on these substrates.

2.8   Synchronization of Functionalization and Transacetalization

Thus far we have synthesized a ligand capable of condensing onto an aldehyde in a stereoselective fashion. We have also been able to perform a functionalization of the sp^2 and sp^3 C–H bonds of the aldehyde substrate when it is bound to our scaffold. In order to achieve our desired transformation (Scheme 2.8.1) we need to be able to synchronize the functionalization of the aldehyde with transacetalization. Having studied the functionalization independently, we now sought to study the transacetalization aspect of our general hypothesis and eventually see if we could achieve synchronization of the two processes.

Scheme 2.8.1. Revisiting the general concept of transacetalization and functionalization
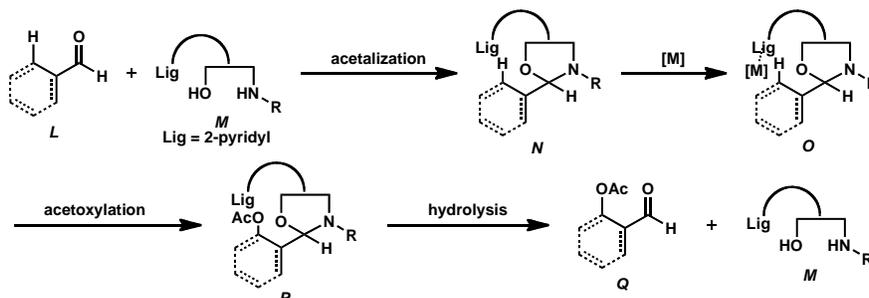
We began this study of transacetalization independent of the functionalization conditions. We wanted to first examine the general ability of the substrate-ligand adduct to exchange with exogenous aldehyde. Subjecting pyridine **129** to various acids in toluene in the presence of superstoichiometric isovaleraldehyde afforded none of the desired exchange product (Scheme 2.8.2). Treating with PTSA in toluene at room temperature provided a small amount of the exchange product (**248**).

Scheme 2.8.2. Initial exchange of isobutyraldehyde
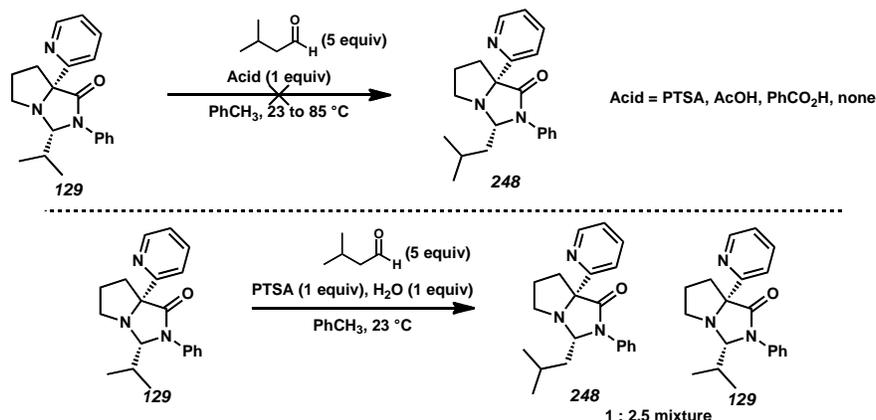
We next tried to exchange aromatic aldehydes employing similar conditions with a variety of solvents (Table 2.8.1). In toluene at 100 °C we observed almost 75% conversion to exchange product **205** as a mixture of diastereomers. Treating with PTSA in THF afforded 75% conversion to the exchange product. In MeOH, near complete exchange was observed with both PTSA and CSA in 24 h. Employing *t*-AmOH as the solvent also provided high levels of conversion with CSA, with slightly diminished reactivity with PTSA. Reactions run in isopropanol and *t*-butanol afforded less exchange product **205**. In AcOH, with no additional acid we saw greater than 75% conversion to the exchange product, which was a promising lead for the acetoxylation reaction, as AcOH is a necessary solvent. Running the exchange reaction in a mixed solvent of AcOH/Ac$_2$O, however, afforded none of the exchange product, and only provided isomerization.

Table 2.8.1. Exchange of aromatic aldehydes

entry	acid	solvent	*202 : 205*	entry	acid	solvent	*202 : 205*
1	CSA	PhCH$_3$	1 : 2.8	6	PTSA	*t*-AmOH	1 : 7.0
2	PTSA	THF	1 : 3.0	7	CSA	*i*-PrOH	1 : 3.8
3	PTSA	MeOH	1 : 11.3	8	CSA	*t*-BuOH	1 : 5.1
4	CSA	MeOH	1 : 11.0	9	-	AcOH	1 : 3.5
5	CSA	*t*-AmOH	1 : 11.5	10	-	AcOH/Ac$_2$O	1 : 0
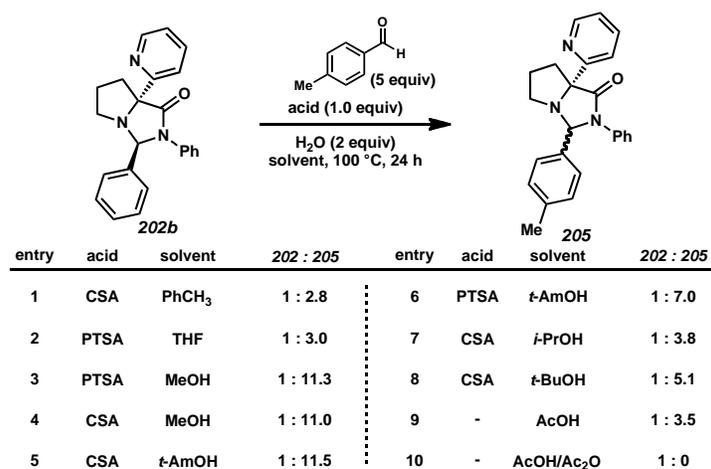
We also tried to employ exchange conditions to substrate **174** (Scheme 2.8.3). Treating functionalized **174** with benzaldehyde and CSA in THF afforded a small amount of exchange product **172**. Additionally, we treated **172** with *p*-tolualdehyde and CSA in alcoholic solvents, but observed only minimal exchange. Furthermore, in AcOH only a trace amount of the exchange product was observed, illustrating the limited exchangeability of this ligand-substrate adduct.

Scheme 2.8.3. Exchange of functionalized product

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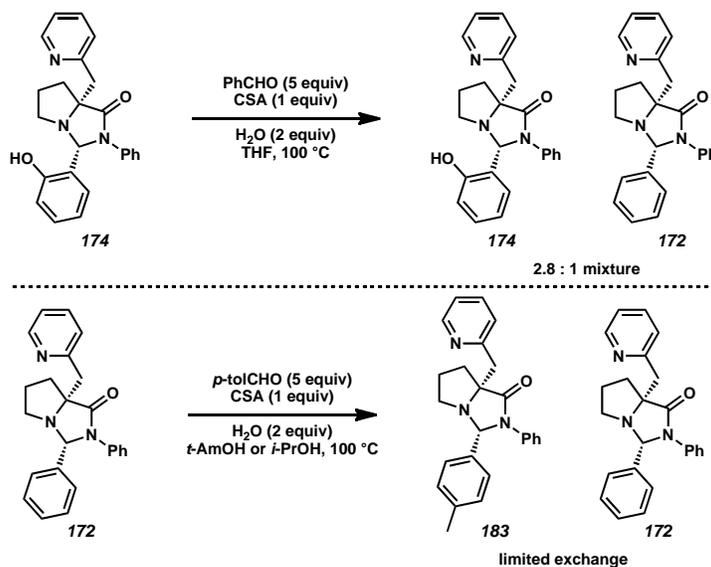
After testing the exchangeability of our ligand-substrate adducts independent of functionalization conditions, we sought to test the synchronization of functionalization and transacetalization with our scaffold. Treating pyridine **172** with acetoxylation conditions with additional benzaldehyde and water afforded functionalized product **173** with only limited amounts of acetoxylated salicyaldehyde (Scheme 2.8.4). We believed the exchange of benzaldehyde for **249** was very slow under these conditions. Additionally, we tried to exchange benzaldehyde with substrate **183** under the acetoxylation conditions. We observed a very small amount of the functionalized exchange product under these conditions.

Scheme 2.8.4. Initial attempts at synchronizing transacetalization with functionalization
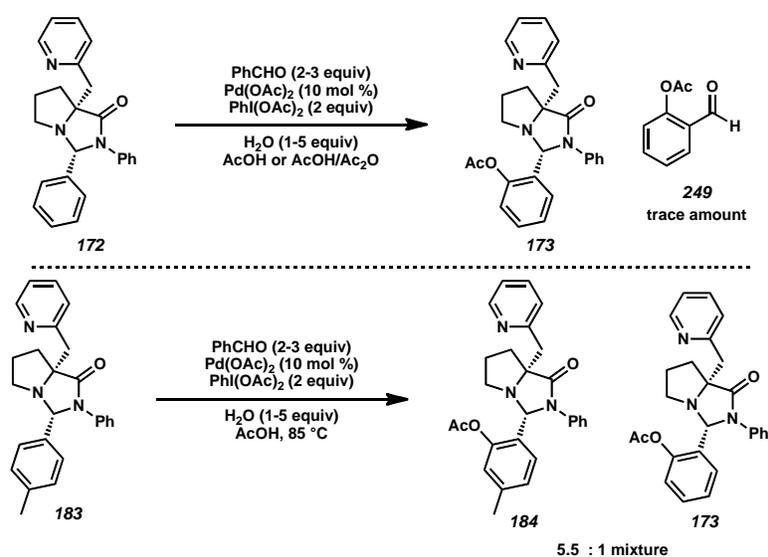
With our limited success with substrate **183**, we sought to examine the electronically modified phenyl amide substrates (Scheme 2.8.5). Subjecting tolualdehyde substrate **187-CF₃** to benzaldehyde and functionalization conditions afforded a 2.2 : 1 mixture of acetoxylated **189-CF₃** to functionalized exchange product **181-CF₃**. Treating the *p*-OMe variant with acetoxylation conditions and benzaldehyde afforded none of the acetoxylated exchange product (**181-OMe**), but good exchange was observed. We attributed this lack of reactivity to the oxidatively sensitive aminal center.

Scheme 2.8.5. Electronically modified pyridines for synchronization

X = CF₃   2.2 : 1 mixture

X = OMe   1 : 0 mixture
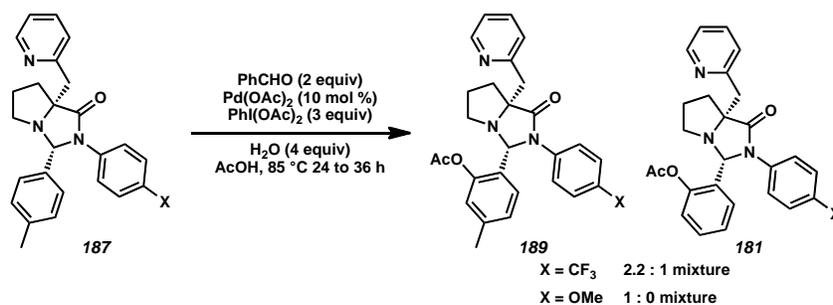
We envisioned that using a two-step addition of catalyst and oxidant might provide higher levels of conversion to the functionalized exchange product. Treating **183** with benzaldehyde, catalyst and oxidant in AcOH at 105 °C for 18 h, followed by more catalyst and oxidant afforded a 1.5 : 1 mixture of **185** to **173** (Scheme 2.8.6). We increased the temperature of the reaction in an effort to see higher levels of exchange, likely trading off for more decomposition.

Scheme 2.8.6. Small amounts of synchronization

1.5 : 1 mixture
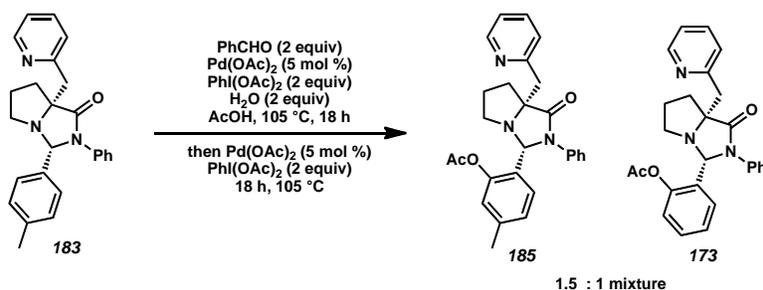
Next we sought to examine the synchronization of functionalization and transacetalization utilizing our isobutyraldehyde ligand adduct **250**. Although exchanging with another aliphatic aldehyde may be unsuccessful, we imagined using an aromatic exchange aldehyde. Treating **250** with benzaldehyde, catalyst and oxidant, again in a two step procedure afforded 30% yield of the exchange product (**179-CF₃**) and 7% yield of the activated exchange product (**181-CF₃**) (Scheme 2.8.7).

Scheme 2.8.7. Synchronization of aliphatic and aromatic aldehydes
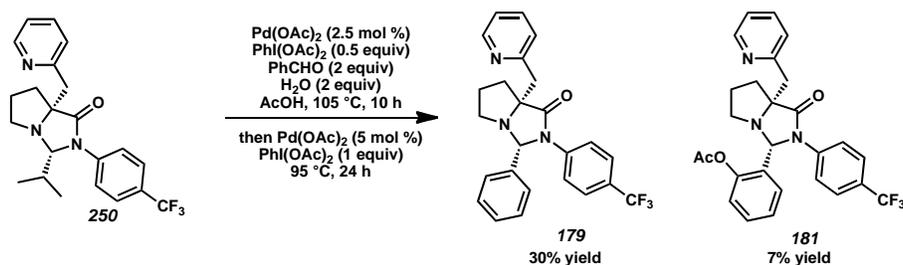
While we had observed some synchronization of functionalization and transacetalization, we wanted to improve the yield of the functionalized exchange product. A literature example showed transacetalization of aminals using an amine in the reaction.[32] We tried this reaction on substrates **129** and **172** with no observed free ligand or exchange product (Scheme 2.8.8). Interestingly, when we used anti isomer **220** in the reaction, we observed good exchange and a small amount of free ligand. While we were unable to use additional amine in the functionalization conditions, due to immediate acetalization, we believed using the anti isomer of **220** may exhibit improved transacetalization.

Scheme 2.8.8. Attempted ligand hydrolysis/exchange
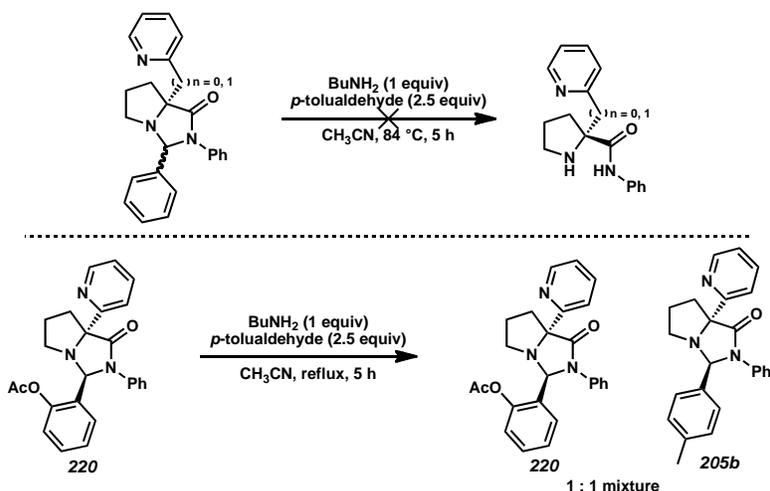
Based on these results, we hypthesized that using **202b** as the starting ligand adduct might facilitate transacetalization and thereby functionalization. Treating anti **202b** with *p*-

toluadehyde, catalyst and oxidant afforded a 1 : 1 mixture of exchange product to starting

material, but no functionalization of either substrate was observed (Scheme 2.8.9).

Scheme 2.8.9. Another attempt at synchronization
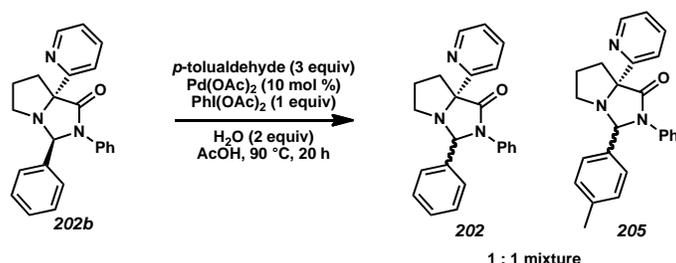
We turned to our two-step procedure in order to improve the yield of the functionalized

exchange product. Treating anti **202b** with *p*-tolualdehyde, catalyst and oxidant for 20 h at 90

°C, followed by the addition of more catalyst, oxidant and Ac₂O afforded 16% yield of the

exchange product (**205**) and 19% yield of the functionalized exchange product (**212**) (Scheme

2.8.10). It is likely that the first step induces the transacetalization, and the addition of more

catalyst and oxidant in the second step performs the functionalization. While not the ideal

synchronized process, this reaction still demonstrates that exchange and functionalization can

occur under similar reactions conditions to afford functionalized exchange product.

Scheme 2.8.10. Successful transacetalization and functionalization
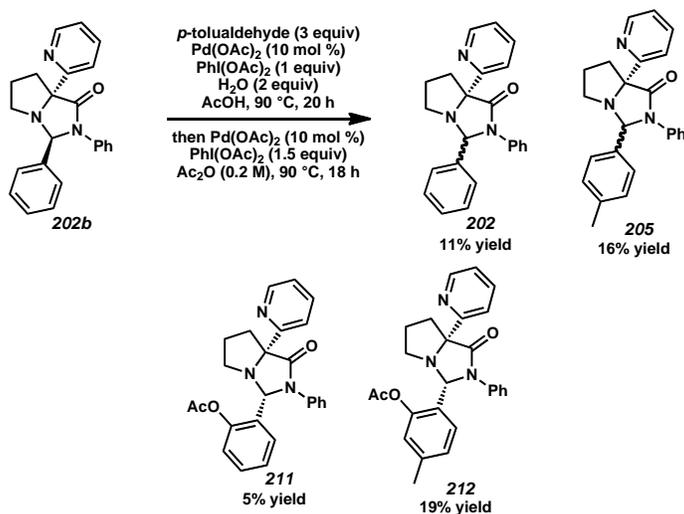
Lastly, we sought to remove the ligand independent of the functionalization conditions. Treating any functionalized product with Lewis acids or exchange acetal components afforded none of the desired acetoxylated aldehyde in either aromatic or aliphatic cases (Scheme 2.8.11). Removing the ligand from the activated isobutyraldehyde proved exceedingly difficult, presumably because elimination could occur to form an acrolein byproduct.

Scheme 2.8.11. Removal of the ligand from functionalized products
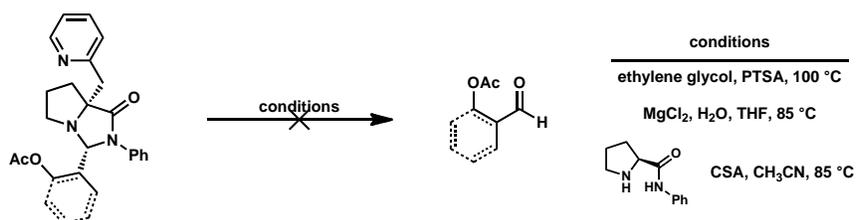
We then turned our attention to removing the ligand from aromatic aldehydes, as we could actually isolate the product without elimination. We imagined testing the unactivated product first, as it should be easier to remove the ligand from **251**. Treating **172** with amino amide **122** and acid afforded a complex mixture of products (Table 2.8.2). We imagined that silica gel was weakly acidic enough to possibly remove the ligand, and we had observed slight decomposition when purifying these *N,N*-aminals by chromatography. At 85 °C in MeOH however, we only observed recovered starting material. We then tried a reductive cleavage using NaBH₄, which also only afforded starting material. Using a stronger reducing agent like LAH reduced the amide to the secondary amine, forming a very labile aminal. Treatment with aqueous acid cleaved the aminal to afford benzaldehyde in 25% yield.

Table 2.8.2. Conditions for ligand removal

entry	conditions	result
1	122, CSA CH₃CN	complex mix
2	silica, MeOH H₂O, 85 °C	172
3	NaBH₄, MeOH 80 °C	172
4	LAH, THF 1 M KHSO₄	25% yield
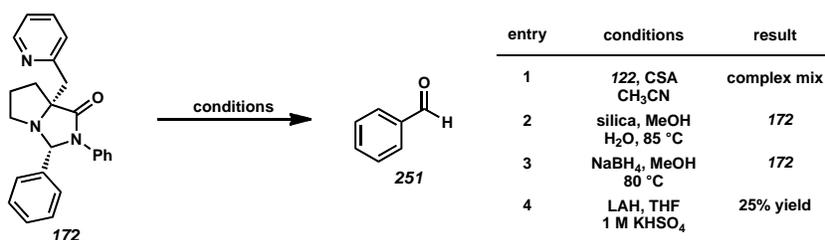
Rather than finding an acetal component which could exchange with the ligand and condense onto the desired aldehyde, we endeavored to use a sacrificial aldehyde which could condense with the ligand and free the desired aldehyde. Again we turned to aromatic aldehydes, as these substrates would avoid elimination and reduce the complexity of the product mixture. Additionally, we used substrate **174** as it was afforded from acetate **173** upon exposure to acid. Using isobutyraldehyde as the sacrificial aldehyde we observed less than 15% liberation of aldehyde as based on crude ^{1}H NMR (Scheme 2.8.12). Changing the solvent to THF afforded much higher conversion (nearly 40%). Using CH₃CN and dioxane as the solvent also increased the conversion of observed salicyladehyde.

Scheme 2.8.12. Transacetalization for ligand removal

entry	conditions	174 : 249
1	MeOH (0.1 M)	6.6 : 1
2	THF (0.4 M)	1.6 : 1
3	CH₃CN (0.25 M)	1.7 : 1
4	dioxane (0.25 M)	1.4 : 1
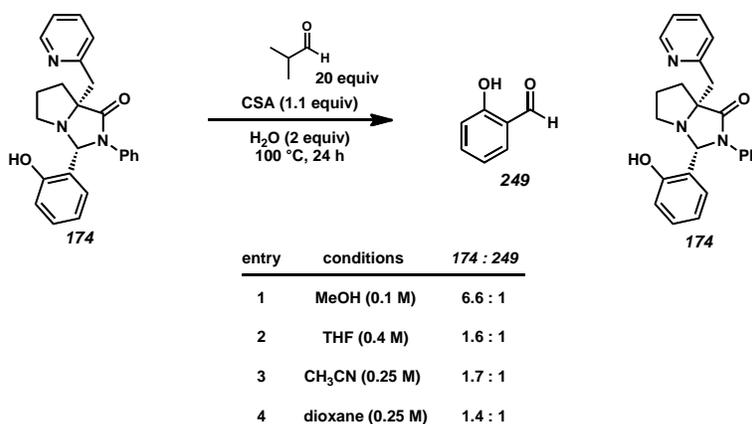
Based on the success of ligand cleavage with isobutyraldehyde, we imagined that using a more electron rich aromatic aldehyde may push the conversion higher. Treating **174** with *p*-tolualdehyde and acid in dioxane afforded nearly 50% conversion by crude ^{1}H NMR (Scheme

2.8.13). Analysis by GC, however, showed only a 16% yield of salicylaldehyde. We attribute

this discrepancy to decomposition under the reaction conditions. Furthermore, we tried a more

electron rich *p*-anisaldehyde as the sacrificial component, but could not improve the yield of **249**.

Scheme 2.8.13. Exchange with aromatic aldehydes
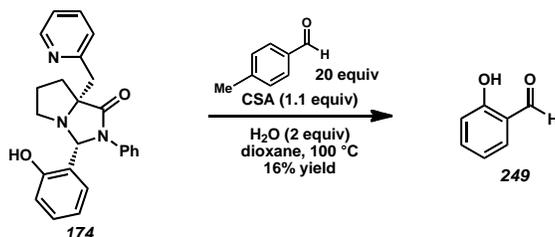
Rather than exchanging an aldehyde or acetal component to free the ligand, we thought

we could use a transamidation reaction.[33] The reaction should create an unstable aminal center,

which could fall apart, yielding the transamidated free ligand and the corresponding imine.

Interestingly, when we subjected substrate **250** to AlCl$_3$ and aniline in DCE we observed near

full conversion to the corresponding free ligand and aniline derived imine **253** (Scheme 2.8.14).

While this is not the transamidation product, it still releases the desired ligand and a cleavable

imine. Likewise, we could perform this cleavage reaction on **179** to afford the free ligand and

imine **254**.

Scheme 2.8.14. Transamidation conditions for ligand removal
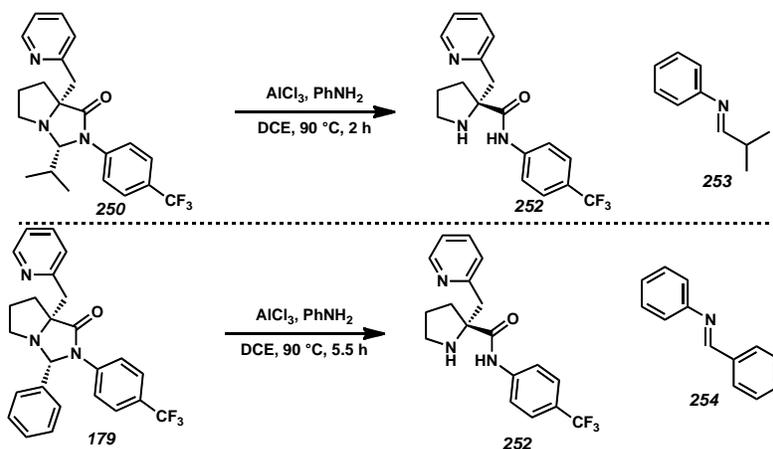
Subjecting the functionalized isobutyraldehyde ligand adduct to the reaction conditions afforded a complex mixture, most likely due to side elimination reactions. Subjecting **181-CF$_3$** to the conditions, however, afforded the phenol imine product (**255**) (Scheme 2.8.15). Cleavage of the imine with aqueous acid afforded salicylaldehyde in 34% over two steps, with 23% recovery of the starting imine.

Scheme 2.8.15. Successful removal of the ligand
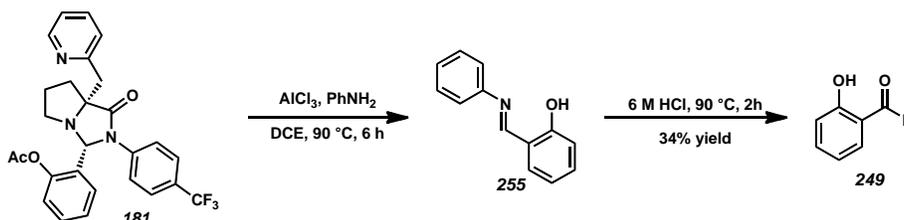
We are now able to perform transacetalization with our ligand under acidic alcoholic conditions. We can also remove the ligand after functionalization using Lewis acidic conditions and a sacrificial amine. Furthermore, we can combine these two processes under our modified acetoxylation conditions and generate a functionalized exchange product. While this process is not yet ideal, it is an initial lead which demonstrates the feasibility of our original hypothesis.

2.9   Conclusions

We have endeavored to combine a regio- and stereoselective functionalization with a removable directing group *in situ*, without the need for extra steps for removal of a ligating group. We have shown the asymmetric development of ligands that can condense onto an aromatic or aliphatic aldehyde and be isolated as a stable aminal. We have shown a highly diastereoselective *sp^3* C–H acetoxylation reaction that generates a single enantiomer. We have also shown that changing the connectivity of the ligating group can impart differential reactivity. Furthermore, we have demonstrated regioselective *sp^2* C–H acetoxylation, again with differential reactivity dependent on ligating group connectivity. Finally, we have shown that

synchronization of transacetalization and functionalization is possible with our scaffold. While we have only displayed modest amounts of this synchronization, the development of new reactions and modification of the scaffold should ultimately lead to the realization of this idea.

2.10   References and Notes

[1] (a) Dick, A. R.; Hull, K. L.; Sanford, M. S. *J. Am. Chem. Soc.* **2004**, *126*, 2300-2301. (b) Desai, L. V.; Hull, K. L.; Sanford, M. S. *J. Am. Chem. Soc.* **2004***, 126*, 9542-9543. (c) Dick, A. R.; Kampf, J. W.; Sanford, M. S. *J. Am. Chem. Soc.* **2005***, 127*, 12790-12791. (d) Kalyani, D.; Deprez, N. R.; Desai, L. V.; Sanford, M. S. *J. Am. Chem. Soc.* **2005***, 127*, 7330-7331. (e) Deprez, N. R.; Sanford, M. S. *J. Am. Chem. Soc.* **2009***, 131*, 11234-11241. (f) Desai, L. V.; Stowers, K. J.; Sanford, M. S. *J. Am. Chem. Soc.* **2008***, 130*, 13285-13293………..

[2] (a) Giri, R.; Liang, J.; Lei, J. –G.; Li, J. –J.; Wang, D. –H.; Chen, X.; Naggar, I. C.; Guo, C.; Foxman, B. M.; Yu, J. –Q. *Angew. Chem. Int. Ed.* **2005***, 44*, 7420-7424. (b) Wang, D. –H.; Wu, D. –F.; Yu, J. –Q. *Org. Lett.* **2006***, 8*, 3387-3390.

[3] (a) Boes, M.; Naef, R.; Schweizer, W. B.; Seebach, D. *J. Am. Chem. Soc.* **1983**, *105*, 5390-5398. (b) Sting, A. R.; Hoffman, M.; Seebach, D. *Angew. Chem., Int. Ed. Engl.* **1996**, *35*, 2708.

[4] Jørgensen, M.; Lee, S.; Liu, X.; Wolkowski, J. P.; Hartwig, J. F. *J. Am. Chem. Soc.* **2002***, 124*, 12557-12565.

[5] Tempo-sodium hypochlorite and a buffered sodium chlorite procedure were attempted with no isolated acid.

[6] Becker, Y.; Eisendstadt, A.; Stille, J. K. *J. Org. Chem.* **1980**, *45*, 2145-2151.

[7] We also converted the alcohol to the corresponding tosylate and mesylate, expecting the amide to cyclize *in situ*, but no oxazoline product was recovered.

[8] Ishihara, M.; Togo, H. *Tetrahedron* **2007***, 63,* 1474.

[9] We employed TFA/CH_2Cl_2 (which led to decomposition), EtOAc/HCl (aq) provided a small amount of deprotected product, and $ZnBr_2$ which was unreactive.

[10] Hwu, J. R.; Jain, M. L.; Shwu-Chen, T.; Hakimelahi, G. H. *Tetrahedron Lett.* **1996**, *37*, 2035-2038.

[11] Lyons, T. W.; Sanford, M. S. *Chem. Rev.* **2010**, *110*, 1147.

[12] Gupta, S. C.; Hu, H.: Berenschot, D. R.; White, K. B.; Moriarty, R. M. *J. Am. Chem. Soc.* **1981**, *103*, 686.

[13] Uozumi, Y.; Mizutani, K.; Nagai, S. *Tetrahedron Lett.* **2001**, *42*, 407.

[14] Patterson, J. W. *J. Org. Chem.* **1995**, *60*, 4542.

[15] Chen, X.; Engle, K. M.; Wang, D. -H.**; Yu, J. -Q.** *Angew. Chem. Int. Ed.* **2009**, *48*, 5094.

[16] Seco, J. M.; Quiñóá, Riguera, R. *Chem. Rev.* **2004***, 104*, 17-118.

[17] Ramón, D. J.; Guillena, G.; Seebach, D. *Helv. Chim. Acta* **1996**, *79*, 875-894.

[18] Zhao, M. M; Li, J.; Mano, E.; Song, Z. J.; Tschaen, D. M. *Organic Syntheses* **2005**, *81*, 195.

[19] Yabuuchi, T.; Kusumi, T. *J. Org. Chem.* **2000***, 65*, 397-404.

[20] *t*-Butyl groups are generally easier to activate because there are three equal methyl groups, one of which will place a C–H bond in the correct position for activation. See reference 1 and 11 for further reading.

[21] Reddy, B. V. S.; Reddy, L. R.; Corey, E. J. *Org. Lett.* **2006**, *8*, 3391-3394.

[22] (a) Kalyani, D.; Dick, A. R.; Anani, W. Q.; Sanford, M. S. *Tetrahedron* **2006**, *62* 11483-11498. (b) Whitfield, S. R.; Sanford, M. S. *J. Am. Chem. Soc.* **2007***, 129*, 15142-15143. (c), Stowers, K. J.; Sanford, M. S. *Org. Lett.* **2009**, *11*, 4584-4587.

[23] Giri, R.; Maugel, N. L.; Li, J. –J.; Wang, D. –H; Brezzano, S. P.; Saunders, L. B.; Yu, J. –Q. *J. Am. Chem. Soc.* **2007**, *129*, 3510.

[24] Zaitsev, V. G.; Shabashov, D.; Daugulis, O. *J. Am. Chem. Soc*. **2005**, *127,* 13154.

[25] Deprez, N. R.; Sanford, M. S. *J. Am. Chem. Soc.* **2009**, *131*, 11234.

[26] Stowers, K. J.; Fortner, K. C.; Sanford, M. S. *J. Am. Chem. Soc.* **2011**, *133*, 6541.

[27] Wasa, M.; Engle, K. M.; Yu, J. –Q. *J. Am. Chem. Soc.* **2010**, *132*, 3680-3681.

[28] Schipper, D. J.; Hutchinson, M.; Fagnou, K. *J. Am. Chem. Soc.* **2010**, *132*, 6910-6911.

[29] Obora, Y.; Ishii, Y. *Molecules* **2010**, *15*, 1487. This example is a non-directed C–H olefination using a polyoxometalate species as a catalytic additive. It is likely that this transformation would be successful on our aromatic systems, but it has not yet been examined.

[30] Matsumura, A.; Mikamiyama, H.; Tsuno, N.; Kyle, J. D.; Shao, B.; Yao, J. *PCT Int. Appl.* WO2008008398, **2008**, 599.

[31] Finger, G. C.; Starr, L. D. *J. Am. Chem. Soc.* **1959***, 81*, 2674.

[32] Li, D.; Zhang, Y.; Xia, C.; Guo, W. *Heterocycles* **2005**, *65*, 1829-1836.

[33] Bon, E.; Bigg, D. C. H.; Bertrand, G. *J. Org. Chem.* **1994***, 59*, 4035-4036.

2.11 Experimental Procedures and Characterization

Materials and Methods. All reactions were performed under an argon atmosphere unless otherwise noted. Tetrahydrofuran, *N,N*-dimethylformamide, dichloromethane, hexanes, and toluene were purified by passing through activated alumina columns. Diisopropylamine was distilled over CaH_2. 2-Fluoropyridine was freshly distilled before use. All other reagents were used as received unless otherwise noted. Commercially available chemicals were purchased from Alfa Aesar (Ward Hill, MA), Sigma-Aldrich (St. Louis, MO), Gelest (Morrisville, PA), Oakwood Products (West Columbia, SC), Strem (Newburport, MA), Mallinckrodt Chemicals (Phillipsburg, NJ), Spectrum (Gardena, CA) Fischer Scientific (Fair Lawn) and TCI America (Portland, OR). Qualitative TLC analysis was performed on 250 mm thick, 60 Å, glass backed, F254 silica (Silicycle, Quebec City, Canada). Visualization was accomplished with UV light and exposure to either *p*-anisaldehyde or $KMnO_4$ solution followed by heating. Flash chromatography was performed using Silicycle silica gel (230-400 mesh). ^{1}H NMR spectra were acquired on either a Varian Mercury 300 (at 300 MHz), a Varian Inova 400 (at 400 MHz), or a Varian 400 MR (at 400 MHz) and are reported relative to $SiMe_4$ (δ 0.00). ^{13}C NMR spectra were acquired on either a Varian Inova 400 (at 100 MHz), a Varian Mercury 300 (at 75 MHz), or a Varian 400 MR (at 100 MHz) and are reported relative to $SiMe_4$ (δ 0.0). All IR spectra were obtained on NaCl plates (film) with either a Nicolet Magna FTIR 760, a Nicolet 380 FTIR, or a Bruker Tensor 27. High resolution mass spectrometry data were acquired by the Colorado State University Central Instrument Facility on an Agilent 6210 TOF LC/MS.

Acetal 69: To a suspension of L-proline (1.09g, 8.69 mmol) and pentane (35 mL) was added pivaladehyde (4.83 mL, 43.5 mmol) and TFA (13µL). The suspension was reflulxed for 48 h, with azeotropic removal of water. The clear reaction mixture was concentrated to afford **69** (1.31 g, 82% yield) as a clear oil.

Pyridine 71: To a solution of diisopropylamine (0.240 mL, 1.67 mmol) and tolune (degassed) (1.00 mL) was added n-BuLi (1.00 mL, 1.61 mmol, 1.6 M in hexanes) at -78 °C. The solution was stirred at -78 °C for 5 min, then at 0 °C for 10 min. A solution of **69** (250 mg, 1.24 mmol) in toluene (degassed) (2.00 mL) was added to the solution at 0 °C, and stirred for 20 min. To a solution of Pd(dba)$_2$ (36.0 mg, 0.062 mmol) and P(t-Bu)$_3$HBF$_4$ (18.0 mg, 0.062 mmol) in toluene (degassed) (1.00 mL) was added 2-bromopyridine (0.120 mL, 1.24 mmol). To this solution was added the enolate solution. The reaction was stirred at 23 °C overnight. The reaction was concentrated under reduced pressure with no product formation by ^{1}H NMR.
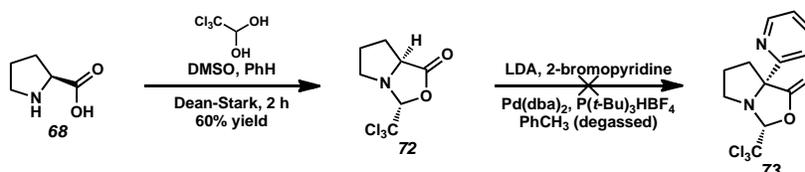
Acetal 72: L-proline (7.00 g, 60.8 mmol) and chloral hydrate (16.6 g, 100 mmol) and DMSO (4.00 mL) were refluxed in benzene (203 mL) for 2 h with azeotropic removal of water. The cooled solution was washed with sat. aqueous NaHCO$_3$ (4 x 75 mL), dried over MgSO$_4$ and concentrated yielding a yellow-orange solid. The product was decolorized over a small plug of silica (5 cm x 5 cm) to yield **72** (8.60 g, 58% yield, R$_f$ = 0.29 in 9:1 hexanes:EtOAc) as a light orange solid.

Pyridine 73: To a solution of diisopropylamine (0.180 mL, 1.25 mmol) and toleune (degassed) (1.00 mL) was added n-BuLi (0.750 mL, 1.21 mmol, 1.6 M in hexanes) at -78 °C. The solution was stirred at -78 °C for 5 min, then at 0 °C for 10 min. A solution of **72** (249 mg, 1.02 mmol)

in toluene (degassed) (2.00 mL) was added to the solution at 0 °C, and stirred for 20 min. To a solution of Pd(dba)$_2$ (27.0 mg, 0.046 mmol) and P(*t*-Bu)$_3$HBF$_4$ (13.0 mg, 0.046 mmol) in toluene (degassed) (1.00 mL) was added 2-bromopyridine (0.0900 mL, 0.927 mmol). To this solution was added the enolate solution. The reaction was stirred at 23 °C overnight. The reaction was concentrated under reduced pressure with no product formation by ^{1}H NMR.
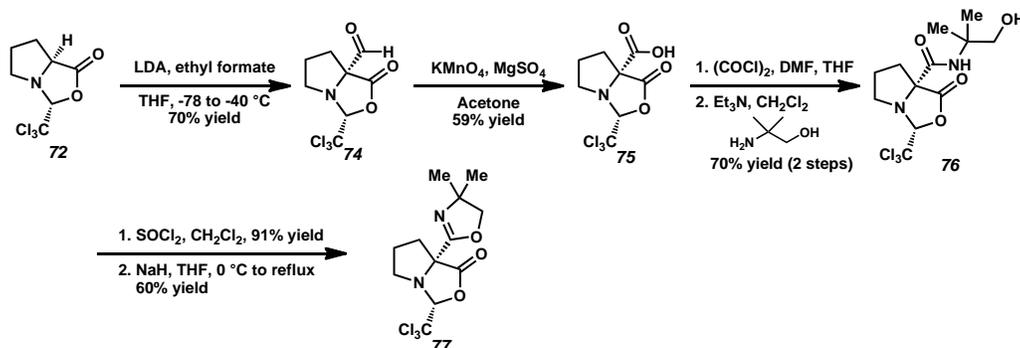
Aldehyde 74: To a solution of diisopropylamine (1.79 mL, 12.7 mmol) and THF (15.4 mL) at -78 °C was added *n*-BuLi (4.9 mL, 12.3 mmol, 2.5 M in hexanes). Stirred for 5 min at -78 °C then warmed to 0 °C for 10 min. To this solution at -78 °C was added a solution of **72** (2.01 g, 8.18 mmol) in THF (8.00 mL) dropwise. This solution was stirred for 30 min at -78 °C. Ethyl formate (2.64 mL, 32.7 mmol) was added slowly at -78 °C. The reaction was allowed to warm to -40 °C over 45 min, then stirred for 10 min at -40 °C. Citric acid (10%, 20 mL) was added slowly. The aqueous layer was extracted with ether (2 x 30 mL). The combined organic layers were washed with brine (30 mL), dried over MgSO$_4$ and concentrated yielding an orange oil. The crude product was purified via chromatography (7:1 to 7:3 hexanes:EtOAc) yielding **74** (1.62 g, 70% yield, R$_f$ = 0.40 in 1:1 hexanes:EtOAc).

Acid 75: To a solution of **74** (1.25 g, 4.59 mmol), MgSO$_4$ (0.912 g, 7.57 mmol), and acetone (77.0 mL) was added KMnO$_4$ (1.02 g, 6.43 mmol) portionwise over 45 min. The reaction was stirred at 23 °C for an additional 30 min. The solvent was removed under reduced pressure. The

remaining brown residue was extracted with hot water (3 x 15 mL). The extract was filtered and decolorized with NaSO$_3$ and extracted with CHCl$_3$ (3 x 15 mL). The aqueous layer was concentrated to 30 mL and acidified to pH ~4 with 1 M HCl. The aqueous layer was extracted with CHCl$_3$ (3 x 20 mL). The organic layer was dried over MgSO$_4$ and concentrated yielding **75** (0.778 g, 59% yield, R_f = 0.2 in 1:1 hexanes:EtOAc) as a white solid.

Acid chloride: To a solution of **75** (0.603 g, 2.09 mmol) in THF (10.5 mL) at 0 °C was added oxalyl chloride (0.912 mL, 10.5 mmol) and DMF (2 drops). The reaction was stirred for 5 min at 0 °C, then 1 h at 23 °C. The reaction mixture was concentrated under reduced pressure and concentrated from benzene (3 x 20 mL) yielding acid chloride (0.528 g, 82% yield, R_f = 0.78 in 1:1 hexanes:EtOAc) as a yellow solid.

Amide 76: To a solution of 3-amino-3-methyl propanol (0.659 mL, 6.88 mmol) Et$_3$N (0.725 mL, 5.16 mmol) and CH$_2$Cl$_2$ (7.50 mL) at 0 °C was added a solution of acid chloride (0.528 g, 1.72 mmol) in CH$_2$Cl$_2$ (8.6 mL) dropwise. The reaction was stirred for 10 min at 0 °C, then at 23 °C for 1 h. The reaction was quenched with 0.5 M HCl (17 mL) and the aqueous layer was extracted with CH$_2$Cl$_2$ (2 x 15 mL). The combined organic layers were washed with sat. aqueous NaHCO$_3$ (20 mL) and brine (20 mL), dried over Na$_2$SO$_4$ and cocnetrated yielding **76** (0.616 g, 99% yield, R_f = 0.60) as an amber oil (yellow amorphous solid under vacuum).

Oxazoline 77: To **76** (0.692 g, 1.93 mmol) in CH$_2$Cl$_2$ (14.1 mL) at 0 °C was added SOCl$_2$ (1.93 mL, 26.5 mmol) dropwise. The reaction was stirred at 0 °C for 10 min, then warmed to 23 °C and stirred for 2 h. The reaction was quenched with H$_2$O (20 mL) and extracted with CH$_2$Cl$_2$ (2 20 mL). The organics were washed with brine (20 mL) dried over Na$_2$SO$_4$ and concentrated to afford the corresponding alkyl chlorde (0.661 g, 91% yield, R_f = 0.62 in 1:1 hexanes:EtOAc). To a suspension of NaH (washed in hexanes) (0.112 g, 2.81 mmol), in THF (5.00 mL) was added

the alkyl chloride (0.529 g, 1.40 mmol) in THF (6.20 mL) at 0 °C. The suspension was stirred at

0 °C for 10 min, then warmed to 23 °C. The reaction was heated to reflux (after adding 3 drops

t-BuOH) for 26 h. The reaction was cooled and quenched with sat. aq. NH4Cl (5 mL) and the

aqueous layer extracted with EtOAc (2 x 10 mL). The organic layers were washed with brine

(15 mL) dried over Na2SO4 and concentrated.   The crude residue was purified via

chromatography (7:3 to 1:1 hexanes:EtOAc) to afford **77** (0.285 g, 60% yield, R_f = 0.38 in 1:1

hexanes:EtOAc) as a light brown solid.
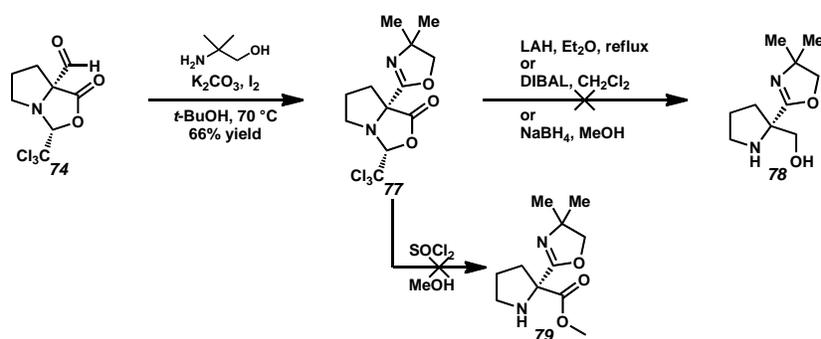
Oxazoline 77: To a solution of **74** (100 mg, 0.367 mmol) in *t*-BuOH (3.70 mL) was added 3-

amino-3-methyl propanol (38.7 µL, 0.404 mmol) at 23 °C. The reaction was stirred at 23 °C for

1 h. K2CO3 (152 mg, 1.10 mmol) and I2 (186 mg, 0.734 mmol) were added at 23 °C. The

reaction was heated to 70 °C for 18 h. The reaction was quenched with Na2SO3 until I2 color

gone. The aqueous was extracted with ether (3 x 10 mL). The organics were washed with sat.

aq. NaHCO3 (10 mL) and brine (10 mL), dried over MgSO4 and concentrated.   The crude

mixture was purified via chromatography (6:1 to 7:3 hexanes:EtOAc) to afford **77** (49.6 mg,

40% yield, R_f = 0.38 in 1:1 hexanes:EtOAc) as a light brown solid.

Ester 79: At 0 °C SOCl2 (0.927 mL, 12.7 mmol) was added slowly to MeOH (2.50 mL). This

solution was stirred at 0 °C for 5 min, then **77** (0.433 g, 1.27 mmol) in MeOH (2.50 mL) was

added slowly. The reaction was warmed to 23 °C and stirred for 5 d. The solution was

concentrated under reduced pressure, the crude residue redissolved in MeOH and concentrated. The residue was taken up in H_2O and neutralized to pH 7 with 1 M NaOH. The aqueous layer was extracted with EtOAc (3 x 10 mL). The combined organics were washed with brine (15 mL) dried over $MgSO_4$ and concentrated under reduced pressure. No product was formed by ^{1}H NMR.

Amino Alcohol 78: To a solution of LAH (51.1 mg, 1.35 mmol) in Et_2O (0.500 mL) was added **77** (70.3 mg, 0.207 mmol) in THF (1.00 mL) and Et_2O (0.500 mL) at 0 °C. The reaction was heated to reflux overnight. The reaction was cooled to 0 °C and H_2O (51.1 µL), 15% NaOH (51.1 µL) and H_2O (153 µL) were added sequentially. The mixture was stirred for 1 h, forming a white precipitate. The precipitate was filtered with EtOAc and the organic layer dried over Na_2SO_4 and concentrated. No product was observed by ^{1}H NMR.

Amino Alcohol 78: To a solution of **77** (62.6 mg, 0.184 mmol) in CH_2Cl_2 (1.00 mL) at -78 °C was added DIBAL (0.368 mL, 0.368 mmol, 1.0 M in hexanes) dropwise. The reaction was stirred at -78 °C for 2 h. The reaction was quenched with EtOH. To the mixture was added sat. aq. Rochelle's salt and EtOAc. The mixture was stirred overnight. The aqueous layer was extracted with EtOAc (2 x 10 mL). The combined organic layers were dried over Na_2SO_4 and concentrated. No product was observed by ^{1}H NMR.

Amino Alcohol 78: To a solution of **77** (49.5 mg, 0.146 mmol) in MeOH (4.90 mL) at 23 °C was added $NaBH_4$ (9.7 mg, 0.256 mmol). The reaction was stirred overnight at 23 °C. The reaction was quenched with sat. aq. NH_4Cl and the solvent removed under reduced pressure. The aqueous layer was extracted with EtOAc (2 x 5 mL), the organic layers dried over Na_2SO_4 and concentrated. No product was observed by ^{1}H NMR.

Base	yield
LDA	19%
KHMDS	0%
LiHMDS	0%
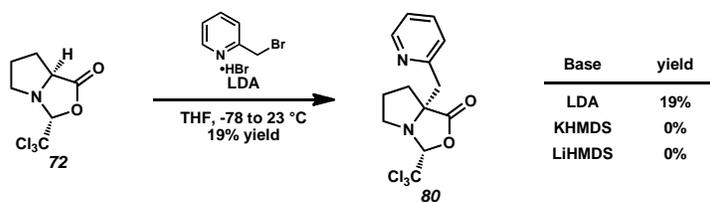
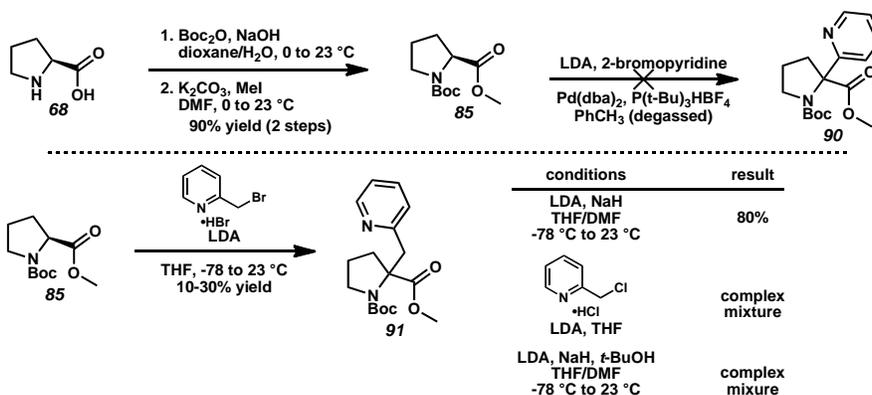**Pyridine 80**: To a solution of diisopropylamine (0.65 mL, 4.62 mmol) in THF (1.00 mL) at -78 °C was added *n*-BuLi (1.81 mL, 4.53 mmol, 2.5 M in hexanes). The solution was stirred for 10 min at -78 °C. This solution was added to a solution of 2-bromomethylpyridine hydrobromide (0.433g, 1.71 mmol) in THF (1.00 mL) at -78 °C. To this suspension was added **72** (0.500 g, 2.05 mmol) in THF (4.00 mL) at -78 °C. The reaction was stirred at -78 °C for 1.5 h. The reaction was quenched with H_2O (10 mL). The aqueous layer was extracted with CH_2Cl_2 (3 x 10 mL). The organic layers were washed with brine (15 mL), dried over $MgSO_4$ and concentrated. The crude mixture was purified via chromatography (4:1 to 2:3 hexanes:EtOAc) to afford **80** (0.114 g, 26% yield, $R_f = 0.29$ in 1:1 hexanes:EtOAc) as a yellow solid.

Pyridine 80: To a suspension of 2-bromomethylpyridine hydrobromide (0.519 g, 2.05 mmol) in THF (1.00 mL) at -78 °C was added KHMDS (409 mg, 2.05 mmol) in THF (1.78 mL). To a solution of KHMDS (0.961 g, 4.82 mmol) in THF (4.20 mL) was added **72** (1.00 g, 4.09 mmol) as a solution in THF (5.00 mL). The electrophile solution was added to the enolate slowly at -78 °C. The reaction was stirred at -78 °C for 1.5 h. The reaction was quenched with H_2O (10 mL). The aqueous layer was extracted with CH_2Cl_2 (3 x 10 mL). The organic layers were washed with brine (15 mL), dried over $MgSO_4$ and concentrated. No product was observed by ^{1}H NMR.

Pyridine 80: To a solution of HMDS (1.46 mL, 6.87 mmol) in THF (3.00 mL) at -78 °C was added *n*-BuLi (2.71 mL, 6.77 mmol, 2.5 M in hexanes). The solution was stirred for 10 min at -78 °C. This solution was added to a solution of 2-bromomethylpyridine hydrobromide (0.517 g, 2.05 mmol) in THF (1.00 mL) at -78 °C. To this suspension was added **72** (1.00 g, 4.09 mmol)

in THF (4.00 mL) at -78 °C. The reaction was stirred at -78 °C for 1.5 h. The reaction was quenched with H$_2$O (10 mL). The aqueous layer was extracted with CH$_2$Cl$_2$ (3 x 10 mL). The organic layers were washed with brine (15 mL), dried over MgSO$_4$ and concentrated. No product was observed by ^{1}H NMR.

Acid 81: HCl (2.89 mL, 2.0 M) was added to **80** (0.114 g, 0.340 mmol) and heated to reflux for 1 h. The reaction mixture was filtered and extracted with CH$_2$Cl$_2$ (3 x 5 mL). The aqueous layer was concentrated under reduced pressure. No product was observed by ^{1}H NMR.

Ester 82: SOCl$_2$ (0.490 mL, 6.78 mmol) was added to MeOH (1.00 mL) at 0 °C and stirred for 5 min at 0 °C. To this solution was added **80** (0.228 g, 0.678 mmol) in MeOH (2.00 mL) at 0 °C. the reaction was warmed to 23 °C and stirred overnight. The mixture was concentrated under reduced pressure. The residue was dissolved in H$_2$O, neutralized with 10% NaOH and extracted with EtOAc (3 x 5 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. No product was observed by ^{1}H NMR.

Amino alcohol 83: To a suspension of LAH (14.6 mg, 0.385 mmol) in THF (0.500 mL) was added **80** (56.5 mg, 0.257 mmol) in THF (1.50 mL) at 0 °C. The mixture was warmed to 23 °C and stirred for 3 h. The reaction was quenched with H$_2$O, acidified with 10% H$_2$SO$_4$ and stirred for 15 min. The mixture was filtered and the aqueous layer extracted with EtOAc (3 x 5 mL).

The combined organic layers were washed with brine (10 mL) dried over MgSO$_4$ and concentrated. No product was observed by ^{1}H NMR.

Prolinol 84: To a solution of NaBH$_4$ (27.0 mg, 0.716 mmol) in MeOH (5.00 mL) was added **72** (100 mg, 0.409 mmol) in MeOH (8.00 mL) at 23 °C. The reaction was stirred at 23 °C for 19 h. The reaction was quenched with acetone (2.70 mL) at 0 °C and stirred for 10 min. The solvent was removed under reduced pressure. The residue was taken up in H$_2$O, then extracted with CH$_2$Cl$_2$ (2 x 10 ml). The organic layers were dried over Na$_2$SO$_4$ and concentrated. No product was observed by ^{1}H NMR.
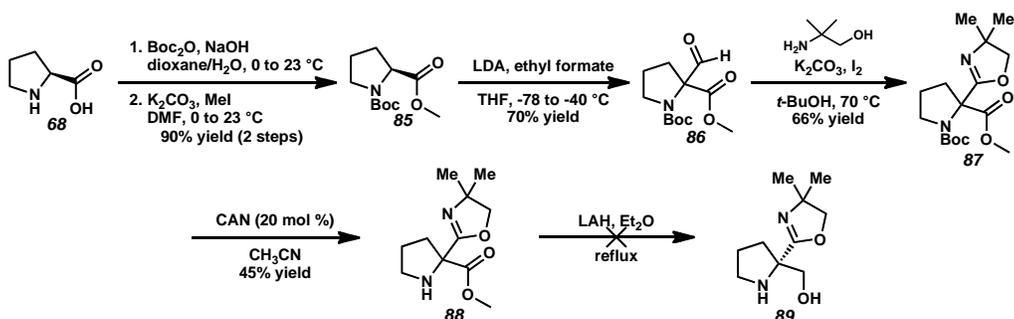
Ester 85: To a solution of (*S*)-proline (**68**) (15.0 g, 130 mmol) in aq. NaOH (1 M, 261 mL) and dioxane (65.2 mL) at 0 °C was added Boc$_2$O (33.1 g, 154 mmol) portionwise over 20 min. The resulting mixture was stirred at 0 °C for 30 min, then allowed to warm to 23 °C and stirred overnight. The organic solvent was removed in vacuo. The remaining aqueous solution was acidified to pH ~2 with 1 M KHSO$_4$. The aqueous solution was extracted with CHCl$_3$ (3 x 150 mL). The combined organic layers were washed with brine (200 mL), dried over Na$_2$SO$_4$ and concentrated to afford carbamate (28.0 g, 99% yield, R$_f$ = 0.17 in 1:1 hexanes/EtOAc) as a white solid.

To the carbamate (2.00 g, 9.29 mmol) in DMF (9.30 mL) was added K$_2$CO$_3$ (1.41 g, 10.2 mmol) at 0 °C. The suspension was stirred 10 min at 0 °C, then MeI (1.20 mL, 18.6 mmol) was added at 0 °C and stirred for an additional 30 min at 0 °C, then 3 h at room temperature. The reaction

mixture was filtered, then partitioned between H$_2$O (50 mL) and EtOAc (50 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na$_2$SO$_4$ and concentrated. The crude mixture was purified via flash chromatography (1:3 EtOAc:hexanes) to give **85** (2.09 g, 98% yield, R$_f$ = 0.60 in 1:1 EtOAc:hexanes) as a clear oil.

Aldehyde 86: To a solution of diisopropylamine (0.365 mL, 2.60 mmol) in THF (2.00 mL) at -78 °C was added *n*-BuLi (1.01 mL, 2.52 mmol, 2.5 M in hexanes). The mixture was stirred for 10 min at -78 °C. To this solution was added **85** (0.385 g, 1.68 mmol) in THF (2.80 mL) at -78 °C. The reaction was stirred an additional 30 min at -78 °C, at which time ethyl formate (0.543 mL, 6.72 mmol) was added at -78 °C. The reaction was warmed to -30 °C over 6 h. The reaction was quenched with citric acid (10 mL, 10%). The aqueous was extracted with EtOAc (2 x 10 mL). The organic layers were washed with brine (15 mL), dried over Na$_2$SO$_4$ and concentrated. The crude mixture was purified via flash chromatography (6:1 to 7:3 hexanes:EtOAc) to afford **86** (0.303 g, 70% yield, R$_f$ = 0.50 in 1:1 hexanes:EtOAc) as a clear oil.

Oxazoline 87: To a solution of **86** (303 mg, 1.18 mmol) in *t*-BuOH (11.8 mL) was added 3-amino-3-methyl propanol (125 µL, 1.30 mmol) at 23 °C and stirred for 1 h. K$_2$CO$_3$ (489 mg, 3.54 mmol) and I$_2$ (599 mg, 2.36 mmol) were added to the reaction at 23 °C, then heated to 70 °C for 18 h. Na$_2$SO$_3$ was added until the I$_2$ color had disappeared. The aqueous was extracted with Et$_2$O (3 x 15 mL). The organic layers were washed with sat. aq. NaHCO$_3$ (20 mL) and brine (20 mL), dried over MgSO$_4$ and concentrated. The crude residue was purified via flash chromatography (4:1 to 3:2 hexanes:EtOAc) to afford **87** (0.254 g, 67% yield, R$_f$ = 0.60 in 1:1 hexanes:EtOAc) as a light yellow solid.

Amine 88: To a solution of **87** (211 mg, 0.645 mmol) in CH$_3$CN (2.60 mL) was added CAN (70.8 mg, 0.129 mmol) and refluxed for 2 h. Additional CAN (94.7 mg, 0.173 mmol) was added

and refluxed overnight. The mixture was concentrated under reduced pressure. The residue was suspended between sat. aq. $NaHCO_3$ and EtOAc. The aqueous layer was extracted with EtOAc (3 x 10 mL). The organic layers were washed with brine (15 mL), dried over $MgSO_4$ and concentrated. The crude mixture was purified via flash chromatography (1:1 hexanes:EtOAc to 90:5:5 EtOAc:MeOH:Et_3N) to afford **88** (66.7 mg, 46% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as a yellow oil.

Amino alcohol 89: To a solution of LAH (28.0 mg, 0.737 mmol) in Et_2O (2.00 mL) was added **88** (66.7 mg, 0.295 mmol) in Et_2O (1.00 mL) at 0 °C. The reaction was stirred at 23 °C overnight. The reaction was cooled to 0 °C and H_2O (28.0 µL), 15% NaOH (28.0 µL) and H_2O (84.0 µL) were added sequentially. The mixture was stirred for 1 h, forming a white precipitate. The precipitate was filtered with EtOAc and the organic layer dried over Na_2SO_4 and concentrated. No product was observed by ^1H NMR.

Pyridine 90: To a solution of diisopropylamine (75.1 µL, 0.535 mmol) in toluene (degassed) (1.00 mL) at -78 °C was added n-BuLi (0.210 mL, 0.515 mmol, 2.5 M in hexanes). After stirring for 10 min at -78 °C, **85** (100 mg, 0.436 mmol) in toluene (degassed) (1.50 mL) was added at -78 °C and stirred for 20 min at 0 °C. This solution was added to $Pd(dba)_2$ (11.4 mg, 0.0198 mmol), $P(t\text{-Bu})_3HBF_4$ (5.7 mg, 0.0198 mmol) and 2-bromopyridine (38.7 µL, 0.396

100

mmol) in toluene (degassed) (0.500 mL) at 23 °C and stirred overnight. The solvent was removed under reduced pressure. No product was observed by ^{1}H NMR.

Pyridine 91: To a solution of freshly distilled diisopropylamine (479 µL, 3.41 mmol) in THF (4 mL) at -78 °C was added *n*-BuLi (1.7 mL, 1.9 M in hexanes, 3.27 mmol). The solution was stirred for 10 min at -78 °C, at which time **85** (750 mg, 3.27 mmol) in THF (4.2 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 °C. To NaH (327 mg, 8.18 mmol) (which was first washed with hexanes (2 x 1.5 mL)), in DMF (7.0 mL) at 0 °C was added 2-bromomethyl pyridine hydrobromide (690 mg, 2.73 mmol). The suspension was stirred at 0 °C for 30 min at which time it was added quantitatively, with additional DMF (1.2 mL), to the enolate solution at -78 °C. The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched with H$_2$O (20 mL) slowly at first at 23 °C. The aqueous layer was extracted with EtOAc (3 x 20 mL). The combined organic layers were washed with brine (2 x 30 mL), dried over Na$_2$SO$_4$, filtered and concentrated. The crude residue was purified by flash chromatography (1:3 to 3:7 EtOAc/Hexanes eluent) to afford **91** (0.684 g, 78% yield, R$_f$ = 0.33 in 1:1 EtOAc:hexanes) as a yellow solid.
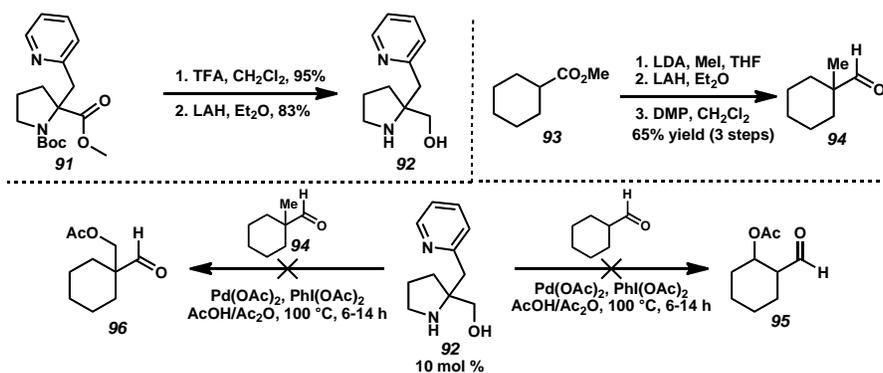
Amino alcohol 92: To a solution of **91** (44.6 mg, 0.139 mmol) in CH$_2$Cl$_2$ (0.300 mL) at 23 °C was added TFA (214 µL, 2.78 mmol). The reaction was stirred at 23 °C for 1 h. The solvent was removed under reduced pressure. The crude residue was neutralized with sat. aq. NaHCO$_3$.

The aqueous layer was extracted with $CHCl_3$ (3 x 5 mL), the organic layers dried over Na_2SO_4 and concentrated to afford amine (35.3 mg, 99% yield, R_f = 0.0 in 1:1 hexanes:EtOAc) as a brown solid.

To a solution of LAH (147 mg, 3.87 mmol) in Et_2O (7.00 mL) was added amine (341 mg, 1.55 mmol) in Et_2O (8.50 mL) at 0 °C. The reaction was stirred at 23 °C for 1 h. The reaction was cooled to 0 °C and H_2O (147 μL), 15% NaOH (147 μL) and H_2O (441 μL) were added sequentially. The mixture was stirred for 1 h, forming a white precipitate. The precipitate was filtered with EtOAc and the organic layer dried over Na_2SO_4 and concentrated. The crude mixture was purified via flash chromatography (90:5:5 EtOAc:Et$_3$N:MeOH) to afford **92** (0.246 g, 83% yield, R_f = 0.0 in 1:1 hexanes:EtOAc) as a clear oil, which began to turn yellow over exposure to air.

Aldehyde 94: To a solution of diisopropylamine (6.13 mL, 43.3 mmol) in THF (32.0 mL) at -78 °C was added *n*-BuLi (26.1 mL, 41.7 mmol, 1.6 M in hexanes). The solution was stirred at -78 C for 5 min, then at 0 °C for 10 min. To this solution at -78 °C was added **93** (5.00 mL, 35.3 mmol). The reaction was stirred at 0 °C for 1 h. To this solution at 0 °C was added MeI (2.0 mL, 32.1 mmol) dropwise. The reaction was stirred at 0 °C for an additional 30 min, then quenched with H_2O (15 mL). The aqueous layer was extracted with ether (4 x 30 mL). The combined organic layers were washed with HCl (50 mL, 10%), brine (50 mL) and H_2O (50 mL), dried over Na_2SO_4 and concentrated to afford the **94a** (4.60 g, 83% yield, Rf = 0.71 in 1:9 hexanes:EtOAc) as a yellow oil.

To a suspension of LAH (1.67 g, 44.1 mmol) and THF (50.0 mL) at 0 °C was added **94a** (4.60 g, 29.4 mmol) in THF (10.0 mL). The reaction was warmed to 23 °C and stirred for 1.5 h. The reaction was cooled to 0 °C and H_2O was added until bubbling stopped. H_2SO_4 (1.00 mL, 10%)

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was added and stirred for 15 min. The mixture was filtered with EtOAc and the organic layer was washed with brine. The aqueous layer was extracted with EtOAc, the combined organic layers dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (9:1 to 6:1 hexanes:EtOAc) to afford **94b** (3.54 g, 94% yield, R_f = 0.17 in 9:1 hexanes:EtOAc) as a clear oil.

To a stirring solution of DMP (3.69 g, 8.59 mmol) in CH_2Cl_2 (40.0 mL) was added **94b** (0.999 g, 7.81 mmol) as a solution in CH_2Cl_2 (20.0 mL). The reaction was stirred at 23 °C for 1 h. Ether was added to the mixture, which was then added to NaOH (50 mL, 1 M) and stirred for 10 min. The ether layer was washed with 1 M NaOH and H_2O. The organic layer was dried over $MgSO_4$ and concentrated. The crude residue was purified via flash chromatography (6:1 hexanes:ether) to afford **95** as a clear oil.

General procedure for functionalization: Aldehyde (1 equiv), ligand **92** (10 mol %), $Pd(OAc)_2$ (10 mol %), $PhI(OAc)_2$ (1.1 equiv) and $AcOH/Ac_2O$ (1:1, 0.12 M) were combined in a vial and stirred at 100 °C for the allotted time. The reaction was filtered through a plug of glass wool and diluted with pentane. The organic layer was washed with H_2O, sat. aq. $NaHCO_3$ and brine, dried over $MgSO_4$ and concentrated. No functionalized aldehydes were observed by ^{1}H NMR.
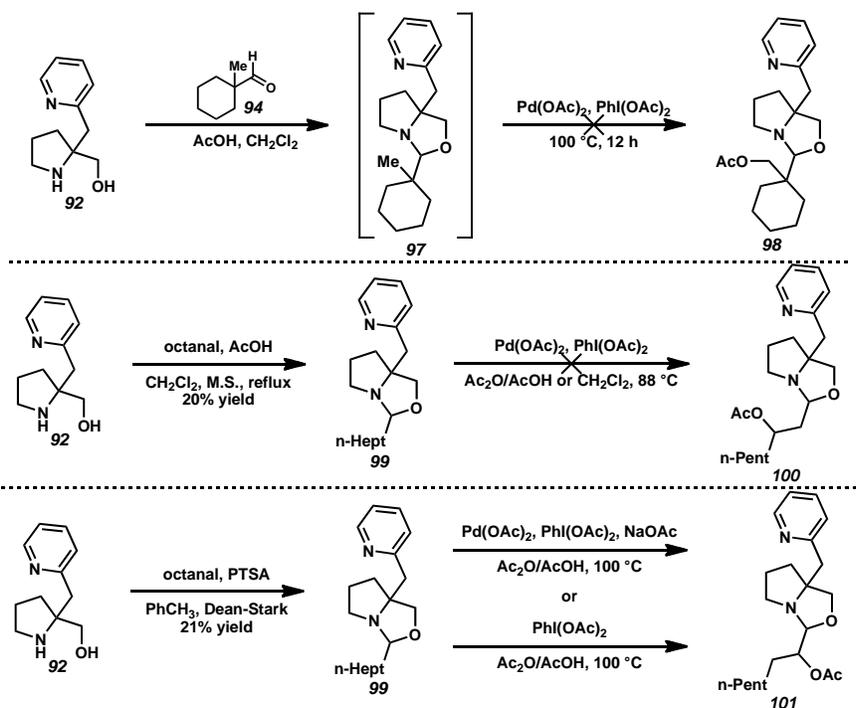
Acetate 98: To **92** (27.7 mg, 0.219 mmol) and **94** (42.1 mg, 0.219 mmol) in CH_2Cl_2 (2.20 mL) was added AcOH (2.0 μL) and heated to 100 °C for 7 h. The reaction was concentrated under reduced pressure, then concentrated from toluene (2 x 2 mL). To $Pd(OAc)_2$ (4.9 mg, 0.0219 mmol) and $PhI(OAc)_2$ (106 mg, 0.329 mmol) was added the crude residue in CH_2Cl_2 (2.20 mL) and AcOH (2.0 μL) and heated to 100 °C for 12 h. The solvent was removed under reduced pressure. No functionalized product was observed by ^1H NMR.

Acetal 99: To a solution of **92** (0.189 g, 0.983 mmol) in CH_2Cl_2 (9.83 mL) and 4 Å molecular sieves was added octanal (0.184 mL, 1.18 mmol) and AcOH (5.6 μL, 0.0983 mmol) and heated to reflux overnight. The reaction was washed with sat. aq. $NaHCO_3$, dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (7:3 to 1:1 hexanes:EtOAc) to afford **99** (59.8 mg, 20% yield, $R_f = 0.20$ in 1:1 hexanes:EtOAc) as a clear oil.

Acetate 100: To Pd(OAc)$_2$ (2.2 mg, 9.89 µmol) and PhI(OAc)$_2$ (35.0 mg, 0.109 mmol) was added **99** (29.9 mg, 0.0989 mmol) and Ac$_2$O/AcOH (1:1, 0.800 mL). The mixture was heated to 88 °C for 13 h. The reaction was filtered through a plug of glass wool and diluted with pentane. The organic layer was washed with H$_2$O, sat. aq. NaHCO$_3$ and brine, dried over MgSO$_4$ and concentrated. No functionalized aldehydes were observed by ^{1}H NMR.

Acetal 99: To a solution of **92** (180 mg, 0.937 mmol) in toluene (4.00 mL) was added PTSA (3.6 mg, 0.0187 mmol) and octanal (0.190 mL, 1.22 mmol). The reaction was heated to reflux with azeotropic removal of water overnight. The reaction was washed with sat. aq. NaHCO$_3$ (2 x 10 mL) and dried over MgSO$_4$ and concentrated to afford **99** (60.6 mg, 21% yield, R$_f$ = 0.20 in 1:1 hexanes:EtOAc) as a clear oil.

Acetate 101: To Pd(OAc)$_2$ (1.1 mg, 4.96 µmol) and PhI(OAc)$_2$ (17.6 mg, 0.0546 mmol) was added **99** (15.0 mg, 0.0496 mmol), NaOAc (2.0 mg, 0.0248 mmol) and Ac$_2$O/AcOH (1:1, 0.400 mL). The mixture was heated to 100 °C for 13 h. The reaction was filtered through a plug of glass wool and diluted with pentane. The organic layer was washed with H$_2$O, sat. aq. NaHCO$_3$ and brine, dried over MgSO$_4$ and concentrated. Acetate **101** was observed by crude ^{1}H NMR.

Acetate 101: To PhI(OAc)$_2$ (17.6 mg, 0.0546 mmol) was added **99** (15.0 mg, 0.0496 mmol) and AcOH/Ac$_2$O (1:1, 0.400 mL) and heated to 100 °C for 4 h. The reaction was filtered through a plug of glass wool and diluted with pentane. The organic layer was washed with H$_2$O, sat. aq. NaHCO$_3$ and brine, dried over MgSO$_4$ and concentrated. Acetate **101** was observed by crude ^{1}H NMR.

aldehyde	yield	funct. conditions	result
i-PrCHO	31%	DCE, 16 h	complex mixture
i-BuCHO	20%	DCE 16 h	complex mixture
CyCHO	31%	Ac$_2$O/AcOH, 12 h	complex mixture
		CH$_3$CN, 17 h	complex mixture
		PhCH$_3$, 17 h	complex mixture
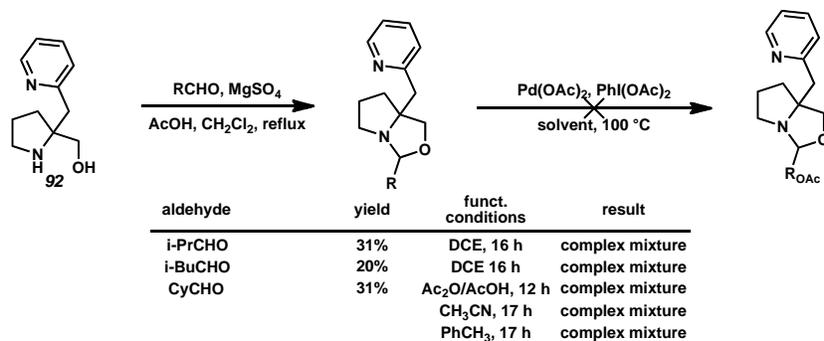
General procedure for forming acetals: To **92** (1 equiv) in CH$_2$Cl$_2$ (0.1 M) was added aldehyde (1.3 equiv) and AcOH (0.1 equiv) and the reaction heated to reflux overnight. The reaction was washed with sat. aq. NaHCO$_3$, dried over Na$_2$SO$_4$ and concentrated.

Acetal 102: 31% yield, R$_f$ = 0.20 in 1:1 hexanes:EtOAc

Acetal 103: 20% yield, R$_f$ = 0.22 in 1:1 hexanes:EtOAc

Acetal 104: 31% yield, R$_f$ = 0.25 in 1:1 hexanes:EtOAc

General procedure for acetoxylation: To acetal (1.0 equiv) was added Pd(OAc)$_2$ (10 mol %), PhI(OAc)$_2$ (1.6 equiv) and solvent (0.12 M) and stirred at 100 °C for the allotted time. The reaction was filtered through a plug of glass wool and diluted with pentane. The organic layer was washed with H$_2$O, sat. aq. NaHCO$_3$ and brine, dried over MgSO$_4$ and concentrated. No acetoxylated product was observed by ^{1}H NMR.
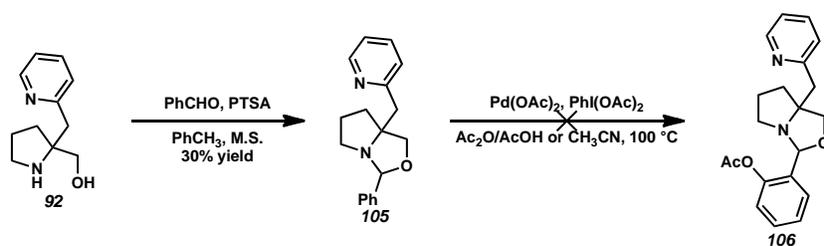
Acetal 105: To a solution of **92** (0.100 g, 0.520 mmol) in toluene (2.10 mL) and 4 Å molecular sieves was added benzaldehyde (68.4 µL, 0.677 mmol) and PTSA (2.0 mg, 0.0104 mmol) and heated to reflux overnight. The reaction was washed with sat. aq. NaHCO$_3$, dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (4:1 to 7:3

hexanes:EtOAc) to afford **105** (19.3 mg, 13% yield, R_f = 0.22 in 1:1 hexanes:EtOAc) as a clear oil.

Acetate 106: To acetal **105** (13.7 mg, 0.0489 mmol) was added Pd(OAc)$_2$ (1.1 mg, 4.90 μmol) PhI(OAc)$_2$ (17.3 mg, 0.0538 mmol) and AcOH/Ac$_2$O or CH$_3$CN (1:1, 0.400 mL) and stirred at 100 °C for 14 h. The reaction was filtered through a plug of glass wool and diluted with pentane. The organic layer was washed with H$_2$O, sat. aq. NaHCO$_3$ and brine, dried over MgSO$_4$ and concentrated. No acetoxylated product was observed by ^{1}H NMR.
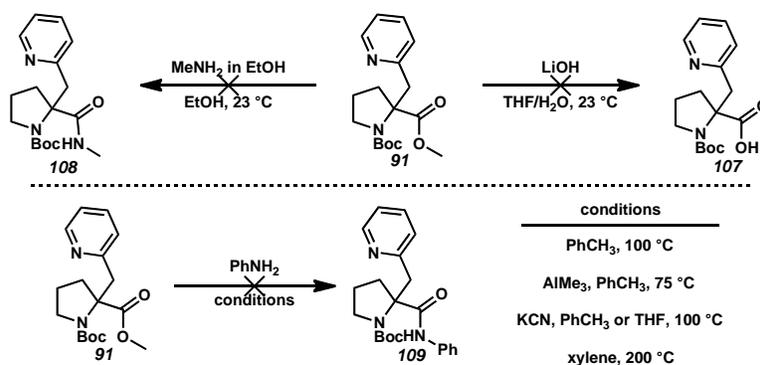
Methyl amide 108: To MeNH$_2$ (0.12 mL, 8.0 M in ethanol) was added **91** (100 mg, 0.312 mmol) in EtOH (1.20 mL) at 23 °C. The reaction was stirred at 23 °C overnight. The solvent was removed under reduced pressure. No product was observed by ^{1}H NMR.

Acid 107: To **91** (1.37 g, 4.28 mmol) in THF (8.56 mL) and H$_2$O (4.30 mL) was added LiOH (103 mg, 4.28 mmol). The reaction was stirred at 23 °C overnight. The solvent was removed under reduced pressure. The solution was acidified to pH 2 with 1 M HCl. The aqueous was extracted with EtOAc (3 x 20 mL), the organic layers dried over Na$_2$SO$_4$ and concentrated. No product was observed by ^{1}H NMR.

Phenyl amide 109: To a solution of aniline (50.6 μL, 0.555 mmol) in toluene (2.00 mL) at 0 °C was added AlMe$_3$ (0.280 mL, 2.0 M in toluene) dropwise over 15 min. Stirred for 1 h at 23 °C. To this solution was added **91** (84.6 mg, 0.264 mmol) in toluene (2.00 mL) at 23 °C. The

reaction was heated to 75 °C for 18 h. The reaction was poured slowly into a mixture of conc.

HCl (1.3 M), ice and EtOAc (1 M). The aqueous layer was extracted with EtOAc (3 x 10 mL),

the organics dried over MgSO$_4$ and concentrated. No product formation by ^{1}H NMR.

Phenyl amide 109: **91** (25.0 mg, 0.0781 mmol), aniline (35.0 µL, 0.390 mmol) and toluene (1.60

mL) were combined in a sealed vial and heated to 100 °C for 24 h. The solvent was removed

under reduced pressure. No product formation by ^{1}H NMR.

Phenyl amide 109: To **91** (30.0 mg, 0.0937 mmol) and KCN (0.6 mg, 3.37 µmol) in toluene or

THF (0.900 mL) was added aniline (42.0 µL, 0.468 mmol) at 23 °C. The mixture was heated to

100 °C overnight. The solvent was removed under reduced pressure. No product formation by

^{1}H NMR.

Phenyl amide 109: **91** (25.0 mg, 0.0781 mmol) aniline (35.0 µL, 0.390 mmol) and xylene (0.800

mL) were combined in a 2-dram vial and heated to 250 °C overnight. The solvent was removed

under reduced pressure. No product formation by ^{1}H NMR.
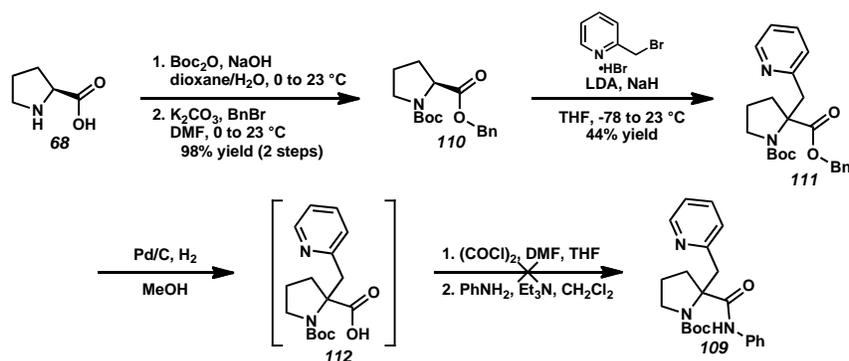
Ester 110: To **119** (2.00 g, 9.29 mmol) in DMF (18.6 mL) was added K$_2$CO$_3$ (1.41 g, 10.2

mmol) at 0 °C. The suspension was stirred 10 min at 0 °C, then BnBr (2.20 mL, 13.9 mmol) was

added at 0 °C and stirred for an additional 30 min at 0 °C, then 3 h at room temperature. The

reaction mixture was filtered, then partitioned between H$_2$O (50 mL) and EtOAc (50 mL). The

organic layer was washed with brine (2 x 30 mL), dried over Na$_2$SO$_4$ and concentrated. The

crude mixture was purified via flash chromatography (1:3 EtOAc:hexanes) to give **110** (2.83 g, 99% yield, R_f = 0.25 in 4:1 EtOAc:hexanes) as a clear oil.

Pyridine 111: To a solution of freshly distilled diisopropylamine (246 µL, 1.75 mmol) in THF (2.00 mL) at -78 ˚C was added *n*-BuLi (0.730 mL, 2.3 M in hexanes, 1.68 mmol). The solution was stirred for 10 min at -78 ˚C, at which time **110** (512 mg, 1.68 mmol) in THF (2.20 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 ˚C. To NaH (168 mg, 4.19 mmol) (which was first washed with hexanes (2 x 1.0 mL)), in DMF (3.20 mL) at 0 ˚C was added 2-bromomethyl pyridine hydrobromide (354 mg, 1.39 mmol). The suspension was stirred at 0 ˚C for 30 min at which time it was added quantitatively, with additional DMF (1.0 mL), to the enolate solution at -78 ˚C. The suspension was warmed to 23 ˚C and stirred overnight. The reaction was quenched with H_2O (10 mL) slowly at first at 23 ˚C. The aqueous layer was extracted with EtOAc (3 x 15 mL). The combined organic layers were washed with brine (2 x 20 mL), dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (1:3 to 3:7 EtOAc/Hexanes eluent) to afford **111** (0.227 g, 41% yield, R_f = 0.35 in 1:1 EtOAc:hexanes) as a yellow oil.

Acid 112: To **111** (325 mg, 0.820 mmol) in MeOH (3.30 mL) was added Pd/C (33.0 mg). The flask was flushed with H_2, then an H_2 balloon was fitted until the reaction was complete by TLC. The reaction mixture was filtered through celite with MeOH. The organics were concentrated to afford **112** (0.296 g) by crude ^{1}H NMR.

Amide 109: To **112** (240 mg, 0.793 mmol) in THF (4.00 mL) was added $(COCl)_2$ (0.346 mL, 3.96 mmol) and DMF (4 drops) at 0 °C. After 5 min the reaction was warmed to 23 °C and stirred for 1 h. The solvent was removed and concentrated from benzene (2 x 5 mL). To aniline (0.289 mL, 3.17 mmol) and triethylamine (0.334 mL, 2.38 mmol) in CH_2Cl_2 (3.40 mL) at 0 °C

was added the acid chloride in CH_2Cl_2 (4.00 mL). The reaction, black in color, was stirred at 23 °C for 1h. The reaction was washed with $KHSO_4$ (10 mL, 1 M), sat. aq. $NaHCO_3$ (10 mL) and brine (10 mL) sequentially. The organic layer was dried over Na_2SO_4 and concentrated. No product was observed via crude ^1H NMR.
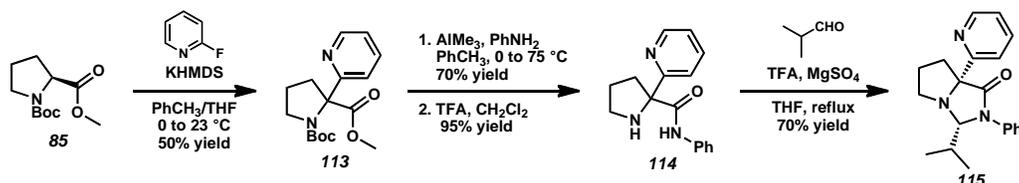
Pyridine 113: To a solution of **85** (1.58 g, 6.90 mmol) and 2-fluoropryidine (0.590 mL, 6.90 mmol) in $PhCH_3$ (23.0 mL) at 0 °C was added KHMDS (13.8 mL, 6.90 mmol, 0.5 M in THF) dropwise over 1 h. The reaction was stirred for 1 h at 0 °C, then 23 °C for 24 h. The reaction was filtered through a plug of silica gel and concentrated. The crude residue was purified via flash chromatography (4:1 to 3:1 hexanes:EtOAc) to afford **113** (1.03g, 48% yield, R_f = 0.38 in 1:1 hexanes:EtOAc) as a yellow oil.

To a solution of aniline (644 µL, 7.07 mmol) in toluene (3.40 mL) at 0 °C was added $AlMe_3$ (3.50 mL, 2.0 M in toluene) dropwise over 15 min. Stirred for 1 h at 23 °C. To this solution was added **113** (1.03 g, 3.37 mmol) in toluene (6.70 mL) at 23 °C. The reaction was heated to 75 °C for 18 h. The reaction was poured slowly into a mixture of conc. HCl (1.3 M), ice and EtOAc (1 M). The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over $MgSO_4$ and concentrated. The crude residue was purified via flash chromatography (7:3 to 3:2 hexanes:EtOAc) to afford amide (0.857 g, 69% yield, R_f = 0.60 in 1:1 hexanes:EtOAc) as thick yellow oil.

Amino amide 114: To amide (0.857 g, 2.33 mmol) in CH_2Cl_2 (4.70 mL) at 23 °C was added TFA (3.60 mL, 46.6 mmol). The reaction was stirred for 1 h at 23 °C. The solvent was removed

under reduced pressure. The crude residue was neutralized with sat. aq. $NaHCO_3$. The aqueous

layer was extracted with EtOAc (3 x 15 mL), dried over Na_2SO_4 and concentrated to afford **114**

(623 mg, 99% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as a pale yellow solid.

Aminal 115: To **114** (100 mg, 0.374 mmol), $MgSO_4$ (67.5 mg, 0.561 mmol) and

isobutyraldehyde (44.4 µL, 0.486 mmol) in THF (4.00 mL) was added TFA (5.8 µL, 0.0748

mmol) at 23 °C. The reaction was heated to reflux overnight. The solvent was removed under

reduced pressure. The residue was taken up in sat. aq. $NaHCO_3$, extracted with CH_2Cl_2 (3 x 10

mL). The organic layers were dried over Na_2SO_4 and concentrated. The crude residue was

purified via flash chromatography (7:3 to 3:2 hexanes:EtOAc with 1% Et_3N) to afford **115** (103

mg, 86% yield, $R_f = 0.31$ in 1:1 hexanes:EtOAc) as a light beige solid.

temp °C	115 : 116
110	1.0 : 0
70	5.9 : 1
75	6.0 : 1
80	2.8 : 1
85	2.7 : 1 (10% yield)

solvent	115 : 116	solvent (1:1)	115 : 116	solvent (1:1)	115 : 116
DCE	1.0 : 0	DCE:Ac$_2$O	1.0 : 0	DCE:AcOH	1.0 : 0
PhCH$_3$	1.0 : 0	PhCH$_3$:Ac$_2$O	1.0 : 0	PhCH$_3$:AcOH	1.0 : 0
PhH	1.0 : 0	PhH:Ac$_2$O	1.0 : 0	CH$_3$CN:AcOH	6.0 : 1
CH$_3$CN	1.0 : 0	CH$_3$CN:Ac$_2$O	1.0 : 0		
CH$_2$Cl$_2$	1.0 : 0	CH$_2$Cl$_2$:Ac$_2$O	1.0 : 0		
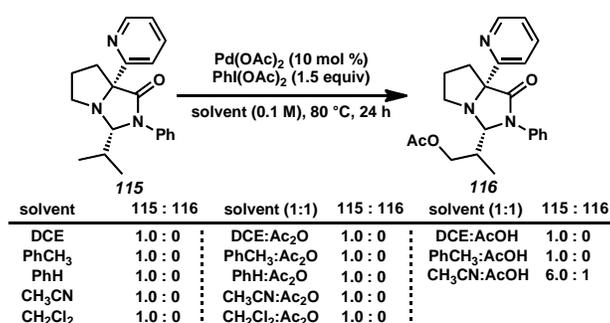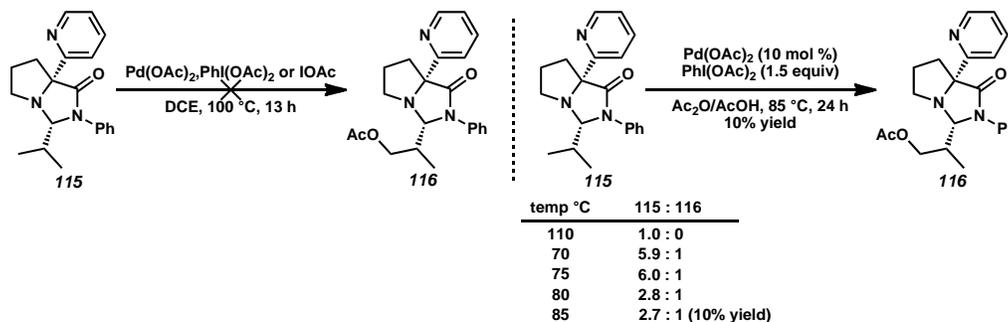
General procedure for acetoxylation: $Pd(OAc)_2$ (10 mol %), $PhI(OAc)_2$ (1.5 equiv) and **115** (1

equiv) were combined with solvent (0.1 M) in a 2-dram vial and heated to the noted temperature

for the allotted time. The reaction was neutralized with sat. aq. $NaHCO_3$. The aqueous was

111

extracted with CH_2Cl_2 (3x), the organics dried over Na_2SO_4 and concentrated. Ratios were assigned by crude ^{1}H NMR.
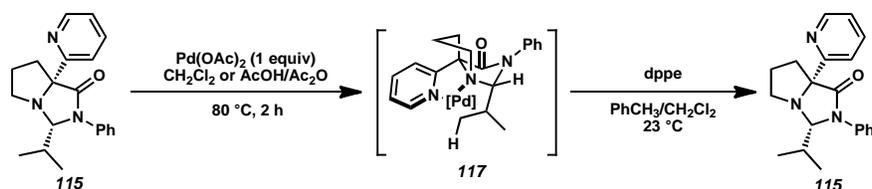
To **115** (10.0 mg, 0.0311 mmol) and Pd(OAc)$_2$ (7.0 mg, 0.0311 mmol) was added $Ac_2O/AcOH$ (1:1, 0.300 mL) in a 2-dram vial, and the reaction was heated to 80 °C for 24 h. The solvent was removed by azeotropic evaporation from heptanes (3 x 2 mL). The residue was treated with dppe (12.4 mg, 0.0311 mmol) and CH_2Cl_2/toluene (1:1, 0.311 mL) at 23 °C for 24 h. The solvent was removed under reduced pressure. ^{1}H NMR revealed the product to be **115**.
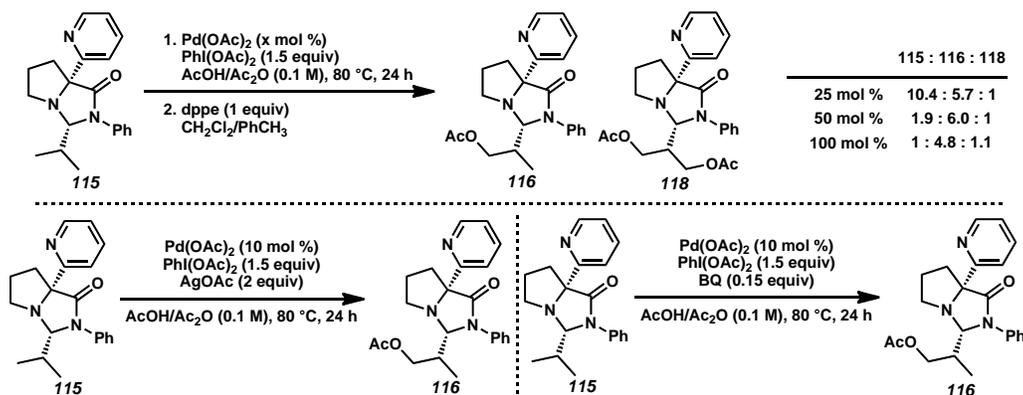
General procedure for acetoxylation: **115** (1 equiv), Pd(OAc)$_2$ (25-100 mol %) and PhI(OAc)$_2$ (1.5 equiv) were combined with $AcOH/Ac_2O$ (1:1, 0.1 M) and heated to 80 °C for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 5 mL). The residue was treated with 1,2-bis(diphenylphosphino)ethane (1 equiv) in $PhCH_3/CH_2Cl_2$ (1:1, 0.1 M) and stirred overnight at 23 °C. The solvent was removed by rotary evaporation, and the

crude residue was purified by flash chromatography (4:1 → 3:1 hexanes/acetone eluent) to afford acetate **116** and diacetate **118**.

Pyridine **116** (200 mg, 0.622 mmol), Pd(OAc)$_2$ (69.8 mg, 0.311 mmol), and PhI(OAc)$_2$ (351 mg, 1.10 mmol) were dissolved in AcOH (3.10 mL) and Ac$_2$O (3.10 mL) in a round-bottomed flask. The flask was capped and heated to 85 °C for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 15 mL). The residue was treated with 1,2-bis(diphenylphosphino)ethane (249 mg, 0.622 mmol) in PhCH$_3$/CH$_2$Cl$_2$ (1:1, 6.20 mL) and stirred overnight at 23 °C. The solvent was removed by rotary evaporation, and the crude residue was purified by flash chromatography (4:1 → 3:1 hexanes/acetone eluent) to afford acetate **120** (156 mg, 66% yield, R$_f$ = 0.45 in 1:1 hexanes/acetone) as a light yellow oil and diacetate **121** (36.0 mg, 13% yield, R$_f$ = 0.43 in 1:1 hexanes/acetone) as a light yellow oil.

Acetate 120: ^{1}H NMR (300 MHz, CDCl$_3$) δ 8.68 (d, *J* = 4.6 Hz, 1H), 7.79 (d, *J* = 7.9 Hz, 1H), 7.68 (t, *J* = 7.7 Hz, 1H), 7.46-7.37 (comp m, 4H), 7.24-7.18 (comp m, 2H), 4.77 (d, *J* = 3.4 Hz, 1H), 3.80 (dd, *J* = 6.5, 3.1 Hz, 2H), 3.53 (dt, *J* = 10.7, 6.4 Hz, 1H), 3.05 (dt, *J* = 11.1, 5.7 Hz, 1H), 2.52 (dt, *J* = 13.3, 6.8 Hz, 1H), 2.39 (dt, *J* = 13.3, 6.7 Hz, 1H), 2.14-1.83 (comp m, 3H), 1.82 (s, 3H), 0.92 (d, *J* = 7.0 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 171.9, 170.6, 149.3, 136.7, 136.5, 129.1, 126.2, 124.1, 122.2, 120.6, 98.4, 84.6, 77.6, 65.5, 59.4, 38.9, 36.2, 25.6, 20.7, 14.0; IR (film) 3061, 2967, 1735, 1701, 1497, 1408, 1237 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{22}$H$_{26}$N$_3$O$_3$]$^+$: 380.1969, found 380.1970.

Diacetate 121: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.67 (d, *J* = 4.5 Hz, 1H), 7.75-7.67 (comp m, 2H), 7.47-7.39 (comp m, 4H), 7.23-7.19 (comp m, 2H), 5.02 (d, *J* = 3.9 Hz, 1H), 4.12 (dd, *J* = 11.3, 7.3 Hz, 1H), 4.04 (dd, *J* = 11.3, 5.5 Hz, 1H), 3.95 (dd, *J* = 11.3, 5.5 Hz, 1H), 3.85 (dd, *J* = 11.3, 7.3 Hz, 1H), 3.53 (dt, *J* = 10.6, 6.5 Hz, 1H), 3.04 (dt, *J* = 11.1, 5.8 Hz, 1H), 2.56 (dt, *J* =

13.3, 6.8 Hz, 1H), 2.40 (dt, J = 13.3, 6.8 Hz, 1H), 2.30 (dddd, J = 7.2, 5.5, 3.9, 1.7 Hz, 1H), 2.03 (s, 3H), 2.02-1.97 (m, 1H), 1.89-1.85 (m, 1H), 1.82 (s, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.0, 170.4, 161.6, 149.4, 136.5, 136.2, 129.1, 126.4, 123.7, 122.3, 120.5, 80.5, 77.8, 62.0, 61.4, 59.1, 41.0, 38.9, 25.7, 20.8, 20.7; IR (film), 2961, 1738, 1703, 1588, 1226, 1039, 753, 697 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{24}$H$_{28}$N$_3$O$_5$]$^+$: 438.2023, found 438.2023.
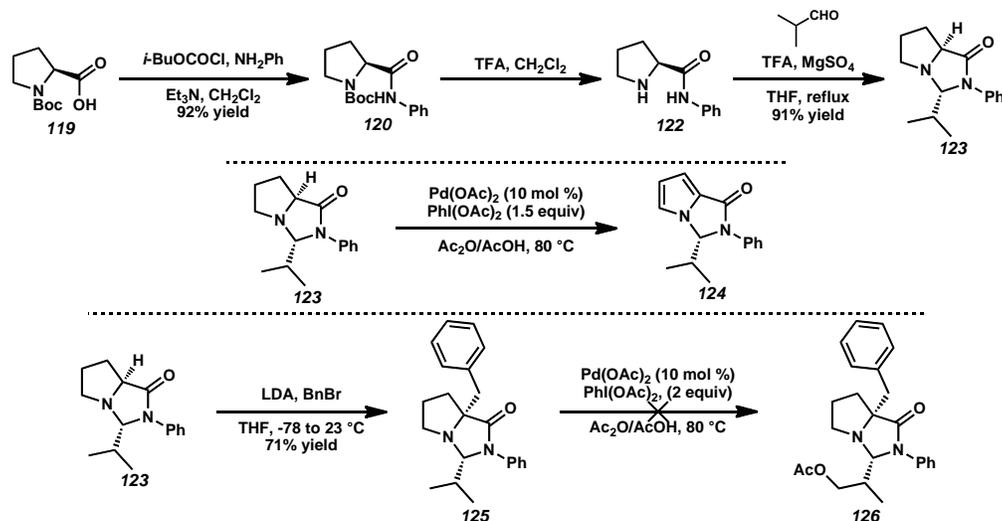
Amide 120: To a solution of **119** (2.50 g, 11.6 mmol) in CH$_2$Cl$_2$ (38.7 mL) at 0 °C was added isobutyl chloroformate (1.67 mL, 12.8 mmol) and triethylamine (1.80 mL, 12.8 mmol). After stirring for 20 minutes at 0 °C, aniline (1.16 mL, 12.8 mmol) was added, and the reaction was warmed to 23 °C and stirred overnight. The reaction was washed sequentially with aq. KHSO$_4$ (1 M, 50 mL), sat. aq. NaHCO$_3$ (50 mL) and brine (50 mL). The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford a pale brown solid. The crude solid was suspended in hexanes (15 mL), cooled to 0 °C, and filtered to afford amide **120** (3.32 g, 98% yield, R$_f$ = 0.52 in 1:1 hexanes/EtOAc) as a light brown solid, which was sufficiently pure to be taken on to the next step.

Amino amide 122: To a solution of amide **120** (3.32 g, 11.4 mmol) in CH$_2$Cl$_2$ (22.8 mL) at 23 °C was added TFA (17.6 mL, 228 mmol). The solution was stirred at 23 °C for 1 h, and the

solvent was removed under reduced pressure. The residue was taken up in CH_2Cl_2 (20 mL) and neutralized with solid Na_2CO_3 until pH ~9. Water (10 mL) was added and the aqueous layer was extracted with CH_2Cl_2 (3 x 50 mL). The combined organic layers were dried over Na_2SO_4 and concentrated to afford amino amide **122** (2.20 g, 99% yield, R_f = 0.05 in 1:1 hexanes/EtOAc) as a light brown solid, which was sufficiently pure to be taken on to the next step.

Aminal 123: To a solution of amino amide **122** (1.50 g, 7.88 mmol) in $PhCH_3$ (26.3 mL) at 23 ˚C was added isobutyraldehyde (1.10 mL, 11.8 mmol), TsOH·H_2O (75.0 mg, 0.394 mmol), and $MgSO_4$ (1.40 g, 11.8 mmol). The suspension was heated to reflux and stirred overnight. Upon cooling to 23 ˚C, the solution was quenched with sat. aq. $NaHCO_3$ (20 mL), and the mixture was extracted with EtOAc (3 x 30 mL). The combined organic layers were dried over Na_2SO_4 and concentrated to afford aminal **123** (1.75 g, 91% yield, R_f = 0.48 in 1:1 hexanes/EtOAc) as a light yellow solid, which was sufficiently pure to be taken on to the next step.

Aminal 123: ^{1}H NMR (400 MHz, $CDCl_3$) δ 7.44 (d, J = 7.6 Hz, 2H), 7.38 (t, J = 7.7 Hz, 2H), 7.19 (t, J = 7.1 Hz, 1H), 4.63 (app. s, 1H), 3.95 (dd, J = 8.0, 5.2 Hz, 1H), 3.31-3.26 (m, 1H), 2.77 (app. q, J = 7.8 Hz, 1H), 2.23-2.15 (m, 1H), 2.06-1.97 (m, 1H), 1.90-1.86 (m, 1H), 1.84-1.78 (comp m, 2H), 0.97 (d, J = 6.7 Hz, 3H), 0.81 (d, J = 6.6 Hz, 3H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.0, 136.8, 129.1, 125.7, 123.4, 87.9, 66.4, 58.5, 31.3, 28.9, 25.1, 18.4, 14.6; IR (film) 2963, 3053, 1683, 1504, 1411, 758, 698 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [$C_{15}H_{20}N_2ONa$]$^+$: 267.1468, found 267.1468.

Pyrrole 124: **123** (10.0 mg, 0.0409 mmol), Pd(OAc)$_2$ (0.9 mg, 4.09 μmol) and PhI(OAc)$_2$ (19.8 mg, 0.0614 mmol) in AcOH/Ac_2O (1:1, 0.400 mL) were heatd to 80 °C overnight. The solvent was removed by azeotropic removal with heptanes (3 x 5 mL). ^{1}H NMR revealed the pyrrole (**124**).

Aminal 125: To a solution of freshly distilled diisopropylamine (167 μL, 1.19 mmol) in THF (5.00 mL) at -78 °C was added *n*-BuLi (0.460 mL, 2.5 M in hexanes, 1.15 mmol) dropwise. The solution was stirred for 10 min at -78 °C, at which time a solution of aminal **123** (200 mg, 0.818 mmol) in THF (3.20 mL) was added, and the resulting mixture was stirred for an additional 30 min at -78 °C. Benzyl bromide (256 μL, 1.64 mmol) was added at -78 °C, and the reaction was warmed to 23 °C and stirred overnight. The reaction was quenched with water (10 mL), and the mixture was extracted with EtOAc (3 x 10 mL). The combined organic phases were washed with brine (10 mL), dried over Na_2SO_4 and concentrated. The crude product was purified by flash chromatography (9:1 to 4:1 hexanes/EtOAc eluent) to afford aminal **125** (195 mg, 71% yield, R_f = 0.74 in 4:1 hexanes/EtOAc) as a white amorphous solid.

Aminal 125: ^{1}H NMR (400 MHz, $CDCl_3$) δ 7.43-7.36 (comp m, 6H), 7.30-7.20 (comp m, 4H), 4.48 (d, *J* = 2.8 Hz, 1H), 3.06 (ABq, *J* = 13.6 Hz, Δν = 76.8 Hz, 2H), 2.73-2.72 (comp m, 2H), 2.08-2.02 (m, 1H), 1.89-1.78 (comp m, 2H), 1.56-1.49 (m, 1H), 1.34-1.32 (m, 1H), 0.93 (d, *J* = 1.1 Hz, 3H), 0.72 (d, *J* = 3.9 Hz, 3H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.7, 137.6, 136.5, 131.0, 129.0, 127.9, 126.4, 126.1, 124.4, 86.9, 75.0, 58.8, 42.8, 34.5, 30.8, 24.8, 18.5, 14.8; IR (film) 3029, 2964, 1700, 1498, 1409, 698 cm[-1]; HRMS (ESI[+]) *m/z* calc'd for (M+H)[+] $[C_{22}H_{27}N_2O]^+$: 335.2118, found 335.2120.

Acetate 126: Aminal **125** (20.0 mg, 0.0598 mmol), $Pd(OAc)_2$ (1.3 mg, 5.98 μmol), and $PhI(OAc)_2$ (28.9 mg, 0.0897 mmol) were dissolved in $AcOH/Ac_2O$ (1:1, 0.600 mL) in a scintillation vial. The vial was heated to 80 °C for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptane (2 x 10 mL). Water (10 mL) was added to the residue, and the mixture was neutralized with solid Na_2CO_3 until pH ~9. The mixture was extracted with

CH$_2$Cl$_2$ (3 x 10 mL), and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. Only starting material was observed by ^{1}H NMR.
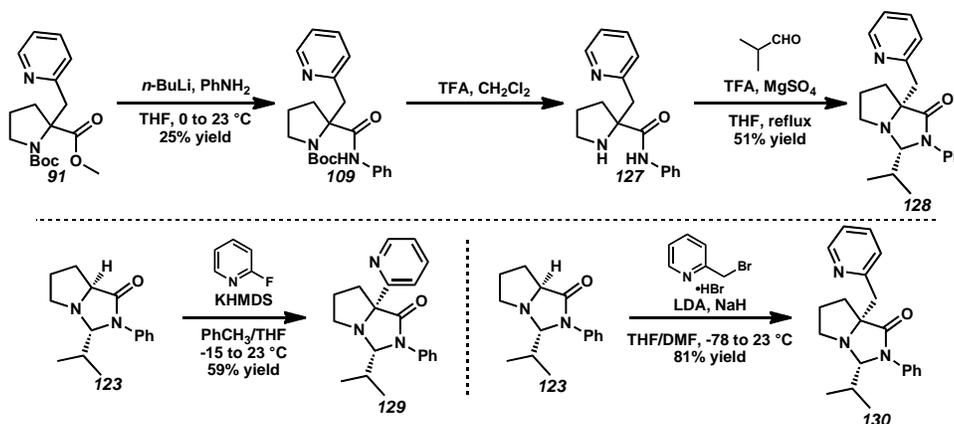
Amide 109: To aniline (210 µL, 2.35 mmol) in THF (15.4 mL) at -78 °C was added *n*-BuLi (0.85 mL, 2.5M in hexanes, 2.14 mmol). Stirred 10 min at -78 °C, then warmed to 0 °C. To this solution was added quickly **91** (684 mg, 2.14 mmol) in THF (6.00 mL) at 0 °C. The reaction was stirred 10 min at 0 °C, then quenched with H$_2$O (25 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organics were washed with brine (40 mL), dried over MgSO$_4$ and concentrated. The crude residue was purified via flash chromatography (1:4 to 7:13 EtOAc:hexanes) to afford **109** (286 mg, 35% yield, R$_f$ = 0.33 in 1:1 EtOAc:hexanes) as a red oil.

Amino amide 127: To **109** (286 mg, 0.75 mmol) in CH$_2$Cl$_2$ (1.50 mL) was added TFA (1.20 mL, 15.0 mmol) at room temperature. Stirred for 1 h at room temperature, then the solvent was removed. The residue was quenched with sat. aq. NaHCO$_3$ solution until pH 9, then the aqueous was extracted with CH$_2$Cl$_2$ (3 x 15 mL). Organics dried over Na$_2$SO$_4$ and concentrated to afford **127** (162 mg, 77% yield, R$_f$ = 0.0 in 1:1 hexanes:EtOAc) as a brown oil with no further purification necessary.

Aminal 128: To **127** (108 mg, 0.384 mmol) in THF (3.80 mL) was added isobutyraldehyde (53.0 µL, 0.576 mmol), TFA (6.0 µL, 0.0768 mmol) and MgSO$_4$ (69.3 mg, 0.576 mmol) at 23 °C. The

suspension was refluxed overnight. Upon cooling, the reaction was quenched with sat. aq.

NaHCO$_3$ solution (10 mL). The aqueous was extracted with EtOAc (3 x 5 mL), dried over

Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (3:1 to 7:3

hexanes:EtOAc) to give **128** (65.6 mg, 51% yield,R$_f$ = 0.74 in 40:1 EtOAc/MeOH) as a light

brown solid.

Pyridine 129: To a solution of aminal **123** (500 mg, 2.04 mmol), 2-fluoropyridine (176 μL, 2.04

mmol) in PhCH$_3$ (10.2 mL) at -15 °C was added KHMDS (408 mg, 2.04 mmol) in THF (4.10

mL) slowly over 1 h. Upon completion of addition, the reaction was allowed to warm to 23 °C

and stirred overnight. The reaction was filtered over a pad of silica (5 x 5 cm, 100 mL EtOAC

eluent) and concentrated. The crude product was purified by flash chromatography (3:1 → 1:1

hexanes/EtOAc eluent) to afford pyridine **129** (385 mg, 59% yield, 148 mg recovered **123**: 83%

yield, corrected, R$_f$ = 0.31 in 1:1 hexanes/EtOAc) as a light beige solid.

Pyridine 129: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.69 (d, *J* = 3.7 Hz, 1H), 7.83 (d, *J* = 7.9 Hz, 1H),

7.64 (td, *J* = 7.7, 1.5 Hz, 1H), 7.41-7.34 (comp m, 4H), 7.21-7.13 (comp m, 2H), 4.64 (d, *J* = 3.1

Hz, 1H), 3.44 (dt, *J* = 10.9, 6.7 Hz, 1H), 3.03 (dt, *J* = 11.2, 5.7 Hz, 1H), 2.48 (app. t, *J* = 7.0 Hz,

2H), 1.93-1.81 (comp m, 3H), 0.93 (d, *J* = 6.9 Hz, 3H), 0.55 (d, *J* = 6.6 Hz, 3H); ^{13}C NMR (100

MHz, CDCl$_3$) δ 171.8, 162.2, 149.4, 136.6, 136.0, 129.0, 126.1, 124.3, 121.8, 120.8, 86.4, 77.7,

59.1, 38.8, 31.1, 25.6, 18.4, 15.1; IR (film) 2962, 1701, 1587, 1497, 1407, 752 cm^{-1}; HRMS

(ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{20}$H$_{24}$N$_3$O]$^+$: 322.1914, found 322.1914.

Pyridine 130: To a solution of freshly distilled diisopropylamine (177 μL, 1.30 mmol) in THF

(3.00 mL) at -78 °C was added *n*-BuLi (0.480 mL, 2.5 M in hexanes, 1.20 mmol) dropwise. The

solution was stirred for 10 min at -78 °C, at which time a solution of aminal **123** (295 mg, 1.20

mmol) in THF (3.10 mL) was added, and the resulting solution was stirred for an additional 30

min at -78 °C. To a suspension of NaH (121 mg, 60% dispersion in mineral oil, 3.00 mmol, washed 2 x 1.0 mL with hexanes) in DMF (5.00 mL) at 0 °C was added 2-(bromomethyl)pyridine hydrobromide (255 mg, 1.00 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.10 mL DMF). The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched by slow addition of H_2O (20 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 20 mL). The combined organic layers were washed with brine (2 x 35 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified by flash chromatography (7:3 → 1:1 hexanes/EtOAc eluent) to afford pyridine **130** (272 mg, 81% yield, R_f = 0.74 in 40:1 EtOAc/MeOH) as a beige solid.

Pyridine 130: ^{1}H NMR (400 MHz, $CDCl_3$) δ 8.56 (d, J = 3.8 Hz, 1H), 7.62 (t, J = 6.8 Hz, 1H), 7.52 (d, J = 7.4 Hz, 1H), 7.40-7.34 (comp m, 4H), 7.23-7.19 (m, 1H), 7.15 (t, J = 5.6 Hz, 1H), 4.47 (d, J = 2.8 Hz, 1H), 3.27 (ABq, J = 13.2 Hz, Δν =18.5 Hz, 2H), 2.76 (app. s, 2H), 2.19-2.06 (comp m, 2H), 1.78-1.74 (m, 1H), 1.64-1.57 (m, 1H), 1.49-1.45 (m, 1H), 0.89 (d, J = 6.9 Hz, 3H), 0.58 (d, J = 6.2 Hz, 3H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.1, 158.1, 148.8, 136.4, 129.0, 126.1, 125.3, 124.4, 121.5, 98.3, 86.4, 74.9, 58.4, 45.4, 34.9, 30.6, 24.7, 18.4, 14.3; IR (film) 2969, 2870, 1676, 1600, 1524, 1443, 755 cm[-1]; HRMS (ESI[+]) *m/z* calc'd for (M+Na)[+] $[C_{21}H_{25}N_3ONa]^+$: 358.1890, found 358.1894.

conditions	130 : 131 : 132
Pd(OAc)₂ (10 mol %) PhI(OAc)₂ (1.0 equiv)	2.7 : 1.5 : 1.0
Pd(OAc)₂ (10 mol %) PhI(OAc)₂ (2.0 equiv) 80 °C 24 h	5.8 : 1.0 : 0
Pd(OAc)₂ (10 mol %) PhI(OAc)₂ (1.0 equiv) 40 °C 24 h	0 : 1.0 : 1.9
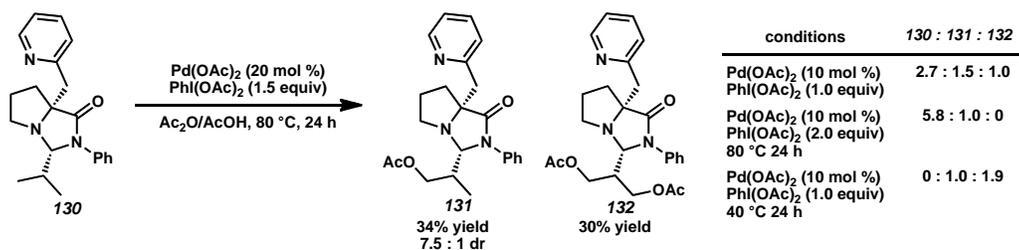
General procedure for acetoxylation: **130** (1 equiv), $Pd(OAc)_2$ (10-20 mol %) and $PhI(OAc)_2$ (1.0-1.5 equiv) in $AcOH/Ac_2O$ was heated to 40-80 °C for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous solution was extracted with CH_2Cl_2 (3 x 15 mL), and the combined organic layers were dried over Na_2SO_4 and concentrated. The crude residue was purified by flash chromatography (3:1 → 1:1 hexanes/acetone eluent) to afford acetate **131** (R_f = 0.48 in 1:1 hexanes/acetone) as a light yellow amorphous solid and diacetate **132** (R_f = 0.35 in 1:1 hexanes/acetone) as a light yellow amorphous solid. Ratios were determined by crude ^{1}H NMR.

time (h)	130 : 131 : 132
8	1 : 4.2 : 2.8
10	1 : 3.4 : 2.5
16	1 : 3.4 : 4.0
18	1 : 2.3 : 2.3
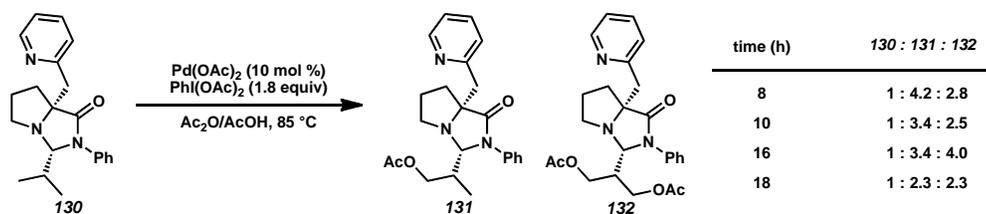
General procedure for acetoxylation: **130** (1 equiv), $Pd(OAc)_2$ (10 mol %) and $PhI(OAc)_2$ (1.8 equiv) in $AcOH/Ac_2O$ was heated to 85 °C for the allotted time. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous solution was extracted with CH_2Cl_2 (3 x 15 mL), and the combined organic layers were dried over Na_2SO_4 and concentrated. Ratios were determined by crude ^{1}H NMR.

time (h)	temp (°C)	131 : 132 (%)	dr
24	55	41 : 13	10 : 1
13	70	48 : 19	5.9 : 1
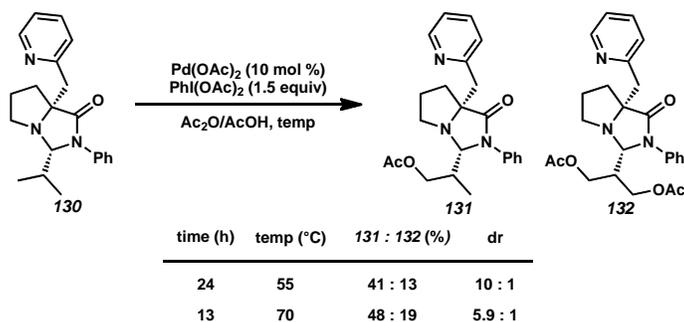
General procedure for acetoxylation: **130** (1 equiv), Pd(OAc)$_2$ (10 mol %) and PhI(OAc)$_2$ (1.5 equiv) in AcOH/Ac$_2$O was heated to 55-70 °C for 13-24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na$_2$CO$_3$ until pH ~9. The aqueous solution was extracted with CH$_2$Cl$_2$ (3 x 15 mL), and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by flash chromatography (3:1 to 1:1 hexanes/acetone eluent) to afford acetate **131** (R$_f$ = 0.48 in 1:1 hexanes/acetone) as a light yellow amorphous solid and diacetate **132** (R$_f$ = 0.35 in 1:1 hexanes/acetone) as a light yellow amorphous solid.

Acetate 131: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.56 (dd, J = 4.9, 0.9 Hz, 1H), 7.62 (td, J = 7.7, 1.8 Hz, 1H), 7.45-7.34 (comp m, 5H), 7.22 (tt, J = 6.8, 1.8 Hz, 1H), 7.15 (ddd, J = 7.4, 5.0, 1.0 Hz, 1H), 4.54 (d, J = 3.5 Hz, 1H), 3.83 (d, J = 6.3 Hz, 2H), 3.23 (ABq, J = 13.2 Hz, Δν = 35.9 Hz, 2H), 2.92-2.85 (m, 1H), 2.83-2.77 (m, 1H), 2.23-2.07 (comp m, 2H), 1.93 (s, 3H), 1.85 (qd, J = 6.6, 3.5 Hz, 1H), 1.68-1.53 (comp m, 2H), 0.83 (d, J = 7.0 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 174.0, 170.7, 157.9, 148.9, 136.5, 135.9, 129.0, 126.3, 125.1, 124.3, 124.0, 121.7, 84.4, 74.9, 65.0, 58.2, 45.6, 36.3, 35.3, 24.7, 20.9, 14.0; IR (film) 3061, 2967, 2881, 1736, 1699, 1594, 1499, 1236, 754 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{23}$H$_{28}$N$_3$O$_3$]$^+$: 394.2125, found 394.2126.

Diacetate 132: ^{1}H NMR (300 MHz, CDCl$_3$) δ 8.55 (d, J = 4.1 Hz, 1H), 7.62 (td, J = 7.6, 1.5 Hz, 1H), 7.43-7.35 (comp m, 5H), 7.23-7.21 (m, 1H), 7.15 (dd, J = 6.8, 5.4 Hz, 1H), 4.78 (d, J = 3.6 Hz, 1H), 3.94 (dd, J = 13.6, 6.1 Hz, 4H), 3.23 (ABq, J = 13.2 Hz, Δν = 33.8 Hz, 2H), 2.93-2.84 (m, 1H), 2.80-2.73 (m, 1H), 2.22-2.08 (comp m, 2H), 1.66-1.57 (comp m, 2H), 2.02 (s, 3H), 1.92 (s, 3H), 1.99-1.92 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.9, 170.5, 170.4, 157.6, 149.0, 136.1, 135.9, 129.1, 126.6, 125.1, 124.0, 121.8, 80.3, 75.0, 61.9, 60.8, 57.9, 45.5, 41.0, 35.6,

24.7, 20.79, 20.76; IR (film) 2965, 1739, 1702, 1593, 1409, 1226 cm^{-1}; HRMS (ESI$^+$) m/z calc'd

for (M+Na)$^+$ [C$_{25}$H$_{29}$N$_3$O$_5$Na]$^+$: 474.1999, 474.1997.

alcohol	Ester (R)-(+)-MTPA	Ester (S)(–)-MTPA
140	141 (δ)^{19}F = -72.10 ppm	142 (δ)^{19}F = -72.12 ppm
134	135 (δ)^{19}F = -72.081 ppm	136 (δ)^{19}F = -72.079 ppm
135	138 (δ)^{19}F = -72.18 ppm	139 (δ)^{19}F = -72.16 ppm
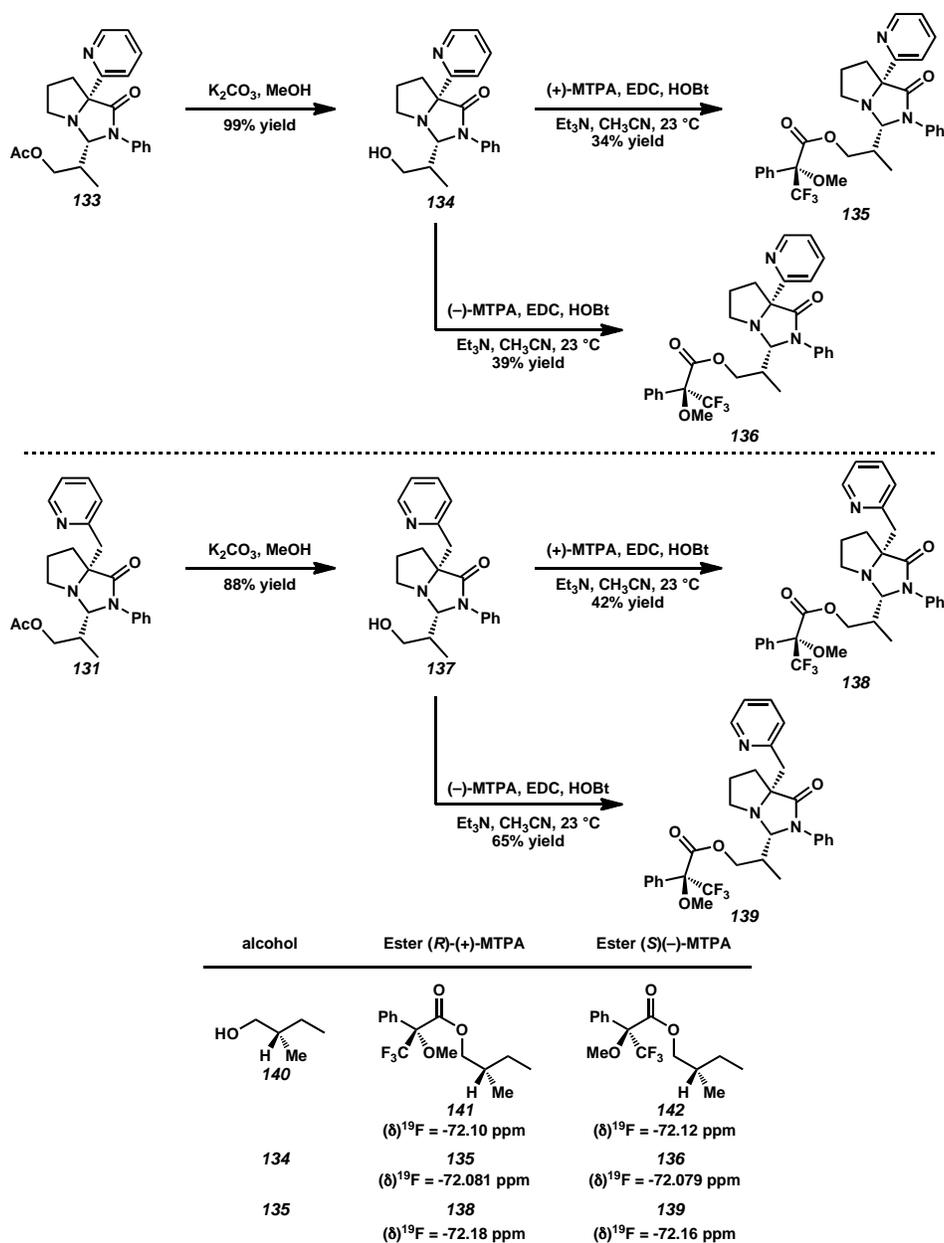
Alcohol 134: To a solution of acetate **133** (127 mg, 0.335 mmol) in MeOH (3.30 mL) was added

K$_2$CO$_3$ (92.5 mg, 0.669 mmol) at 23 °C, and the resulting mixture was stirred 24 h. The reaction

was partitioned between water (10 mL) and EtOAc (10 mL), and the aqueous layer was extracted

with EtOAc (2 x 15 mL). The combined organic phases were dried over Na$_2$SO$_4$ and

concentrated in vacuo to afford alcohol **134** (113 mg, 99% yield, R_f = 0.34 in 1:1 hexanes/acetone) as a white solid. The alcohol was sufficiently pure to be taken on to the next step.

Alcohol 134: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.57 (d, J = 4.9 Hz, 1H), 7.78-7.73 (comp m, 2H), 7.39 (t, J = 7.9 Hz, 2H), 7.31 (d, J = 7.6 Hz, 2H), 7.27-7.22 (comp m, 2H), 4.72 (d, J = 3.5 Hz, 1H), 3.60 (dd, J = 12.1, 1.1 Hz, 2H), 3.44 (dt, J = 11.1, 6.7 Hz, 1H), 3.33 (dd, J = 12.1, 5.8 Hz, 1H), 3.09 (dt, J = 11.3, 5.8 Hz, 1H), 2.56 (dt, J = 13.2, 6.9 Hz, 1H), 2.33 (dt, J = 13.2, 6.7 Hz, 1H), 1.91 (app. quintet, J = 6.7 Hz, 2H), 1.82-1.78 (m, 1H), 1.04 (d, J = 7.2 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 171.9, 160.8, 148.2, 137.1, 136.3, 129.2, 126.7, 124.9, 122.8, 121.8, 86.4, 78.2, 61.6, 58.1, 38.8, 37.6, 25.4, 13.9; IR (film) 3333, 2964, 1701, 1591, 1407, 751 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{20}$H$_{24}$N$_3$O$_2$]$^+$: 338.1863, found 138.1871.

Ester 135: To a solution of alcohol **134** (48.7 mg, 0.144 mmol), (+)-MTPA (33.8 mg, 0.144 mmol), EDC (33.2 mg, 0.173 mmol), HOBt (6.6 mg, 0.0432 mmol) in CH$_3$CN (1.40 mL) at 23 °C was added Et$_3$N (21.3 μL, 0.152 mmol). The reaction was stirred overnight at 23 °C. The solvent was removed, and the residue was partitioned between water (10 mL) and EtOAc (10 mL). The organic layer was dried over Na$_2$SO$_4$ and concentrated in vacuo. The crude residue was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford ester **135** (27.0 mg, 34% yield, R_f = 0.52 in 1:1 hexanes/acetone) as a white solid.

Ester 135: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.67 (d, J = 4.3 Hz, 1H), 7.74-7.65 (comp m, 2H), 7.47-7.30 (comp m, 9H), 7.26-7.16 (comp m, 2H), 4.70 (d, J = 5.3 Hz, 1H), 4.19-4.11 (comp m, 2H), 3.52-3.44 (m, 1H), 3.44 (s, 3H), 2.96 (dt, J = 10.5, 6.5 Hz, 1H), 2.59-2.52 (m, 1H), 2.44-2.37 (m, 1H), 2.03-1.94 (comp m, 2H), 1.92-1.82 (m, 1H), 0.79 (d, J = 7.0 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.5, 166.1, 161.84, 161.82, 149.4, 136.9, 136.5, 132.1, 129.6, 129.2,

128.4, 127.3, 126.5, 124.4, 122.2, 120.7, 83.4, 77.9, 68.1, 58.5, 55.3, 55.2, 38.4, 25.6, 13.8; ^{19}F NMR (300 MHz, CDCl$_3$) δ -72.18; IR (film) 2968, 1748, 1705, 1588, 1169, 1122, 696 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{30}$H$_{30}$F$_3$N$_3$O$_4$Na]$^+$: 576.2081, found 576.2069.

Ester 136: To a solution of alcohol **134** (53.2 mg, 0.158 mmol), (–)-MTPA (36.9 mg, 0.158 mmol), EDC (36.3 mg, 0.189 mmol), HOBt (7.2 mg, 0.0473 mmol) in CH$_3$CN (1.60 mL) at 23 ˚C was added Et$_3$N (23.3 µL, 0.166 mmol). The reaction was stirred overnight at 23 ˚C. The solvent was removed by rotary evaporation, and the residue was partitioned between water (10 mL) and EtOAc (10 mL). The organic layer was dried over Na$_2$SO$_4$ and concentrated in vacuo. The crude residue was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford ester **136** (33.7 mg, 39% yield, R$_f$ = 0.52 in 1:1 hexanes/acetone) as a white solid.

Ester 136: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.68 (d, *J* = 4.5 Hz, 1H), 7.75-7.66 (comp m, 2H), 7.43-7.29 (comp m, 9H), 7.26-7.15 (comp m, 2H), 4.70 (d, *J* = 5.4 Hz, 1H), 4.28 (dd, *J* = 10.8, 6.8 Hz, 1H), 4.02 (dd, *J* = 10.8, 4.1 Hz, 1H), 3.46-3.41 (m, 1H), 3.43 (d, *J* = 0.7 Hz, 3H), 2.93 (dt, *J* = 10.5, 6.5 Hz, 1H), 2.58-2.51 (m, 1H), 2.43-2.36 (m, 1H), 2.03-1.93 (comp m, 2H), 1.91-1.82 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.5, 166.2, 161.9, 149.3, 136.9, 136.6, 132.2, 129.6, 129.2, 128.4, 127.2, 126.5, 124.7, 124.4, 122.3, 121.9, 120.7, 83.2, 77.6, 68.1, 58.4, 55.3, 38.4, 38.3, 25.6, 13.9; ^{19}F NMR (300 MHz, CDCl$_3$) δ -72.16; IR (film) 2969, 1749, 1708, 1588, 1273, 1169, 1023 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{30}$H$_{30}$F$_3$N$_3$O$_4$Na]$^+$: 576.2081, found 576.2080.

Alcohol 137: To a solution of acetate **131** (118 mg, 0.300 mmol) in MeOH (3.00 mL) at 23 °C was added K$_2$CO$_3$ (83.0 mg, 0.600 mmol), and the resulting mixture was stirred overnight. The reaction was partitioned between water (10 mL) and EtOAc (10 mL). The aqueous layer was extracted with EtOAc (2 x 15 mL). The combined organic layers were dried over Na$_2$SO$_4$ and

concentrated to afford alcohol **137** (93.1 mg, 88% yield, R_f = 0.34 in 1:1 hexanes/acetone) as a white solid, which was sufficiently pure to be taken on to the next step.

Alcohol 137: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.55 (dd, J = 4.8, 0.8 Hz, 1H), 7.62 (td, J = 7.7, 1.8 Hz, 1H), 7.41-7.35 (comp m, 2H), 7.30-7.28 (comp m, 3H), 7.26-7.21 (m, 1H), 7.16 (ddd, J = 7.5, 4.9, 0.9 Hz, 1H), 4.51 (d, J = 3.4 Hz, 1H), 3.71 (dd, J = 12.0, 1.6 Hz, 1H), 3.35 (dd, J = 12.0, 5.6 Hz, 1H), 3.26 (ABq, J = 13.2 Hz, Δν = 12.8 Hz, 2H), 2.82-2.77 (m, 1H), 2.72-2.65 (m, 1H), 2.16-2.11 (m, 1H), 2.01-1.94 (m, 1H), 1.82-1.74 (m, 1H), 1.65-1.54 (m, 1H), 1.38-1.31 (m, 1H), 0.97 (d, J = 7.3 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 174.3, 156.8, 149.0, 136.2, 129.2, 126.7, 125.2, 124.9, 123.7, 121.7, 86.5, 74.4, 62.0, 57.7, 45.1, 37.7, 34.3, 24.6, 14.1; IR (film) 3332, 2961, 1696, 1594, 1476, 753, 698 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{21}$H$_{26}$N$_3$O$_2$]$^+$: 352.2020, found 352.2024.

Ester 138: To a solution of alcohol **137** (45.0 mg, 0.128 mmol), (*R*)-(+)-MTPA (30.0 mg, 0.128 mmol), EDC (29.4 mg, 0.154 mmol), HOBt (6.0 mg, 0.0380 mmol) in CH$_3$CN (1.30 mL) at 23 ˚C was added Et$_3$N (19.0 μL, 0.134 mmol). The reaction was stirred 24 h at 23 ˚C. The solvent was removed, and the residue partitioned between water (10 mL) and EtOAc (10 mL). The organic layer was dried over Na$_2$SO$_4$ and concentrated in vacuo. The crude residue was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford ester **138** (30.5 mg, 42% yield, R_f = 0.52 in 1:1 hexanes/acetone) as a white solid.

Ester 138: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.55 (dd, J = 4.9, 0.9 Hz, 1H), 7.62 (td, J = 7.7, 1.8 Hz, 1H), 7.42-7.34 (comp m, 8H), 7.26-7.21 (comp m, 3H), 7.11 (ddd, J = 7.4, 5.0, 1.0 Hz, 1H), 4.40 (d, J = 4.7 Hz, 1H), 3.95 (dd, J = 10.8, 4.0 Hz, 1H), 3.86 (dd, J = 10.8, 7.5 Hz, 1H), 3.43 (d, J = 0.9 Hz, 3H), 3.39 (d, J = 13.2 Hz, 1H), 3.06 (d, J = 13.2 Hz, 1H), 3.05-3.01 (m, 1H), 2.77 (dt, J = 11.4, 5.7 Hz, 1H), 2.16-2.05 (comp m, 2H), 1.73-1.66 (comp m, 2H), 1.43-1.37 (m, 1H), 0.65 (d,

$J = 7.0$ Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 174.0, 166.0, 157.9, 149.1, 149.0, 136.4, 136.0, 132.2, 129.6, 129.2, 128.3, 127.2, 126.8, 125.0, 124.9, 121.8, 83.7, 75.2, 67.3, 57.7, 55.2, 46.0, 37.9, 36.4, 24.7, 13.9; ^{19}F NMR (300 MHz, CDCl$_3$) δ -72.081; IR (film) 2967, 1749, 1704, 1592, 1169, 1024, 720, 698 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{31}$H$_{33}$F$_3$N$_3$O$_4$]$^+$: 568.2418, found 568.2420.

Ester 139: To alcohol **137** (70.6 mg, 0.201 mmol), (*S*)-(–)-MTPA (47.0 mg, 0.201 mmol), EDC (46.2 mg, 0.241 mmol), HOBt (27.7 mg, 0.0181 mmol) in CH$_3$CN (2.00 mL) at 23 ˚C was added Et$_3$N (29.6 µL, 0.211 mmol). The reaction was stirred 24 h at 23 ˚C. The solvent was removed, and the residue dissolved in water (10 mL) and EtOAc (10 mL). The organic layer was separated, then dried over Na$_2$SO$_4$, filtered and concentrated. The crude residue was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford ester **139** (74.5 mg, 65% yield, R$_f$ = 0.52 in 1:1 hexanes/acetone) as a white solid.

Ester 139: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.56 (dd, $J = 4.9$, 0.9 Hz, 1H), 7.62 (td, $J = 7.7$, 1.8 Hz, 1H), 7.43-7.32 (comp m, 8H), 7.26-7.21 (comp m, 3H), 7.12 (ddd, $J = 7.4$, 5.0, 0.9 Hz, 1H), 4.41 (d, $J = 4.9$ Hz, 1H), 4.03 (dd, $J = 10.8$, 7.4 Hz, 1H), 3.82 (dd, $J = 10.8$, 3.8 Hz, 1H), 3.43 (d, $J = 0.8$ Hz, 3H), 3.39 (d, $J = 13.2$ Hz, 1H), 3.07 (d, $J = 13.2$ Hz, 1H), 2.97 (dt, $J = 11.3$, 7.0 Hz, 1H), 2.71 (dt, $J = 11.3$, 5.7 Hz, 1H), 2.17-2.06 (comp m, 2H), 1.71-1.61 (comp m, 2H), 1.48-1.39 (m, 1H), 0.65 (d, $J = 7.0$ Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 174.1, 166.1, 157.9, 149.1, 136.5, 136.0, 132.2, 129.6, 129.2, 128.3, 127.22, 127.21, 126.7, 124.9, 121.8, 83.6, 75.1, 67.5, 57.6, 55.2, 46.0, 38.0, 36.2, 24.7, 13.9; ^{19}F NMR (300 MHz, CDCl$_3$) δ -72.079; IR (film) 3063, 2967, 2881, 1749, 1703, 1592, 1170, 735 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{31}$H$_{33}$F$_3$N$_3$O$_4$]$^+$: 568.2418, found 568.2421.

137 → **143**

TEMPO, buffer
NaOCl$_2$, NaOCl
CH$_3$CN/H$_2$O, 35 °C
58% yield

(*R*)-PGME, EDC, HOBt
Et$_3$N, CH$_3$CN, 54% yield → **144**

(*S*)-PGME, EDC, HOBt
Et$_3$N, CH$_3$CN, 74% yield → **145**
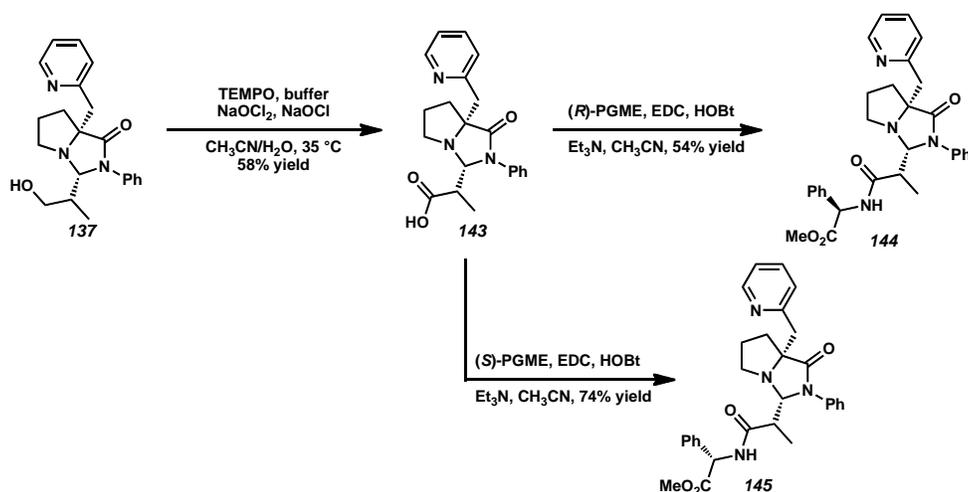
Acid 143: To **137** (122 mg, 0.347 mmol) in CH$_3$CN (1.70 mL) with TEMPO (3.8 mg, 0.0243 mmol), NaH$_2$PO$_4$/Na$_2$HPO$_4$ buffer (1.30 mL, 0.67 M) and NaOCl$_2$ (78.5 mg, 0.694 mmol) at 35 °C was added NaOCl (9.0 µL, 6.94 µmol, 6% solution in H$_2$O) slowly over 1 h. The reaction was stirred at 35 °C overnight (10 h). Upon cooling to 23 °C, water was added and the solution set to pH 8 with 1 M NaOH. The mixture was poured into sat. aq. Na$_2$SO$_3$ and stirred for 30 min. EtOAc was added and separated. The aqueous layer was acidified to pH 2 with 1 M KHSO$_4$ then extracted with EtOAc (2 x 10 mL). The organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (1:3 to 1:1 hexanes:acetone) to afford **143** (74.0 mg, 58% yield, R_f = 0.1 in 1:1 hexanes:acetone) as a white amorphous solid.

Amide 145: To acid **143** (24.5 mg, 0.0670 mmol), (*S*)-(+)-PGME (13.5 mg, 0.0670 mmol), EDC (16.7 mg, 0.0871 mmol), HOBt (10.3 mg, 0.0670 mmol) in CH$_3$CN (0.700 mL) at 23 ˚C was added Et$_3$N (19.3 µL, 0.137 mmol). The reaction was stirred 24 h at 23 ˚C. The solvent was removed, and the residue dissolved in water (10 mL) and EtOAc (10 mL). The organic layer was separated, then dried over Na$_2$SO$_4$, filtered and concentrated. The crude residue was

purified by flash chromatography (4:1 hexanes/acetone eluent) to afford amide **145** (18.4 mg, 54% yield, R_f = 0.55 in 1:1 hexanes/acetone) as a white solid.

Amide 144: To acid **143** (38.1 mg, 0.104 mmol), (*R*)-(-)-PGME (21.0 mg, 0.104 mmol), EDC (26.0 mg, 0.136 mmol), HOBt (16.0 mg, 0.104 mmol) in CH_3CN (1.00 mL) at 23 ˚C was added Et_3N (30.0 µL, 0.214 mmol). The reaction was stirred 24 h at 23 ˚C. The solvent was removed, and the residue dissolved in water (10 mL) and EtOAc (10 mL). The organic layer was separated, then dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford amide **144** (39.4 mg, 74% yield, R_f = 0.54 in 1:1 hexanes/acetone) as a white solid.
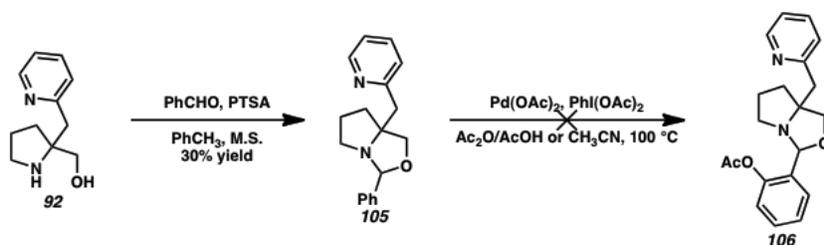
To a solution of alcohol **137** (87.2 mg, 0.248 mmol), *p*-nitrobenzoic acid (45.6 mg, 0.273 mmol), EDC (57.0 mg, 0.298 mmol), HOBt (38.0 mg, 0.248 mmol) in CH_3CN (2.50 mL) at 23 °C was added Et_3N (38.0 µL, 0.273 mmol). The reaction was stirred at 23 °C for 3 d. The volatile organic solvent was removed, and the residue was partitioned between H_2O (15 mL) and EtOAc (15 mL). The organic layer was dried over Na_2SO_4 and concentrated. The crude product was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford benzoate **146** (119 mg, 96% yield, R_f = 0.47 in 1:1 hexanes/acetone) as a white solid. The solid was crystallized by a layering technique with CH_2Cl_2 and hexanes.

Ester 146: ^{1}H NMR (400 MHz, $CDCl_3$) δ 8.62 (dd, *J* = 5.0, 0.9 Hz, 1H), 8.31-8.21 (comp m, 2H), 8.07-7.99 (comp m, 2H), 7.72 (td, *J* = 7.7, 1.7 Hz, 1H), 7.54 (d, *J* = 7.8 Hz, 1H), 7.39-7.29 (comp m, 4H), 7.25-7.20 (comp m, 2H), 4.59 (d, *J* = 3.4 Hz, 1H), 3.99 (dd, *J* = 10.9, 4.9 Hz, 1H),

3.93 (dd, J = 10.9, 8.0 Hz, 1H), 3.34 (ABq, J = 13.2 Hz, $\Delta\nu$ = 67.7 Hz, 2H), 3.10 (dt, J = 11.5,

7.3 Hz, 1H), 2.88 (dt, J = 11.4, 5.7 Hz, 1H), 2.14 (app. t, J = 7.2 Hz, 2H), 2.01-1.95 (m, 1H),

1.79-1.69 (comp m, 2H), 0.93 (d, J = 7.0 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.7, 164.2,

157.3, 150.4, 148.2, 137.0, 136.1, 135.5, 130.5, 129.2, 126.6, 125.5, 124.3, 123.4, 122.3, 84.1,

75.2, 66.2, 57.8, 44.9, 36.4, 35.9, 24.7, 14.3; [IR (film) 2968, 1723, 1721, 1527, 1276, 1103, 720

cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{28}$H$_{29}$N$_4$O$_5$]$^+$: 500.2132, found 500.2137; mp 110-

116 °C.
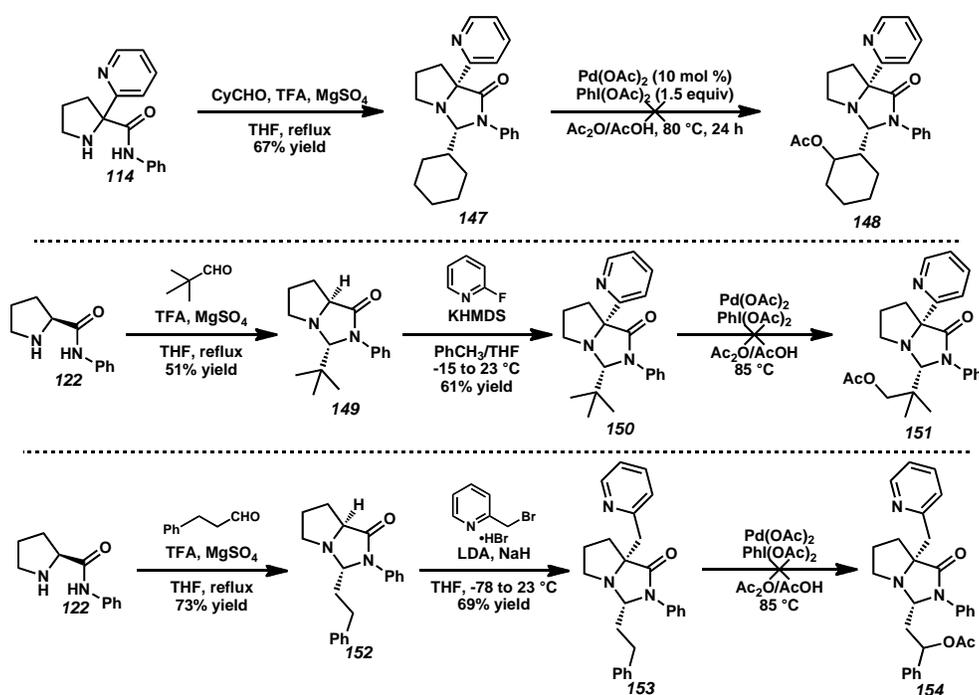
General procedure for condensation: To amino amide (1 equiv) in THF (0.1 M) was added

aldehyde (1.3 equiv), TFA (10 mol %) and MgSO$_4$ (1.5 equiv) at 23 °C. The reaction was

heated to reflux overnight. Upon cooling, sat. aq. NaHCO$_3$ was added. The aqueous layer was

extracted with EtOAc (3x) and the organic layers dried over Na$_2$SO$_4$ and concentrated. The

crude residue was purified via column chromatography (4:1 to 1:1 hexanes:EtOAc) to afford the

pure aminal.

Aminal 147: 67% yield, R$_f$ = 0.25 in 1:1 hexanes:EtOAc

Aminal 149: 51% yield, $R_f = 0.30$ in 1:1 hexanes:EtOAc

Aminal 152: 73% yield, $R_f = 0.35$ in 1:1 hexanes:EtOAc

Pyridine 150: To a solution of aminal **149** (809 mg, 3.13 mmol), 2-fluoropyridine (269 μL, 3.13 mmol) in PhCH$_3$ (10.4 mL) at -15 °C was added KHMDS (625 mg, 3.13 mmol) in THF (6.30 mL) slowly over 1 h. Upon completion of addition, the reaction was allowed to warm to 23 °C and stirred overnight. The reaction was filtered over a pad of silica (5 x 5 cm, 100 mL EtOAC eluent) and concentrated. The crude product was purified by flash chromatography (3:1 to 1:1 hexanes/EtOAc eluent) to afford pyridine **150** (244 mg, 23% yield, $R_f = 0.25$ in 1:1 hexanes/EtOAc) as a light beige solid.

Pyridine 153: To a solution of freshly distilled diisopropylamine (552 μL, 3.93 mmol) in THF (4.40 mL) at -78 °C was added *n*-BuLi (1.51 mL, 2.5 M in hexanes, 3.77 mmol) dropwise. The solution was stirred for 10 min at -78 °C, at which time a solution of aminal **152** (1.16 g, 3.77 mmol) in THF (5.00 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 °C. To a suspension of NaH (377 mg, 60% dispersion in mineral oil, 9.42 mmol, washed 2 x 2.0 mL with hexanes) in DMF (8.00 mL) at 0 °C was added 2-(bromomethyl)pyridine hydrobromide (795 mg, 3.14 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.40 mL DMF). The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched by slow addition of H$_2$O (20 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 20 mL). The combined organic layers were washed with brine (2 x 35 mL), dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by flash chromatography (1:3 to 2:3 hexanes/EtOAc eluent) to afford pyridine **153** (865 mg, 69% yield, $R_f = 0.18$ in 1:1 hexanes:EtOAc) as a light orange solid.

General procedure for acetoxylation: Pyridine (1 equiv), Pd(OAc)$_2$ (10-50 mol %) and PhI(OAc)$_2$ (1.5-2.0 equiv) were combined in AcOH/Ac$_2$O (1:1, 0.1 M) in a 2-dram vial and heated to 80-85 °C for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na$_2$CO$_3$ until pH ~9. The aqueous solution was extracted with CH$_2$Cl$_2$ (3 x 15 mL), and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. Crude ^{1}H NMR revealed no product formation.
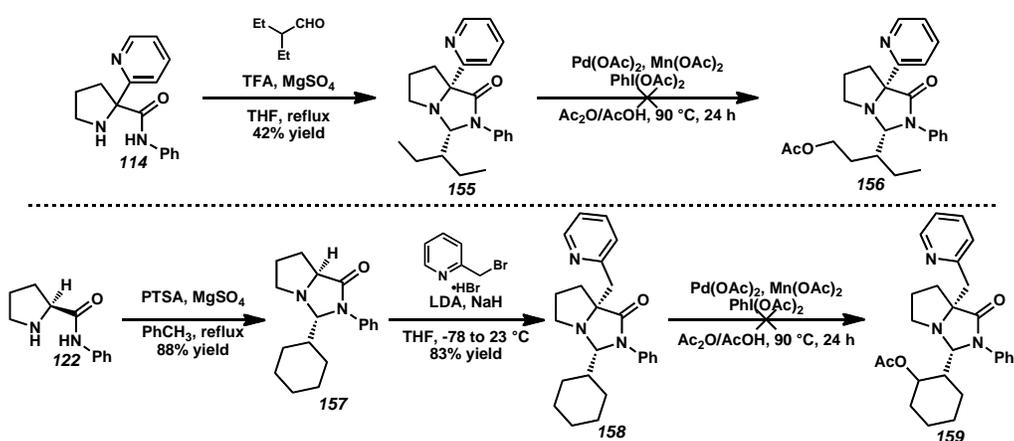
General procedure for condensation: To amino amide (1 equiv) in THF or toluene (0.1 M) was added aldehyde (1.3 equiv), TFA (10 mol %) or PTSA (5 mol %) and MgSO$_4$ (1.5 equiv) at 23 °C. The reaction was heated to reflux overnight. Upon cooling, sat. aq. NaHCO$_3$ was added. The aqueous layer was extracted with EtOAc (3x) and the organic layers dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via column chromatography (4:1 to 1:1 hexanes:EtOAc) to afford the pure aminal.

Aminal 155: 42% yield, R$_f$ = 0.25 in 1:1 hexanes:EtOAc

Aminal 157: 88% yield, R$_f$ = 45 in 1:1 hexanes:EtOAc

Pyridine 158: To a solution of freshly distilled diisopropylamine (337 μL, 2.40 mmol) in THF (3.70 mL) at -78 °C was added n-BuLi (0.920 mL, 2.5 M in hexanes, 2.31 mmol) dropwise. The

solution was stirred for 10 min at -78 ˚C, at which time a solution of aminal **157** (654 mg, 2.30 mmol) in THF (4.00 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 ˚C. To a suspension of NaH (230 mg, 60% dispersion in mineral oil, 5.75 mmol, washed 2 x 2.0 mL with hexanes) in DMF (6.00 mL) at 0 ˚C was added 2-bromomethylpyridine hydrobromide (485 mg, 1.92 mmol). The suspension was stirred at 0 ˚C for 30 min, at which time it was added to the enolate solution at -78 ˚C (flask rinsed with additional 1.70 mL DMF). The suspension was warmed to 23 ˚C and stirred overnight. The reaction was quenched by slow addition of H_2O (20 mL) at 23 ˚C, and the resulting mixture was extracted with EtOAc (3 x 20 mL). The combined organic layers were washed with brine (2 x 35 mL), dried over Na_2SO_4 and concentrated.   The crude residue was purified by flash chromatography (1:3 to 3:2 hexanes/EtOAc eluent) to afford pyridine **158** (865 mg, 83% yield, R_f = 0.18 in 1:1 hexanes:EtOAc) as a light yellow solid.

General procedure for acetoxylation: Pyridine (1 equiv), $Pd(OAc)_2$ (10 mol %), $Mn(OAc)_2$ (1.2 equiv) and $PhI(OAc)_2$ (4.0 equiv) were combined in $AcOH/Ac_2O$ (1:1, 0.1 M) in a 2-dram vial and heated to 90 °C for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptanes (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous solution was extracted with CH_2Cl_2 (3 x 15 mL), and the combined organic layers were dried over Na_2SO_4 and concentrated. Crude ^{1}H NMR revealed no product formation.
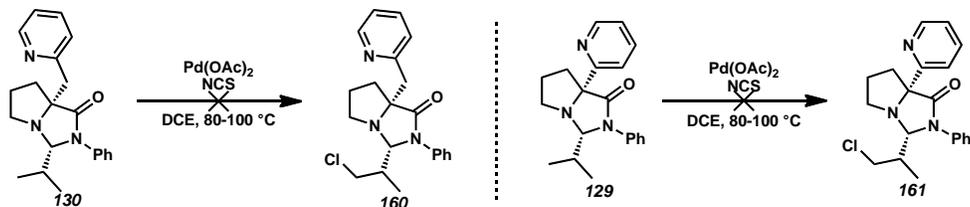
General procedure for chlorination: Pyridine (1 equiv), Pd(OAc)$_2$ (10-50 mol %) and NCS (1-2 equiv) were combined in DCE in a 2-dram vial and heated to 80-100 °C for 24 h. Solvent was removed under reduced pressure. Crude ^{1}H NMR revealed no product formation.
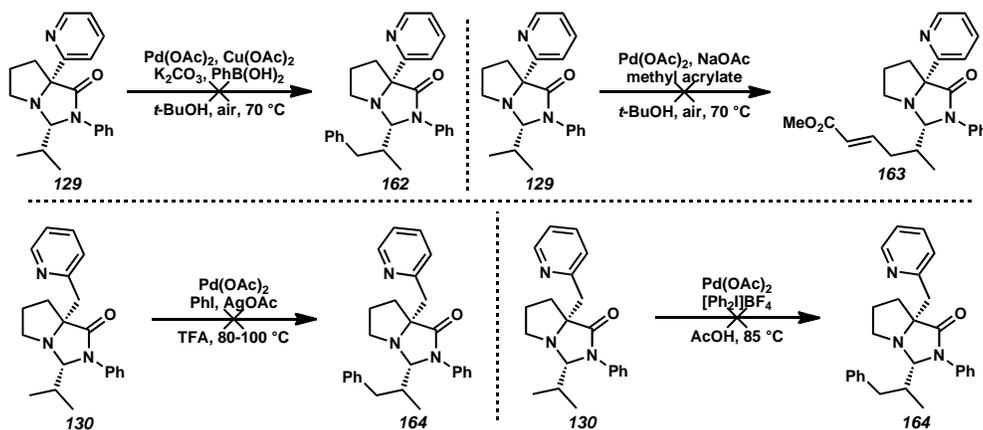
General procedure for olefination: Pyridine (1 equiv), Pd(OAc)$_2$ (10 mol %), H$_4$[PMo$_{11}$VO$_{40}$] (3 mol %), methyl acrylate (4 equiv) and NaOAc (1.1 equiv) were combined in AcOH or DCE (0.1 M) in a 2-dram vial and heated to 110 °C for 24 h. The reaction was filtered through a plug of celite with CH$_2$Cl$_2$ and the solvent was removed under reduced pressure. Product ratios determined by crude ^{1}H NMR.

Pd(OAc)₂ (10 mol %)
H₄[PMo₁₁VO₄₀] (3 mol %)
methyl acryate (4 equiv)
solvent, 110 °C

129 → **168**

entry	solvent	base	additive	% conversion
1	DCE	NaOAc (2 equiv)	none	18
2	t-AmOH	NaOAc (2 equiv)	none	0
3	CH₃CN	NaOAc (2 equiv)	none	17
4	DCE	Na₂CO₃ (1 equiv)	NaOPiv (0.2 equiv)	trace
5	DCE	K₂CO₃ (1 equiv)	none	0
6		NaOAc (1.1 equiv)	none	23
7	DMF	NaOAc (1.1 equiv)	none	0
8	DCE	NaOAc (1.0 equiv)	acac (10 mol %)	0
9	CF₃CH₂OH	NaOAc (1.0 equiv)	AcOH (1.0 equiv)	trace

General procedure for olefination: Pyridine (1 equiv), Pd(OAc)₂ (10 mol %), H₄[PMo₁₁VO₄₀] (3 mol %), methyl acrylate (4 equiv), additive (0-1.0 equiv) and base (1-2 equiv) were combined in solvent (0.1 M) in a 2-dram vial and heated to 110 °C for 24 h. The reaction was filtered through a plug of celite with CH₂Cl₂ and the solvent was removed under reduced pressure. Product ratios determined by crude ¹H NMR.

Pd catalyst
H₄[PMo₁₁VO₄₀] (3 mol %)
methyl acryate (4 equiv)
NaOAc (1.0 equiv)
solvent, 110 °C, 20 h

129 → **168**

entry	catalyst	solvent	% conversion
1	Pd(CH₃CN)₄(BF₄)₂	CF₃CH₂OH	20
2	Pd(CH₃CN)₄(BF₄)₂	DCE	18
3	Pd(OAc)₂	DCE	28
4	PdCl₂	DCE	0
5	Pd(TFA)₂	DCE	14
6	Pd(acac-F₆)₂	CF₃CH₂OH	11

General procedure for olefination: Pyridine (1 equiv), palladium catalyst (10 mol %), H₄[PMo₁₁VO₄₀] (3 mol %), methyl acrylate (4 equiv), and NaOAc (1.0 equiv) were combined in

DCE or CF₃CH₂OH (0.1 M) in a 2-dram vial and heated to 110 °C for 24 h. The reaction was filtered through a plug of celite with CH₂Cl₂ and the solvent was removed under reduced pressure. Product ratios determined by crude ^{1}H NMR.

CF₃CH₂OH	33% conversion
DCE	18% conversion
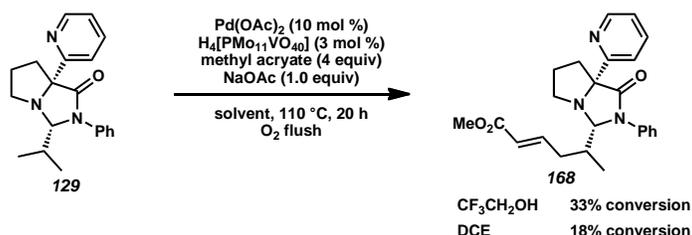
General procedure for olefination: Pyridine (1 equiv), Pd(OAc)₂ (10 mol %), H₄[PMo₁₁VO₄₀] (3 mol %), methyl acrylate (4 equiv), and NaOAc (1.0 equiv) were combined in DCE or CF₃CH₂OH (0.1 M) in a 2-dram vial fitted with a septum. The vial was flushed with O₂ from a balloon, then sealed and heated to 110 °C for 20 h. The reaction was filtered through a plug of celite with CH₂Cl₂ and the solvent was removed under reduced pressure. Product ratios determined by crude ^{1}H NMR.
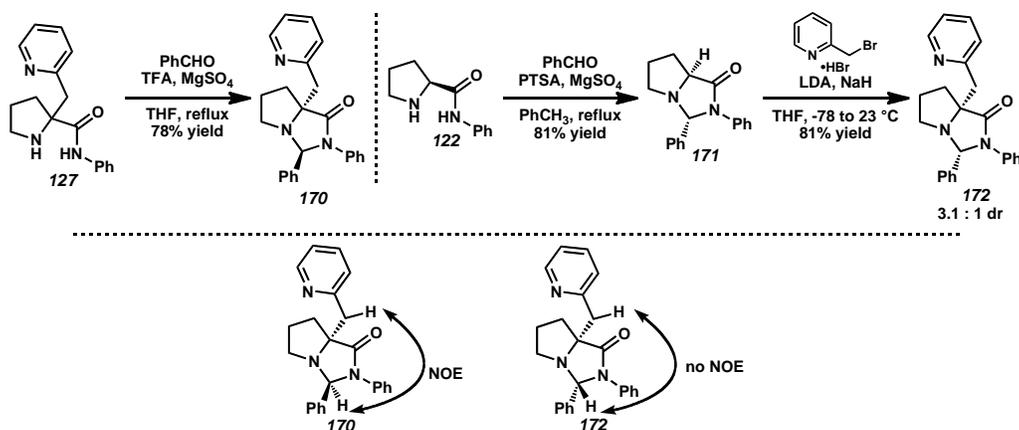
Pyridine 170: To **127** (71.3 mg, 0.253 mmol) in THF (2.50 mL) was added PhCHO (33.3 μL, 0.329 mmol), TFA (3.9 μL, 0.12 mmol) and MgSO₄ (61.0 mg, 0.507 mmol) at 23 °C. The suspension was refluxed overnight. Upon cooling, the reaction was quenched with sat. aq.

NaHCO$_3$ solution (10 mL). The aqueous was extracted with EtOAc (3 x 5 mL), dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (1:4 to 1:1 EtOAc:hexanes) to give **170** (73.6 mg, 78% yield, R$_f$ = 0.21 in 40:1 EtOAc:MeOH) as a light brown solid.

Aminal 171: To a solution of amino amide **122** (1.50 g, 7.90 mmol) in PhCH$_3$ (26.3 mL) at 23 °C was added benzaldehyde (1.00 mL, 10.2 mmol), TsOH·H$_2$O (75.0 mg, 0.395 mmol), and MgSO$_4$ (1.40 g, 11.8 mmol). The suspension was heated to reflux overnight. Upon cooling to 23 °C, the solution was quenched with sat. aq. NaHCO$_3$ (20 mL). The aqueous layer was extracted with EtOAc (3 x 30 mL), and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The dark brown residue was purified by flash chromatography (7:3 hexanes/EtOAc eluent) to afford aminal **171** (1.77 g, 81% yield, R$_f$ = 0.24 in 1:1 hexanes/EtOAc) as a light brown solid.

Aminal 171: ^{1}H NMR (400 MHz, CDCl$_3$) δ 7.47 (d, J = 8.2 Hz, 2H), 7.36-7.25 (comp m, 7H), 7.11-7.07 (m, 1H), 5.67 (s, 1H), 4.03 (app. t, J = 6.6 Hz, 1H), 3.46-3.41 (m, 1H), 2.88 (app. q, J = 8.3 Hz, 1H), 2.20 (app. q, J = 8.3 Hz, 2H), 1.92-1.87 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.1, 139.6, 137.9, 129.3, 129.2, 128.7, 126.2, 125.3, 121.3, 83.8, 64.5, 56.2, 27.7, 25.0; IR (film) 2969, 3032, 1699, 1598, 1498, 1384, 757 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H$^+$) [C$_{18}$H$_{19}$N$_2$O]$^+$: 279.1492, found 279.1492.

Pyridine 172: To a solution of freshly distilled diisopropylamine (933 µL, 6.60 mmol) in THF (8.00 mL) at -78 °C was added n-BuLi (2.60 mL, 2.5 M in hexanes, 6.40 mmol) dropwise. The solution was stirred for 10 min at -78 °C, at which time a solution of aminal **171** (1.77 g, 6.40 mmol) in THF (7.90 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 °C. To a suspension of NaH (638 mg, 60% dispersion in mineral oil, 15.9 mmol,

washed 2 x 1.5 mL with hexanes) in DMF (14.0 mL) at 0 °C was added 2-bromomethylpyridine hydrobromide (1.34 g, 5.30 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.90 mL DMF). The reaction mixture was warmed to 23 °C and stirred overnight. The reaction was quenched slowly with H_2O (40 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 30 mL). The combined organic layers were washed with brine (2 x 50 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified by flash chromatography (2:3 hexanes/EtOAc eluent) to afford pyridine **172** (1.38 g, 71% yield, R_f = 0.46 in 40:1 EtOAc/MeOH) as a beige solid and pyridine **172b** (458 mg, 23% yield, R_f = 0.21 in 40:1 EtOAc/MeOH) as a light beige solid.

Pyridine 172: ^1H NMR (400 MHz, $CDCl_3$) δ 8.53 (d, J = 3.6 Hz, 1H), 7.53 (app. t, J = 7.2 Hz, 1H), 7.34 (d, J = 7.7 Hz, 2H), 7.23 (t, J = 7.9 Hz, 2H), 7.18-7.13 (comp m, 4H), 7.07 (t, J = 8.1 Hz, 2H), 6.89-6.88 (comp m, 2H), 5.44 (s, 1H), 3.20 (ABq, J = 13.1 Hz, Δv = 80.5 Hz, 2H), 3.18-3.14 (m, 1H), 2.99 (dt, J = 11.2, 5.7 Hz, 1 H), 2.33-2.26 (m, 1H), 2.23-2.16 (m, 1H), 1.83-1.76 (m, 1H), 1.63-1.57 (m, 1H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.9, 157.8, 148.6, 140.1, 137.0, 135.8, 128.7, 128.3, 128.2, 127.0, 125.4, 125.3, 122.5, 121.5, 83.1, 75.1, 55.8, 45.2, 35.5, 24.6; IR (film) 3052, 2968, 1702, 1592, 1499, 1391, 747, 702 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [$C_{24}H_{23}N_3ONa$]$^+$: 392.1733, found 392.1732.

Pyridine 172b: ^1H NMR (400 MHz, $CDCl_3$) δ 8.57 (d, J = 4.0 Hz, 1H), 7.50 (td, J = 7.7, 1.7 Hz, 1H), 7.24 (d, J = 7.8 Hz, 1H), 7.20-7.15 (comp m, 6H), 7.11-7.09 (comp m, 2H), 7.02 (d, J = 7.4 Hz, 1H), 6.98 (d, J = 7.6 Hz, 2H), 5.34 (s, 1H), 3.48 (d, J = 13.1 Hz, 1H), 3.12 (d, J = 13.1 Hz, 1H), 2.57-2.50 (m, 1H), 2.45-2.36 (comp m, 2H), 2.10 (dt, J = 12.7, 6.1 Hz, 1H), 1.66-1.60 (comp m, 2H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 177.5, 157.6, 149.1, 137.5, 136.1, 134.3, 128.5,

128.42, 128.40, 128.2, 124.9, 124.6, 122.1, 121.7, 78.5, 75.0, 51.2, 46.2, 35.8, 24.5; IR (film) 2967, 1707, 1591, 1377, 1301, 746, 703 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{24}$H$_{24}$N$_3$O]$^+$: 370.1914, found 370.1917.
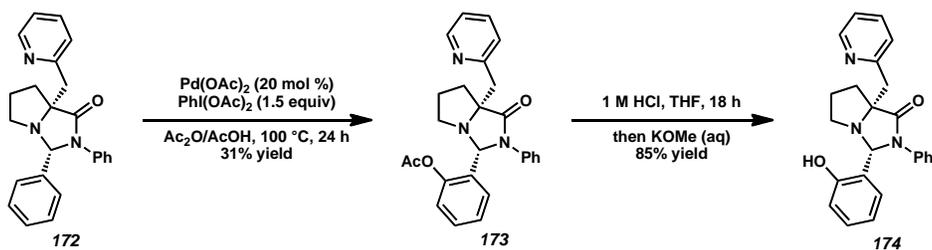
Acetate 173: Pyridine **172** (50.0 mg, 0.141 mmol), Pd(OAc)$_2$ (6.3 mg, 0.0283 mmol), and PhI(OAc)$_2$ (68.0 mg, 0.211 mmol) were dissolved in AcOH (0.700 mL) and Ac$_2$O (0.700 mL) in a round-bottomed flask. The flask was capped and heated to 100 °C for 24 h. The reaction was filtered through a plug of silicaand the solvent was removed by azeotropic removal with heptanes (3 x 15 mL).   The crude residue was purified by flash chromatography (17:3 → 4:1 hexanes/acetone eluent) to afford acetate **173** (18.9 mg, 31% yield, R$_f$ = 0.50 in 1:1 hexanes/acetone) as a beige solid.

Acetate 173: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.47 (dd, J = 4.9, 0.9 Hz, 1H), 7.52 (dd, J = 8.7, 1.0 Hz, 2H), 7.45 (td, J = 7.7, 1.8 Hz, 1H), 7.30-7.25 (comp m, 4H), 7.14 (dd, J = 8.1, 1.0 Hz, 1H), 7.10-7.06 (comp m, 2H), 7.01 (td, J = 7.6, 0.8 Hz, 1H), 6.85-6.81 (comp m, 2H), 5.80 (s, 1H), 3.17 (dt, J = 10.3, 6.2 Hz, 1 H), 3.08 (s, 2H), 2.95 (dt, J = 10.2, 6.6 Hz, 1H), 2.38 (s, 3H), 2.21 (app. t, J = 7.2 Hz, 2H), 1.71-1.62, (m, 1H), 1.37-1.28 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.8, 168.7, 157.3, 148.9, 148.3, 137.3, 135.8, 132.0, 129.0, 128.9, 126.6, 126.0, 125.9, 124.9, 122.8, 121.5, 120.7, 98.4, 77.7, 75.0, 57.4, 44.5, 34.7, 24.7, 21.1; IR (film) 3062, 2966, 1767, 1702, 1497, 1385, 1199 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H$^+$) [C$_{26}$H$_{26}$N$_3$O$_3$]$^+$: 428.1969, found 428.1976.

Phenol 174: Acetate **173** (122 mg, 0.284 mmol) was dissolved in aq. HCl (1 M, 2.80 mL) and THF (5.70 mL), and the resulting solution was heated to reflux overnight. Upon cooling the reaction was quenched with solid Na_2CO_3 until pH ~9-10. The mixture was then extracted with EtOAc (3 x 20 mL). The combined organic layers were dried over Na_2SO_4, filtered and concentrated. The residue was taken up in ether (20 mL) and extracted with Claisen's alkali (17.5 g KOH dissolved in 12.5 mL H_2O, then 37.5 mL MeOH added, 3 x 15 mL). The combined aqueous layers were acidified to pH ~9-10 and extracted with EtOAc (3 x 20 mL). The combined organic layers were dried over Na_2SO_4, filtered and concentrated to afford phenol **174** (98.0 mg, 89% yield, R_f = 0.42 in 1:1 acetone:hexanes) as a beige solid.

Phenol 174: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.60 (d, *J* = 4.8 Hz, 1H), 7.63 (td, *J* = 7.7, 1.8 Hz, 1H), 7.29-7.25 (m, 1H), 7.23-7.20 (comp m, 3H), 7.17-7.10 (comp m, 2H), 7.06-7.04 (comp m, 2H), 6.78 (d, *J* = 7.8 Hz, 1H), 6.65 (dd, *J* = 7.6, 1.6 Hz, 1H), 6.57 (td, *J* = 7.4, 1.0 Hz, 1H), 5.34 (s, 1H), 3.32 (ABq, *J* = 14.1 Hz, Δν = 93.0 Hz, 2H), 3.10 (dt, *J* = 12.4, 8.0 Hz, 1H), 2.99 (ddd, *J* = 12.4, 7.9, 4.6 Hz, 1H), 2.38-2.24 (comp m, 2H), 2.16-2.07 (m, 1H), 1.95-1.85 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.5, 157.5, 156.8, 149.2, 136.6, 135.8, 130.7, 130.6, 128.9, 127.0, 125.7, 124.9, 122.0, 119.7, 118.7, 117.8, 82.0, 74.5, 50.8, 43.0, 33.1, 23.9; IR (film) 3061, 2965, 1699, 1597, 1499, 1399, 755 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{24}$H$_{24}$N$_3$O$_2$]$^+$: 386.1863, found 386.1865.

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solvent	temp (°C)	yield	solvent	temp (°C)	yield
AcOH	100	<10%	AcOH:Ac₂O	80	50
Ac₂O	100	0%	AcOH	80	25
AcOH:CH₂Cl₂	100	<10%	AcOH:Ac₂O	90	36
AcOH:Ac₂O	23	0%	AcOH	90	18
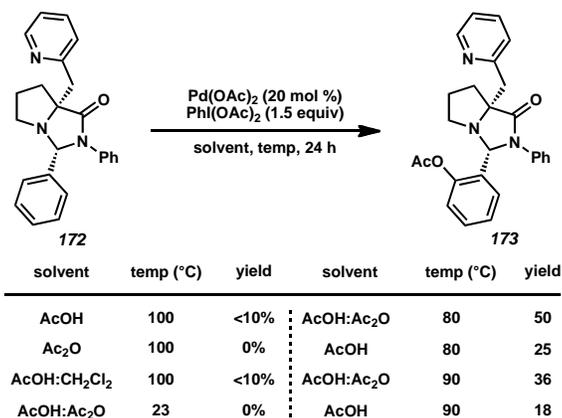
General procedure for acetoxylation: Pyridine **172** (1 equiv), Pd(OAc)$_2$ (20 mol %), PhI(OAc)$_2$ (1.5 equiv) in solvent (0.1 M) were heated to 80-100 °C for 24 h. The reaction was filtered through a plug of silicaand the solvent was removed by azeotropic removal with heptanes (3 x 15 mL). The crude residue was purified by flash chromatography (17:3 → 4:1 hexanes/acetone eluent).

solvent	concentration (M)	*172 : 173*	yield (%, crude)
Ac₂O/AcOH	0.05	1.0 : 2.0	40
Ac₂O/AcOH	0.1	1.0 : 3.0	62
Ac₂O/AcOH	0.2	1.0 : 8.3	50
AcOH	0.05	2.3 : 1.0	12
AcOH	0.1	1.3 : 1.0	18
AcOH	0.2	1 : 1	27
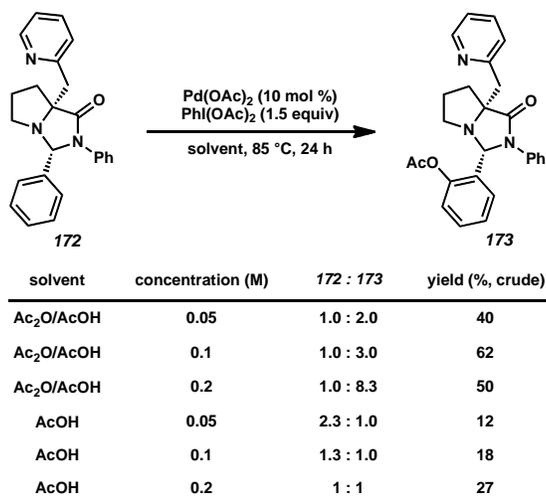
General procedure for acetoxylation: Pyridine **172** (1 equiv), Pd(OAc)$_2$ (10 mol %), PhI(OAc)$_2$ (1.5 equiv) in solvent (0.05-0.2 M) were heated to 85 °C for 24 h. Upon cooling, the solvent was removed by azeotropic removal with heptanes (3 x 15 mL). The yields were calculated based on the crude mass recovery.

Pd catalyst (mol %)	172 : 173
PdCl₂ (10)	1.0 : 1.5
Pd(TFA)₂ (10)	1.2 : 1.0
Pd(OAc)₂ (5)	1.7 : 1
Pd(OAc)₂ (10)	1.0 : 8.0
Pd(OAc)₂ (15)	1.0 : 2.0
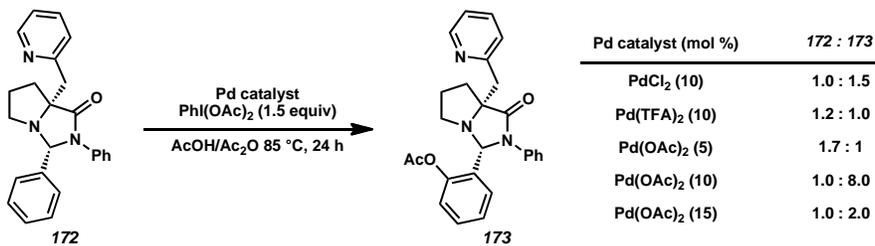
General procedure for acetoxylation: Pyridine **172** (1 equiv), palladium catalyst (5-15 mol %), PhI(OAc)₂ (1.5 equiv) in AcOH/Ac₂O (0.2 M) were heated to 85 °C for 24 h. Upon cooling, the solvent was removed by azeotropic removal with heptanes (3 x 15 mL). The product ratios were determined by crude ¹H NMR.

entry	oxidant	yield (%)
1	PhI(OAc)₂ (1.5 equiv)	62
2	PhI(OAc)₂ (2 equiv)	65
3	PhI(OAc)₂ (3 equiv)	41
4		<10
5	oxone	decomp
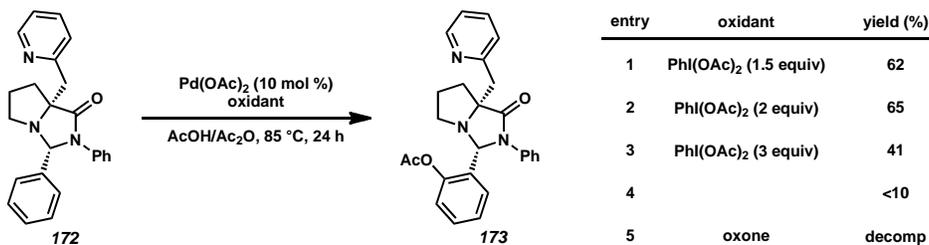
General procedure for acetoxylation: Pyridine **172** (1 equiv), Pd(OAc)₂ (10 mol %), oxidant (1.5 -3 equiv) in AcOH/Ac₂O (0.2 M) were heated to 85 °C for 24 h. Upon cooling, the solvent was removed by azeotropic removal with heptanes (3 x 15 mL). The yields were calculated based on the crude mass recovery.

entry	oxidant	yield (%)
1	PhI(OAc)₂ (1.5 equiv)	62
2	PhI(OAc)₂ (2 equiv)	65
3	PhI(OAc)₂ (3 equiv)	41
4		<10
5	oxone	decomp
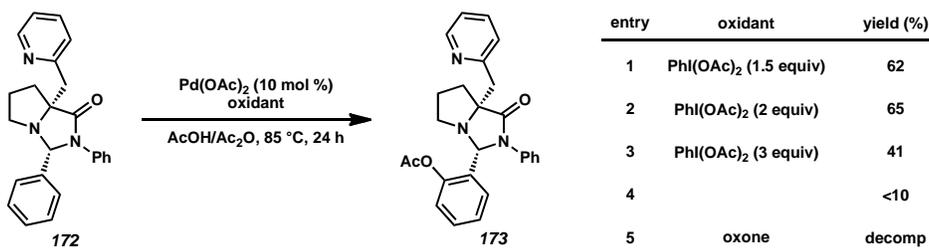
General procedure for acetoxylation: Pyridine **172** (1 equiv), Pd(OAc)₂ (10 mol %), PhI(OAc)₂ (1.5 equiv) in solvent (0.2 M) were heated to 85 °C for 24 h. Upon cooling, the solvent was

removed by azeotropic removal with heptanes (3 x 15 mL). The product ratios were determined by crude ^{1}H NMR. In entry 4, K_2CO_3 (1 equiv) and PivOH (20 mol %) were added to the reaction with the other reagents.
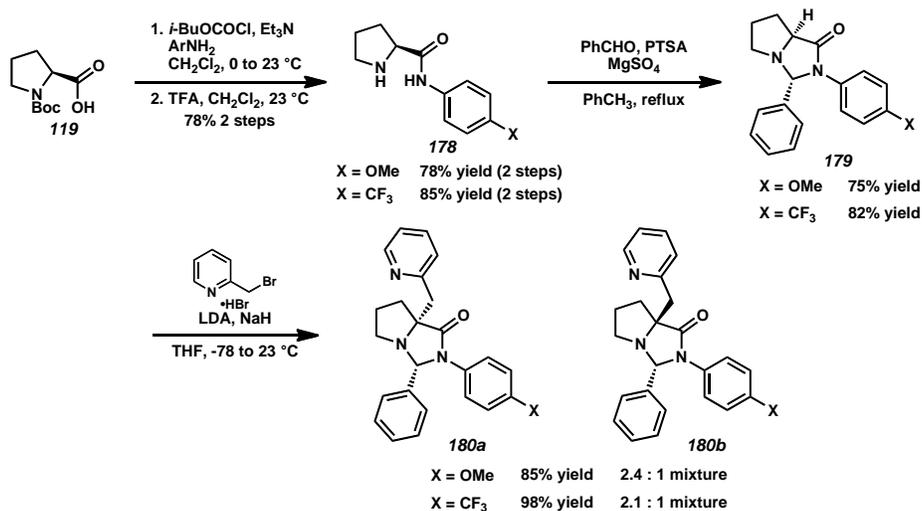
Aminal 179-OMe: To a solution of (*S*)-*N*-Boc proline (0.500 g, 2.32 mmol) in CH_2Cl_2 (11.6 mL) at 0 °C was added isobutyl chloroformate (0.334 mL, 2.56 mmol) and Et_3N (0.359 mL, 2.56 mmol). After stirring for 20 minutes at 0 °C, *p*-anisidine (315 mg, 2.56 mmol) was added, and the reaction was allowed to warm to 23 °C and stirred overnight. The reaction mixture was washed sequentially with aq. $KHSO_4$ (1 M, 15 mL), sat. aq. $NaHCO_3$ (15 mL), and brine (15 mL). The organic layer was dried over Na_2SO_4 and concentrated to afford a pale brown solid. The crude solid was suspended in hexanes (5 mL), cooled to 0 °C and filtered to afford amide (750 mg, 99% yield, R_f = 0.41 in 1:1 hexanes/EtOAc) as a light beige solid, which was sufficiently pure to be taken on to the next step.

To a solution of amide (3.72 g, 11.6 mmol) in CH_2Cl_2 (23.2 mL) at 23 °C was added TFA (18.0 mL, 232 mmol). The resulting solution was stirred at 23 °C for 1 h, and the solvent was removed under reduced pressure. The residue was taken up in CH_2Cl_2 (20 mL) and neutralized with solid Na_2CO_3 until pH ~9-10. Water (10 mL) was added, and the mixture was extracted with CH_2Cl_2

(3 x 50 mL). The combined organic layers were dried over Na_2SO_4 and concentrated to afford amino amide (2.03 g, 79% yield, R_f = 0.0 in 1:1 hexanes/EtOAc) as a white solid, which was sufficiently pure to be taken on to the next step.

To a solution of amino amide (250 mg, 1.14 mmol) in $PhCH_3$ (5.70 mL) at 23 °C was added benzaldehyde (0.150 mL, 1.48 mmol), TsOH·H_2O (11.0 mg, 0.0578 mmol), and $MgSO_4$ (205 mg, 1.70 mmol). The suspension was heated to reflux overnight. Upon cooling to 23 °C, the solution was quenched with sat. aq. $NaHCO_3$ (10 mL), and the mixture was extracted with EtOAc (3 x 10 mL). The combined organic layers were dried over Na_2SO_4 and concentrated. The dark brown residue was purified by flash chromatography (7:3 → 1:1 hexanes/EtOAc eluent) to afford aminal **179-OMe** (264 mg, 75% yield R_f = 0.22 in 1:1 hexanes/EtOAc) as a light yellow solid.

Aminal 179-OMe: ^{1}H NMR (400 MHz, $CDCl_3$) δ 7.36-7.26 (comp m, 6H), 6.83-6.78 (comp m, 3H), 5.56 (s, 1H), 4.07 (t, J = 6.8 Hz, 1H), 3.74 (s, 3H), 3.42 (dt, J = 9.6, 5.3 Hz, 1H), 2.92-2.86 (m, 1H), 2.23-2.17 (comp m, 2H), 1.93-1.87 (comp m, 2H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.4, 157.0, 139.6, 130.4, 128.9, 128.5, 126.2, 123.5, 114.2, 84.4, 64.4, 56.2, 55.3, 27.6, 24.8; IR (film) 2966, 1513, 1248, 1031, 831, 700 cm[-1]; HRMS (ESI[+]) m/z calc'd for (M+H)[+] $[C_{19}H_{21}N_2O_2]^+$: 309.1598, found 309.1595.

Pyridine 180-OMe: To a solution of freshly distilled diisopropylamine (484 μL, 3.45 mmol) in THF (4.10 mL) at -78 °C was added *n*-BuLi (1.32 mL, 2.5 M in hexanes, 3.31 mmol). The solution was stirred for 10 min at -78 °C, at which time a solution of aminal **179-OMe** (1.02 g, 3.31 mmol) in THF (7.00 mL) was added, and the resulting mixture was stirred for an additional 30 min at -78 °C. To a suspension of NaH (331 mg, 60% dispersion in mineral oil, 8.27 mmol, washed 2 x 1.5 mL with hexanes) in DMF (9.10 mL) at 0 °C was added 2-

(bromomethyl)pyridine hydrobromide (697 mg, 2.76 mmol). The suspension was stirred at 0 ˚C

for 30 min, at which time it was added to the enolate solution at -78 ˚C (flask rinsed with

additional 2.00 mL DMF). The suspension was warmed to 23 ˚C and stirred overnight. The

reaction was quenched slowly with H_2O (30 mL) at 23 ˚C, and the mixture was extracted with

EtOAc (3 x 25 mL). The combined organic layers were washed with brine (2 x 50 mL), dried

over Na_2SO_4 and concentrated. The crude residue was purified by flash chromatography (2:3

hexanes/EtOAc eluent) to afford pyridine **180a-OMe** (663 mg, 60% yield, R_f = 0.38 in 40:1

EtOAc/MeOH) as a beige solid and pyridine **180b-OMe** (278 mg, 25% yield, R_f = 0.22 in 40:1

EtOAc/MeOH) as a yellow oil.

Pyridine 180a-OMe: ^{1}H NMR (400 MHz, $CDCl_3$) δ 8.54 (dt, *J* = 4.4, 1.5 Hz, 1H), 7.51 (td, *J* =

7.7, 1.9 Hz, 1H), 7.17-7.11 (comp m, 7H), 6.83 (dd, *J* = 7.8, 1.7 Hz, 2H), 6.76-6.72 (comp m,

2H), 5.30 (s, 1H), 3.69 (s, 3H), 3.36 (d, *J* = 13.2 Hz, 1H), 3.10-3.03 (comp m, 2H), 2.96 (ddd, *J* =

11.4, 6.5, 5.0 Hz, 1H), 2.30 (ddd, *J* = 13.5, 8.2, 5.5 Hz, 1H), 2.15 (dt, *J* = 13.3, 7.6 Hz, 1H), 1.83-

1.76 (m, 1H), 1.66-1.60 (m, 1H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.5, 158.1, 157.2, 148.9,

140.1, 135.6, 129.7, 128.24, 128.20, 127.3, 125.3, 124.8, 121.4, 114.0, 83.7, 75.0, 55.30, 55.27,

45.6, 35.6, 24.6; IR (film) 2958, 1700, 1589, 1513, 1249, 749, 702 cm[-1]; HRMS (ESI[+]) *m/z*

calc'd for (M+H)[+] $[C_{25}H_{26}N_3O_2]^+$: 400.2020, found 400.2024.

Pyridine 181b-OMe: ^{1}H NMR (400 MHz, $CDCl_3$) δ 8.60 (dd, *J* = 4.9, 0.9 Hz, 1H), 7.52 (td, *J* =

7.7, 1.9 Hz, 1H), 7.26-7.08 (comp m, 6H), 6.89-6.85 (comp m, 2H), 6.73-6.69 (comp m, 2H),

5.30 (s, 1H), 3.71 (s, 3H), 3.49 (d, *J* = 13.1 Hz, 1H), 3.13 (d, *J* = 13.1 Hz, 1H), 2.54 (td, *J* = 9.1,

6.7 Hz, 1H), 2.44-2.37 (comp m, 2H), 2.14-2.05 (m, 1H), 1.68-1.59 (comp m, 2H); ^{13}C NMR

(100 MHz, $CDCl_3$) δ 193.5, 177.4, 157.6, 156.7, 136.1, 134.4, 128.54, 128.47, 128.2, 124.7,

123.7, 121.7, 113.8, 78.8, 75.1, 55.3, 51.1, 46.2, 35.9, 24.6; IR (film) 2961, 1703, 1512, 1248,

1032, 830, 702 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{25}$H$_{26}$N$_3$O$_2$]$^+$: 400.2020, found 400.2016.

Aminal 179-CF$_3$: To a solution of (*S*)-*N*-Boc proline (2.50 g, 11.6 mmol) in CH$_2$Cl$_2$ (33.2 mL) at 0 ˚C was added isobutyl chloroformate (1.67 mL, 12.8 mmol) and Et$_3$N (1.80 mL, 12.8 mmol). After stirring for 20 minutes at 0 ˚C, *p*-trifluoromethylaniline (1.59 mL, 12.8 mmol) was added and the reaction was warmed to 23 ˚C and stirred overnight. The reaction was washed sequentially with aq. KHSO$_4$ (1 M, 50 mL), sat. aq. NaHCO$_3$ (50 mL) and brine (50 mL). The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford a pale brown solid. The crude solid was suspended in hexanes (15 mL), cooled to 0 ˚C and filtered to afford amide (4.44 g, 89% yield, R$_f$ = 0.59 in 1:1 hexanes/EtOAc) as a light beige solid, which was sufficiently pure to be taken on to the next step.

To a solution of amide (3.70 g, 10.3 mmol) in CH$_2$Cl$_2$ (20.7 mL) at 23 ˚C was added TFA (15.9 mL, 207 mmol). The solution was stirred at 23 ˚C for 1 h, at which point the solvent was removed under reduced pressure. The residue was taken up in CH$_2$Cl$_2$ (20 mL) and neutralized with solid Na$_2$CO$_3$ until pH ~9. Water (10 mL) was added, and the mixture was extracted with CH$_2$Cl$_2$ (3 x 50 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated to afford amino amide (2.57 g, 96% yield, R$_f$ = 0.00 in 1:1 hexanes/EtOAc) as a white solid, which was sufficiently pure to be taken on to the next step.

To a solution of amino amide (948 mg, 3.70 mmol) in PhCH$_3$ (18.3 mL) at 23 ˚C was added benzaldehyde (0.482 mL, 4.80 mmol), TsOH·H$_2$O (34.9 mg, 0.184 mmol), and MgSO$_4$ (663 mg, 5.50 mmol). The suspension was heated to reflux overnight. Upon cooling to 23 ˚C, the solution was quenched with sat. aq. NaHCO$_3$ (20 mL). The mixture was extracted with EtOAc (3 x 30 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The dark brown

residue was purified by flash chromatography (4:1 → 7:3 hexanes/EtOAc eluent) to afford

aminal **179-CF$_3$** (1.04 g, 82% yield, R$_f$ = 0.45 in 1:1 hexanes/EtOAc) as a light yellow solid.

Aminal 179-CF$_3$: ^{1}H NMR (400 MHz, CDCl$_3$) δ 7.67 (d, J = 8.8 Hz, 2H), 7.53 (d, J = 8.6 Hz,

2H), 7.40-7.34 (comp m, 3H), 7.33-7.28 (comp m, 2H), 5.74 (s, 1H), 4.03 (t, J = 6.7 Hz, 1H),

3.47 (dt, J = 9.6, 5.0 Hz, 1H), 2.87 (app. q, J = 8.6 Hz, 1H), 2.22 (dt, J = 8.1, 6.6, 2H), 1.95-1.87

(comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.5, 140.8, 138.6, 129.2, 128.8, 126.1 (q, J =

3.8 Hz), 125.8, 120.0, 83.1, 64.2, 56.0, 27.4, 24.8; IR (film) 3034, 2971, 1711, 1615, 1522, 1380,

1327, 1124 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{19}$H$_{19}$F$_3$N$_2$O]$^+$: 347.1366, found

347.1365.

Pyridine 180-CF$_3$: To a solution of freshly distilled diisopropylamine (634 µL, 4.50 mmol) in

THF (4.8 mL) at -78 ˚C was added n-BuLi (1.73 mL, 2.5 M in hexanes, 4.30 mmol) dropwise.

The solution was stirred for 10 min at -78 ˚C, at which time a solution of aminal **179-CF$_3$** (1.50

g, 4.30 mmol) in THF (6.00 mL) was added, and the resulting mixture was stirred for an

additional 30 min at -78 ˚C. To a suspension of NaH (433 mg, 60% dispersion in mineral oil,

10.8 mmol, washed 2 x 1.5 mL with hexanes) in DMF (8.80 mL) at 0 ˚C was added 2-

(bromomethyl)pyridine hydrobromide (913 mg, 3.60 mmol). The suspension was stirred at 0 ˚C

for 30 min, at which time it was added to the enolate solution at -78 ˚C (flask rinsed with

additional 2.00 mL DMF). The suspension was warmed to 23 ˚C and stirred overnight. The

reaction was quenched slowly with H$_2$O (30 mL) at 23 ˚C, and the mixture was extracted with

EtOAc (3 x 30 mL). The combined organic layers were washed with brine (2 x 50 mL), dried

over Na$_2$SO$_4$ and concentrated. The crude residue was purified by flash chromatography (2:3

hexanes/EtOAc eluent) to afford pyridine **180a-CF$_3$** (1.04 g, 66% yield, R$_f$ = 0.58 in 40:1

EtOAc/MeOH) as a beige solid and pyridine **180b-CF₃** (498 mg, 32% yield, R$_f$ = 0.19 in 40:1 EtOAc/MeOH) as a light beige solid.

Pyridine 180a-CF₃ : ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.51 (dd, J = 4.9, 0.8 Hz, 1H), 7.55-7.47 (comp m, 5H), 7.22-7.19 (comp m, 3H), 7.17-7.14 (m, 1H), 7.01 (d, J = 7.8 Hz, 1H), 6.88 (dd, J = 7.6, 1.8 Hz, 2H), 5.49 (s, 1H), 3.19 (ABq, J = 13.2 Hz, Δν = 69.6 Hz, 2H), 3.21 (dt, J = 11.1, 6.8 Hz, 1H), 3.00 (dt, J = 11.4, 5.8 Hz, 1H), 2.31-2.19 (comp m, 2H), 1.84-1.74 (m, 1H), 1.65-1.55 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.4, 157.3, 148.3, 140.12, 140.11, 139.5, 136.2, 128.6, 125.9 (q, J = 3.8 Hz), 125.5, 125.2, 121.7, 121.4, 82.6, 75.1, 56.0, 44.8, 35.6, 24.6; IR (film) 3063, 2967, 1710, 1614, 1326, 1122 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{25}$H$_{22}$F$_3$N$_3$ONa]$^+$: 460.1607, found 460.1614.

Pyridine 181b-CF₃: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.50 (dd, J = 4.9, 0.9 Hz, 1H), 7.52 (td, J = 7.7, 1.8 Hz, 1 H), 7.34 (d, J = 8.6 Hz, 2H), 7.24-7.21 (comp m, 4H), 7.17-7.14 (comp m, 3H), 7.09-7.07 (comp m, 2H), 5.32 (s, 1H), 3.50 (d, J = 13.2 Hz, 1H), 3.12 (d, J = 13.2 Hz, 1H), 2.54 (td, J = 9.0, 6.9 Hz, 1H), 2.46-2.37 (comp m, 2H), 2.16-2.10 (m, 1H), 1.70-1.62 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 177.9, 149.1, 140.7, 136.1, 133.8, 128.8, 128.5, 128.2, 125.6 (q, J = 3.8 Hz), 124.5, 121.8, 121.4, 78.5, 74.9, 51.3, 46.3, 36.0, 24.5; IR (film) 2967, 1713, 1614, 1324, 1166, 1119, 844, 703 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{25}$H$_{23}$F$_3$N$_3$O]$^+$: 438.1788, found 438.1793.
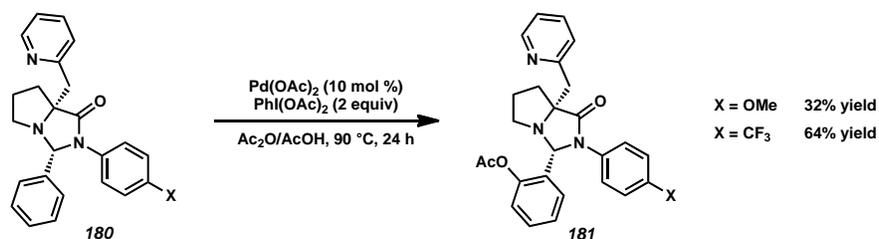
Acetate 181-OMe: Pyridine **180-OMe** (200 mg, 0.501 mmol), Pd(OAc)$_2$ (11.2 mg, 0.0501 mmol), and PhI(OAc)$_2$ (161 mg, 0.501 mmol) were dissolved in AcOH (3.50 mL) and Ac$_2$O (3.50 mL) in a round-bottomed flask. The flask was capped and heated to 95 °C for 8 h, at which time PhI(OAc)$_2$ (161 mg, 0.501 mmol) was added. The reaction was stirred an additional 16 h at 95 °C. Upon cooling the solvent was removed by azeotropic evaporation with heptanes (3 x 15 mL). Water (10 mL) was added, and the mixture was treated with solid Na$_2$CO$_3$ until pH ~9. The mixture was extracted with CH$_2$Cl$_2$ (3 x 15 mL), and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by flash chromatography (17:3 to 4:1 hexanes/acetone eluent) to afford acetate **181-OMe** (73.4 mg, 32% yield, R$_f$ = 0.30 in 1:1 hexanes/acetone) as a light yellow residue.

Acetate 181-OMe: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.47 (dd, J = 4.9, 0.9 Hz, 1H), 7.45 (td, J = 7.7, 1.9 Hz, 1H), 7.38-7.34 (comp m, 2H), 7.26-7.22 (m, 1H), 7.11-7.06 (comp m, 2H), 6.99 (td, J = 7.6, 1.1 Hz, 1H), 6.89 (d, J = 7.8 Hz, 1H), 6.81-6.74 (comp m, 3H), 5.70 (s, 1H), 3.72 (s, 3H), 3.14-3.05 (comp m, 3H), 2.93 (dt, J = 10.5, 6.4 Hz, 1H), 2.34 (s, 3H), 2.27-2.15 (comp m, 2H), 1.75-1.61 (m, 1H), 1.42-1.31 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.2, 168.7, 157.3, 156.8, 148.9, 148.6, 135.6, 132.0, 130.3, 129.0, 126.9, 126.0, 125.7, 122.7, 121.4, 114.1, 77.7, 75.0, 57.0, 55.3, 44.9, 34.8, 24.7, 21.1; IR (film) 2959, 1766, 1513, 1249, 1199, 832 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{27}$H$_{28}$N$_3$O$_4$]$^+$: 458.2074, found 458.2080.

Acetate 181-CF$_3$: Pyridine **180-CF$_3$** (300 mg, 0.686 mmol), Pd(OAc)$_2$ (15.4 mg, 0.0686 mmol), and PhI(OAc)$_2$ (221 mg, 0.686 mmol) were dissolved in AcOH (3.50 mL) and Ac$_2$O (3.50 mL) in a round-bottomed flask. The flask was capped and heated to 95 °C for 12 h, at which time PhI(OAc)$_2$ (221 mg, 0.686 mmol) was added. The reaction was stirred an additional 12 h at 95 °C. Upon cooling, the solvent was removed by azeotropic evaporation with heptane (3 x 15 mL).

Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous mixture was extracted with CH_2Cl_2 (3 x 15 mL), and the combined organic phases were dried over Na_2SO_4 and concentrated. The crude residue was purified by flash chromatography (17:3 → 4:1 hexanes/acetone eluent) to afford acetate **181-CF₃** (218 mg, 64% yield, R_f = 0.53 in 1:1 hexanes/acetone) as a light yellow residue.

Acetate 181-CF₃: ^{1}H NMR (400 MHz, CDCl₃) δ 8.46 (d, *J* = 4.6 Hz, 1H), 7.69 (d, *J* = 8.8 Hz, 2H), 7.53 (d, *J* = 8.8 Hz, 2H), 7.47 (td, *J* = 7.7, 1.8 Hz, 1H), 7.30 (td, *J* = 8.4, 1.6 Hz, 1H), 7.17 (d, *J* = 7.3 Hz, 1H), 7.10 (dd, *J* = 6.9, 5.4 Hz, 1H), 7.01 (t, *J* = 7.5 Hz, 1H), 6.80 (d, *J* = 7.8 Hz, 1H), 6.71 (dd, *J* = 7.8, 1.3 Hz, 1H), 5.80 (s, 1H), 3.21 (dt, *J* = 10.2, 6.1 Hz, 1H), 3.08 (ABq, *J* = 13.2 Hz, Δν =12.2 Hz, 2H), 2.95 (dt, *J* = 10.2, 6.7 Hz, 1H), 2.39 (s, 3H), 2.26-2.19 (comp m, 2H), 1.71-1.64 (m, 1H), 1.38-1.31 (m, 1H); ^{13}C NMR (100 MHz, CDCl₃) δ 176.3, 168.7, 156.9, 148.9, 148.2, 140.3, 136.0, 131.5, 129.3, 128.6, 126.22, 126.16, 126.1 (q, *J* = 3.8 Hz), 125.9, 123.1, 121.6, 119.9, 77.6, 75.0, 57.6, 44.3, 34.8, 24.7, 21.1; IR (film) 2968, 1768, 1712, 1379, 1326, 1199, 843, 736 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [$C_{27}H_{25}F_3N_3O_3$]$^+$ 496.1843, found 496.1843.
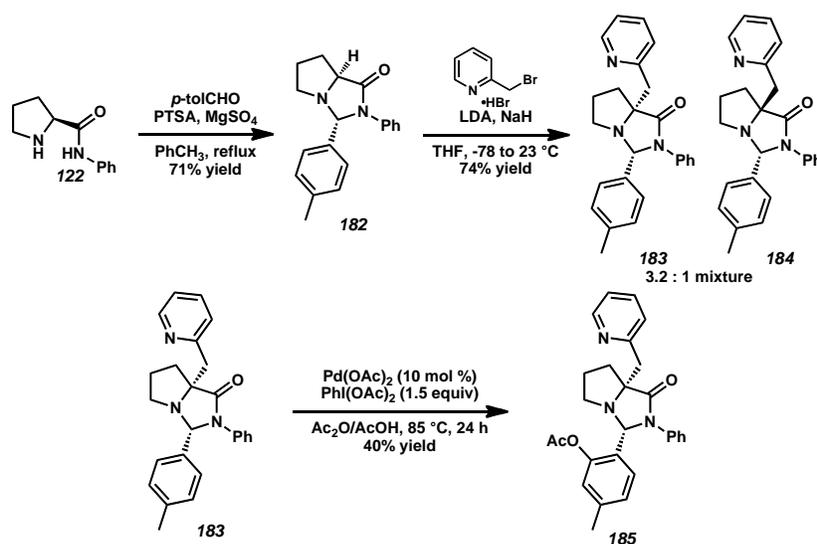
To a solution of amide **122** (684 mg, 3.60 mmol) in PhCH$_3$ (17.9 mL) at 23 °C was added *p*-tolualdehyde (0.553 mL, 4.67 mmol), TsOH·H$_2$O (34.2 mg, 0.180 mmol) and MgSO$_4$ (562 mg, 4.67 mmol). The mixture was heated to reflux and stirred 16 h. Upon cooling to 23 °C, the solution was quenched with sat. aq. NaHCO$_3$ (15 mL). The mixture was extracted with EtOAc (3 x 25 mL), and the combined organic layers were dried over Na$_2$SO$_4$, filtered and concentrated. The crude product was purified via flash chromatography (7:3 → 3:2 hexanes/EtOAc eluent) to afford aminal **182** (764 mg, 73% yield, R$_f$ = 0.34 in 1:1 hexanes/EtOAc) as a beige solid.

Aminal 182: ^{1}H NMR (400 MHz, CDCl$_3$) δ 7.49 (dd, *J* = 8.7, 1.0 Hz, 2H), 7.31-7.26 (comp m, 2H), 7.21-7.14 (comp m, 4H), 7.11-7.07 (m, 1H), 5.65 (s, 1H), 4.03 (t, *J* = 6.7 Hz, 1H), 3.42 (dt, *J* = 9.7, 5.1 Hz, 1H), 2.86 (app. q, *J* = 8.5 Hz, 1H), 2.31 (s, 3H), 2.20 (app. q, *J* = 7.3 Hz, 2H), 1.92-1.85 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.0, 138.3, 137.8, 136.5, 129.7, 128.9, 125.9, 124.9, 121.0, 83.5, 64.3, 55.9, 27.4, 24.8, 21.1; IR (film) 2967, 1701, 1498, 1382, 757, 691 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{19}$H$_{21}$N$_2$O]$^+$: 293.1648, found 293.1650.

To a solution of freshly distilled diisopropylamine (354 μL, 2.52 mmol) in THF (4.00 mL) at -78 °C was added *n*-BuLi (0.970 mL, 2.5 M in hexanes, 2.42 mmol). The solution was stirred for 10 min at -78 °C, at which time a solution of aminal **182** (707 mg, 2.42 mmol) in THF (4.10 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 °C. To a suspension of NaH (242 mg, 60% dispersion in mineral oil, 6.05 mmol, washed with 2 x 1.5 mL hexanes) in DMF (7.00 mL) at 0 °C was added 2-(bromomethyl)pyridine hydrobromide (510 mg, 2.02 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.10 mL DMF). The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched slowly with H$_2$O (30 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 25 mL). The combined organic

layers were washed with brine (2 x 50 mL), dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (2:3 hexanes/EtOAc eluent) to afford pyridine **183** (545 mg, 71% yield, R_f = 0.40 in 40:1 EtOAc/MeOH) as a beige solid and pyridine **184** (210 mg, 27% yield, R_f = 0.09 in 40:1 EtOAc/MeOH) as a light beige solid.

Pyridine 183: ^{1}H NMR (400 MHz, $CDCl_3$) δ 8.54 (dd, J = 4.9, 0.9 Hz, 1H), 7.51 (td, J = 7.7, 1.8 Hz, 1H), 7.34 (dd, J = 8.6, 1.0 Hz, 2H), 7.23 (t, J = 8.0 Hz, 2H), 7.15-7.12 (m, 1H), 7.10-7.04 (comp m, 2H), 6.97 (d, J = 7.9 Hz, 2H), 6.79 (d, J = 8.0 Hz, 2H), 5.40 (s, 1H), 3.19 (ABq, J = 13.2 Hz, $\Delta\nu$ = 88.2 Hz, 2H), 3.13-3.07 (m, 1H), 2.96 (dt, J = 11.3, 5.8 Hz, 1H), 2.32-2.26 (m, 1H), 2.26 (s, 3H), 2.17 (dt, J = 13.7, 7.0 Hz, 1H), 1.83-1.72 (m, 1H), 1.62-1.52 (m, 1H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 174.9, 158.0, 148.8, 137.9, 137.2, 137.0, 135.5, 129.0, 128.7, 127.0, 125.4, 125.3, 122.6, 121.3, 82.9, 75.0, 55.6, 45.4, 35.4, 24.6, 21.1; IR (film) 3048, 2965, 2878, 1701, 1592, 1499, 1387, 754, 693 cm^{-1}; HRMS (ESI[+]) m/z calc'd for (M+H)[+] $[C_{25}H_{26}N_3O]^+$: 384.2070, found 384.2068.

Pyridine 184: ^{1}H NMR (400 MHz, $CDCl_3$) δ 8.57 (dd, J = 4.8, 0.9 Hz, 1H), 7.50 (td, J = 7.7, 1.8 Hz, 1H), 7.23 (d, J = 7.8 Hz, 1H), 7.19-7.15 (comp m, 3H), 7.04-6.96 (comp m, 7H), 5.31 (s, 1H), 3.48 (d, J = 13.1 Hz, 1H), 3.12 (d, J = 13.1 Hz, 1H), 2.55 (dt, J = 9.6, 7.6 Hz, 1H), 2.44-2.37 (comp m, 2H), 2.25 (s, 3H), 2.10 (dt, J = 12.6, 6.1 Hz, 1H), 1.66-1.59 (comp m, 2H); ^{13}C NMR (100 MHz, $CDCl_3$) δ 177.5, 157.6, 149.1, 138.3, 137.5, 136.0, 131.3, 128.9, 128.4, 128.3, 124.8, 124.6, 122.2, 121.7, 78.4, 75.0, 51.1, 46.2, 35.8, 24.5, 21.1; IR (film) 3047, 2967, 2870, 1707, 1597, 1501, 1379, 734, 694 cm^{-1}; HRMS (ESI[+]) m/z calc'd for (M+Na)[+] $[C_{25}H_{25}N_3ONa]^+$: 406.1890, found 406.1895.

Acetate 185: Pyridine **183** (100 mg, 0.261 mmol), $Pd(OAc)_2$ (5.9 mg, 0.0261 mmol), and $PhI(OAc)_2$ (168 mg, 0.522 mmol) were dissolved in AcOH (1.30 mL) and Ac_2O (1.30 mL) in a

round-bottomed flask. The flask was capped and heated to 90 ˚C, and stirred for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptane (3 x 15 mL). Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous mixture was extracted with CH_2Cl_2 (3 x 15 mL), and the combined organic phases were dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (17:3 → 4:1 hexanes/acetone eluent) to afford acetate **185** (51.3 mg, 45% yield, R_f = 0.44 in 1:1 hexanes/acetone) as a light yellow residue.

Acetate 185: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.48 (dd, *J* = 4.1, 0.7 Hz, 1H), 7.54-7.44 (comp m, 3H), 7.29-7.23 (comp m, 2H), 7.10-7.04 (comp m, 2H), 6.93-6.89 (comp m, 2H), 6.81 (d, *J* = 8.1 Hz, 1H), 6.69 (d, *J* = 7.9 Hz, 1H), 5.74 (s, 1H), 3.12 (dt, *J* = 10.3, 6.2 Hz, 1H), 3.08 (s, 2H), 2.92 (dt, *J* = 10.3, 6.6 Hz, 1H), 2.36 (s, 3H), 2.29 (s, 3H), 2.20 (app. t, *J* = 7.2 Hz, 2H), 1.69-1.62 (m, 1H), 1.34-1.27 (m, 1H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.8, 168.8, 157.5, 148.7, 148.5, 139.3, 137.4, 135.5, 128.93, 128.85, 126.9, 126.5, 125.8, 124.9, 123.3, 121.3, 120.7, 98.5, 77.6, 74.9, 57.2, 44.7, 34.6, 24.7, 21.1 21.0; IR (film) 2925, 1767, 1703, 1383, 1200, 753, 692 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{27}$H$_{28}$N$_3$O$_3$]$^+$: 442.2125, found 442.2130.

178

186
X = OMe   60% yield
X = CF$_3$ 92% yield

187
X = OMe   88% yield   2.1 : 1 mixture
X = CF$_3$ 85% yield   2.1 : 1 mixture

188

Pd(OAc)$_2$ (10 mol %)
PhI(OAc)$_2$ (1.5 equiv)

Ac$_2$O/AcOH, 85 °C, 24 h

189
X = OMe   21% yield
X = CF$_3$ 50% yield
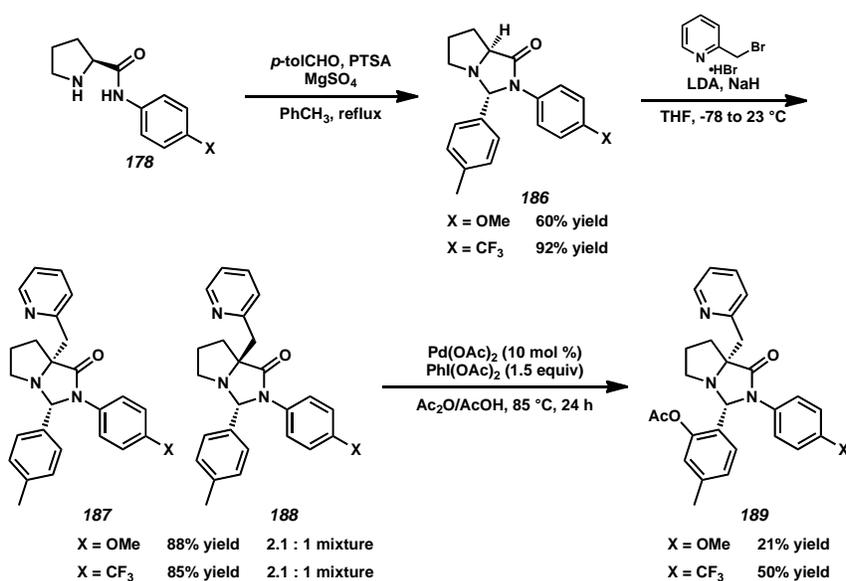
Aminal 186-OMe: To a solution of amide **178** (500 mg, 2.27 mmol) in $PhCH_3$ (11.3 mL) at 23 °C was added *p*-tolualdehyde (0.348 mL, 2.95 mmol), $TsOH \cdot H_2O$ (21.6 mg, 0.114 mmol) and $MgSO_4$ (410 mg, 3.40 mmol). The mixture was heated to reflux and stirred 16 h. Upon cooling to 23 °C, the solution was quenched with sat. aq. $NaHCO_3$ (15 mL). The mixture was extracted with EtOAc (3 x 25 mL), and the combined organic layers were dried over Na_2SO_4, filtered and concentrated. The crude product was purified via flash chromatography (7:3 → 3:2 hexanes/EtOAc eluent) to afford aminal **186-OMe** (439 mg, 60% yield, R_f = 0.30 in 1:1 hexanes/EtOAc 1 % Et_3N) as a beige solid.

Pyridine 187: To a solution of freshly distilled diisopropylamine (199 μL, 1.42 mmol) in THF (2.8 mL) at -78 °C was added *n*-BuLi (0.540 mL, 2.5 M in hexanes, 1.36 mmol). The solution was stirred for 10 min at -78 °C, at which time a solution of aminal **186-OMe** (439 mg, 1.36 mmol) in THF (4.00 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 °C. To a suspension of NaH (136 mg, 60% dispersion in mineral oil, 3.40 mmol, washed with 2 x 1.0 mL hexanes) in DMF (5.80 mL) at 0 °C was added 2-(bromomethyl)pyridine hydrobromide (287 mg, 1.13 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.00 mL DMF). The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched slowly with H_2O (30 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 25 mL). The combined organic layers were washed with brine (2 x 50 mL), dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (2:3 hexanes/EtOAc eluent) to afford pyridine **187-OMe** (206 mg, 54% yield, R_f = 0.40 in 40:1 EtOAc/MeOH) as a beige solid and pyridine **188-OMe** (90.4 mg, 31% yield, R_f = 0.09 in 40:1 EtOAc/MeOH) as a light beige solid.

Acetate 189-OMe: Pyridine **187-OMe** (15.0 mg, 0.0363 mmol), $Pd(OAc)_2$ (0.8 mg, 3.63 μmol), and $PhI(OAc)_2$ (17.5 mg, 0.0544 mmol) were dissolved in AcOH (0.180 mL) and Ac_2O (0.180 mL) in a round-bottomed flask. The flask was capped and heated to 90 ˚C, and stirred for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptane (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous mixture was extracted with CH_2Cl_2 (3 x 10 mL), and the combined organic phases were dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (17:3 → 4:1 hexanes/acetone eluent) to afford acetate **189-OMe** (3.8 mg, 21% yield, $R_f = 0.40$ in 1:1 hexanes/acetone) as a light yellow residue.

Aminal 186-CF₃: To a solution of amide **178** (500 mg, 1.93 mmol) in $PhCH_3$ (9.70 mL) at 23 ˚C was added p-tolualdehyde (0.297 mL, 2.52 mmol), $TsOH \cdot H_2O$ (18.4 mg, 0.0968 mmol) and $MgSO_4$ (350 mg, 2.90 mmol). The mixture was heated to reflux and stirred 16 h. Upon cooling to 23 ˚C, the solution was quenched with sat. aq. $NaHCO_3$ (15 mL). The mixture was extracted with EtOAc (3 x 25 mL), and the combined organic layers were dried over Na_2SO_4, filtered and concentrated. The crude product was purified via flash chromatography (7:3 to 3:2 hexanes/EtOAc eluent) to afford aminal **186-CF₃** (645 mg, 92% yield, $R_f = 0.40$ in 1:1 hexanes/EtOAc 1 % Et_3N) as a beige solid.

Pyridine 187-CF₃: To a solution of freshly distilled diisopropylamine (262 μL, 1.86 mmol) in THF (4.00 mL) at -78 ˚C was added n-BuLi (0.720 mL, 2.5 M in hexanes, 1.79 mmol). The solution was stirred for 10 min at -78 ˚C, at which time a solution of aminal **186-CF₃** (645 mg, 1.79 mmol) in THF (5.00 mL) was added, and the resulting solution was stirred for an additional 30 min at -78 ˚C. To a suspension of NaH (179 mg, 60% dispersion in mineral oil, 4.47 mmol, washed with 2 x 1.0 mL hexanes) in DMF (8.00 mL) at 0 ˚C was added 2-

(bromomethyl)pyridine hydrobromide (377 mg, 1.49 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.00 mL DMF). The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched slowly with H_2O (30 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 25 mL). The combined organic layers were washed with brine (2 x 50 mL), dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (2:3 hexanes/EtOAc eluent) to afford pyridine **187-CF₃**(387 mg, 70% yield, R_f = 0.45 in 40:1 EtOAc/MeOH) as a beige solid and pyridine **188-CF₃** (183 mg, 15% yield, R_f = 0.15 in 40:1 EtOAc/MeOH) as a light beige solid.

Acetate 189-OMe: Pyridine **187-CF₃** (15.0 mg, 0.0332 mmol), $Pd(OAc)_2$ (0.7 mg, 3.32 μmol), and $PhI(OAc)_2$ (16.1 mg, 0.0498 mmol) were dissolved in AcOH (0.170 mL) and Ac_2O (0.170 mL) in a round-bottomed flask. The flask was capped and heated to 90 °C, and stirred for 24 h. Upon cooling, the solvent was removed by azeotropic evaporation with heptane (3 x 5 mL). Water (10 mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9. The aqueous mixture was extracted with CH_2Cl_2 (3 x 10 mL), and the combined organic phases were dried over Na_2SO_4, filtered and concentrated. The crude residue was purified by flash chromatography (17:3 to 4:1 hexanes/acetone eluent) to afford acetate **187-CF₃** (8.8 mg, 50% yield, R_f = 0.45 in 1:1 hexanes/acetone) as a light yellow residue.
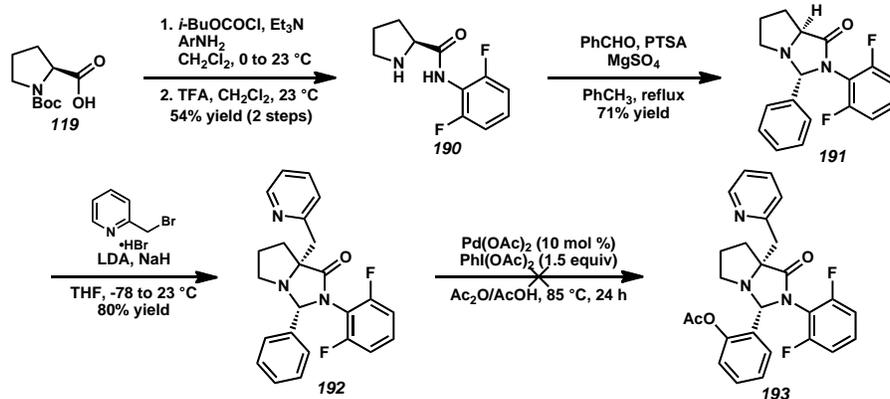
Aminal 191: To a solution of (*S*)-*N*-Boc proline (500 mg, 2.32 mmol) in CH$_2$Cl$_2$ (15.5 mL) at 0 °C was added isobutyl chloroformate (0.334 mL, 2.56 mmol) and Et$_3$N (0.359 mL, 2.56 mmol). After stirring for 20 minutes at 0 °C, 2,6-difluoroaniline (0.275 mL, 2.56 mmol) was added and the reaction was warmed to 23 °C and stirred overnight. The reaction was washed sequentially with aq. KHSO$_4$ (1 M, 20 mL), sat. aq. NaHCO$_3$ (20 mL) and brine (20 mL). The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford a pale brown solid. The crude solid was suspended in hexanes (15 mL), cooled to 0 °C and filtered to afford amide (736 g, 97% yield, R$_f$ = 0.59 in 1:1 hexanes/EtOAc) as a white solid, which was sufficiently pure to be taken on to the next step.

To a solution of amide (736 g, 2.26 mmol) in CH$_2$Cl$_2$ (4.50 mL) at 23 °C was added TFA (3.50 mL, 45.1 mmol). The solution was stirred at 23 °C for 1 h, at which point the solvent was removed under reduced pressure. The residue was taken up in CH$_2$Cl$_2$ (20 mL) and neutralized with solid Na$_2$CO$_3$ until pH ~9. Water (10 mL) was added, and the mixture was extracted with CH$_2$Cl$_2$ (3 x 20 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated to afford amino amide **190** (284 mg, 56% yield, R$_f$ = 0.00 in 1:1 hexanes/EtOAc) as a white solid, which was sufficiently pure to be taken on to the next step.

To a solution of amino amide **190** (283 mg, 1.25 mmol) in PhCH$_3$ (8.30 mL) at 23 ˚C was added benzaldehyde (0.164 mL, 1.63 mmol), TsOH·H$_2$O (11.9 mg, 0.0625 mmol), and MgSO$_4$ (226 mg, 1.88 mmol). The suspension was heated to reflux overnight. Upon cooling to 23 ˚C, the solution was quenched with sat. aq. NaHCO$_3$ (20 mL). The mixture was extracted with EtOAc (3 x 20 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The dark brown residue was purified by flash chromatography (4:1 to 7:3 hexanes/EtOAc eluent) to afford aminal **191** (304 mg, 77% yield, R$_f$ = 0.45 in 1:1 hexanes/EtOAc) as a light yellow solid.

Pyridine 192: To a solution of freshly distilled diisopropylamine (233 µL, 1.66 mmol) in THF (2.00 mL) at -78 ˚C was added *n*-BuLi (0.640 mL, 2.5 M in hexanes, 1.59 mmol) dropwise. The solution was stirred for 10 min at -78 ˚C, at which time a solution of aminal **191** (500 g, 1.59 mmol) in THF (3.30 mL) was added, and the resulting mixture was stirred for an additional 30 min at -78 ˚C. To a suspension of NaH (160 mg, 60% dispersion in mineral oil, 3.98 mmol, washed 2 x 1.5 mL with hexanes) in DMF (4.30 mL) at 0 ˚C was added 2-(bromomethyl)pyridine hydrobromide (335 mg, 1.33 mmol). The suspension was stirred at 0 ˚C for 30 min, at which time it was added to the enolate solution at -78 ˚C (flask rinsed with additional 1.00 mL DMF). The suspension was warmed to 23 ˚C and stirred overnight. The reaction was quenched slowly with H$_2$O (20 mL) at 23 ˚C, and the mixture was extracted with EtOAc (3 x 20 mL). The combined organic layers were washed with brine (2 x 30 mL), dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by flash chromatography (2:3 hexanes/EtOAc eluent) to afford pyridine **192** (428 g, 80% yield, R$_f$ = 0.58 in 40:1 EtOAc/MeOH) as a light yellow amorphous solid.

Acetate 193: Pyridine **192** (25.0 mg, 0.0617 mmol), Pd(OAc)$_2$ (1.4 mg, 6.17 µmol), and PhI(OAc)$_2$ (29.8 mg, 0.0925 mmol) were dissolved in AcOH (0.310 mL) and Ac$_2$O (0.310 mL)

in a round-bottomed flask. The flask was capped and heated to 90 ˚C, and stirred for 24 h. Upon

cooling, the solvent was removed by azeotropic evaporation with heptane (3 x 5 mL). Water (10

mL) was added, and the mixture was treated with solid Na_2CO_3 until pH ~9.   The aqueous

mixture was extracted with CH_2Cl_2 (3 x 10 mL), and the combined organic phases were dried

over Na_2SO_4, filtered and concentrated. No acetoxylated product was observed by crude [1]H

NMR.
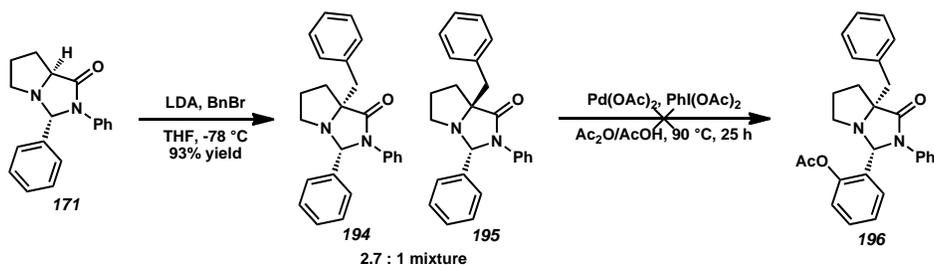
Aminal 194: To a solution of freshly distilled diisopropylamine (146 µL, 1.04 mmol) in THF

(4.00 mL) at -78 ˚C was added *n*-BuLi (0.400 mL, 2.5 M in hexanes, 1.01 mmol). The solution

was stirred for 10 min at -78 ˚C, at which time **171** (250 mg, 0.720 mmol) in THF (3.20 mL) was

added, and the resulting solution was stirred for an additional 30 min at -78 ˚C. Benzyl bromide

(225 µL, 1.44 mmol) was added at -78 ˚C, and the reaction was warmed to 23 ˚C, and stirred

overnight. The reaction was quenched with water (10 mL). The aqueous was extracted with

EtOAc (3 x 10 mL). The combined organics were washed with brine, dried over Na_2SO_4,

filtered and concentrated.   The crude product was purified by flash chromatography (7:1

hexanes:EtOAc to 4:1 hexanes:EtOAc) to afford **194:195** as a 2.7 : 1 ratio of inseparable

diastereomers (225 mg, 85% yield, R_f = 0.29 in 4:1 hexanes:EtOAc) as a white amorphous solid.

Aminal 194: ^{1}H NMR (400 MHz, CDCl$_3$) (see NMR spectrum for details); ^{13}C NMR (100 MHz,

CDCl$_3$) δ 178.0, 174.9, 140.1, 137.5, 137.4, 137.1, 136.7, 134.5, 131.1, 130.4, 128.7, 128.4,

128.38, 128.24, 128.17, 128.1, 127.8, 127.3, 126.7, 126.3, 125.4, 125.0, 122.9, 122.6, 83.4, 78.6,

75.7, 75.4, 55.1, 51.1, 44.0, 43.2, 36.1, 35.7, 24.7, 24.6; IR (film) 3031, 2965, 2877, 1704, 1599, 1497, 1385, 700 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{25}$H$_{25}$N$_2$O]$^+$: 369.1961, found 369.1968.

time (h)	temp (°C)	131 : 132 (%)	dr
24	55	41 : 13	10 : 1
13	70	48 : 19	5.9 : 1
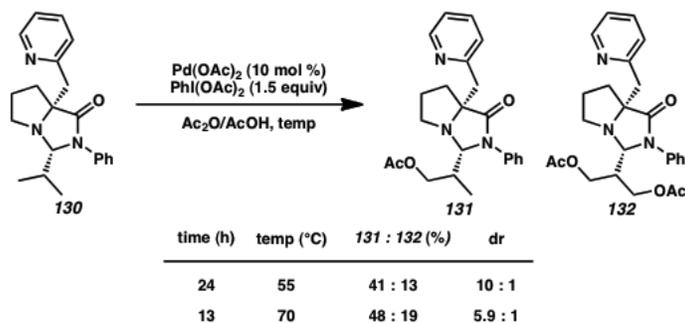
Pyridine **172** (10.0 mg, 0.0271 mmol) and Pd(OAc)$_2$ (6.1 mg, 0.0271 mmol) were dissolved in AcOH (0.270 mL) in a scintillation vial. The vial was sealed and heated to 85 °C for 1 h. The reaction was cooled to 23 °C, and the organic solvent was removed azeotropically with heptane (3 x 5 mL) to afford palladacycle **197** (17.9 mg, 99% yield) as a light brown solid. The solid was crystallized by a layering technique with CH$_2$Cl$_2$ and hexanes.

Palladacycle 197: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.93 (dd, J = 5.5, 1.4 Hz, 1H), 7.85 (td, J = 7.7, 1.4 Hz, 1H), 7.43-7.37 (comp m, 5H), 7.09-6.96 (comp m, 4H), 6.71 (t, J = 7.3 Hz, 1H), 6.17 (d, J = 7.5 Hz, 1H), 5.43 (s, 1H), 4.51 (d, J = 14.5 Hz, 1H), 3.60 (d, J = 14.5 Hz, 1H), 3.55 (dd, J = 13.0, 5.9 Hz, 1H), 2.83-2.75 (m, 1H), 2.61 (dd, J = 12.2, 7.6 Hz, 1H), 2.04 (s, 3H), 1.98-1.84 (m, 1H), 1.82-1.71 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 170.9, 154.3, 151.7, 146.7, 146.3, 138.5, 134.0, 133.3, 129.4, 128.9, 128.7, 127.4, 126.5, 124.3, 124.2, 123.7, 93.3, 75.8, 62.2, 46.7, 33.6, 24.6; IR (film) 3051, 2970, 1712, 1598, 1402, 730, 702 cm^{-1}; mp 250 °C dec.
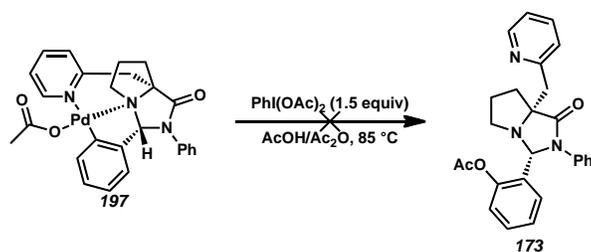
Acetate 173: To palladacycle **197** (14.5 mg, 0.0271 mmol) in AcOH/Ac$_2$O (1:1, 0.270 mL) was added PhI(OAc)$_2$ (13.1 mg, 0.0406 mmol) and heated to 85 °C overnight. Upon cooling to 23 °C, dppe (21.6 mg, 0.0541 mmol) was added and the mixture stirred overnight at 23 °C. The solvent was removed via azeotropic removal with heptanes (3 x 10 mL). No acetoxylated product was observed by crude ^{1}H NMR.
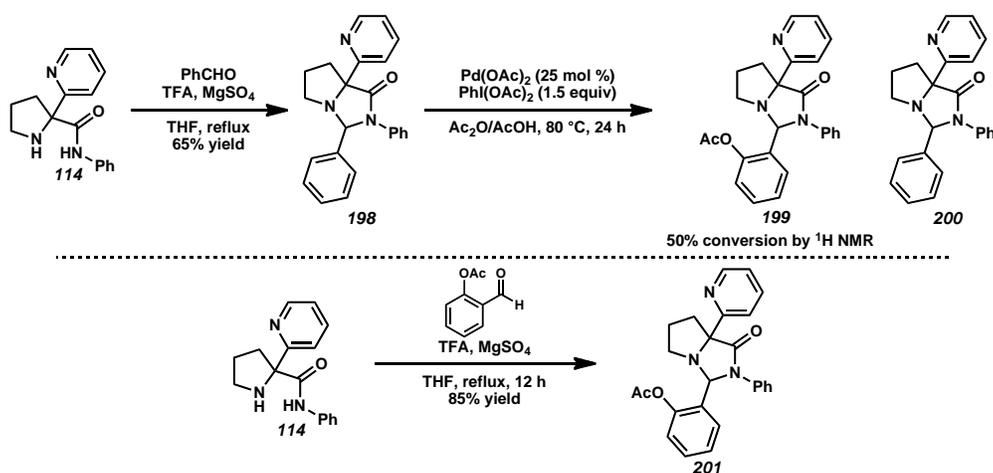
Aminal 198: To a solution of amino amide **114** (100 mg, 0.374 mmol) in THF (3.70 mL) at 23 °C was added benzaldehyde (49.2 µL, 0.486 mmol), TFA (5.8 µL, 0.0748 mmol) and MgSO$_4$ (67.5 mg, 0.561 mmol). The suspension was heated to 75 °C for 12 h. Upon cooling to 23 °C, the solution was quenched with sat. aq. NaHCO$_3$ (10 mL), and the mixture was extracted with EtOAc (3 x 10 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by column chromatography (7:3 to 1:1 hexanes/EtOAc eluent) to afford aminal **198** (85.8 mg, 65% yield, R$_f$ = 0.10 in 1:1 hexanes:EtOAc) as a beige solid.

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Acetate 199: Pyridine **198** (10.0 mg, 0.0281 mmol), Pd(OAc)$_2$ (1.6mg, 7.03 μmol), and PhI(OAc)$_2$ (13.6 mg, 0.0422 mmol) were dissolved in AcOH (0.140 mL) and Ac$_2$O (0.140 mL) in a 2-dram vial. The vial was sealed with a Teflon cap and heated to 80 °C for 24 h. Upon cooling, CH$_2$Cl$_2$ (10 mL) and water (10 mL) were added and the mixture was neutralized with Na$_2$CO$_3$ until pH ~9. The mixture was extracted with CH$_2$Cl$_2$ (2 x 15 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. To the crude mixture was added heptane (10 mL) and concentrated to ensure removal of residual Ac$_2$O. The crude ^{1}H NMR showed 50% conversion to acetate **199**.

Aminal 201: To a solution of amino amide **114** (50 mg, 0.187 mmol) in THF (1.87 mL) at 23 °C was added acetylated salicylaldehyde (39.9 mg, 0.243 mmol), TFA (1.4 μL, 0.0187 mmol) and MgSO$_4$ (33.8 mg, 0.281 mmol). The suspension was heated to 75 °C for 12 h. Upon cooling to 23 °C, the solution was quenched with sat. aq. NaHCO$_3$ (10 mL), and the mixture was extracted with EtOAc (3 x 10 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by column chromatography (7:3 to 1:1 hexanes/EtOAc eluent) to afford acetate **201** (65.6 mg, 85% yield, R$_f$ = 0.50 in EtOAc) as a beige solid.

Aminal 201: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.60 (dd, *J* = 4.1, 0.8 Hz, 1H), 7.69-7.62 (comp m, 2H), 7.33-7.17 (comp m, 6H), 7.13 (d, *J* = 8.0 Hz, 1H), 7.07-7.02 (comp m, 3H), 6.54 (s, 1H), 2.83 (dt, *J* = 13.5, 8.9 Hz, 1H), 2.74 (td, *J* = 9.5, 6.7 Hz, 1H), 2.59-2.55 (m, 1H), 2.42 (ddd, *J* = 12.9, 8.4, 4.0 Hz, 1H), 2.35 (s, 3H), 1.90-1.79 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.9, 169.4, 160.3, 149.7, 149.3, 137.8, 136.4, 129.6, 129.0, 128.5, 126.9, 125.6, 124.7, 123.3, 122.4, 121.8, 120.9, 77.9, 72.6, 50.1, 36.0, 24.7, 20.8; IR (film) 2967, 1766, 1711, 1587, 1369,

1198, 753 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{25}$H$_{23}$N$_3$O$_3$Na]$^+$: 436.1632, found 436.1632.

acid	solvent	temp (°C), time (h)	a : b
TFA (10 mol %)	THF	72, 18 h	0 : 1.0
PTSA (5 mol %)	PhCH$_3$	110, 24 h	0 : 1.0
AcOH (1 M)	PhCH$_3$	110 °C, 24 h	4.3 : 1.0
AcOH (0.5 M)	PhCH$_3$	110 °C, 24 h	6.3 : 1.0
AcOH (0.1 M)	PhCH$_3$	110 °C, 24 h	6.0 : 1.0
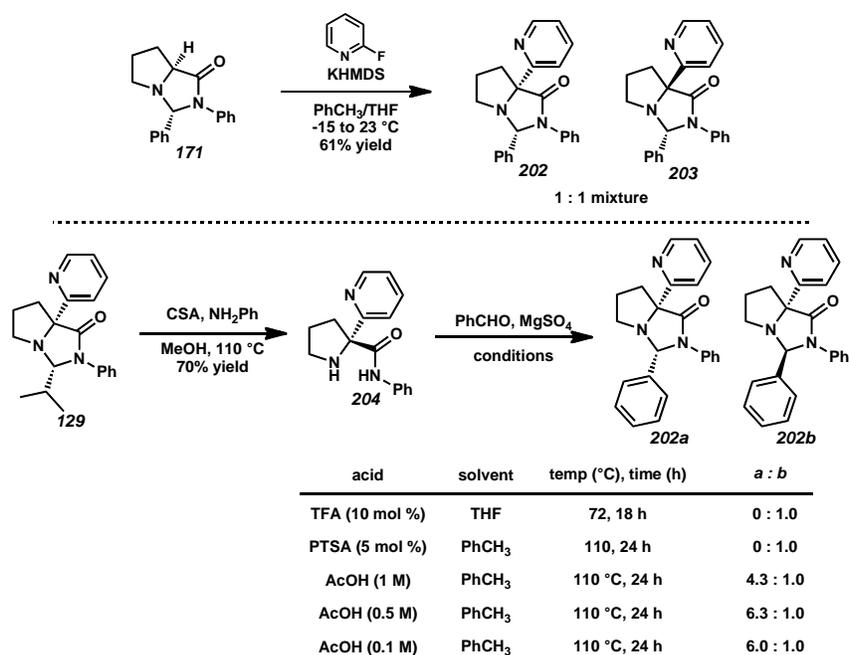
Pyridine 202: To a solution of aminal **171** (585 mg, 2.10 mmol), 2-fluoropyridine (181 µL, 2.10 mmol) in PhCH$_3$ (7.01 mL) at -15 °C was added KHMDS (419 mg, 2.10 mmol) in THF (4.20 mL) slowly over 1 h. Upon completion of addition, the reaction was allowed to warm to 23 °C and stirred overnight. The reaction was filtered over a pad of silica (5 x 5 cm, 100 mL EtOAC eluent) and concentrated. The crude product was purified by flash chromatography (3:1 to 1:1 hexanes/EtOAc eluent) to afford a 1 to 1 mixture of diastereomers of pyridine **202** (459 mg, 61% yield, R$_f$ = 0.10 in 1:1 hexanes/EtOAc) as a light beige solid.

Pyridine 202: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.61 (ddd, J = 4.8, 1.7, 0.9 Hz, 1H), 7.53-7.48 (comp m, 4H), 7.23 (d, J = 8.6 Hz, 2H), 7.20-7.15 (comp m, 5H), 7.10-7.04 (comp m, 2H), 5.68 (s, 1H), 3.48 (dt, J = 10.6, 6.4 Hz, 1H), 3.20 (dt, J = 10.6, 6.6 Hz, 1H), 2.70 (dt, J = 13.2, 7.6 Hz, 1H), 2.57 (ddd, J = 13.4, 7.4, 6.2 Hz, 1H), 2.04-1.91 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.9, 161.1, 149.2, 139.74, 139.69, 137.2, 136.0, 128.8, 128.3, 128.2, 127.0, 125.2,

122.1, 121.9, 121.0, 83.2, 77.9, 56.7, 37.6, 25.2; IR (film) 2966, 1704, 1587, 1495, 1382, 1299, 753, 692 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{23}$H$_{21}$N$_3$ONa]$^+$: 378.1577, found 378.1574.

Amino Amide 204: To pyridine **129** (750 mg, 2.33 mmol) in a screw cap vial with Teflon cap was added CSA (542 mg, 2.33 mmol), NH$_2$Ph (106 μL, 1.17 mmol) and MeOH (4.66 mL). The reaction was heated to 110 °C for 24 h. Upon cooling, the reaction mixture was concentrated. To the residue was added sat. aq. NaHCO$_3$ (20 mL). The mixture was extracted with EtOAc (3 x 15 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by column chromatography (3:1 to 3:2 hexanes/EtOAc eluent) to afford amino amide **204** (874 mg, 70% yield (97% yield borsm) R$_f$ = 0.05 in 1:1 hexanes/EtOAc) as a beige solid.

Amide 204: ^{1}H NMR (400 MHz, CDCl$_3$) δ 10.45 (bs, 1H), 8.47 (ddd, J = 4.8, 1.9, 1.0 Hz, 1H), 7.89 (dt, J = 8.1, 1.0 Hz, 1H), 7.69 (td, J = 7.8, 1.8 Hz, 1H), 7.61-7.58 (comp m, 2H), 7.31-7.26 (comp m, 2H), 7.19 (ddd, J = 7.4, 4.9, 1.1 Hz, 1H), 7.08-7.03 (m, 1H), 4.29 (bs, 1H), 3.19 (dt, J = 10.3, 7.0 Hz, 1H), 3.11 (ddd, J = 10.3, 6.9, 5.6 Hz, 1H), 2.81 (ddd, J = 12.6, 6.9, 5.6 Hz, 1H), 2.11 (dt, J = 12.6, 7.9 Hz, 1H), 1.91-1.72 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 190.8, 173.2, 160.3, 147.4, 138.0, 136.7, 128.8, 123.8, 122.5, 122.4, 119.1, 74.2, 47.1, 39.1, 26.9; IR (film) 3262, 2968, 2869, 1682, 1601, 1516, 1441, 1312, 751, 692 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{16}$H$_{18}$N$_3$O]$^+$: 268.1444, found 268.1445.

General procedure for condensation reaction: To amide **204** (1 equiv) in PhCH$_3$ or THF (0.1 M) was added benzaldehyde (1.3 equiv), MgSO$_4$ (1.5 equiv) and acid (amount indicated) and refluxed overnight. Upon cooling, the reaction was quenched with sat. aq. NaHCO$_3$ (10 mL),

and the mixture was extracted with EtOAc (3 x 10 mL). The combined organic layers were dried over Na_2SO_4 and concentrated. Product ratios were determined by crude ^{1}H NMR.

ArCHO	product	dr (a : b)	yield (%)	ArCHO	product	dr (a : b)	yield (%)
Ph	202	4.0 : 1	89	p-OMe	208	5.8 : 1	78
p-Me	205	6.0 : 1	86	p-Cl	209	3.8 : 1	67
m-Me	206	5.8 : 1	76		210	7.3 : 1	80
o-Me	207	1 : 1.5	87				

General procedure for condensation reaction: To amide **204** (1 equiv) in $PhCH_3$ or THF (0.1 M) was added benzaldehyde (1.3 equiv), $MgSO_4$ (1.5 equiv) and acid (amount indicated) and refluxed overnight. Upon cooling, the reaction was quenched with sat. aq. $NaHCO_3$ (10 mL), and the mixture was extracted with EtOAc (3 x 10 mL). The combined organic layers were dried over Na_2SO_4 and concentrated. Product ratios were determined by crude ^{1}H NMR.

Pyridine 205: According to the general procedure, amino amide **204** (120 mg, 0.449 mmol), *p*-tolualdehyde (69.0 μL, 0.584 mmol), $MgSO_4$ (81.1 mg, 0.674 mmol), and $PhCH_3$/AcOH (5:1, 2.99 mL) were heated to 110 °C for 24 h. Aminal **205** was isolated as a 6.0:1 mixture of diastereomers (143 mg, 86 % yield, R_f = 0.53 in 40:1 EtOAc:MeOH eluent) as a beige solid. The major diastereomer could be further purified for characterization analysis.

Pyridine 205: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.63 (ddd, J = 4.8, 1.8, 0.9 Hz, 1H), 7.56 (dt, J = 8.0, 1.1 Hz, 1H), 7.50 (td, J = 7.7, 1.8 Hz, 1H), 7.45-7.42 (comp m, 2H), 7.25-7.21 (comp m, 2H), 7.12-7.04 (comp m, 4H), 7.00 (d, J = 7.9 Hz, 2H), 5.63 (s, 1H), 3.42 (dt, J = 10.7, 6.5 Hz, 1H), 3.18 (dt, J = 10.7, 6.5 Hz, 1H), 2.72-2.57 (comp m, 2H), 2.25 (s, 3H), 2.03-1.89 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.7, 161.2, 149.2, 138.0, 137.1, 136.8, 136.0, 129.1, 128.7, 126.9, 125.2, 122.3, 121.9, 121.0, 83.1, 77.9, 56.4, 37.5, 25.2, 21.1; IR (film) 2925, 1707,

1598, 1499, 1381, 1313, 753, 693 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{24}$H$_{24}$N$_3$O]$^+$: 370.1914, found 370.1915.

Pyridine 206: According to the general procedure, amino amide **204** (150 mg, 0.561 mmol), *m*-tolualdehyde (85.7 μL, 0.729 mmol), MgSO$_4$ (101 mg, 0.842 mmol), and PhCH$_3$/AcOH (5:1, 3.74 mL) were heated to 110 °C for 24 h. Aminal **206** was isolated as a 5.8:1 mixture of diastereomers (157 mg, 76% yield, R$_f$ = 0.53 in 40:1 EtOAc:MeOH eluent). The major diastereomer could be further purified for characterization analysis.

Pyridine 206: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.61 (dd, *J* = 4.8, 0.7 Hz, 1H), 7.53-7.44 (comp m, 4H), 7.23 (d, *J* = 8.2 Hz, 2H), 7.10-7.03 (comp m, 3H), 6.98-6.94 (comp m, 2H), 5.63 (s, 1H), 3.49 (dt, *J* = 10.5, 6.2 Hz, 1H), 3.19 (dt, *J* = 10.5, 6.6 Hz, 1H), 2.70 (dt, *J* = 13.3, 7.6 Hz, 1H), 2.55 (dt, *J* = 13.4, 6.7 Hz, 1H), 2.19 (s, 3H), 2.01-1.93 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.0, 161.2, 149.1, 139.6, 138.0, 137.3, 135.9, 128.9, 128.7, 128.2, 127.6, 125.1, 122.0, 121.9, 121.0, 83.3, 78.0, 56.9, 37.7, 25.3, 21.3; IR (film) 2924, 1707, 1587, 1496, 1381, 1300, 752, 692 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{24}$H$_{23}$N$_3$ONa]$^+$: 392.1733, found 392.1732.

Pyridine 207: According to the general procedure, amino amide **204** (102 mg, 0.382 mmol), *o*-tolualdehyde (57.6 μL, 0.496 mmol), MgSO$_4$ (69.0 mg, 0.573 mmol), and PhCH$_3$/AcOH (5:1, 2.55 mL) were heated to 110 °C for 24 h. Aminals **207a** and **207b** were isolated as a 1:1.5 mixture of inseparable diastereomers (123 mg, 87% yield, R$_f$ = 0.58 in 40:1 EtOAc:MeOH eluent).

Pyridine 207a (syn): ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.54 (dd, *J* = 4.1, 0.8 Hz, 1H), 7.56 (d, *J* = 7.9 Hz, 2H), 7.31-6.94 (comp m, 8H), 6.73-6.64 (comp m, 2H), 5.91 (s, 1H), 3.67 (dt, *J* = 9.6, 5.2 Hz, 1H), 3.18 (td, *J* = 8.8, 6.9 Hz, 1H), 2.87-2.73 (m, 1H), 2.57 (s, 3H), 2.35 (ddd, *J* = 13.2,

7.9, 5.3 Hz, 1H), 2.01-1.90 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.7, 161.1, 148.8, 138.0, 136.6, 135.7, 135.5, 130.6, 128.8, 128.3, 127.7, 126.2, 124.6, 121.6, 120.8, 120.5, 79.9, 77.7, 57.5, 38.3, 25.1, 19.4; IR (film) 3061, 2968, 1710, 1598, 1498, 1375, 1303, 748, 693 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{24}$H$_{23}$N$_3$O$_3$Na]$^+$: 392.1733, found 392.1735.

Pyridine 207b (anti): ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.65 (dt, *J* = 4.7, 1.4 Hz, 1H), 7.72-7.65 (comp m, 2H), 7.30-7.14 (comp m, 7H), 7.04-6.95 (comp m, 3H), 6.61 (s, 1H), 2.84 (dt, *J* = 13.2, 8.8 Hz, 1H), 2.76 (td, *J* = 9.4, 6.8 Hz, 1H), 2.62 (s, 3H), 2.51-2.42 (comp m, 2H), 1.86-1.79 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 175.7, 160.3, 149.7, 137.1, 136.6, 132.4, 130.6, 128.4, 128.3, 127.9, 125.4, 122.3, 121.6, 120.7, 77.7, 75.0, 50.5, 36.3, 24.8, 19.0; IR (film) 2968, 1711, 1597, 1367, 1321, 747, 694 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{24}$H$_{23}$N$_3$ONa]$^+$: 392.1733, found 392.1730.

Pyridine 208: According to the general procedure, amino amide **204** (109 mg, 0.408 mmol), *p*-anisaldehyde (64.5 μL, 0.530 mmol), MgSO$_4$ (73.7 mg, 0.612 mmol), and PhCH$_3$/AcOH (5:1, 2.72 mL) were heated to 110 °C for 24 h. Pyridine **208** was isolated as a 5.8:1 mixture of diastereomers (123 mg, 78% yield, R$_f$ = 0.50 in 40:1 EtOAc:MeOH eluent). The major diastereomer could be further purified for characterization analysis.

Pyridine 208: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.62 (d, *J* = 4.3 Hz, 1H), 7.57 (d, *J* = 7.9 Hz, 1H), 7.51 (td, *J* = 7.7, 1.6 Hz, 1H), 7.42 (d, *J* = 8.1 Hz, 2H), 7.24 (t, *J* = 8.0 Hz, 2H), 7.13-7.04 (comp m, 4H), 6.71 (d, *J* = 8.6 Hz, 2H), 5.62 (s, 1H), 3.72 (s, 3H), 3.42 (dt, *J* = 10.7, 6.5 Hz, 1H), 3.17 (dt, *J* = 10.8, 6.4 Hz, 1H), 2.73-2.54 (comp m, 2H), 2.06-1.89 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.8, 161.2, 159.4, 149.2, 137.1, 136.0, 131.9, 128.7, 128.3, 125.2, 122.4, 121.9, 121.0, 113.7, 82.9, 77.9, 56.3, 55.2, 37.5, 25.2; IR (film) 2958, 1707, 1587, 1384, 1248, 753, 693 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{24}$H$_{24}$N$_3$O$_2$]$^+$: 386.1863, found 386.1849.

Pyridine 209: According to the general procedure, amino amide **204** (106 mg, 0.397 mmol), *p*-chlorobenzaldehyde (72.4 mg, 0.515 mmol), MgSO$_4$ (71.7 mg, 0.596 mmol), and PhCH$_3$/AcOH (5:1, 2.65 mL) were heated to 110 °C for 24 h. Pyridine **209** was isolated as a 3.8:1 mixture of diastereomers (104 mg, 67% yield, R$_f$ = 0.57 in 40:1 EtOAc:MeOH eluent). The major diastereomer could be further purified for characterization analysis.

Pyridine 209: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.60 (d, *J* = 4.7 Hz. 1H), 7.51 (d, J = 3.7 Hz, 2H), 7.42 (d, *J* = 8.0 Hz, 2H), 7.28-7.24 (m, 1H), 7.16-7.07 (comp m, 6H), 5.65 (s, 1H), 3.49 (dt, *J* = 10.6, 6.3 Hz, 1H), 3.17 (dt, *J* = 10.6, 6.6 Hz, 1H), 2.70 (dt, *J* = 13.3, 7.5 Hz, 1H), 2.57-2.54 (dt, *J* = 13.4, 6.8 Hz, 1H), 2.02-1.94 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.8, 161.0, 149.2, 138.3, 137.0, 136.1, 128.9, 128.5, 128.4, 125.4, 122.1, 122.0, 120.8, 82.6, 77.9, 56.8, 37.6, 25.3; IR (film) 2959, 2360, 1707, 1597, 1382, 1089, 753 cm[1]; HRMS (ESI[+]) *m/z* calc'd for (M+H)[+] [C$_{23}$H$_{21}$N$_3$OCl][+]: 390.1368, found 390.1370.

Pyridine 210: According to the general procedure, amino amide **204** (101 mg, 0.378 mmol), 2-naphthaldehyde (76.7 mg, 0.491 mmol), MgSO$_4$ (68.2 mg, 0.567 mmol), and PhCH$_3$/AcOH (5:1, 2.52 mL) were heated to 110 °C for 24 h. Pyridine **210** was isolated as a 7.3:1 mixture of diastereomers (123 mg, 80% yield, R$_f$ = 0.50 in 40:1 EtOAc:MeOH eluent). The major diastereomer could be further purified for characterization analysis.

Pyridine 210: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.60 (d, *J* = 4.7 Hz, 1H), 7.76-7.73 (m, 1H), 7.71 (d, *J* = 8.6 Hz, 1H), 7.66-7.65 (m, 1H), 7.58 (s, 1H), 7.55 (d, *J* = 7.9 Hz, 1H), 7.49 (d, *J* = 7.9 Hz, 2H), 7.45-7.35 (comp m, 4H), 7.22 (t, *J* = 7.9 Hz, 2H), 7.04 (t, *J* = 6.9 Hz, 2H), 5.82 (s, 1H), 3.49 (dt, *J* = 10.6, 6.4 Hz, 1H), 3.26 (dt, *J* = 10.7, 6.5 Hz, 1H), 2.74 (dt, *J* = 13.3, 7.6 Hz, 1H), 2.64-2.57 (m, 1H), 2.08-1.92 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.9, 161.2, 149.2, 137.13, 137.08, 133.2, 132.8, 128.8, 128.6, 127.9, 127.6, 126.4, 126.2, 126.1, 125.3, 124.4,

122.2, 121.9, 120.9, 83.5, 78.0, 56.6, 37.7, 25.3; IR (film) 2967, 1707, 1597, 1499, 1380, 1317, 749 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{27}$H$_{23}$N$_3$ONa]$^+$: 428.1733, found 428.1729.

ArCHO	product	yield (%) Ar$_{ox}$	ArCHO	product	yield (%) Ar$_{ox}$
Ph	211	55[a]	p-OMe	215	37[c]
p-Me	212	47[b]	p-Cl	216	39
m-Me	213	72		217	58
o-Me	214	85			

[a] 6% yield diacetoxylated product. [b] 16% diacetoxylated product.
[c] 9% diacetoxylated product
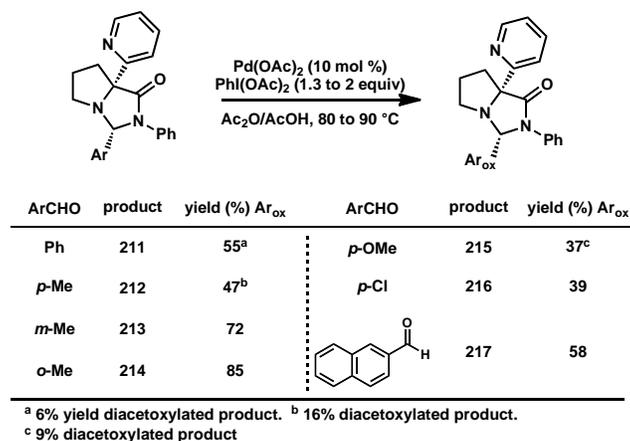
Representative acetoxylation procedure: Pyridine **202** (84.7 mg, 0.238 mmol), Pd(OAc)$_2$ (5.3 mg, 0.0238 mmol), and PhI(OAc)$_2$ (115 mg, 0.357 mmol) were dissolved in AcOH (1.76 mL) and Ac$_2$O (1.76 mL) in a 2-dram vial. The vial was sealed with a Teflon cap and heated to 80 °C for 15.5 h, at which time PhI(OAc)$_2$ (38.3 mg, 0.119 mmol) was added. The reaction was heated at 85 °C for an additional 9.5 h. Upon cooling, CH$_2$Cl$_2$ (10 mL) and water (10 mL) were added and the mixture was neutralized with Na$_2$CO$_3$ until pH ~9. The mixture was extracted with CH$_2$Cl$_2$ (2 x 15 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. To the crude mixture was added heptane (10 mL) and concentrated to ensure removal of residual Ac$_2$O. The crude residue was purified by flash chromatography (4:1 → 7:3 hexanes/acetone eluent) to afford acetate **211** (54.2 mg, 55% yield (61% borsm), R$_f$ = 0.45 in 1:1 hexanes/acetone) as a beige solid and the corresponding diacetate (6.7 mg, 6% yield, R$_f$ = 0.26 in 1:1 hexanes/acetone) as a beige solid.

Acetate 212: According to the general procedure, pyridine **205** (125 mg, 0.338 mmol), Pd(OAc)$_2$ (7.6 mg, 0.0338 mmol), PhI(OAc)$_2$ (109 mg, 0.338 mmol) and AcOH/Ac$_2$O (1:1, 3.38 mL) were stirred at 85 °C for 8 h. PhI(OAc)$_2$ (32.7 mg, 0.101 mmol) was added, and the mixture stirred an

additional 10.5 h at 85 °C. Acetate **212** was isolated as a beige solid (68.2 mg, 47% yield, R_f = 0.50 in 1:1 hexanes/acetone) and diacetate product was isolated as a beige solid (26.0 mg, 16% yield, R_f = 0.27 in 1:1 hexanes/acetone).

Acetate 212: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.58 (ddd, J = 4.8, 1.7, 1.0 Hz, 1H), 7.58-7.55 (comp m, 2H), 7.38-7.30 (comp m, 3H), 7.28-7.23 (m, 1H), 7.08-7.00 (comp m, 2H), 6.90 (d, J = 0.6 Hz, 1H), 6.68-6.60 (comp m, 2H), 5.89 (s, 1H), 3.53 (dt, J = 9.9, 5.8 Hz, 1H), 3.15 (dt, J = 9.9, 7.1 Hz, 1H), 2.73 (dt, J = 13.2, 7.8 Hz, 1H), 2.45-2.38 (m, 1H), 2.40 (s, 3H), 2.24 (s, 3H), 1.98-1.85 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.2, 169.0, 161.0, 148.9, 148.5, 139.3, 137.5, 135.9, 128.8, 127.1, 126.3, 123.2, 121.7, 121.1, 120.4, 77.8, 77.6, 57.5, 38.1, 25.2, 21.1, 21.0; IR (film) 2968, 1766, 1708, 1497, 1373, 1200, 733, 692 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{26}$H$_{25}$N$_3$O$_3$Na]$^+$: 450.1788, found 450.1796.

Diacetate: ^{1}H NMR (300 MHz, CDCl$_3$) δ 8.71 (dd, J = 4.8, 1.8, 0.9 Hz, 1H), 7.72-7.76 (m, 1H), 7.53-7.47 (comp m, 3H), 7.25-7.19 (comp m, 3H), 7.17-7.11 (m, 1H), 7.07-7.01 (m, 1H), 6.76 (s, 2H), 5.98 (s, 1H), 3.35 (dt, J = 10.6, 6.6 Hz, 1H), 3.12 (dt, J = 10.6, 6.5 Hz, 1H), 2.69-2.50 (comp m, 2H), 2.28 (s, 3H), 2.07-1.89 (comp m, 8H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 172.0, 168.4, 160.7, 149.3, 149.0, 139.7, 136.9, 135.9, 128.74, 128.68, 124.8, 121.8, 121.3, 120.8, 119.8, 119.6, 75.3, 56.6, 39.1, 25.0, 21.2, 21.0; IR (film) 2968, 1769, 1709, 1371, 1181, 1045, 753, 692 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{28}$H$_{27}$N$_3$O$_5$Na]$^+$: 508.1843, found 508.1852.

Acetate 213: According to the general procedure, pyridine **206** (124 mg, 0.336 mmol), Pd(OAc)$_2$ (7.5 mg, 0.0336 mmol), PhI(OAc)$_2$ (162 mg, 0.504 mmol), and AcOH/Ac$_2$O (1:1, 3.36 mL) were stirred at 80 °C for 15.5 h. PhI(OAc)$_2$ (54.1 mg, 0.168 mmol) was added, and the reaction stirred

at 85 °C for an additional 9.5 h. Acetate **213** (104 mg, 72% yield, R_f = 0.48 in 1:1 hexanes/acetone) was isolated as a beige solid .

Acetate 213: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.57 (ddd, J = 4.8, 1.7, 0.9 Hz, 1H), 7.60-7.57 (comp m, 2H), 7.36-7.24 (comp m, 4H), 7.08-7.04 (m, 1H), 7.00 (ddd, J = 7.1, 4.9, 1.5 Hz, 1H), 6.95 (d, J = 1.2 Hz, 2H), 6.53 (s, 1H), 5.87 (s, 1H), 3.56 (dt, J = 9.7, 5.7 Hz, 1H), 3.16-3.10 (m, 1H), 2.75 (dt, J = 13.3, 7.9 Hz, 1H), 2.38 (s, 3H), 2.38-2.32 (m, 1H), 1.91 (s, 3H), 1.97-1.90 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.5, 169.1, 161.2, 148.7, 146.4, 137.7, 135.7, 135.2, 130.6, 129.4, 128.9, 127.6, 124.7, 122.4, 121.7, 121.1, 120.2, 77.8, 77.7, 57.5, 38.1, 25.2, 21.1, 20.6; IR (film) 3061, 2968, 1762, 1709, 1496, 1378, 1190, 755, 693 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+H)$^+$ [C$_{26}$H$_{26}$N$_3$O$_3$]$^+$: 428.1969, found 428.1974.

Acetate 214: According to the general procedure, pyridines **207a** and **207b** (123 mg, 0.333 mmol), Pd(OAc)$_2$ (7.5 mg, 0.0333 mmol), PhI(OAc)$_2$ (107 mg, 0.333 mmol) and AcOH/Ac$_2$O (1:1, 3.33 mL) were stirred at 85 °C for 10 h. PhI(OAc)$_2$ (53.6 mg, 0.167 mmol) was added, and the reaction stirred an additional 10.5 h at 85 °C. Acetate **214** (128 mg, 85% yield, R_f = 0.42 in 1:1 hexanes/acetone) was isolated as a beige solid. The ^{1}H NMR spectrum featured highly broadened peaks, complicating characterization. Acetate **214** was therefore hydrolyzed to the phenol for characterization analysis.

Acetate 214: HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{26}$H$_{25}$N$_3$O$_3$Na]$^+$: 450.1788, found 450.1786.

Phenol 214b: Acetate **214** (33.7 mg, 0.0789 mmol) was dissolved in aq. HCl (1 M, 0.789 mL) and THF (1.47 mL), and the resulting solution was heated to reflux overnight. Upon cooling the reaction was quenched with solid Na$_2$CO$_3$ until pH ~9. The mixture was then extracted with EtOAc (3 x 5 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The

crude residue was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford phenol **214b** (14.1 mg, 46% yield, R_f = 0.28 in 1:1 hexanes/acetone) as a light yellow oil.

Phenol 214b: ^{1}H NMR (400 MHz, CDCl$_3$) δ 12.13 (bs, 1H), 8.67 (dt, J = 4.8, 1.4 Hz, 1H), 7.77-7.75 (comp m, 2H), 7.29-7.24 (m, 1H), 7.23-7.19 (comp m, 3H), 7.10 (t, J = 7.8 Hz, 1H), 6.95-6.90 (comp m, 2H), 6.81 (d, J = 8.1 Hz, 1H), 6.45 (d, J = 7.5 Hz, 1H), 5.73 (s, 1H), 3.18 (dt, J = 12.5, 8.0 Hz, 1H), 3.08 (ddd, J = 12.4, 7.8, 4.4 Hz, 1H), 2.73-2.65 (comp m, 2H), 2.23-2.16 (m, 1H), 1.99-1.91 (m, 1H), 1.86 (s, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 171.5, 159.1, 148.5, 138.5, 136.9, 135.2, 130.7, 128.8, 128.0, 127.8, 122.9, 122.0, 121.3, 117.8, 116.8, 78.8, 50.9, 44.9, 34.8, 25.0, 19.4; IR (film) 3061, 2959, 1709, 1586, 1471, 1397, 1123, 749, 702 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{24}$H$_{23}$N$_3$O$_2$Na]$^+$: 408.1682, found 408.1690.

Acetate 215: According to the general procedure, pyridine **208** (123 mg, 0.319 mmol), Pd(OAc)$_2$ (7.2 mg, 0.0319 mmol), PhI(OAc)$_2$ (134 mg, 0.415 mmol) and AcOH/Ac$_2$O (1:1, 3.19 mL) were stirred at 80 °C for 13 h. PhI(OAc)$_2$ (51.4 mg, 0.160 mmol) was added, and the mixture was stirred for an additional 5 h at 85 °C. Acetate **215** was isolated as a beige solid (73.1 mg, 37% yield, R_f = 0.40 in 1:1 hexanes/acetone), as well as the corresponding diacetoxylation product (14.0 mg, 9% yield, R_f = 0.23 in 1:1 hexanes/acetone) as a beige solid.

Acetate 215: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.58 (ddd, J = 4.8, 1.8, 1.0 Hz, 1H), 7.58-7.54 (comp m, 2H), 7.41-7.33 (comp m, 2H), 7.30-7.23 (m, 2H), 7.09-7.01 (comp m, 2H), 6.70 (d, J = 8.7 Hz, 1H), 6.64 (d, J = 2.5 Hz, 1H), 6.35 (dd, J = 8.7, 2.5 Hz, 1H), 5.85 (s, 1H), 3.70 (s, 3H), 3.52 (dt, J = 9.9, 5.9 Hz, 1H), 3.14 (dt, J = 9.9, 7.1 Hz, 1H), 2.73 (dt, J = 13.3, 7.8 Hz, 1H), 2.44-2.38 (m, 1H), 2.40 (s, 3H), 1.98-1.86 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.2, 168.8, 161.1, 159.9, 149.5, 148.9, 137.5, 135.9, 128.8, 128.0, 124.8, 123.3, 121.8, 121.0, 120.5, 111.1, 108.7, 77.8, 77.5, 57.3, 55.4, 38.0, 25.2, 21.1; IR (film) 2922, 1765, 1708, 1501, 1375,

1201, 753 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{26}$H$_{25}$N$_3$O$_4$Na]$^+$: 466.1737, found 466.1734.

Acetate 216: According to the general procedure, pyridine **209** (104 mg, 0.267 mmol), Pd(OAc)$_2$ (6.0 mg, 0.0267 mmol), PhI(OAc)$_2$ (172 mg, 0.534 mmol) and AcOH/Ac$_2$O (1:1, 2.67 mL) were stirred at 90 °C for 24 h with acetate **216** isolated as a beige solid (47.0 mg, 39% yield (51% borsm), R$_f$ = 0.48 in 1:1 hexanes/acetone).

Acetate 216: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.56 (ddd, *J* = 4.8, 1.8, 0.9 Hz, 1H), 7.56-7.53 (comp m, 2H), 7.44-7.37 (comp m, 2H), 7.30-7.24 (comp m, 2H), 7.13 (d, *J* = 1.8 Hz, 2H), 7.04 (ddd, *J* = 7.4, 4.8, 1.2 Hz, 1H), 6.77 (dd, *J* = 8.4, 2.0 Hz, 1H), 6.71 (d, *J* = 8.4 Hz, 1H), 5.89 (s, 1H), 3.56 (dt, *J* = 9.8, 5.8 Hz, 1H), 3.14 (dt, *J* = 9.7, 7.2 Hz, 1H), 2.75 (dt, *J* = 13.3, 7.8 Hz, 1H), 2.42 (s, 3H), 2.36 (ddd, *J* = 13.2, 7.5, 5.7 Hz, 1H), 2.02-1.90 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 173.3, 168.4, 160.8, 148.9, 137.3, 136.0, 134.0, 129.9, 129.0, 128.1, 125.6, 125.0, 123.3, 122.0, 121.9, 121.0, 120.3, 77.8, 57.6, 38.1 25.2, 21.0; IR (film) 2959, 1769, 1701, 1598, 1376, 1193, 754, 692 cm^{-1}; HRMS (ESI$^+$) *m/z* calc'd for (M+Na)$^+$ [C$_{25}$H$_{22}$N$_3$O$_3$ClNa]$^+$: 470.1242, found 470.1247.

Acetate 217: According to the general procedure, pyridine **210** (123 mg, 0.303 mmol), Pd(OAc)$_2$ (6.8 mg, 0.0303 mmol), PhI(OAc)$_2$ (127 mg, 0.394 mmol) and AcOH/Ac$_2$O (1:1, 3.03 mL) were stirred at 80 °C for 13 h. PhI(OAc)$_2$ (48.8 mg, 0.151 mmol) was added, and the reaction was stirred at 85 °C for an additional 5 h. Acetate **217** was isolated as a beige solid (81.5 mg, 58% yield, R$_f$ = 0.43 in 1:1 hexanes/acetone).

Acetate 217: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.48 (ddd, *J* = 4.8, 1.8, 0.9 Hz, 1H), 7.72-7.69 (comp m, 3H), 7.58 (s, 1H), 7.42-7.37 (m, 1H), 7.33 (d, *J* = 7.4 Hz, 1H), 7.29-7.24 (comp m, 3H), 7.18 (s, 1H), 7.15 (dt, *J* = 8.0, 1.0 Hz, 1H), 7.08-7.02 (comp m, 2H), 6.78 (ddd, *J* = 7.4, 4.8,

1.2 Hz, 1H), 6.04 (s, 1H), 3.67 (dt, J = 9.5, 5.5 Hz, 1H), 3.24-3.18 (m, 1H), 2.82 (dt, J = 13.3, 8.0

Hz, 1H), 2.46 (s, 3H), 2.33 (ddd, J = 13.2, 7.7, 5.4 Hz, 1H), 2.01-1.91 (comp m, 2H); ^{13}C NMR

(100 MHz, CDCl$_3$) δ 173.8, 169.1, 161.1, 148.6, 146.3, 138.0, 135.7, 133.0, 130.5, 129.7, 128.9,

127.9, 127.0, 126.7, 125.8, 124.6, 121.5, 120.9, 120.3, 120.0, 79.4, 77.9, 57.9, 38.4, 25.2, 21.2;

IR (film) 2968, 1763, 1708, 1376, 1198, 752 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$

[C$_{29}$H$_{26}$N$_3$O$_3$]$^+$: 464.1969, found 464.1969.

ArCHO	product	dr ($a : b$)	yield (%)	ArCHO	product	dr ($a : b$)	yield (%)
Ph	202	0 : 1.0	87	1-naphthaldehyde	218	0 : 1.0	85
p-Me	205	1 : 3.0	83				
m-Me	206	1 : 5.0	82				
p-OMe	208	1 : 3.9	69	2-naphthaldehyde	210	0 : 1.0	72
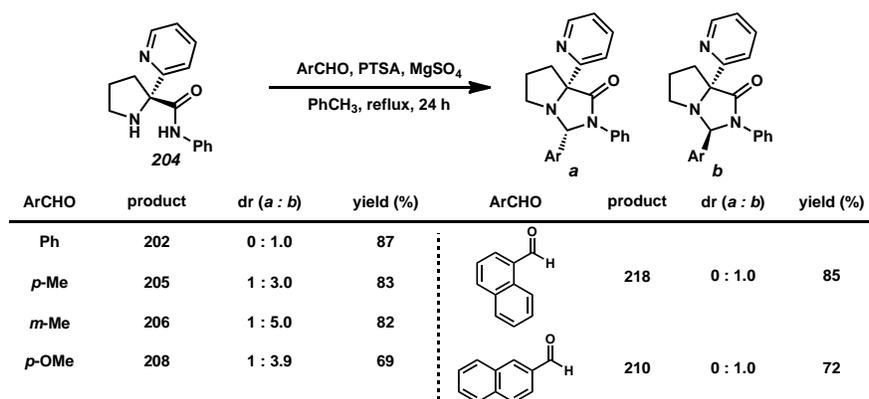
Aminal 202b: According to the general procedure, aminal **202b** was isolated as a beige solid

(87% yield).

Aminal 205b: According to the general procedure, aminal **205b** was isolated as a beige solid

(143 mg, 83% yield, R$_f$ = 0.19 in EtOAc).

Aminal 205b: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.65 (ddd, J = 4.8, 1.8, 1.0 Hz, 1H), 7.69-7.60

(comp m, 2H), 7.33-7.30 (comp m, 2H), 7.24-7.17 (comp m, 5H), 7.08 (d, J = 7.8 Hz, 2H), 7.07-

7.02 (m, 1H), 6.44 (s, 1H), 2.85-2.77 (comp m, 2H), 2.59-2.54 (m, 1H), 2.50-2.44 (m, 1H), 2.29

(s, 3H), 1.93-1.80 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 190.8, 176.0, 160.2, 149.6,

138.5, 137.9, 136.6, 131.3, 129.0, 128.5, 128.4, 124.7, 122.4, 122.1, 121.1, 78.2, 78.0, 50.5, 36.2,

25.0, 21.2; IR (film) 2967, 1709, 1597, 1377, 1303, 752, 693 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for

(M+H)$^+$ [C$_{24}$H$_{24}$N$_3$O]$^+$: 370.1914, found 370.1917.

Aminal 206b: According to the general procedure, aminal **206b** was isolated as a beige solid (109 mg, 82% yield, R_f = 0.50 in 9:1 CH_2Cl_2/MeOH).

Aminal 206b: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.65 (dd, J = 4.8, 0.7 Hz, 1H), 7.69-7.58 (comp m, 2H), 7.32 (dd, J = 8.6, 1.0 Hz, 2H), 7.23-7.20 (comp m, 4H), 7.14-7.02 (comp m, 4H), 6.42 (s, 1H), 2.85-2.77 (comp m, 2H), 2.55 (ddd, J = 9.3, 5.8, 3.5 Hz, 1H), 2.50-2.44 (m, 1H), 2.29 (s, 3H), 1.92-1.81 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 190.1, 176.0, 160.1, 149.6, 138.1, 137.9, 136.6, 134.3, 129.5, 129.4, 128.4, 128.1, 125.5, 124.8, 122.4, 122.0, 121.1, 78.3, 78.0, 50.5, 36.2, 25.0, 21.3; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{24}$H$_{23}$N$_3$ONa]$^+$: 392.1733, found 392.1738.

Aminal 207b: According to the general procedure, aminal **207b** was isolated as a beige solid (166 mg, 88% yield, R_f = 0.59 in 40:1 EtOAc/MeOH).

Aminal 208b: According to the general procedure, aminal **208b** was isolated as a beige solid (135 mg, 69% yield, R_f = 0.52 in 9:1 CH_2Cl_2/MeOH).

^{1}H NMR (400 MHz, CDCl$_3$) δ 8.65 (ddd, J = 4.8, 1.8, 0.9 Hz, 1H), 7.69-7.59 (comp m, 3H), 7.32-7.29 (comp m, 2H), 7.23 (d, J = 8.4 Hz, 4H), 7.06-7.02 (m, 1H), 6.82-6.79 (comp m, 2H), 6.42 (s, 1H), 3.76, (s, 3H), 2.85-2.77 (comp m, 2H), 2.58 (ddd, J = 9.4, 6.0, 3.7, 1H), 2.50-2.44 (m, 1H), 1.93-1.81 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 176.0, 159.7, 149.6, 137.8, 136.7, 129.9, 128.7, 128.5, 128.3, 126.3, 124.8, 122.4, 122.2, 121.1, 113.7, 82.9, 78.0, 55.1, 50.4, 36.2, 25.0; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{24}$H$_{23}$N$_3$O$_2$Na]$^+$: 408.1682, found 408.1684.

Aminal 218: According to the general procedure, aminal **218** was isolated as a beige solid (262 mg, 85% yield, R_f = 0.63 in EtOAc).

^{1}H NMR (400 MHz, CDCl$_3$) δ 8.73 (dt, J = 4.7, 1.4 Hz, 1H), 8.53 (d, J = 8.4 Hz, 1H), 7.90 (d, J = 8.1 Hz, 1H), 7.79 (d, J = 7.9 Hz, 1H), 7.76-7.74 (comp m, 2H), 7.69 (ddd, J = 8.4, 7.0, 1.4 Hz,

1H), 7.57 (td, $J = 7.5$, 1.0 Hz,1H), 7.33-7.16 (comp m, 8H) 7.04-7.00 (m, 1H), 2.89 (dt, $J = 13.2$, 8.8 Hz, 1H), 2.78 (td, $J = 9.6$, 6.8 Hz, 1H), 2.53-2.47 (m, 1H), 2.40-2.34 (m, 1H), 1.88-1.75 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 176.0, 160.4, 149.6, 138.4, 136.7, 133.7, 131.5, 129.6, 129.0, 128.7, 128.5, 127.0, 126.1, 126.0, 124.63, 124.60, 123.5, 122.5, 121.7, 121.2, 78.2, 75.2, 50.4, 36.1, 24.7; IR (film) 2967, 1710, 1597, 1499, 1368, 1301 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{27}$H$_{24}$N$_3$O]$^+$: 406.1914, found 406.1914.

Aminal 210b: According to the general procedure, aminal **210b** was isolated as a beige solid (174 mg, 72% yield, R$_f$ = 0.22 in EtOAc).

Aminal 210b: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.69 (ddd, $J = 4.8$, 1.6, 0.9 Hz, 1H), 7.87 (s, 1H), 7.81-7.74 (comp m, 3H), 7.72-7.64 (comp m, 2H), 7.48-7.46 (comp m, 2H), 7.40-7.39 (comp m, 3H), 7.23-7.17 (comp m, 3H), 7.03-7.00 (m, 1H), 6.66 (s, 1H), 2.90-2.82 (comp m, 2H), 2.58-2.47 (comp m, 2H), 1.92-1.82 (comp m, 2H); ^{13}C NMR (100 MHz, CDCl$_3$) δ176.1, 160.1, 149.6, 137.8, 136.7, 133.3, 132.9, 132.0, 128.7, 128.5, 128.1, 127.6, 126.6, 126.4, 125.5, 124.9, 122.5, 122.1, 121.2, 78.5, 78.1, 50.6, 36.2, 25.1; HRMS (ESI$^+$) m/z calc'd for (M+H)$^+$ [C$_{27}$H$_{24}$N$_3$O]$^+$: 406.1914, found 406.1902.
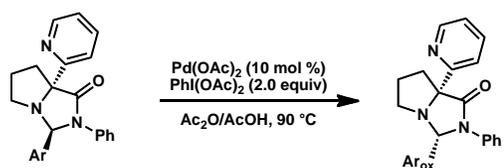
ArCHO	product	yield (%) Ar$_{ox}$	ArCHO	product	yield (%) Ar$_{ox}$
Ph	211	42			
p-Me	212	31 (28)[a]		219	no rxn
m-Me	213	51			
p-OMe	215	23[b]		217	44

[a] yield in parentheses corresponds to observed diacetoxylated product.
[b] Run at 80 °C for 20 h.

General acetoxylation procedure: Pyridine (1 equiv), Pd(OAc)$_2$ (10 mol %), and PhI(OAc)$_2$ (2 equiv) were dissolved in AcOH/Ac$_2$O (1:1, 0.1 M) in a 2-dram vial. The vial was sealed with a Teflon cap and heated to 90 °C for 24 h. Upon cooling, CH$_2$Cl$_2$ (10 mL) and water (10 mL) were added and the mixture was neutralized with Na$_2$CO$_3$ until pH ~9. The mixture was extracted with CH$_2$Cl$_2$ (2 x 15 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. To the crude mixture was added heptane (10 mL) and concentrated to ensure removal of residual Ac$_2$O. The crude residue was purified by flash chromatography (4:1 → 7:3 hexanes/acetone eluent) to afford acetates **211-217** in the yields indicated.
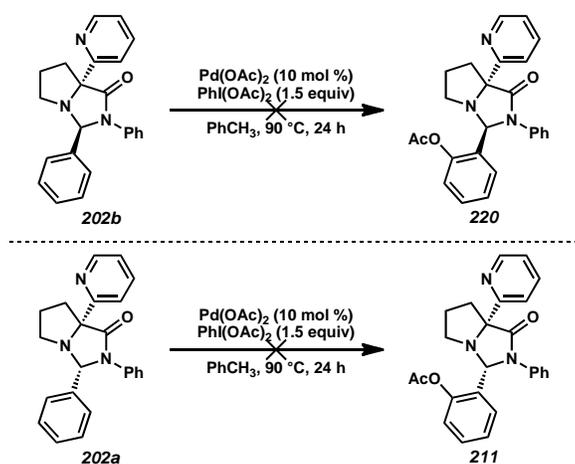
General procedure for functionalization: Pyridine (1 equiv), Pd(OAc)$_2$ (10 mol %), and PhI(OAc)$_2$ (2 equiv) were dissolved in toluene (0.1 M) in a 2-dram vial. The vial was sealed with a Teflon cap and heated to 90 °C for 24 h. Upon cooling, the solvent was removed. No acetoxylated product was observed by crude ^{1}H NMR.
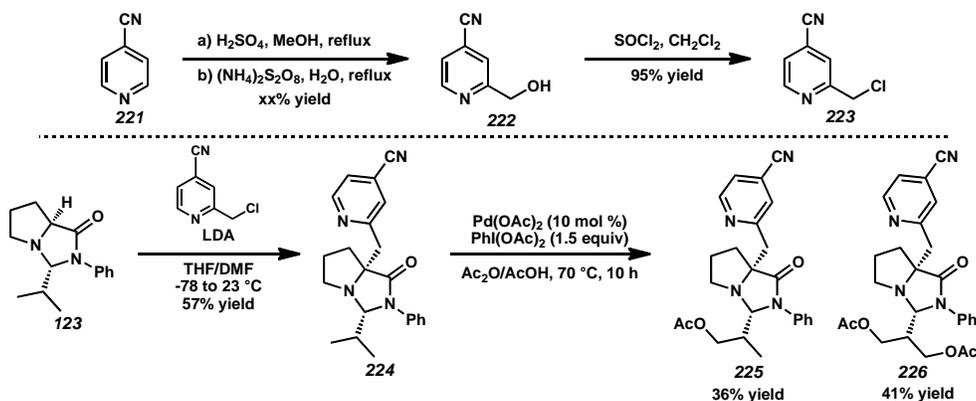
Pyridine 224: 4-cyanopyridine (2.00 g, 19.2 mmol) and H_2SO_4 (0.200 mL, conc.) were refluxed in MeOH (28.6 mL) for 30 min. $(NH_4)_2S_2O_8$ (7.01 g, 30.7 mmol) in H_2O (12.8 mL) was added to the solution over 30 min at reflux. The reaction was refluxed an additional 1 h. After cooling to 23 °C, the resultant precipitate was filtered, and the sovent was removed under reduced pressure. The residue was neutralized with sat. aq. Na_2CO_3. The aqueous was extracted with $CHCl_3$ (4 x 30 mL), the organics dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (2:3 hexanes:EtOAc to EtOAc) to afford **222** (994 mg, 39% yield, R_f = 0.15 in 1:1 hexanes:EtOAc) as a beige solid.

To pyridine (990 mg, 7.38 mmol) in CH_2Cl_2 (29.5 mL) at 23 °C was added $SOCl_2$ (0.639 mL, 8.86 mmol) slowly. The reaction was stirred for 1 h at 23 °C. The reaction was quenched with sat. aq. Na_2CO_3, then the aqueous was extracted with CH_2Cl_2 (40 mL). The organic layer was dried over Na_2SO_4 and concentrated to afford pyridine **223** (1.07 g, 95% yield, R_f = 0.60 in 1:1 hexanes:EtOAc) as a red oil.

To diisopropylamine (0.250 mL, 1.78 mmol) in THF (5.20 mL) at -78 °C was added *n*-BuLi (0.690 mL, 1.72 mmol, 2.5 M in hexanes) slowly. The reaction was stirred at -78 °C for 10 min, then aminal **123** (750 mg, 3.07 mmol) in THF (7.00 mL) was added at -78 °C and stirred for 30 min. To this solution was added pyridine **223** (225 mg, 1.47 mmol) in DMF (2.90 mL) at -78 °C. The solution was warmed to 23 °C and stirred overnight. The reaction was quenched with

H_2O (10 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (3:2 to 1:1 hexanes:EtOAc) to afford pyridine **224** (250 mg, 57% yield, $R_f = 0.22$ in 1:1 hexanes:EtOAc) as a yellow solid.
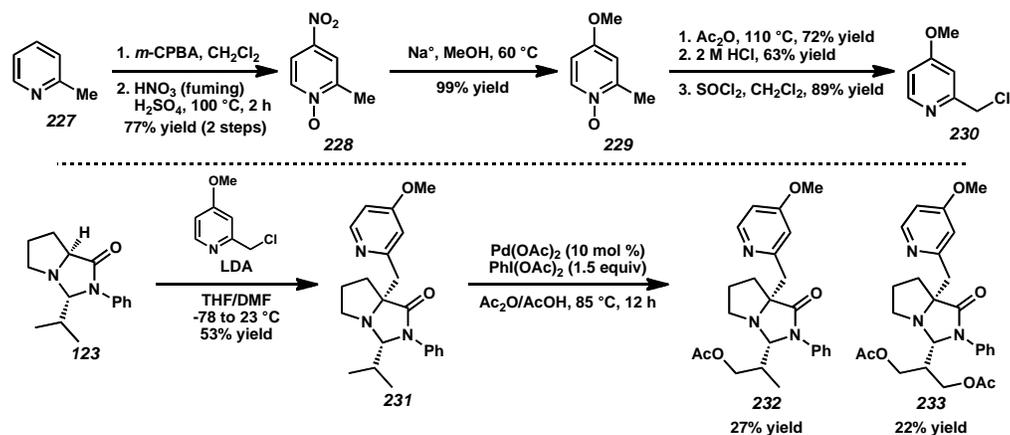
Pyridine 230: To **227** (1.00 mL, 10.1 mmol) in CH_2Cl_2 (25.3mL) was added *m*-CPBA (4.54 g, 20.3 mmol) at 23 °C. The reaction was capped and stirred at 23 °C overnight. The suspension was washed with NaOH (3 x 10 ml, 10% aq.). The aqueous was extracted with CH_2Cl_2 (5 x 20 mL). The organic layer was dried over Na_2SO_4 and concentrated to afford the *N*-oxide (1.06 g, 95% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as a yellow oil.

To *N*-oxide (1.06 g, 9.67 mmol) in conc. H_2SO_4 (3.70 mL) at 0 °C was added fuming HNO_3 (3.3 mL) dropwise. The mixture was heated to 100 °C for 2 h. The reaction was cooled to 23 °C, then neutralized with sat. aq. Na_2CO_3. The aqueous was extracted with CH_2Cl_2 (4 x 20 mL) and dried over Na_2SO_4 and concentrated to afford pyridine **228** (1.34 g, 90% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as a yellow solid.

To **228** (3.00 g, 19.5 mmol) in MeOH (194 mL) at 60 °C was added a solution of Na° (0.460 g, 19.9 mmol) in MeOH (62.0 mL). The reaction was stirred 20 min at 60 °C, then cooled to 23 °C. The solvent was removed under reduced pressure. The residue was taken up in H_2O (100 mL)

and extracted with CH_2Cl_2 (4 x 100 mL). The organic layers were dried over Na_2SO_4 and concentrated to afford **229** (2.79 g, 99% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as an orange oil.

N-oxide (19.5 mmol) and Ac_2O (7.50 mL) were heated to 110 °C for 2 h. The reaction was cooled to 23 °C, then neutralized with sat. aq. Na_2CO_3. The aqueous was extracted with CH_2Cl_2 (4 x 75 mL). The organic layers were dried over Na_2SO_4 and concentrated to afford pyridine (2.53 g, 72% yield, $R_f = 0.20$ in 1:1 hexanes:EtOAc) as an orange oil.

Pyridine (2.53, 14.0 mmol) in HCl (17.5 mL, 2 M (aq)) was heated to 70 °C for 2 h. The mixture was cooled to 23 °C, then quenched with sat. aq. Na_2CO_3. The aqueous was extracted with CH_2Cl_2 (4 x 20 mL). The organics were dried over Na_2SO_4 and concentrated to afford the product (1.23 g, 63% yield) as a light yellow solid.

To pyridine (1.23 g, 8.82 mmol) in CH_2Cl_2 (34.0 mL) at 23 °C was added $SOCl_2$ (0.763 mL, 10.6 mmol) slowly. The reaction was stirred for 1 h at 23 °C. The reaction was quenched with sat. aq. Na_2CO_3, then the aqueous was extracted with CH_2Cl_2 (40 mL). The organic layer was dried over Na_2SO_4 and concentrated to afford pyridine **230** (1.24 g, 89% yield, $R_f = 0.45$ in 1:1 hexanes:EtOAc) as a red oil.

Pyridine 231: To diisopropylamine (0.625 mL, 4.45 mmol) in THF (7.30 mL) at -78 °C was added *n*-BuLi (1.72 mL, 4.29 mmol, 2.5 M in hexanes) slowly. The reaction was stirred at -78 °C for 10 min, then aminal **123** (750 mg, 3.07 mmol) in THF (8.00 mL) was added at -78 °C and stirred for 30 min. To this solution was added pyridine **230** (580 mg, 3.68 mmol) in DMF (7.40 mL) at -78 °C. The solution was warmed to 23 °C and stirred overnight. The reaction was quenched with H_2O (10 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na_2SO_4 and concentrated. The crude

residue was purified via flash chromatography (1:4 to 3:7 hexanes:EtOAc) to afford pyridine **231** (599 mg, 53% yield, R_f = 0.20 in 1:1 hexanes:EtOAc) as a yellow solid.
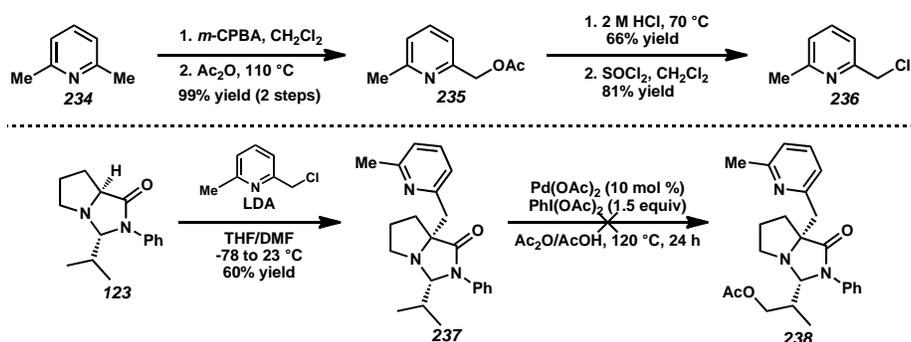
Pyridine 236: To **234** (1.00 mL, 8.59 mmol) in CH_2Cl_2 (21.5 mL) was added *m*-CPBA (3.85 g, 17.2 mmol) at 23 °C. The reaction was capped and stirred at 23 °C overnight. The suspension was washed with NaOH (3 x 10 ml, 10% aq.). The aqueous was extracted with CH_2Cl_2 (5 x 20 mL). The organic layer was dried over Na_2SO_4 and concentrated to afford the *N*-oxide (1.47 g, 99% yield, R_f = 0.0 in 1:1 hexanes:EtOAc) as a yellow oil.

N-oxide (8.59 mmol) and Ac_2O (3.3 mL) were heated to 110 °C for 2 h. The reaction was cooled to 23 °C, then neutralized with sat. aq. Na_2CO_3. The aqueous was extracted with CH_2Cl_2 (4 x 20 mL). The organic layers were dried over Na_2SO_4 and concentrated to afford pyridine **235** (1.71 g, 99% yield, R_f = 0.75 in 40:1 EtOAc:MeOH) as a yellow oil.

Pyridine **235** (1.71, 8.59 mmol) in HCl (10.7 mL, 2 M (aq)) was heated to 70 °C for 2 h. The mixture was cooled to 23 °C, then quenched with sat. aq. Na_2CO_3. The aqueous was extracted with CH_2Cl_2 (4 x 20 mL). The organics were dried over Na_2SO_4 and concentrated to afford the product (697 mg, 66% yield) as a light yellow solid.

To pyridine (697 mg, 5.66 mmol) in CH_2Cl_2 (23.6 mL) at 23 °C was added $SOCl_2$ (0.490 mL, 6.79 mmol) slowly. The reaction was stirred for 1 h at 23 °C. The reaction was quenched with sat. aq. Na_2CO_3, then the aqueous was extracted with CH_2Cl_2 (40 mL). The organic layer was

dried over Na_2SO_4 and concentrated to afford pyridine **236** (648 mg, 81% yield, $R_f = 0.70$ in 1:1 hexanes:EtOAc) as an orange oil.

Pyridine 237: To diisopropylamine (0.287 mL, 2.04 mmol) in THF (4.00 mL) at -78 °C was added *n*-BuLi (0.790 mL, 1.96 mmol, 2.5 M in hexanes) slowly. The reaction was stirred at -78 °C for 10 min, then aminal **123** (400 mg, 1.64 mmol) in THF (5.90 mL) was added at -78 °C and stirred for 30 min. To this solution was added pyridine **236** (278 mg, 1.96 mmol) in DMF (3.90 mL) at -78 °C. The solution was warmed to 23 °C and stirred overnight. The reaction was quenched with H_2O (10 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (1:4 to 3:7 hexanes:EtOAc) to afford pyridine **237** (343 mg, 60% yield, $R_f = 0.20$ in 1:1 hexanes:EtOAc) as a yellow solid.
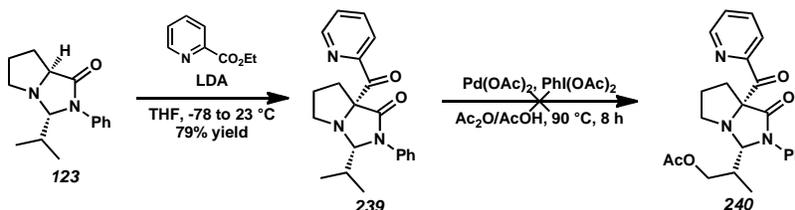
Pyridine 239: To diisopropylamine (0.417 mL, 2.96 mmol) in THF (6.60 mL) at -78 °C was added *n*-BuLi (1.15 mL, 1.96 mmol, 2.5 M in hexanes) slowly. The reaction was stirred at -78 °C for 10 min, then aminal **123** (500 mg, 2.04 mmol) in THF (7.00 mL) was added at -78 °C and stirred for 30 min. To this solution was added 2-ethylnicotinate (0.414 mL, 3.07 mmol) at -78 °C. The solution was warmed to 23 °C and stirred overnight. The reaction was quenched with H_2O (10 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (1:4 to 1:3 hexanes:EtOAc) to afford pyridine **240** (233 mg, 79% yield, $R_f = 0.40$ in 1:1 hexanes:EtOAc) as a yellow solid.
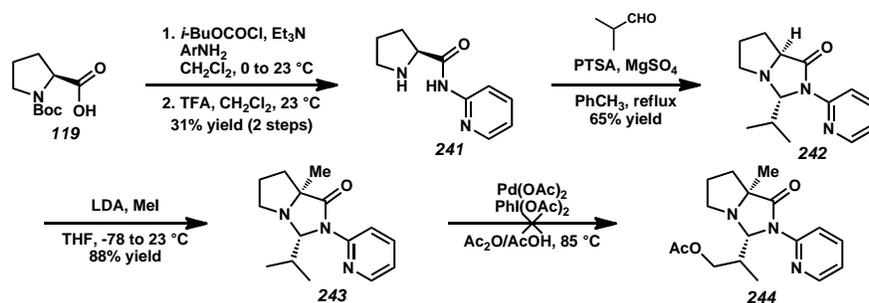
Aminal 243: To **119** (2.00 g, 9.29 mmol) in CH_2Cl_2 (31.0 mL) at 0 °C was added *i*-BuOCOCl

(1.34 mL, 10.2 mmol) and Et_3N (1.44 mL, 10.2 mmol). The suspension was stirred at 0 °C for

15 min, at which time 2-aminopyridine (962 mg, 10.2 mmol) was added at 0 °C. The reaction

was warmed to 23 °C and stirred overnight. The reaction was washed with 1 M $KHSO_4$ (20

mL), sat. aq. $NaHCO_3$ (20 mL) and brine (20 mL) sequentially. Dried over Na_2SO_4 and

concentrated to afford the amide (1.09 g, 40% yield, $R_f = 0.40$ in 1:1 hexanes:EtOAc) as a white

solid.

To the amide (1.09 g, 3.73 mmol) in CH_2Cl_2 (7.40 mL) at 23 °C was added TFA (5.70 mL, 74.5

mmol). The reaction was stirred for 1 h at 23 °C. The solvent was removed under reduced

pressure. The residue was neutralized with solid Na_2CO_3. H_2O was added, and then the aqueous

was extracted with CH_2Cl_2 (3 x 20 mL). The organic layers were dried over Na_2SO_4 and

concentrated to afford amide **241** (559 mg, 78% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as an

orange oil.

To **241** (559 mg, 2.92 mmol) in toluene (14.6 mL) was added isobutyraldehyde (0.400 mL, 4.38

mmol), PTSA (27.8 mg, 0.146 mmol) and $MgSO_4$ (528 mg, 4.38 mmol) and the suspension

heated to reflux overnight. Upon cooling, sat. aq. $NaHCO_3$ (20 mL) was added. The aqueous

was extracted with EtOAc (3 x 20 mL). The organics were dried over Na_2SO_4 and concentrated.

The crude residue was purfied via flash chromatography (4:1 hexanes:EtOAc) to afford **242** (466

mg, 65% yield, $R_f = 0.60$ in 1:1 hexanes:EtOAc) as a light yellow oil.

To diisopropylamine (0.387 mL, 2.76 mmol) in THF (4.00 mL) at -78 °C was added *n*-BuLi (1.06 mL, 2.66 mmol, 2.5 M in hexanes) slowly. The reaction was stirred at -78 °C for 10 min, then aminal **242** (466 mg, 1.90 mmol) in THF (5.50 mL) was added at -78 °C and stirred for 30 min. To this solution was added MeI (0.237 mL, 3.80 mmol) at -78 °C. The solution was warmed to 23 °C and stirred overnight. The reaction was quenched with H_2O (10 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (1:4 hexanes:EtOAc) to afford pyridine **243** (434 mg, 88% yield, $R_f = 0.70$ in 1:1 hexanes:EtOAc) as a yellow oil.
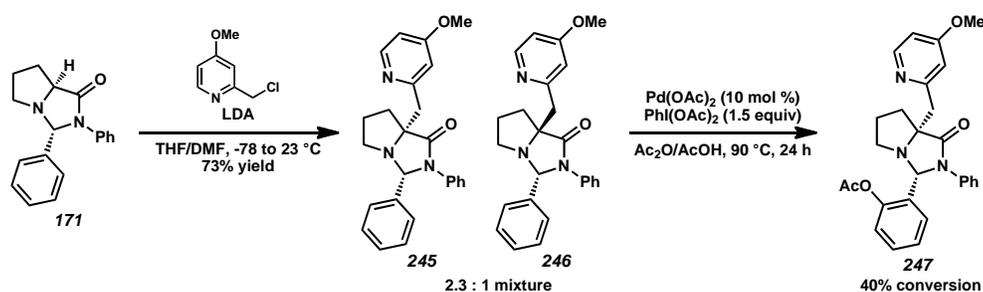
Pyridine 245: To diisopropylamine (0.384 mL, 2.73 mmol) in THF (5.50 mL) at -78 °C was added *n*-BuLi (1.05 mL, 2.64 mmol, 2.5 M in hexanes) slowly. The reaction was stirred at -78 °C for 10 min, then aminal **171** (524 mg, 1.88 mmol) in THF (7.00 mL) was added at -78 °C and stirred for 30 min. To this solution was added pyridine **236** (356 mg, 2.26 mmol) in DMF (4.50 mL) at -78 °C. The solution was warmed to 23 °C and stirred overnight. The reaction was quenched with H_2O (10 mL). The aqueous was extracted with EtOAc (3 x 20 mL). The organic layer was washed with brine (2 x 30 mL), dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (1:7 to 3:7 hexanes:acetone) to afford pyridine **245** (551 mg, 73% yield, $R_f = 0.51$ in 1:1 hexanes:acetone) as an amorphous yellow solid.

Acetate 247: Pyridine **245** (15.0 mg, 0.0376 mmol), Pd(OAc)$_2$ (0.8 mg, 3.76 μmol) and

PhI(OAc)$_2$ (18.1 mg, 0.0563 mmol) were heated in AcOH/Ac$_2$O (1:1, 0.380 mL) to 90 °C for 24

h. The solvent was removed by azeotropic removal with heptanes (3 x 10 mL). Crude ^{1}H NMR

revealed 40% conversion to product, with no purification.
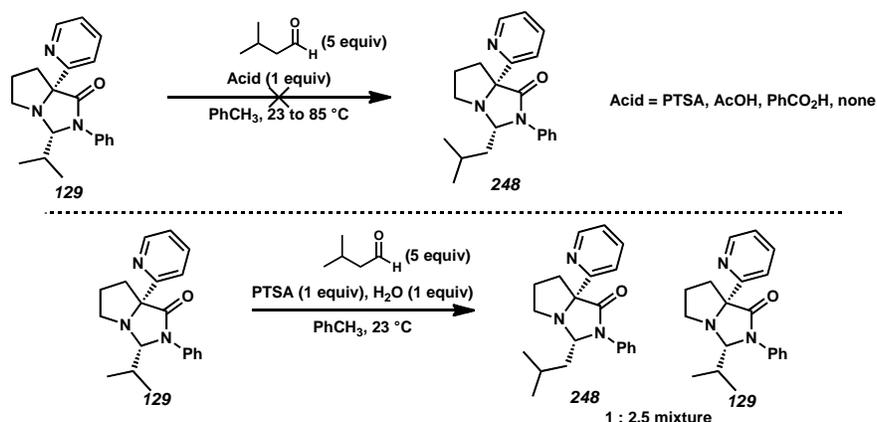
General exchange procedure: Pyridine **129** (1 equiv), isovaleraldehyde (5 equiv), acid (1 equiv)

and H$_2$O (1 equiv) were combined in toluene (0.1 M) and heated to the temperature indicated for

the time indicated. Solvent was removed under reduced pressure. Analysis by crude ^{1}H NMR

afforded exchange ratios.

entry	acid	solvent	202 : 205	entry	acid	solvent	202 : 205
1	CSA	PhCH$_3$	1 : 2.8	6	PTSA	t-AmOH	1 : 7.0
2	PTSA	THF	1 : 3.0	7	CSA	i-PrOH	1 : 3.8
3	PTSA	MeOH	1 : 11.3	8	CSA	t-BuOH	1 : 5.1
4	CSA	MeOH	1 : 11.0	9	-	AcOH	1 : 3.5
5	CSA	t-AmOH	1 : 11.5	10	-	AcOH/Ac$_2$O	1 : 0
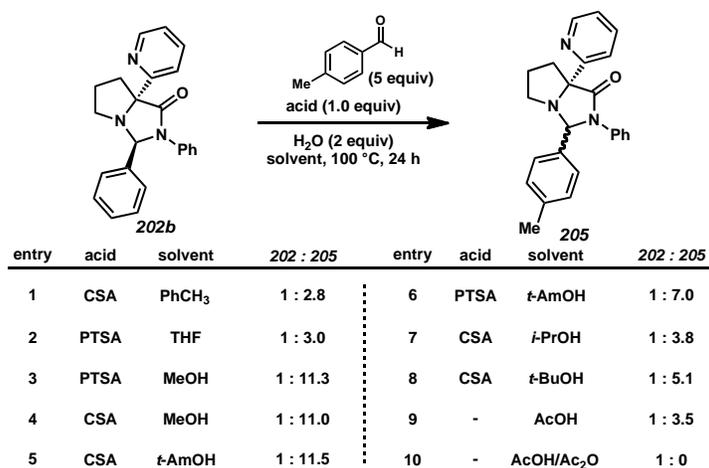
General exchange procedure: Pyridine **202b** (1 equiv), *p*-tolualdehyde (5 equiv), acid (1 equiv) and H_2O (2 equiv) were combined in solvent (0.1 M) and heated to 100 °C for 24 h. The solvent was removed under reduced pressure. Analysis by crude ^{1}H NMR afforded exchange ratios.
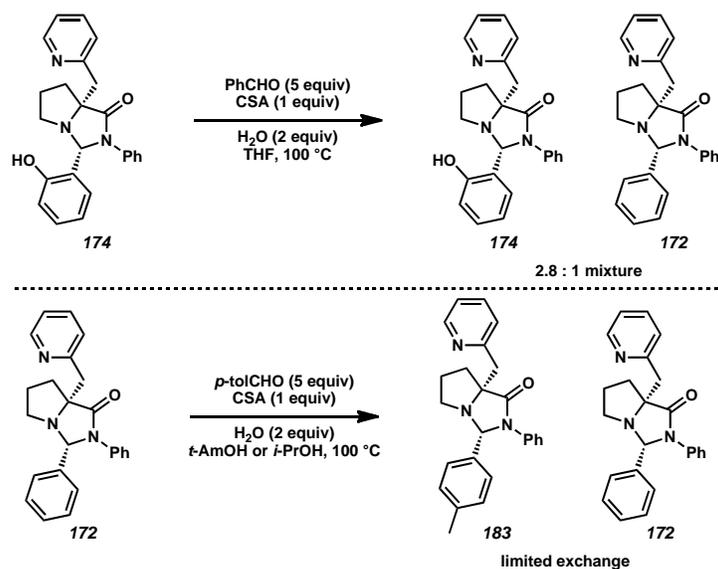
General exchange procedure: Pyridine (1 equiv), benzaldehyde (5 equiv), CSA (1 equiv) and H_2O (2 equiv) were combined in solvent (0.5 M) and heated to 100 °C for 24 h. The solvent was removed under reduced pressure. Analysis by crude ^{1}H NMR afforded exchange ratios.
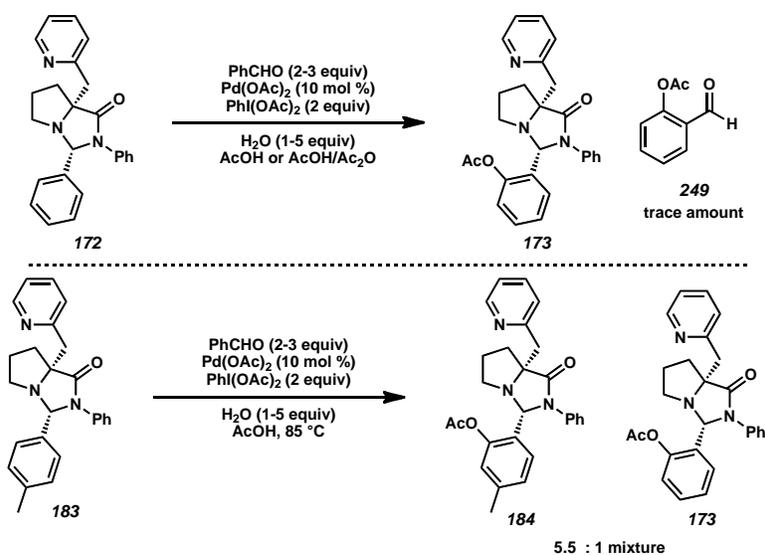
General procedure for synchronization: Pyridine (1 equiv), PhCHO (2 equiv), Pd(OAc)$_2$ (10 mol %), PhI(OAc)$_2$ (2 equiv) and H$_2$O (5 equiv) were combined in AcOH or AcOH/Ac$_2$O (1:1) (0.2 M) and heated to 85 °C for 24 h. The solvent was removed via azeotropic removal with heptanes (3 x 10 mL). Analysis by crude ^{1}H NMR afforded exchange ratios.
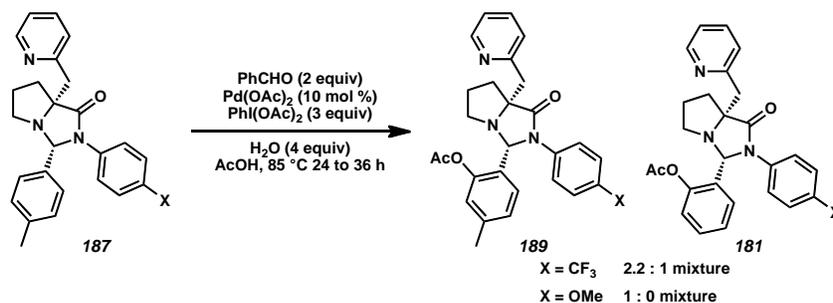
General procedure for synchronization: Pyridine (1 equiv), PhCHO (2 equiv), Pd(OAc)$_2$ (10 mol %), PhI(OAc)$_2$ (3 equiv) and H$_2$O (4 equiv) were combined in AcOH (0.1 M) and heated to 85 °C for 24-36 h. The solvent was removed via azeotropic removal with heptanes (3 x 10 mL). Analysis by crude ^{1}H NMR afforded exchange ratios.
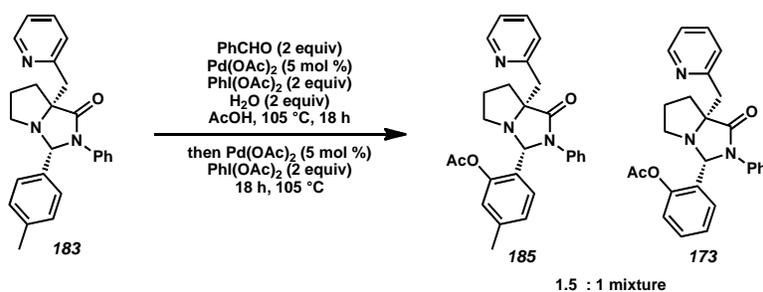
General procedure for synchronization: Pyridine (1 equiv), PhCHO (2 equiv), Pd(OAc)$_2$ (5 mol %), PhI(OAc)$_2$ (2 equiv) and H$_2$O (2 equiv) were combined in AcOH (0.1 M) and heated to 105 °C for 18 h. Then Pd(OAc)$_2$ (5 mol %) and PhI(OAc)$_2$ (2 equiv) were added and heated to 105 °C for an additional 18 h. The solvent was removed via azeotropic removal with heptanes (3 x 10 mL). Analysis by crude ^{1}H NMR afforded exchange ratios.

Reaction scheme with conditions: Pd(OAc)$_2$ (2.5 mol %), PhI(OAc)$_2$ (0.5 equiv), PhCHO (2 equiv), H$_2$O (2 equiv), AcOH, 105 °C, 10 h; then Pd(OAc)$_2$ (5 mol %), PhI(OAc)$_2$ (1 equiv), 95 °C, 24 h. Starting material **250** gives **179** (30% yield) and **181** (7% yield).
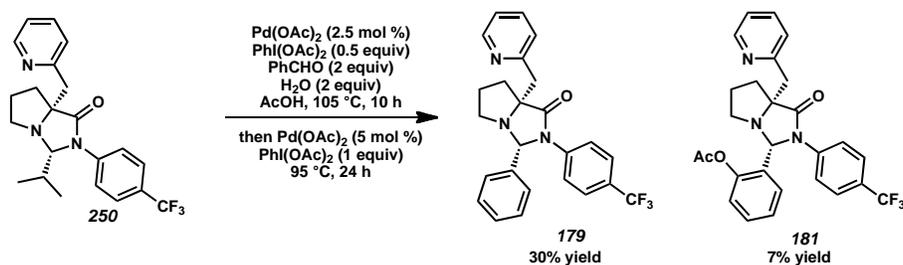
Pyridine 250: To a solution of amino amide **178-CF$_3$** (1.00 g, 3.87 mmol) in PhCH$_3$ (19.4 mL) at 23 °C was added isobutyraldehyde (0.530 mL, 5.81 mmol), TsOH·H$_2$O (37.0 mg, 0.194 mmol), and MgSO$_4$ (0.700 g, 5.81 mmol). The mixture was heated to reflux overnight. Upon cooling to 23 °C, the solution was quenched with sat. aq. NaHCO$_3$ (20 mL), and the resulting mixture was extracted with EtOAc (3 x 25 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude product was purified via flash chromatography (4:1 hexanes/EtOAc eluent) to give the aminal (1.12 g, 92% yield, R_f = 0.49 in 1:1 hexanes/EtOAc) as a light yellow oil.

To a solution of freshly distilled diisopropylamine (516 μL, 3.68 mmol) in THF (5.00 mL) at -78 °C was added *n*-BuLi (1.41 mL, 2.5 M in hexanes, 3.52 mmol) dropwise. The solution was stirred for 10 min at -78 °C, at which time a solution of aminal (1.10 g, 3.52 mmol) in THF (6.80 mL) was added, and the resulting mixture was stirred for an additional 30 min at -78 °C. To a suspension of NaH (368 mg, 60% dispersion in mineral oil, 9.19 mmol, washed 2 x 1.5 mL) in DMF (10.0 mL) at 0 °C was added 2-(bromomethyl)pyridine hydrobromide (775 mg, 3.06 mmol). The suspension was stirred at 0 °C for 30 min, at which time it was added to the enolate solution at -78 °C (flask rinsed with additional 1.80 mL DMF). The suspension was warmed to 23 °C and stirred overnight. The reaction was quenched slowly with H$_2$O (25 mL) at 23 °C, and the resulting mixture was extracted with EtOAc (3 x 25 mL). The combined organic layers were washed with brine (2 x 35 mL), dried over Na$_2$SO$_4$ and concentrated. The crude residue was

purified by flash chromatography (7:3 → 1:1 hexanes/EtOAc eluent) to afford pyridine **250** (905 mg, 73% yield, R_f = 0.21 in 1:1 hexanes/EtOAc) as a light yellow solid.

Pyridine 250: ^{1}H NMR (400 MHz, CDCl$_3$) δ 8.55 (dd, J = 4.9, 0.9 Hz, 1H), 7.65-7.59 (comp m, 3H), 7.53 (d, J = 8.5 Hz, 2H), 7.48 (d, J = 7.8 Hz, 1H), 7.15 (ddd, J = 7.4, 5.0, 1.0 Hz, 1H), 4.53 (d, J = 3.0 Hz, 1H), 3.24 (ABq, J = 12.2 Hz, Δv = 20.5 Hz, 2H), 2.82-2.75 (comp m, 2H), 2.20-2.15 (m, 1H), 2.10-2.03 (m, 1H), 1.79 (septet of doublets, J = 6.8, 3.0 Hz, 1H), 1.65-1.54 (m, 1H), 1.51-1.42 (m, 1H), 0.91 (d, J = 6.9 Hz, 3H), 0.55 (d, J = 6.6 Hz, 3H); ^{13}C NMR (100 MHz, CDCl$_3$) δ 174.4, 157.9, 148.8, 139.6, 135.8, 126.1 (q, J = 3.8 Hz), 125.2, 123.8, 121.6, 85.9, 74.8, 58.5, 45.4, 35.0, 30.5, 24.6, 18.3, 14.3; IR (film) 2966, 1704, 1614, 1325, 1124, 845, 748 cm^{-1}; HRMS (ESI$^+$) m/z calc'd for (M+Na)$^+$ [C$_{22}$H$_{25}$N$_3$OF$_3$Na]$^+$: 426.1764, found 426.1766.

Acetate 181: Pyridine **250** (100 mg, 0.248 mmol), Pd(OAc)$_2$ (2.8 mg, 0.0124 mmol), PhI(OAc)$_2$ (80.0 mg, 0.248 mmol), PhCHO (50.0 μL, 0.496 mmol), and H$_2$O (18.0 μL, 0.992 mmol) were dissolved in AcOH (2.50 mL) in a scintillation vial. The vial was capped and heated to 105 °C for 10 h. The reaction was cooled to 95 °C and Pd(OAc)$_2$ (5.6 mg, 0.0248 mmol) and PhI(OAc)$_2$ (160 mg, 0.496 mmol) were added, and the reaction was stirred for another 24 h at 95 °C. Upon cooling, the solvent was removed, and the resulting mixture was neutralized with solid Na$_2$CO$_3$ and water (10 mL). The mixture was extracted with CH$_2$Cl$_2$ (3 x 10 mL), and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude mixture was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford pyridine **179-CF$_3$** (33.0 mg, 30% yield) and acetate **181-CF$_3$** (8.6 mg, 7% yield).
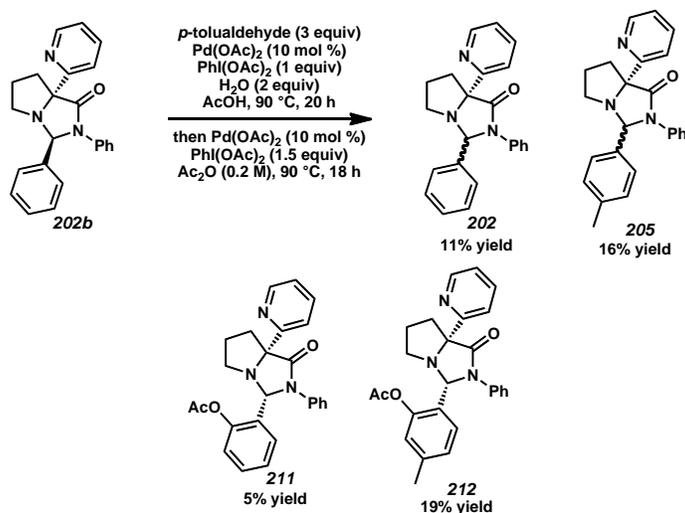
Acetate 212: Pyridine **202b** (63.4 mg, 0.178 mmol), Pd(OAc)$_2$ (4.0 mg, 0.0178 mmol), PhI(OAc)$_2$ (57.3 mg, 0.178 mmol), *p*-tolualdehyde (63.0 µL, 0.534 mmol), and H$_2$O (6.4 µL, 0.356 mmol) were dissolved in AcOH (0.890 mL) in a 2-dram vial. The vial was capped and heated to 90 °C for 20 h. Pd(OAc)$_2$ (4.0 mg, 0.0178 mmol), PhI(OAc)$_2$ (86.0 mg, 0.267 mmol) and Ac$_2$O (0.890 mL) were added and the reaction heated for an additional 18 h at 90 °C. Upon cooling, the solvent was removed and the resulting mixture was neutralized with solid Na$_2$CO$_3$ and water (10 mL). The mixture was extracted with CH$_2$Cl$_2$ (3 x 10 mL) and the combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude mixture was purified by flash chromatography (4:1 hexanes/acetone eluent) to afford pyridine **202** (7.0 mg, 11% yield), acetate **211** (3.6 mg, 5% yield), pyridine **205** (10.8 mg, 16% yield) and acetate **212** (14.1 mg, 19% yield).

189
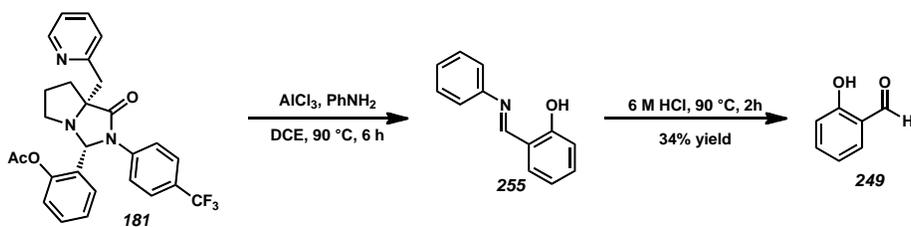
To a solution of phenol **181** (76.0 mg, 0.198 mmol) and AlCl$_3$ (66.9 mg, 0.502 mmol) in DCE (1.30 mL) at 23 °C was added PhNH$_2$ (69.0 μL, 0.753 mmol). The resulting mixture was heated to 90 °C and stirred for 8 h. Upon cooling, the reaction mixture was poured into water, and sat. Rochelle's salt (10 mL) was added. The aqueous was extracted with EtOAc (3 x 15 mL), the organics washed with brine (20 mL) and dried over Na$_2$SO$_4$ and concentrated. To the crude mixture was added aq. HCl (6 M, 2.50 mL) and heated to 70 °C for 3 h. Upon cooling, the mixture was poured into water and the aqueous extracted with EtOAc (3 x 10 mL), dried over Na$_2$SO$_4$ and concentrated to afford a mixture of salicylaldehyde (**249**) (9.3 mg, 38% yield) and imine **255** (5.2 mg, 13% yield).

Chapter 3

Alternative Ligand Scaffolds

In addition to our acetoxylation of sp^2 and sp^3 C-H bonds using an amino amide scaffold, we have synthesized alternative ligand scaffolds that we imagined could perform our desired chelate-assisted functionalization (Scheme 3.0.1). Examining our desired reaction profile, we wanted to design a scaffold that contains both an acetalization component and a ligating group. The acetalization component can condense onto an aldehyde to generate the substrate ligand adduct. Upon treatment with a transition metal, the ligating group can coordinate the metal and place it in position to undergo functionalization of an unreactive bond. Finally, hydrolysis will release the functionalized product and regenerate the ligand. Again we envisioned a *cis* relationship between the ligating group and the substrate to ensure the close proximity needed to induce the desired functionalization.

Scheme 3.0.1. General Concept
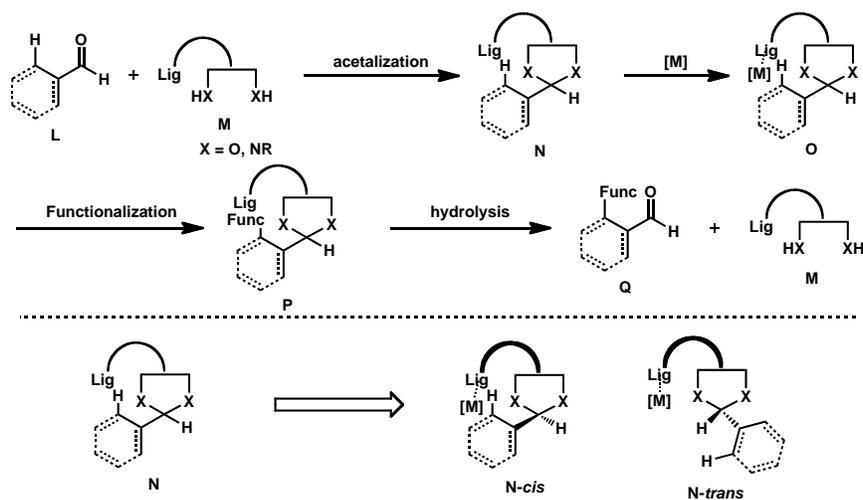
3.1 Examining Amino Acid Derived Scaffolds

Our first entry into examining alternative scaffolds began with derivatization of our proline scaffold. Our original amino alcohol (**92**) scaffold proved much too labile to perform the

desired chemistry (Scheme 3.1.1). We anticipated that by changing the electronics of the condensation components we might be able to introduce more stability into the *N,O*-acetal. By changing the amine into an amide and the primary alcohol into a tertiary alcohol, we envisioned a much more robust acetal. From pyroglutamic acid we converted acid **256** to the methyl ester, followed by amide protection to afford **257** in two steps. Alkylation with 2-bromomethyl pyridine, however, was unsuccessful in affording **258**.

Scheme 3.1.1. Acetalization component redesign
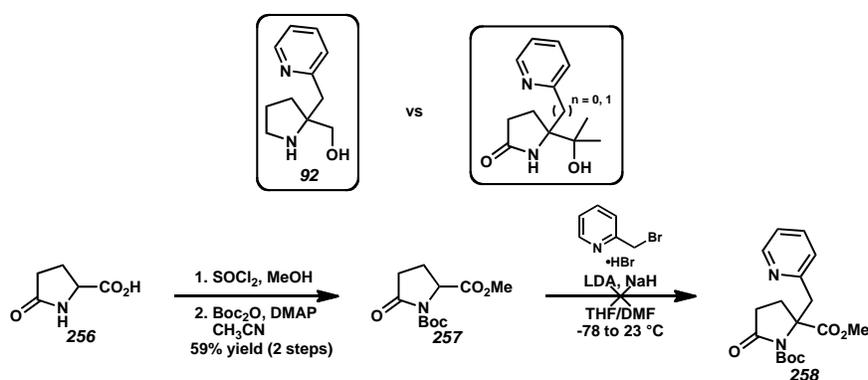
Unable to install the ligating group after the amide was in place, we conceptualized first installing the pyridyl ligating group, then oxidizing the amine to the amide. We treated substrate **259** with several oxidizing conditions but were unable to isolate the desired amide (Scheme 3.1.2). In most cases we observed a single oxidation to form a hemiacetal, which could then form the aldehyde, preventing any further oxidation.

Scheme 3.1.2. Initial attempted oxidation to the amide
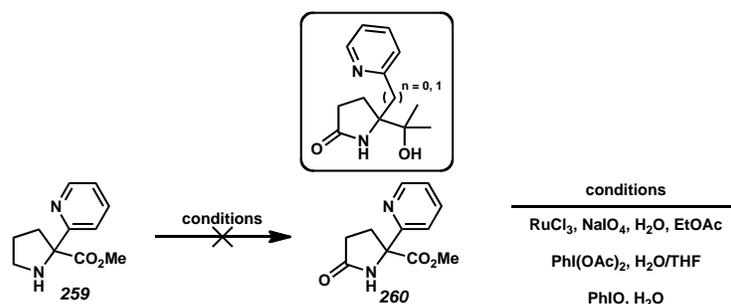
We next attempted the oxidation of the amine to the amide on the protected substrate. When **113** or **91** was reacted with RuO_2 and $NaIO_4$ in a miscible t-BuOH/H_2O mixture none of the desired amide was generated (Scheme 3.1.3). Using a two-phase solvent system, however, afforded amide **261** in 50% yield and amide **258** in 70% yield.[1] Deprotection afforded amide **262** in 84% yield, and amide **263** in quantitative yield. Treatment of amide **263** with MeMgBr afforded the tertiary alcohol (**264**) in moderate yield. Under identical conditions, amide **263** afforded the desired alcohol **265** in less than 10% yield. Employing MeLi instead of MeMgBr did not improve the conversion to alcohol **265**.

Scheme 3.1.3. Oxidation of pyrrolidine to pyrrolidinone
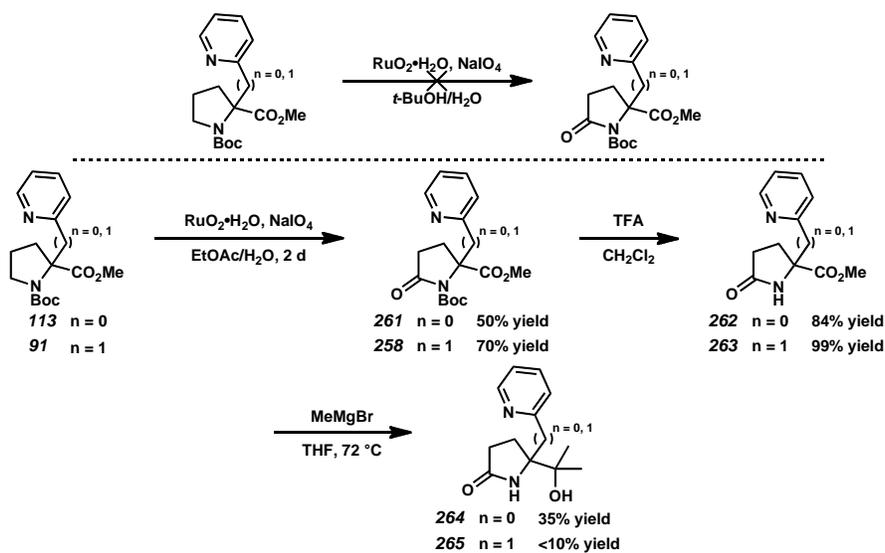
With amide alcohol **264**, we attempted to condense the ligand onto different aldehydes. Any attempted condensation with aromatic aldehydes was unsuccessful at producing the *N,O*-acetal (Scheme 3.1.4). Ligand **264** did condense with aliphatic aldehydes in the presence of PTSA and $MgSO_4$ to afford *N,O*-acetal **267**. Due to the difficulty of the synthesis of the ligand, we abandoned this ligand scaffold for the amino amide scaffold.

Scheme 3.1.4. Attempted condensation of the amide alcohol
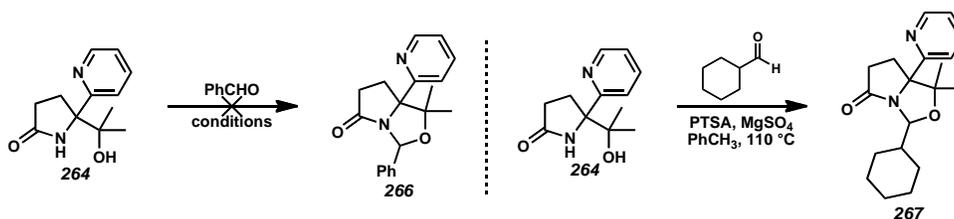
As an alternative we began investigating a scaffold based on L-serine. We imagined using the amino alcohol as the acetalization component, while converting the acid moiety into an oxazoline-ligating group (Scheme 3.1.5). Based on Seebach's work with amino acids, we believed we would achieve a *syn* relationship between the ligating group and aldehyde substrate.[2] From **268** we introduced a Boc protecting group and converted the acid into corresponding methyl ester **269**. Protecting the acetalization components with an acetonide allowed us to manipulate methyl ester **270**. Saponification in quantitative yield followed by conversion into the acid chloride afforded **272**. Treatment of the acid chloride with 2,2-dimethyl-1-aminoethanol afforded amide **273** in quantitative yield. Protecting the alcohol TsCl generated the tosylate, which quickly closed down under the basic conditions to afford oxazoline **274**.

Scheme 3.1.5. Oxazoline scaffold from serine
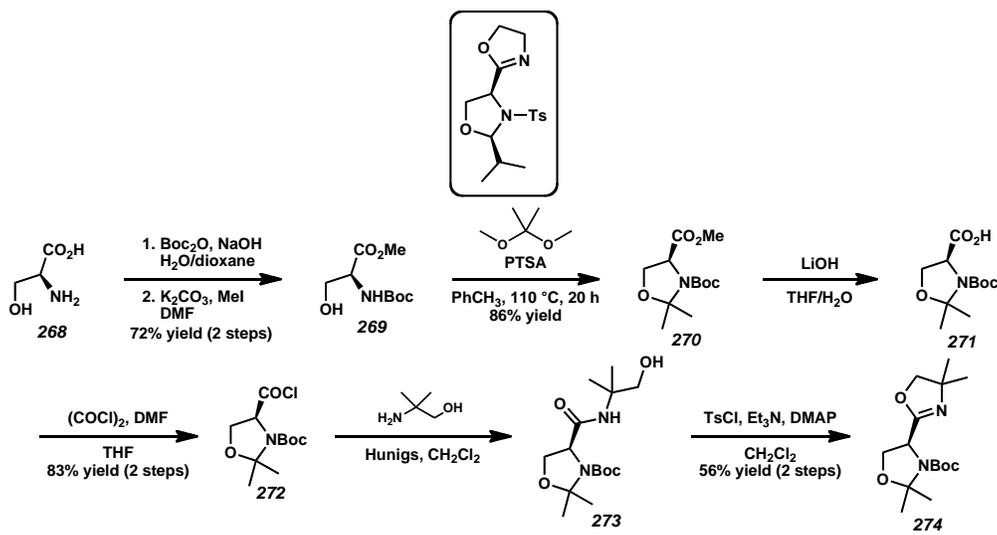
To confirm product **274** by ^{1}H NMR, which proved difficult due to methyl rotamers, we imagined using a slightly different procedure to generate the oxazoline product. Treating amide alcohol **273** with MsCl and Et$_3$N afforded product **274**, which we believed to be the oxazoline (Scheme 3.1.6). Removal of the acetonide generated amino alcohol **275**. Removal of the amine protecting group with TFA, however, did not afford amino alcohol **276**. This was likely due to the propensity of alcohols to act as nucleophiles during this reaction. Rather than continue to pursue the deprotection of the amine, we attempted to examine the functionalization potential of **274**. Treatment with Pd(OAc)$_2$ and PhI(OAc)$_2$ in DCE afforded none of the desired acetoxylated product (**277**).

Scheme 3.1.6. Another route to the oxazoline
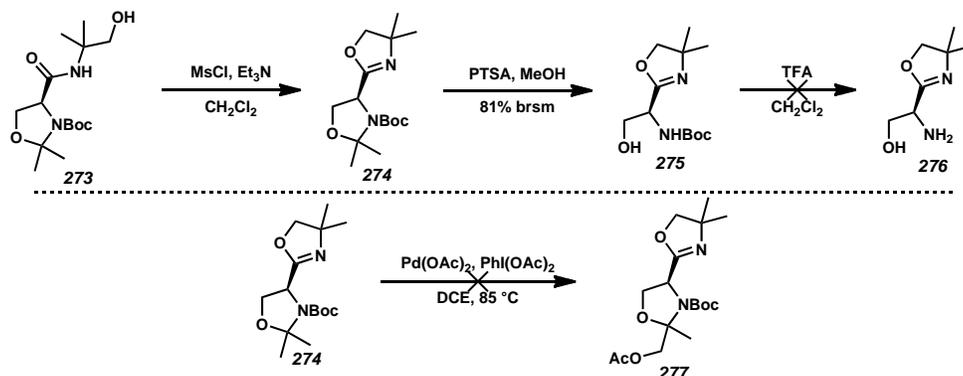
Due to unsuccessful formation of **276**, we discontinued our examination of this ligand. Based on our work with the amino amide ligand in Chapter 2, it is probable that a more straightforward synthesis of this ligand substrate adduct could be achieved (Scheme 3.1.7). Rather than forming an acetonide to protect the amino alcohol component, **278** could be condensed onto the aldehyde to be functionalized. This would eliminate the need for further manipulation after the oxazoline is installed. If the protecting group proved problematic due to its size, presumably a different protecting group could be installed prior to condensation.

195

Subjecting acetal **280** to functionalization conditions could yield the desired functionalized product.

Scheme 3.1.7. Possible route to the desired ligand
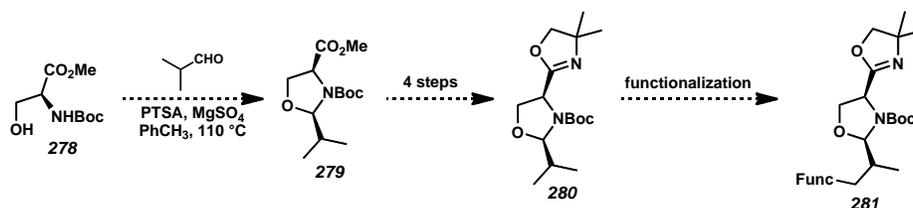
In the synthesis of oxazoline **276** we imagined one of the difficulties to be the steric bulk of the Boc protecting group, which could inhibit both condensation and functionalization. Alternatively, we imagined locking the amino alcohol as an oxazolidinone, installing the ligating group α to the methyl ester, and then forming the alcohol to generate ligand **284** (Scheme 3.1.8). Formation of the methyl ester followed by synthesis of the oxazolidinone afforded **282**. We imagined that protection of the amine would be necessary for alkylation, so we installed a Boc group to give **283**. Any attempts at alkylation, however, afforded none of the desired product. We attributed this to the propensity of anion **285** to liberate CO_2 before alkylation, leading to decomposition product **286**.

Scheme 3.1.8. Attempted synthesis of an oxazolidinone ligand
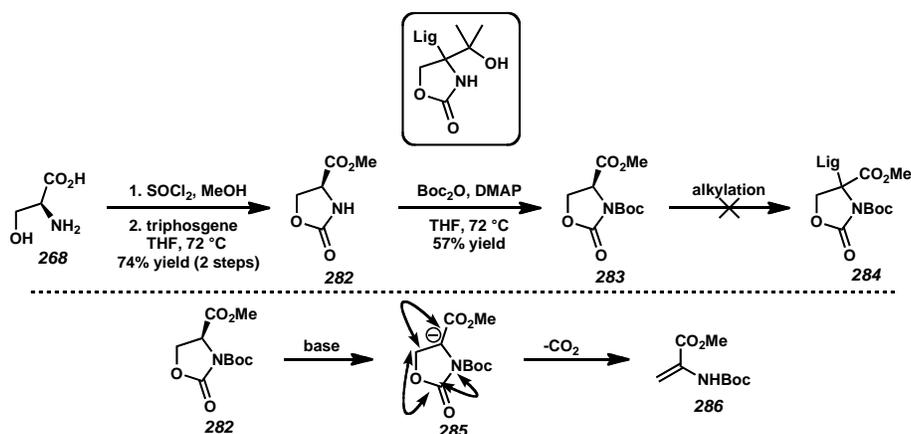
Rather than installing a heterocylic directing group, which has shown to involve numerous steps for ligand synthesis, we envisioned using Yu's work with Boc ligating groups.[3] Starting from L-phenylalanine, we reduced the acid to the corresponding alcohol, then protected the amine to afford amino alcohol **288** (Scheme 3.1.9). Condensation onto cyclohexanecarboxaldehyde afforded *N,O*-acetal **289** in good yield. Treatment with acetoxylation conditions, however, only afforded amide **290** via an oxidation of the acetal center. Subjecting **289** to just PhI(OAc)$_2$ also afforded **290** as the only product. Presumably the Boc group cannot direct the functionalization under these conditions; therefore, the uncatalyzed oxidation is the only transformation that takes place.

Scheme 3.1.9. Ligand design based on phenylalanine
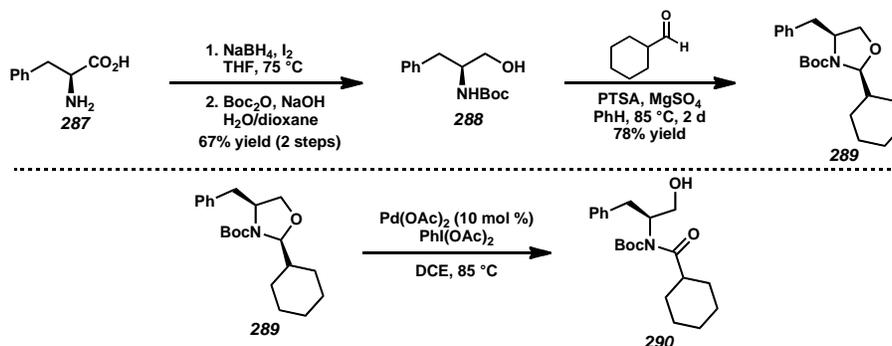
While none of these ligands has shown any success in a C–H functionalization reaction, further exploration is needed. With an improved synthesis of serine derived oxazoline **280**, functionalization may be successful. Additionally, we have already demonstrated that the *N,O*-acetal can be easily cleaved with PTSA in MeOH. Furthermore, the amide alcohol ligand derivatives have not been examined fully and may exhibit potential for C–H functionalization.

3.2 Serine Derived Scaffolds for C–C Bond Formation

We have examined amino acid scaffolds with traditional cyclic heteroatom ligating groups for C–H functionalization. We also sought to examine non-cyclic heteroatom ligating

groups on the L-serine acetalization framework. We began with a carboxylic acid ligating group. Treating L-serine with anhydrous HCl in MeOH followed by TsCl and Et$_3$N afforded the *N*-Ts methyl ester (**291**) (Scheme 3.2.1). Condensation onto benzaldehyde generated *N,O*-acetal **292** as a single diastereomer. We reacted the methyl ester under functionalization conditions to ascertain the directing group ability of the methyl ester. Treatment with PhB(OH)$_2$ in the presence of Pd(OAc)$_2$, K$_2$CO$_3$, benzoquinone, and Cu(OAc)$_2$ afforded an unknown product with no acetal.[4] Acetoxylation of **292** under our standard conditions resulted in recovered starting material.[5] Additionally, subjecting **292** to PhI in the presence of Pd(OAc)$_2$ and K$_2$CO$_3$ resulted in decomposition of the starting material.[6]

Scheme 3.2.1. Attempted functionalization of benzaldehyde

conditions	X	result
Pd(OAc)$_2$, K$_2$CO$_3$, BQ Cu(OAc)$_2$, PhB(OH)$_2$ *t*-AmOH, 90 °C	Ph	no acetal
Pd(OAc)$_2$, PhI(OAc)$_2$ AcOH/Ac$_2$O, 85 °C	OAc	SM
Pd(OAc)$_2$, PhI, K$_2$CO$_3$ *t*-AmOH, 90 °C	Ph	decomp
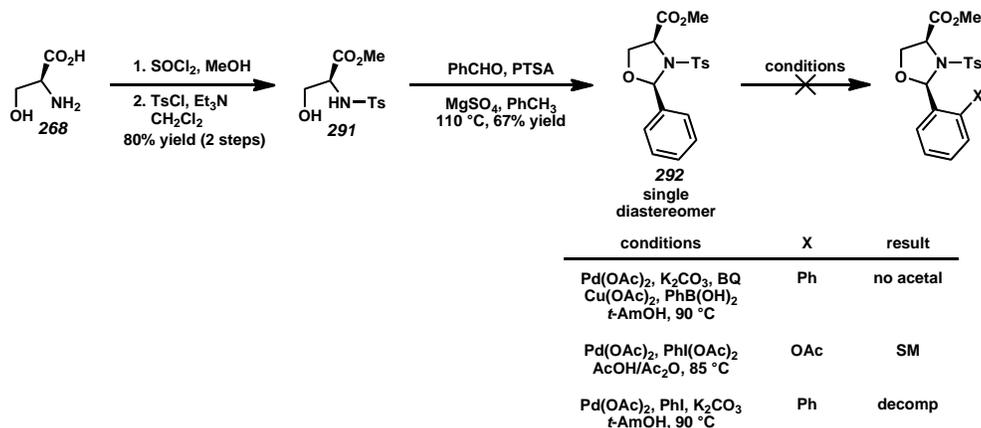
We wanted to examine the success of a carboxylic acid ligating group for C–H functionalization as demonstrated by Yu and coworkers.[7] Saponification of the methyl ester to the corresponding acid **293** occurred in good yield (Scheme 3.2.2). Treatment of **293** to a variety of functionalization conditions afforded none of the desired C–C bond formation. Only cleavage of the acetal was observed, with no recovered starting material.

Scheme 3.2.2. Acid directed functionalization reactions

conditions	X	result
Pd(OAc)$_2$, [Ph$_2$I]BF$_4$ AcOH, 80 °C	Ph	no acetal
Pd(OAc)$_2$, K$_2$HPO$_4$ Cu(OAc)$_2$, BQ, PhB(OH)$_2$ t-BuOH, 100 to 120 °C	Ph	no acetal
Pd(OAc)$_2$, methyl acrylate Cu(OAc)$_2$, DMF 100 to 120 °C	CO$_2$Me	no acetal
Pd(OAc)$_2$, K$_2$HPO$_4$ Cu(OAc)$_2$, BQ PhBF$_3$K, t-BuOH 100 to 120 °C	Ph	no acetal
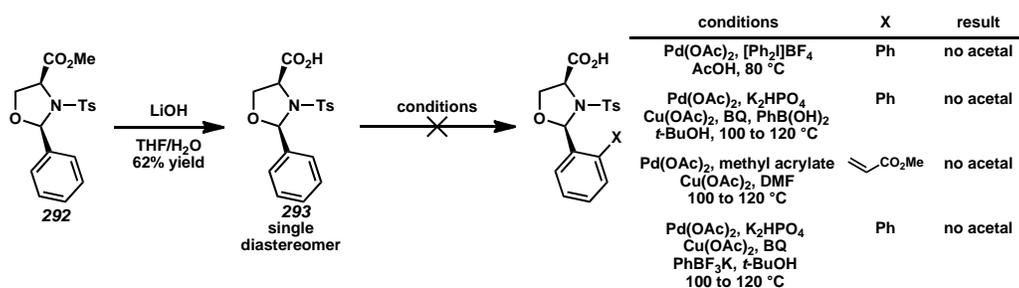
We were able to convert acid **293** directly into the dimethylamide to probe its directing group ability (Scheme 3.2.3). Functionalization conditions utilizing a diaryliodonium salt resulted in no observed arylated product.[7] Treatment with acetoxylation conditions resulted in complete recovery of starting material. Utilizing RhCl(PPh$_3$)$_3$ and methyl acrylate resulted in no observed olefinated product, with only starting material being recovered.[8]

Scheme 3.2.3. Generation of a dimethyl amide ligating group

conditions	X
Pd(OAc)$_2$ [Ph$_2$I]BF$_4$ AcOH, 80 °C	Ph
Pd(OAc)$_2$ PhI(OAc)$_2$ AcOH/Ac$_2$O, 85 °C	OAc
RhCl(PPh$_3$)$_3$ methyl acrylate PhCH$_3$, 150 °C	CO$_2$Me
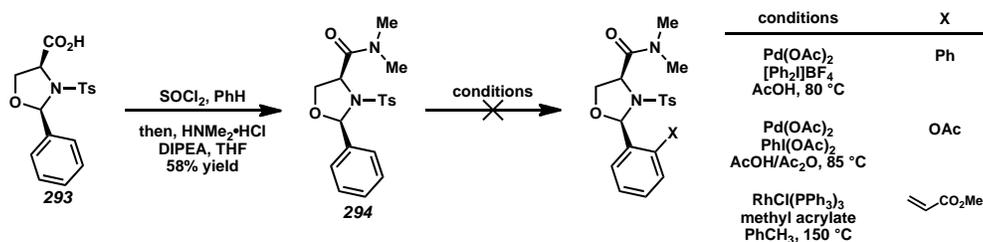
Yu and coworkers have demonstrated very electron poor aryl amides to be effective directing groups for C–H olefination.[9] We converted acid **293** into the pentafluorophenyl amide (**295**) in modest yield (Scheme 3.2.4). Employing Yu's conditions afforded none of the desired olefinated product and only resulted in decomposition. Additionally, subjecting **295** to PhB(OH)$_2$ under palladium catalysis only decomposed the starting material.

Scheme 3.2.4. Installation of a pentafluorophenyl amide ligating group

conditions	X	result
Pd(OAc)₂, benzyl acrylate LiCl, Cu(OAc)₂, DMF 100 to 110 °C	CO₂Bn	decomp
Pd(OAc)₂, K₂CO₃ Cu(OAc)₂, BQ, PhB(OH)₂ t-BuOH, 100 °C	Ph	decomp
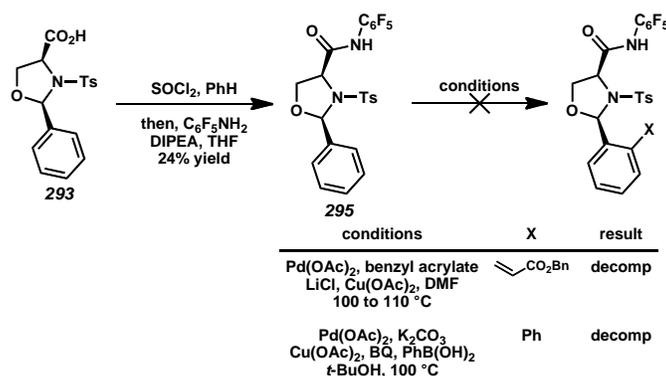
We wanted to investigate sp^3 C–H functionalization with amide or acid ligating groups. Condensing **291** with 2-ethylbutyraldehyde afforded *N,O*-acetal **296** in good yield (Scheme 3.2.5). Saponification of the methyl ester afforded acid **297**, which was converted into the electron deficient amide (**298**) via the acid chloride. Under functionalization conditions, substrate **298** produced none of the desired product and only resulted in decomposition.

Scheme 3.2.5. Attempted functionalization of sp^3 C–H bonds
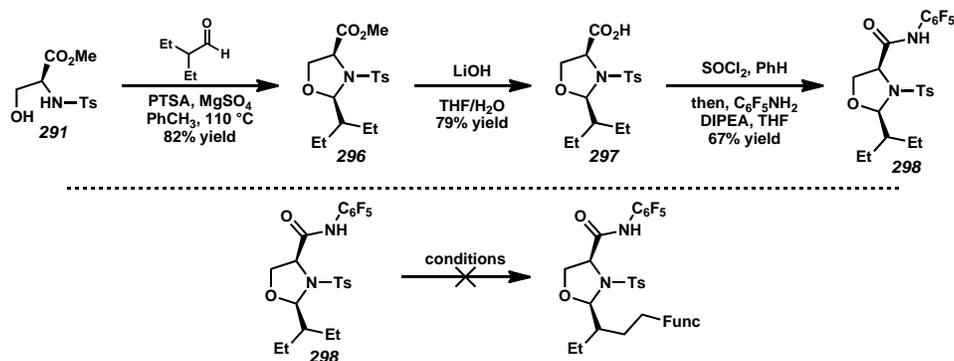
We subjected carboxylic acid **297** to a variety of functionalization conditions with no success. When we subjected **297** to Pd(OAc)₂, Cu(OAc)₂, and methyl acrylate in DMF at 110 °C, however, we observed the formation of olefin **299** in modest conversion (Scheme 3.2.6). We imagined that the product arose from a coupling between the *p*-tolyl moiety of the *N*-Ts and methyl acrylate. This transformation was also possible utilizing substrate **301**, which was

synthesized in a similar fashion. Treating **301** with Pd(OAc)$_2$, Cu(OAc)$_2$, methyl acrylate and

NaOAc afforded the olefin product in modest yield.

Scheme 3.2.6. Formation of C–C bond under palladium catalysis
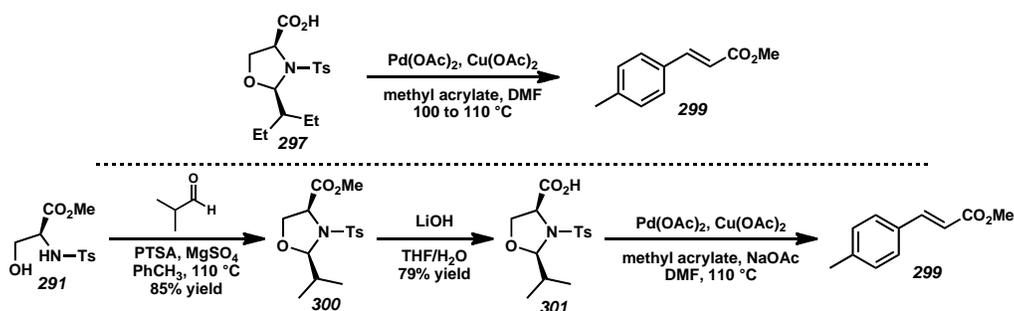
We ran control experiments to ascertain information about the transformation. Treating

297 or **301** to the reaction conditions in the absence of palladium afforded none of the coupled

product, suggesting that the transformation was a palladium catalyzed event. We also removed

the carboxylic acid to determine its ligating role in this reaction. From aminoethanol (**302**)

tosylation of the amine and then condensation onto isobutyraldehyde provided *N,O*-acetal **303**

(Scheme 3.2.7). Treatment of **303** with the coupling conditions afforded none of the olefin

product.   Additionally, *N,N*-dimethyltoluenesulfonamide was subjected to the reaction

conditions, but furnished none of the desired product. We believed the role of the carboxylic

acid to be essential for this transformation.

Scheme 3.2.7. Control experiments for C–C bond formation
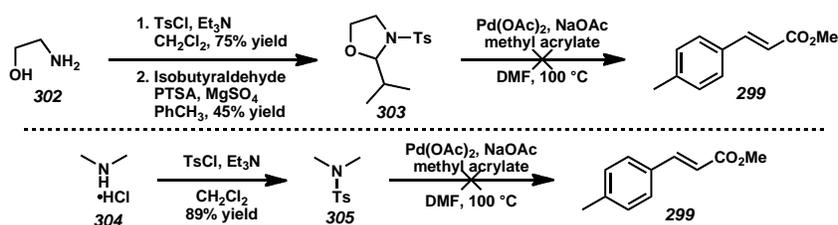
We sought to examine this transformation further by optimization. Treating **301** with the

conditions and copper afforded a new product, resulting from sulfinate addition into the Michael

acceptor (Scheme 3.2.8). We examined the role of copper in this reaction and the corresponding

product distribution. Treating **301** with Pd(OAc)$_2$ and 2 equiv of Cu(OAc)$_2$ gave 10% of olefin

product **299** and 4% of sulfone product **306**. Running the reaction in the absence of Cu(OAc)$_2$

affords 15% yield of the olefin product and 32% of the sulfone product. Changing the catalyst to

Pd(dba)$_2$, a source of Pd^0, and in the absence of Cu(OAc)$_2$, the reaction conditions afforded only

8% yield of the olefin product and 43% yield of the sulfone product. In the absence of Pd^{II}, it is

apparent that the nucleophilic addition into methyl acrylate is the dominant transformation.

Scheme 3.2.8. Initial optimization yields a new product

Pd (10 mol %)	Cu(OAc)$_2$	yield (%) 299 : 306
Pd(OAc)$_2$	2.0 equiv	10 : 4
Pd(OAc)$_2$	none	15 : 32
Pd(dba)$_2$	none	8 : 43
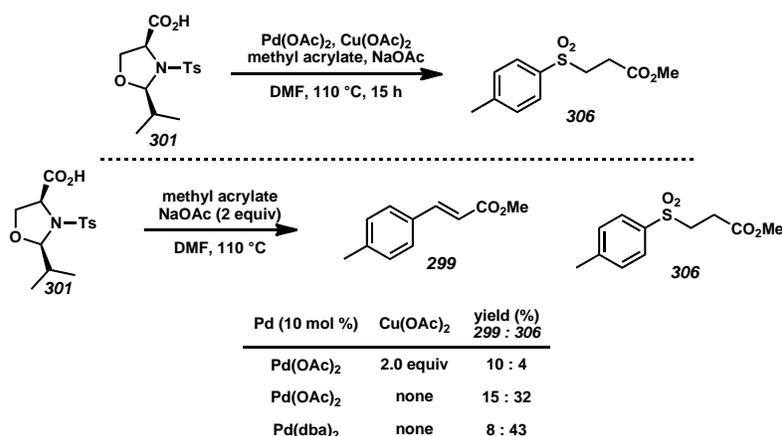
We next investigated solvent optimization and different bases (Table 3.2.1). We decided

to employ methyl ester **300** for optimization purposes, as using the corresponding acid likely

resulted in additional decomposition pathways and lower yields due to the acidic proton.

Employing the reaction conditions in DMF for 15 h afforded 35% of olefin product **299** and 32%

of sulfone product **306**. Employing different solvents afforded only trace amounts of product.

Modifying the base to K$_2$CO$_3$ resulted in the formation of a new product, oxazoline **307**. In

DMF, 36% of the olefin product and 45% of the oxazoline product was observed, with no

sulfone product. In CH$_3$CN, equal amounts of olefin and sulfone products were obtained, with

the majority of the product arising from oxazoline **307**.

Table 3.2.1. Base and solvent screen

base	solvent	yield (%) 299 : 306 : 307
NaOAc (2 equiv)	DMF	35 : 32 : 0
NaOAc (2 equiv)	*t*-AmOH	trace
NaOAc (2 equiv)	CH$_3$CN	none
NaOAc (2 equiv)	DCE	none
K$_2$CO$_3$ (1 equiv)	DMF	36 : 0 : 45
K$_2$CO$_3$ (1 equiv)	CH$_3$CN	15 : 15 : 46
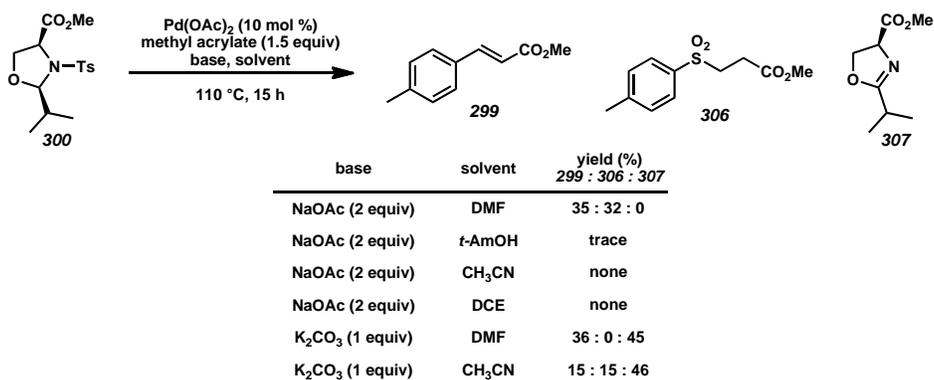
We continued our optimization examining the palladium catalyst, base and solvent (Table 3.2.2). Treating substrate **300** with Pd(OAc)$_2$ and K$_2$CO$_3$ in DMSO afforded 26% of olefin **299** and 41% of oxazoline **307**. Modifying the palladium catalyst to Pd(dba)$_2$, a source of Pd^0, afforded only 10% of sulfone product when K$_2$CO$_3$ was employed in CH$_3$CN. When K$_2$HPO$_4$ or Na$_2$CO$_3$ were employed, no products were isolated. It was apparent that a source of Pd^{II} was necessary for the formation of olefin **299**. Employing PdCl$_2$ in DMF, 17% of the olefin product and 40% of the oxazoline product were isolated, with no observable sulfone. Employing the same conditions in DMSO afforded 87% yield of the olefin product and 72% of the oxazoline product. It seemed evident that the oxazoline product and olefin were arising from the same starting material. Employing Na$_2$CO$_3$ or K$_2$HPO$_4$ provided less than 30% of the olefin product, with no sulfone or oxazoline isolated from the reaction. The use of a more soluble palladium catalyst, PdCl$_2$(CH$_3$CN)$_2$ or PdCl$_2$(PPh$_3$)$_2$, did not lead to an improvement in yield, with only 53% and 38% of olefin product isolated, respectively.

Table 3.2.2. Catalyst and base screen

Pd (10 mol %)	base	solvent	yield (%) 299 : 306 : 307
Pd(OAc)$_2$	K$_2$CO$_3$	DMSO	26 : 0 : 41
Pd(dba)$_2$	K$_2$CO$_3$	CH$_3$CN	0 : 10 : 0
Pd(dba)$_2$	K$_2$HPO$_4$	CH$_3$CN	0 : 0 : 0
Pd(dba)$_2$	Na$_2$CO$_3$	CH$_3$CN	0 : 0 : 0
PdCl$_2$	K$_2$CO$_3$	DMF	17 : 0 : 40
PdCl$_2$	K$_2$CO$_3$	DMSO	87 : 0 : 72
PdCl$_2$	Na$_2$CO$_3$	DMSO	29 : 0 : 0
PdCl$_2$	K$_2$HPO$_4$	DMSO	10 : 0 : 0
PdCl$_2$(CH$_3$CN)$_2$	K$_2$CO$_3$	DMSO	53 : 0 : 51
PdCl$_2$(PPh$_3$)$_2$	K$_2$CO$_3$	DMSO	38 : 0 : 54

We wanted to further investigate the role of base in this transformation. While NaOAc and K$_2$CO$_3$ appeared to work well in most reactions, several different salts of the acetate or carbonate bases did not work at all. Using KHCO$_3$ as the base, we obtained 33% yield of the olefin product and 40% yield of oxazoline **307** (Table 3.2.3). Employing hydroxide salts like barium and calcium resulted in no reaction. After increasing the equiv of NaOAc we observed an improvement in yield of olefin product to 71%, with nearly 50% of the oxazoline product. KOAc also functioned well in this transformation, providing almost 60% yield of the olefin product when 2 equiv were used. LiOAc also afforded nearly 60% yield of the olefin, with a comparable amount of oxazoline product. Employing different carbonate bases gave mixed results, with strong bases causing decomposition and weaker salts providing no reaction. The use of different carboxylate salts was mostly ineffective, with sodium pivalate providing the greatest yield at 54%.

Table 3.2.3. Extensive examination of bases

PdCl$_2$ (10 mol %)
methyl acrylate (1.5 equiv)
base, DMSO (0.1 M)

90 or 100 °C, 2-6 h

300 → **299** **306** **307**

base	yield (%) 299 : 306 : 307	base	yield (%) 299 : 306 : 307
KHCO$_3$ (2 equiv)	33 : 0 : 40	LiOAc (2 equiv)	59 : 0 : 52
Ba(OH)$_2$ (0.5 equiv)	none	Cs$_2$CO$_3$ (1 equiv)	complex
Ca(OH)$_2$ (0.5 equiv)	none	Li$_2$CO$_3$ (1 equiv)	none
NaOAc (4 equiv)	71 : 0 : 46	t-BuCO$_2$Na (2 equiv)	54 : 0 : 43
KOAc (2 equiv)	59 : 0 : 37	Cl$_3$CCO$_2$Na (2 equiv)	10 : 0 : 0
KOAc (1.5 equiv)	46 : 0 : 41	PhCO$_2$Na (2 equiv)	29 : 0 : 9

We began assembling our evidence to get a mechanistic picture of the coupling reaction. It was evident that a Pd^{II} salt was the active catalyst, as Pd^0 catalysts were ineffective at generating olefin product **299**. PdCl$_2$ was the best palladium catalyst examined, while DMSO was the most effective solvent. There also appeared to be a significant dependence on the type of base used in the reaction. It seemed essential that either an acetate anion or potassium cation be present for significant product formation. Additionally, we had not confirmed whether olefin **299** and oxazoline **307** were originating from the same starting material.

We examined the literature for any background information that could help us understand the transformation. There have been some examples of C–S bond activation of sulfones via nickel catalysis[10] and sulfonyl chlorides via palladium catalysis.[11] One such example demonstrates the C–S bond activation of a sulfonamide via a Kumada coupling with nickel. More recently, Deng and coworkers discovered a desulfitative Heck reaction employing sodium sulfinates and activated olefins to generate coupled olefin products (Scheme 3.2.9).[12] Treating **308** with methyl acrylate under palladium catalysis and O$_2$ afforded the olefin product and only trace amounts of sulfone **310**. The sulfinate adds into Pd^{II} with loss of SO$_2$ to afford the activated Pd^{II} species. Migratory insertion followed by β-hydride elimination afforded the olefin

product in good yield. The use of oxygen as a terminal oxidant was essential for regenerating the

Pd^{II} catalyst.

Scheme 3.2.9. Desulfitative Heck reaction
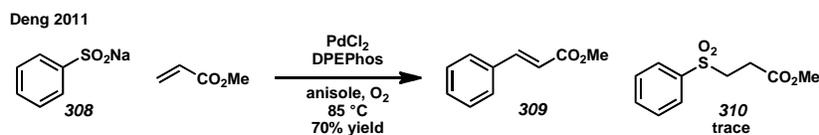
Based on this example, we imagined we might be generating the sulfinate salt, which

could then undergo the desulfitative Heck coupling to afford the olefin product. Subjecting

sodium sulfinate **311** to our reaction conditions only afforded 30% yield of the coupled olefin

product, suggesting that our reaction was not going through a discrete sodium sulfinate (Scheme

3.2.10).

Scheme 3.2.10
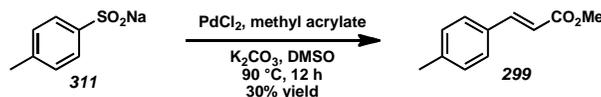
We wanted to further investigate the structural aspects that were necessary for the

coupling reaction to occur.   Starting from L-proline, the amine was tosyl protected and

subsequent esterification afforded the methyl ester (**312**) (Scheme 3.2.11). Subjecting **312** to the

reaction conditions afforded none of the desired olefin product. We also wanted to examine the

elimination of the cyclic structure, and ascertain its role in the coupling reaction. We tosylated

glycine methyl ester to afford **314**. Methylation with MeI and Cs$_2$CO$_3$ afforded acyclic substrate

315. Treating this substrate with the reaction conditions containing NaOAc afforded none of the

coupled product. Under the same conditions with K$_2$CO$_3$ as the base, however, product **299** was

isolated in 38% yield. It is interesting that the proline derivative is unreactive under our coupling

conditions, while the parent substrate, cyclic **300**, affords product in up to 90% yield.

Additionally, acyclic product **315** is converted to the coupled product in nearly 40% yield. Evidently there are structural aspects to this transformation that have yet to be elucidated.

Scheme 3.2.11. Synthesis of additional substrates
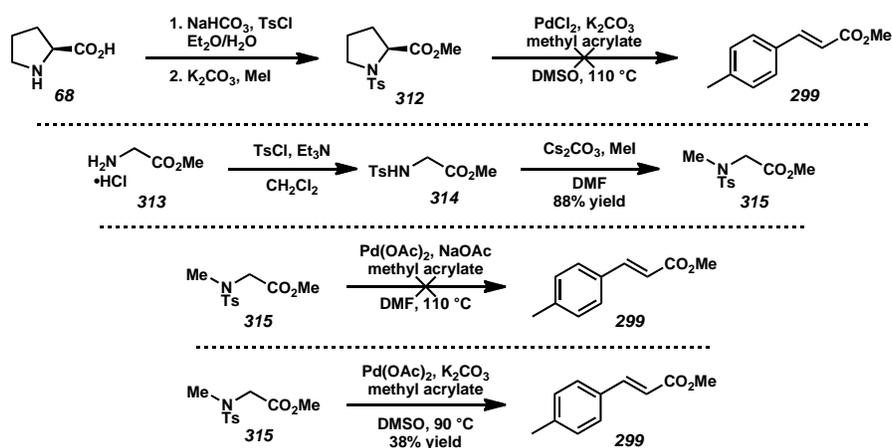
Based on the conversion of glycine derived substrate **315** to olefin **299**, we sought to examine the role of the methyl ester. Treating amine **316** with TsCl followed by alkylation afforded the tertiary amine (**317**) (Scheme 3.2.12). Reacting substrate **317** with the coupling conditions afforded less than 10% yield of the olefin product, demonstrating the necessity of the methyl ester, potentially as a ligating group. Similarly, we sought to examine a more sterically hindered alanine derived substrate. Subjecting **318** to anhydrous HCl in MeOH, followed by tosylation and alkylation with MeI afforded substrate **319**. Subjecting **319** to the coupling conditions afforded only 21% yield of the olefin product, signifying a decrease in reactivity, likely due to the α-methyl group. Furthermore, we wanted to eliminate the possibility of the generation of a sulfinate salt. Methyl alanine methyl ester hydrochloride was reacted with TsCl and subsequent methylation with MeI afforded sulfonamide **321**. Employing coupling conditions, however, afforded none of the desired olefin product. This result may signify that the mechanism of the coupling reaction proceeds through a sulfinate salt, although the increased steric hinderance cannot be discredited.

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Scheme 3.2.12. Further structural evaluation
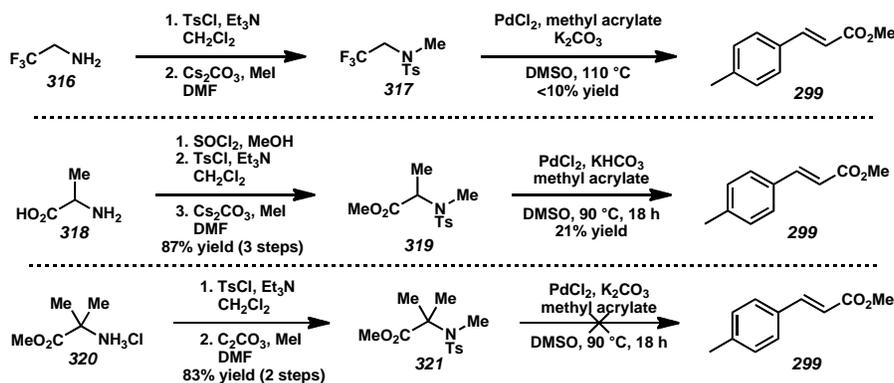
We speculated that the reaction could be initiated by an elimination of the sulfinate salt, followed by a desulfitative Heck coupling as demonstrated by Deng and coworkers. To further probe this pathway, we imagined generating acetonide **322** to the conditions (Scheme 3.2.13). With no acetal proton available for elimination to form the oxazoline, we imagined the reaction could not occur if it were proceeding via an elimination of the sulfinate. Treating acetonide **322** with the coupling conditions afforded a 1 to 1.7 mixture of coupled product to oxazoline **323**. It is possible that an elimination via the α-proton occurs first, followed by isomerization to the oxazoline **323**. This result again indicates that the reaction may be proceeding through an elimination to a sulfinate.

Scheme 3.2.13. Formation of an acetonide
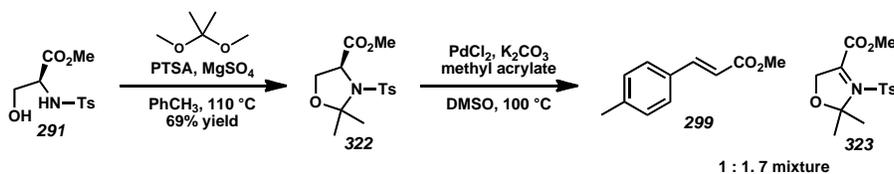
Wanting to improve the yield of this transformation, we examined catalyst turnover more closely. Based on the work by Deng and coworkers, we imagined that the addition of a terminal oxidant might assist the conversion of Pd^0 to the active Pd^{II} species. Employing K_2CO_3 or KOAc in the presence of air afforded low conversion to olefin and sulfone products, with

significant oxazoline formation (Table 3.2.4). Employing K_2CO_3 or KOAc in the presence of

$Cu(OAc)_2$ afforded only a small amount of olefin product, and in the absence of base there was

no reaction. Using K_2CO_3 and benzoquinone afforded a complex mixture of products, while

employing KOAc afforded almost 50% yield of olefin product **299**. Utilizing KOAc in the

presence of an oxygen balloon and catalytic $Cu(OAc)_2$ afforded a 63% yield of the olefin

product. Removal of $Cu(OAc)_2$ and only using an oxygen balloon afforded 67% of the desired

olefin product, with 30% of the isolated oxazoline. Treating **300** with catalytic amounts of BQ,

$NaNO_2$ or AgOAc in the presence of KOAc afforded less than a 40% yield of the desired olefin

product.

Table 3.2.4. Additional optimization of C–C bond formation

base	oxidant	yield (%) 299 : 306 : 307	base	oxidant	yield (%) 299 : 306 : 307
K_2CO_3 (1 equiv)	air	18 : 5 : 54	KOAc (2 equiv)	O_2 balloon $Cu(OAc)_2$ (20 mol %)	63 : 0 : 4
KOAc (2 equiv)	air	25 : 38 : 39			
K_2CO_3 (1 equiv)	$Cu(OAc)_2$ (2 equiv)	18 : 7 : 0	KOAc (2 equiv)	O_2 balloon	67 : 0 : 29
none	$Cu(OAc)_2$ (2 equiv)	none	KOAc (2 equiv)	BQ (25 mol %)	36 : 0 : 35
KOAc (2 equiv)	$Cu(OAc)_2$ (2 equiv)	20 : 0 : 0	KOAc (2 equiv)	$NaNO_2$ (25 mol %)	27 : 0 : 7
K_2CO_3	BQ (1 equiv)	complex	KOAc (2 equiv)	AgOAc (25 mol %)	27 : 16 : 29
KOAc	BQ (1 equiv)	47 : 31 : 63			

Continuing with our optimization we wanted to reduce the catalyst loading for the

coupling reaction, and based on the results in Table 3.2.4 the use of O_2 should increase catalyst

turnover. Utilizing 3 equivialents of NaOAc afforded 92% yield of the olefin product and only

18% of the oxazoline product (Table 3.2.5). Reducing the amount of base to 1.5 equiv afforded

only 52% yield of the olefin product, with no byproducts. Utilizing Na_2CO_3 with catalytic

AcOH afforded a 70% yield of the desired product. Additionally, we found that using pulverized

NaOAc gave a 90% yield of the desired olefin product.

Table 3.2.5. Reducing catalyst loading

base	yield (%) 299 : 306 : 307
NaOAc (3 equiv)	92 : 0 : 18
NaOAc (1.5 equiv)	52 : 0 : 0
Na$_2$CO$_3$ (1.5 equiv) AcOH (10 mol %)	70 : 0 : 7
NaOAc (3 equiv) (pulverized)	89: 0 : 0
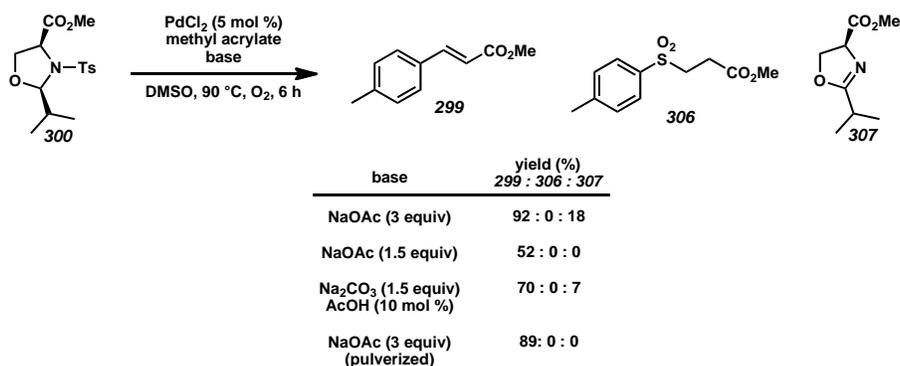
Throughout our optimization, we found that the yields of the corresponding products was inconsistent from reaction to reaction. We ground the NaOAc to make it more soluble and therefore reactive in the coupling reaction. We also tried slow addition of base, which still provided inconsistent results. Furthermore, we were concerned with the oxygen transfer to the reaction vessel, and attempted to modify the mode of oxygen transfer to increase consistency, but were unsuccessful. Lastly, we set up replicate experiments to test the reliability of the transformation (Table 3.2.6). At 1 and 2 equiv of NaOAc, the yields of olefin **299** and oxazoline **307** were relatively constant. Increasing to 3 and 4 equiv of NaOAc, however, afforded very inconsistent results. We attribute this irregularity to the possible rate of elimination of the sulfinate salt. With small amounts of base only small amounts of the sodium sulfinate can be generated, which leads to consistent conversion to product. Larger amounts of base can eliminate the sulfinate quickly, which may lead to other decomposition byproducts rather than the olefin product.

Table 3.2.6. Examination of reaction consistency

equivalents NaOAc	yield (%) 299 : 306 : 307	equivalents NaOAc	yield (%) 299 : 306 : 307
1.0	44 : 0 : 8	3.0	18 : 22 : 12
1.0	44 : 0 : 2	3.0	65 : 0 : 12
2.0	62 : 0 : 13	4.0	54 : 5 : 0
2.0	62 : 0 : 5	4.0	76 : 0 : 13
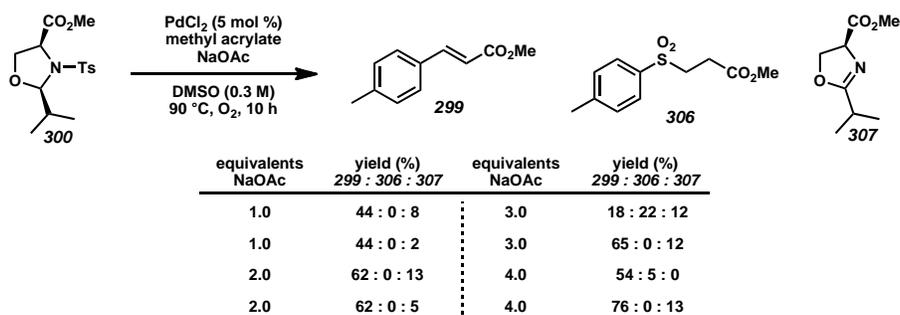
Based on our results we believed that the reaction was occurring via a controlled elimination of the sulfinate salt and then undergoing a desulfitative Heck reaction as demonstrated by Deng and coworkers. Further examination of this transformation and structural analysis is necessary to confirm the sulfinate mechanism. We imagined we could expand the substrate scope, however, which thus far has been limited to activated olefins and a limited number of sulfinate salts. We began our scope evaluation with electron deficient *p*-NO$_2$-sulfonylchloride to form sulfonamide **325** (Scheme 3.2.14). Condensation onto isobutyraldehyde afforded *N,O*-acetal **326**. Treatment with the coupling conditions afforded the coupled product in 31% yield. Additionally, we utilized benzene sulfonyl chloride to generate sulfonamide **328**. Condensation onto isobutyraldehyde afforded substrate **329**, which when subjected to the coupling conditions afforded the desired olefin in 68% yield. Furthermore, we wanted to determine if we could exploit this coupling to form hindered C–C bonds. Coupling with mesitylene sulfonylchloride to from sulfonamide **330** occurred in good yield. Condensation with isobutyraldehyde afforded the desired acetal, that when subjected to the coupling conditions afforded the desired olefin product in 22% yield.

Scheme 3.2.14. Expansion of substrate scope
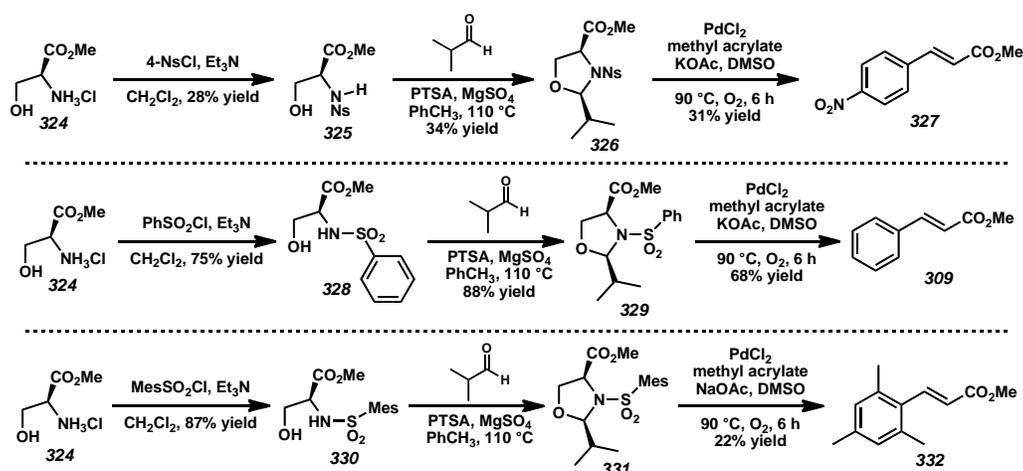
Ultimately, we envisioned our substrate scope extending to triflates, which when utilized in our coupling conditions could install a trifluormethyl group, a functional group of great interest recently. We began with the mesyl derivative to test the feasibility of this reaction with non-aryl coupling partners (Scheme 3.2.15). To install the methane sulfonamide we employed a two-step procedure to generate the sulfonamide and TBS-protect the alcohol.[13] Removal of the TBS group afforded amino alcohol **334**. Condensation onto isobutyraldehyde afforded the desired acetal, that when subjected to the coupling conditions only resulted in decomposition.

Scheme 3.2.15. Attempted coupling of a methane sulfonamide
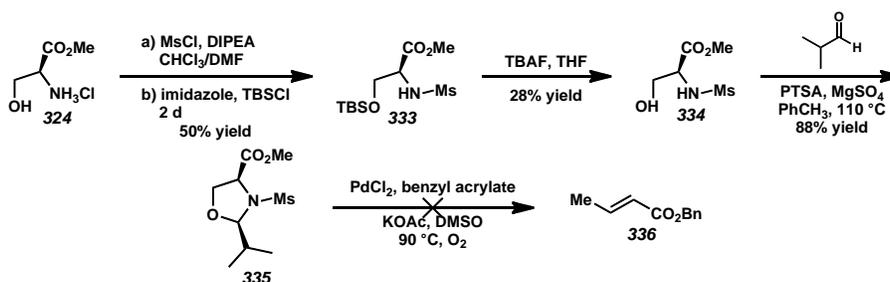
We imagined that performing a Heck reaction with a methyl-Pd^{II} species might prove difficult. Instead we envisioned performing a desulfitative Suzuki coupling, which should prove more facile. Subjecting **300** to our coupling conditions in the presence of PhB(OH)$_2$ afforded

phenyl toluene (**337**) in moderate yield (Scheme 3.2.16).   Further optimization of this transformation needs to be examined to determine its feasibility in coupling with methane or triflouromethane sulfonamides.

Scheme 3.2.16. Coupling with boronic acids
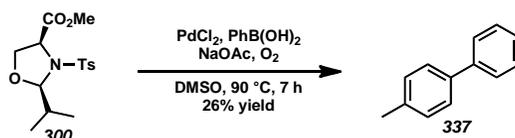
While attempting to develop a methodology to perform C–H functionalization reactions with a transient directing group, we have discovered a relatively unexplored transformation. The extent of this new method to form C–C bonds is not yet known, but could extend to simple methyl or trifluoromethyl Heck couplings with olefins or Suzuki couplings with boronic acids. More investigation into the scope of this transformation could lead to higher utility and mechanistic insight.

3.3   Non-Amino Acid Derived Scaffolds

Looking more closely at the work done with transient directing groups, we re-examined our approach.   Tan and coworkers have used aromatic amino phosphines as acetalization components.[14] We imagined using amino alcohol acetalization components in a similar manner. In our design, we envisioned installing the directing group on the amine component, either directly attached to the aromatic ring, or benzylic to the aromatic ring (Scheme 2.3.1). Before installing a directing group we wanted to determine the feasibility of the condensation reaction and transacetalization.   Amino phenol **338** was condensed with benzaldehyde to form the corresponding imine, then reduced to the secondary amine with $NaBH_4$ to afford **339** in quantitative yield over 2 steps. Condensation of amino phenol **339** with benzaldehyde afforded the *N,O*-acetal in quantitative yield. The product was not stable to silica gel chromatography,

speaking to its high lability. We also wanted to examine the exchangeability of this framework.

Treating **340** with *p*-tolualdehyde with CSA and H$_2$O in dioxane overnight at 50 °C afforded

80% conversion to exchange product **341**.

Scheme 3.3.1. New catalyst design
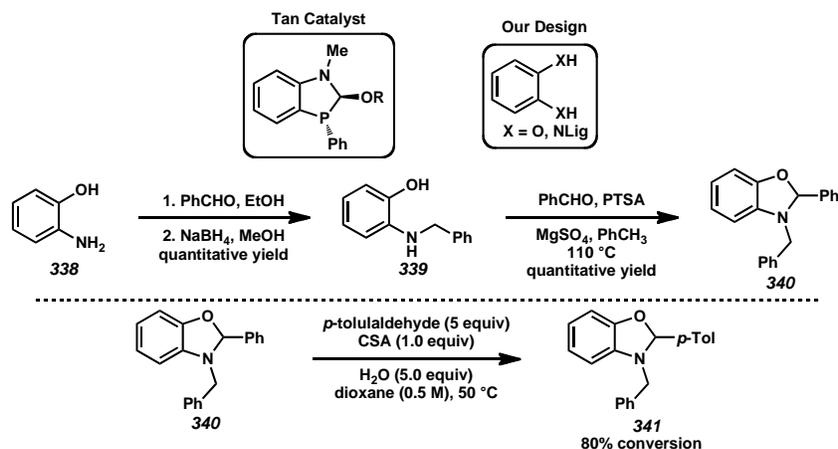
Encouraged by these preliminary experiments we sought to install a ligating group to

examine the possibility of functionalization of the desired C–H bond. From amino phenol **338**

we were able to form the corresponding imine with 2-pyridinecarboxaldehyde (Scheme 3.3.2).

Treatment with NaBH$_4$ afforded amino phenol **342** in 97% yield over two steps. Condensation

with benzaldehyde only proceeded to 50% conversion to afford *N,O*-acetal **344**. When the crude

mixture was reacted under acetoxylation conditions, a complex mixture of products, with

observed trace amounts of acetoxylated benzaldehyde were obtained.

Scheme 3.3.2. Attempts at acetoxylation
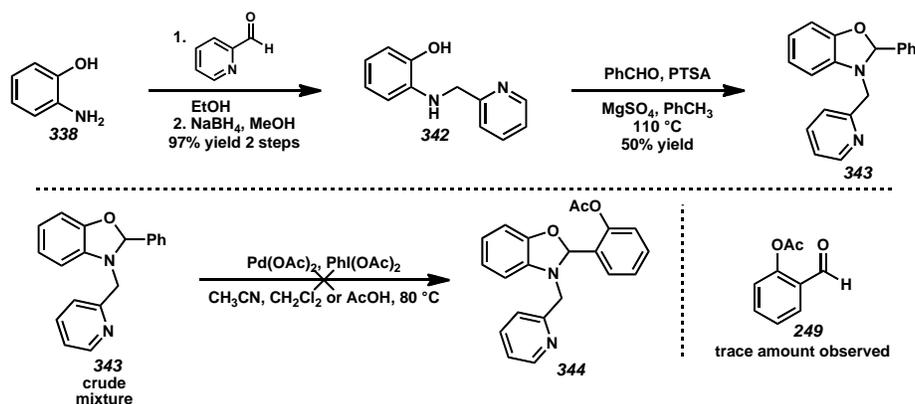
We were concerned about the conformational flexibility of substrate ligand adduct **343** and its inability to direct C–H functionalization regioselectively. We sought to inhibit this flexibility by attaching the pyridyl ligating group directly to the nitrogen via Buchwald's arylation chemistry.[15] Coupling with iodopyridine and amino phenol **338** was unsuccessful (Scheme 3.3.3). Alternatively we imagined installing a carbonyl moiety at the benzylic position to reduce flexibility of the ligating group. Amide coupling with picolinic acid afforded amide **346**. Unfortunately, condensation with aromatic or aliphatic aldehydes was unsuccessful at generating the desired substrate ligand adduct. Additionally, substitution of the 2-pyridyl ligating group for any other amide did not improve the condensation onto any aldehyde.

Scheme 3.3.3. Altering pyridyl connectivity
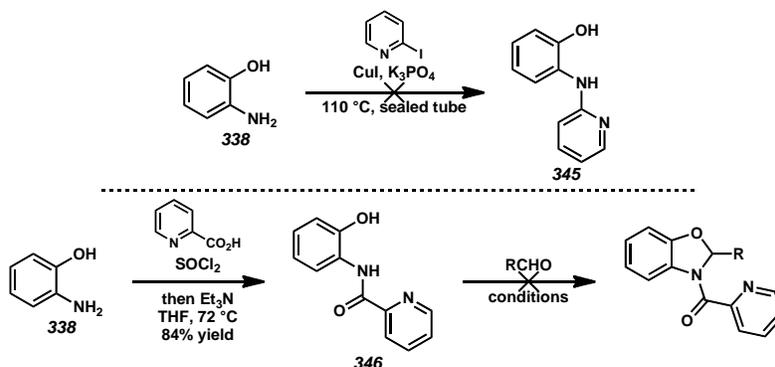
Next we sought to alter the connectivity of the acetalization components and positioning of the ligating group. Starting from amino alcohol **347**, we formed *N,O*-acetal **348** via condensation with 2-pyridylcarboxaldehyde (Scheme 3.3.4).[16] Treatment with LAH afforded the hydroamination product, followed by condensation with benzaldehyde to afford *N,O*-acetal **349** in good yield, which was stable to chromatography. Subjecting **349** to acetoxylation conditions afforded an acetoxylated product in 20% yield, but it was unclear whether the desired C–H bond had been activated to afford **350**, rather than acetate **351**. We subjected the product of the acetoxylation to an exchange reaction with benzaldehyde, but no salicylaldehyde or acylated salicylaldehyde was observed in the reaction mixture.

Scheme 3.3.4. Altering the connectivity of acetalization components
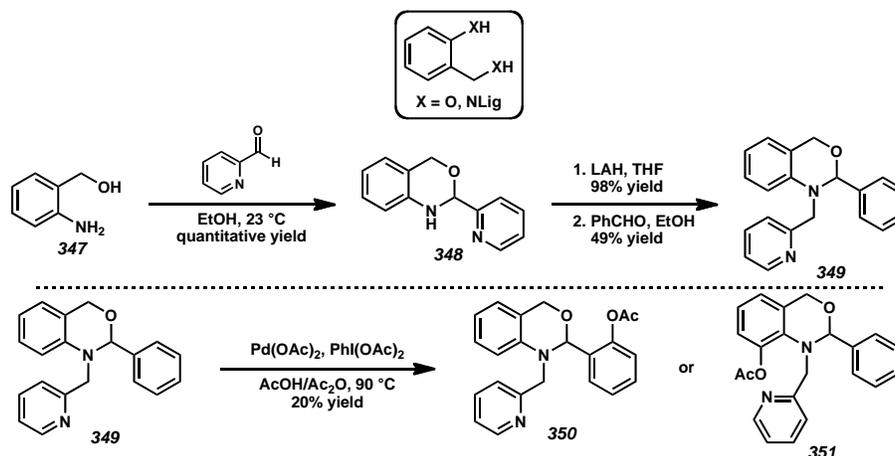
We next tried to form a C–C bond via C–H functionalization utilizing substrate ligand adduct **349** (Scheme 3.3.5). Treating **349** with PhI in the presence of palladium and silver resulted in decomposition with no observable product **352**. Treatment with PhB(OH)$_2$ in the presence of palladium, copper, K$_2$CO$_3$ and benzoquinone only resulted in recovery of starting material. Likewise, coupling with PhI in the presence of palladium and K$_2$CO$_3$ only provided recovered staring material. We also wanted to investigate the functionalization of *sp^3* C–H bonds using ligand **348**. Condensation onto isobutyraldehyde afforded *N,O*-acetal **353** in

moderate yield, again stable to chromatography. Subjecting **353** to acetoxylation conditions, however, resulted only in decomposition.

Scheme 3.3.5. Other C–H functionalization reactions
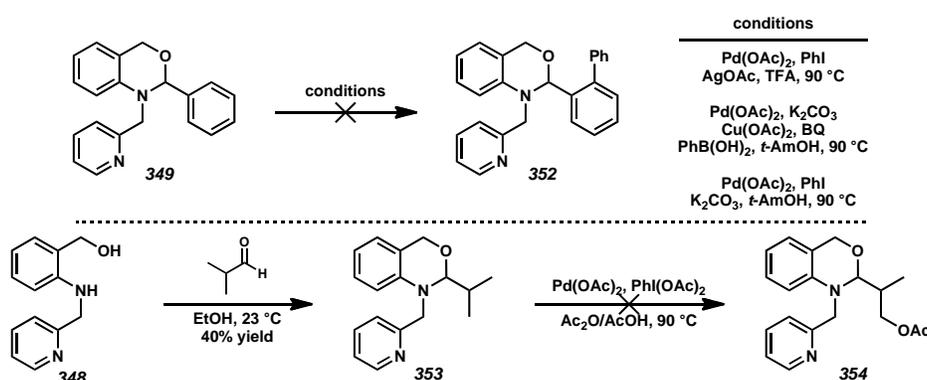
Lastly, we sought to alter the positioning of the alcohol and amine components. From amino alcohol **355** we installed the ligating group in two steps in excellent yield (Scheme 3.3.6). Condensation onto benzaldehyde, however, did not occur to afford *N,O*-acetal **357**. We then condensed the ligand onto isobutyraldehyde to afford *N,O*-acetal **358**. Subjecting **358** to palladium and a diaryliodonium salt to perform a C–H arylation was unsuccessful and resulted in decomposition.

Scheme 3.3.6. Alteration of acetalization components
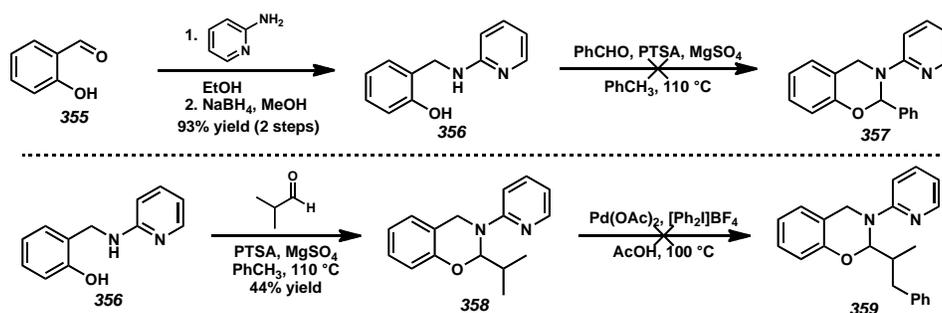
The problems associated with these non-amino acid derived ligands mostly lie with the instability of the *N,O*-acetal or poor regioselectivity of the functionalization. Using the aryl ring as the backbone lends several possible *sp²* C–H bonds to be functionalized. Alteration of the aryl

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ring by placing substituents in the ortho positions may resolve the issue of regioselectivity, but could result in other complications, particularly ease of ligand synthesis.

3.4   Redesigning Substrate-Ligand Relationship

We wondered if a different substrate ligand relationship may facilitate reaction development. Rather than the ligand coordinating through an aldehyde, we envisioned forming the substrate ligand adduct via an ether linkage. Employing a vinylogous ester scaffold, we may achieve good levels of exchangeability with C–H functionalization (Scheme 3.4.1). We first sought to install the ligating group via alkylation. Treatment of dione **360** with 2-bromomethyl pyridine under various conditions did not afford the desired enol product. We attempted a Knovenagel condensation followed by *in situ* reduction with the Hanztch ester to give the enol product.[17] We were able to form the desired product, but it was inseparable from the byproducts of the reaction. Employing a two-step Knovenagel/hydrogenation procedure afforded the desired enol product (**361**).[18] From this product, formation of the vinylogous ester with BnOH and Dean-Stark conditions should be possible to achieve the desired test substrate.

Scheme 3.4.1. Installation of a pyridyl ligating group
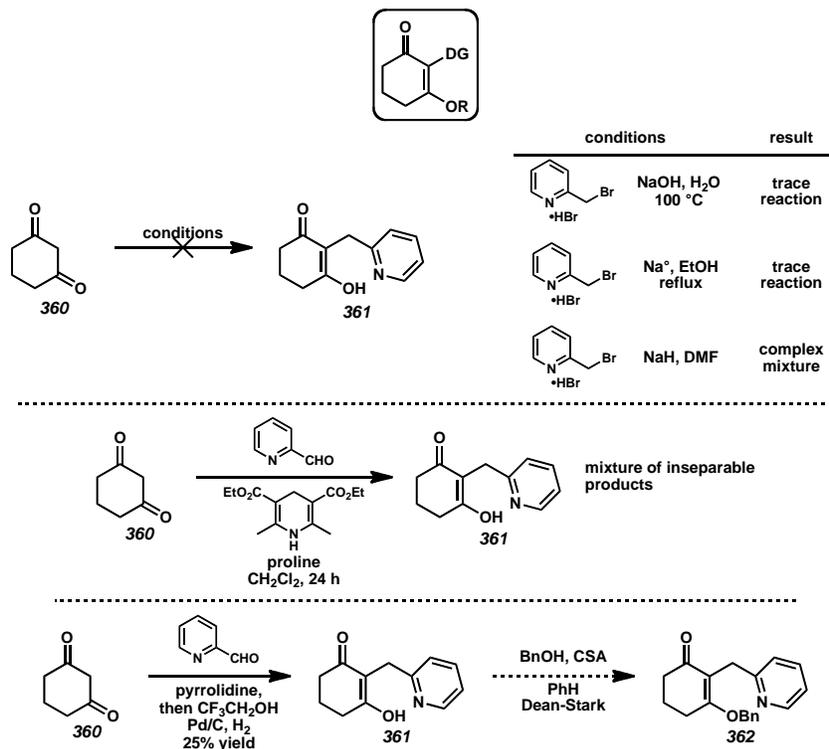
Because of the difficulties associated with installing a pyridyl ligating group, we elected to examine other ligating groups. Installation of an allyl group via alkylation occurred in moderate yield (Scheme 3.4.2).[19] Esterification with BnOH under Dean-Stark conditions afforded the benzyl ester. Oxidative cleavage and condensation with hydroxyl amine would generate the oxime ligand (**366**).

Scheme 3.4.2. Installation of an oxime ligating group
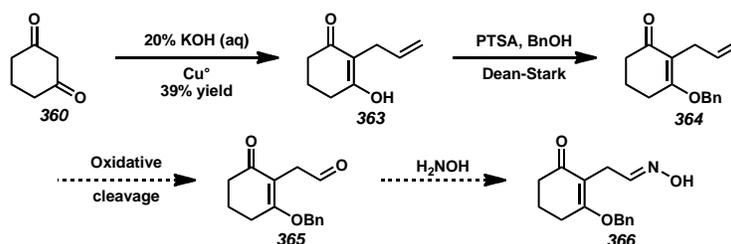
These initial ligand syntheses for alcohol exchange need more synthetic exploration. Upon synthesis of **366**, both exchangeability and directing group capability should be examined.

Furthermore, the exploration of other ligating groups may improve ligand synthesis and/or exchangeability of alcohol substrates.

3.5   Conclusions

Our investigations into ligands for C–H functionalization have revealed *N,O* acetals as promising scaffolds. Modulating the ligating groups have exhibited promising results. These studies suggest that further investigations into the serine derived scaffold could be beneficial. These explorations have also revealed new reactivity, which upon further examination may be advantageous to the synthetic community.

3.6   References and Notes

[1] Yoshifuji, S.; Tanaka, K.; Kawal, T.; Nitta, Y. *Chem. Pharm. Bull.* **1986**, *34*, 3873-3881.

[2] (a) Boes, M.; Naef, R.; Schweizer, W. B.; Seebach, D. *J. Am. Chem. Soc.* **1983**, *105*, 5390-5398. (b) Sting, A. R.; Hoffman, M.; Seebach, D. *Angew. Chem., Int. Ed. Engl.* **1996**, *35*, 2708.

[3] Wang, D. -H.; Wu, D. -F.; Yu, J. -Q. *Org. Lett.* **2006**, *8*, 3387.

[4] Chen, X.; Li, J. –J.; Hao, X. –S.; Goodhue, C. E.; Yu, J. –Q. *J. Am. Chem. Soc.* **2006**, *128*, 78.

[5] Dick, A. R.; Hull, K. L.; Sanford, M. S. *J. Am. Chem. Soc.* **2004**, *126*, 2300.

[6] Satoh, T.; Kawamura, Y; Miura, M.; Nomura, M. *Angew. Chem., Int. Ed. Engl.* **1997**, *36*, 1740.

[7] (a) Deprez, N. R.; Sanford, M. S. *J. Am. Chem. Soc.* **2009**, *131*, 11234. (b) Chen, X.; Engle, K. M.; Wang, D. –H.; Yu, J. –Q. *Angew. Chem. Int. Ed.* **2009**, *48*, 5094-5115, and references therein.

[8] Colby, D. A.; Bergman, R. G.; Ellman, J. A. *Chem. Rev.* **2010**, *110,* 624-655, and references therein.

[9] Wasa, M.; Engle, K. M.; Yu, J. –Q. *J. Am. Chem. Soc.* **2010**, *132*, 3680-3681.

[10] (a) Dubbaka, S. R.; Vogel, P. *Angew. Chem. Int. Ed.* **2005***, 44*, 7674. (b) Volla, C. M.; Vogel, P. *Angew. Chem. Int. Ed.* **2008**, *47*, 1305.

[11] Zhao, X.; Dimitrijevic, E.; Dong, V. M. *J. Am. Chem. Soc.* **2009**, *131,* 3466-3467, and references therein.

[12] Zhou, X.; Luo, J.; Liu, J.; Peng, S.; Deng, G –J. *Org. Lett.* **2011**, *13*, 1432-1435.

[13] Zhong, F.; Wang, Y.; Han, X.; Huang, K. –W.; Lu, Y. *Org. Lett.* **2011**, *13*, 1310-1313.

[14] (a) Lightburn, T. E.; Dombrowski, M. T.; Tan, K. L. *J. Am. Chem. Soc.* **2008**, *130*, 9210. (b) Worthy, A. D.; Joe, C. L.; Lightburn, T. E.; Tan, K. L. *J. Am. Chem. Soc.* **2010**, *132*, 14757.

[15] Maiti, D.; Buchwald, S. L. *J. Am. Chem. Soc.* **2009**, *131*, 17423-17429.

[16] Yotphan, S.; Bergman, R. G.; Ellman, J. A. *Org. Lett.* **2009**, *11*, 1511-1514.

[17] Ramachary, D. B.; Kishor, M. *J. Org. Chem.* **2007**, *72*, 5056-5068.

[18] Kennedy, J. W. J.; Veietrich, S.; Weinmann, H.; Brittain, D. E. A. *J. Org. Chem.* **2008**, *73*, 5151-5154.

[19] Prakash, C.; Mohanakrishnan, A. K. *Eur. J. Org. Chem.* **2008**, 1535-1543.

3.7   Experimental Procedures

Materials and Methods.   All reactions were performed under an argon atmosphere unless otherwise noted.   Tetrahydrofuran, *N,N*-dimethylformamide, dichloromethane, hexanes, and toluene were purified by passing through activated alumina columns.   Diisopropylamine was distilled over CaH_2. 2-Fluoropyridine was freshly distilled before use. All other reagents were used as received unless otherwise noted.   Commercially available chemicals were purchased from Alfa Aesar (Ward Hill, MA), Sigma-Aldrich (St. Louis, MO), Gelest (Morrisville, PA), Oakwood Products (West Columbia, SC), Strem (Newburport, MA), Mallinckrodt Chemicals (Phillipsburg, NJ), Spectrum (Gardena, CA) Fischer Scientific (Fair Lawn) and TCI America (Portland, OR). Qualitative TLC analysis was performed on 250 mm thick, 60 Å, glass backed, F254 silica (Silicycle, Quebec City, Canada).   Visualization was accomplished with UV light and exposure to either *p*-anisaldehyde or $KMnO_4$ solution followed by heating.   Flash chromatography was performed using Silicycle silica gel (230-400 mesh). ^{1}H NMR spectra were acquired on either a Varian Mercury 300 (at 300 MHz), a Varian Inova 400 (at 400 MHz), or a Varian 400 MR (at 400 MHz) and are reported relative to $SiMe_4$ (δ 0.00). ^{13}C NMR spectra were acquired on either a Varian Inova 400 (at 100 MHz), a Varian Mercury 300 (at 75 MHz), or a Varian 400 MR (at 100 MHz) and are reported relative to $SiMe_4$ (δ 0.0).   All IR spectra were obtained on NaCl plates (film) with either a Nicolet Magna FTIR 760, a Nicolet 380 FTIR, or a Bruker Tensor 27. High resolution mass spectrometry data were acquired by the Colorado State University Central Instrument Facility on an Agilent 6210 TOF LC/MS.

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To **256** (500 mg, 3.87 mmol) in MeOH (3.90 mL) at 0 °C was added $SOCl_2$ (0.390 mL, 5.34 mmol) slowly. Stirred an additional 5 min at 0 °C, then 23 °C overnight. $NaHCO_3$ (77.4 mg) was added, then the mixture was filtered through celite. The celite was washed with hot MeOH (3 x 15 mL). The organic solvent was removed under reduced pressure to afford the methyl ester (554 mg, 99% yield) as a murky yellow oil, with no further purification.

To methyl ester (19.1 mmol) and DMAP (468 mg, 3.83 mmol) in CH_3CN (19.1 mL) was added Boc_2O (4.22 g, 19.3 mmol) at 23 °C. The reaction was stirred at 23 °C overnight, at which point the solvent was removed under reduced pressure. To the residue was added 1 M $KHSO_4$ and EtOAc. The aqueous layer was extracted with EtOAc (3 x 20 mL). The organics were washed with brine, dried over $MgSO_4$ and concentrated. The crude residue was purified via flash chromatography (1:1 hexanes:EtOAc) to afford **257** (2.75 g, 59% yield, R_f = 0.30 in 1:1 hexanes:EtOAc) as a white solid.
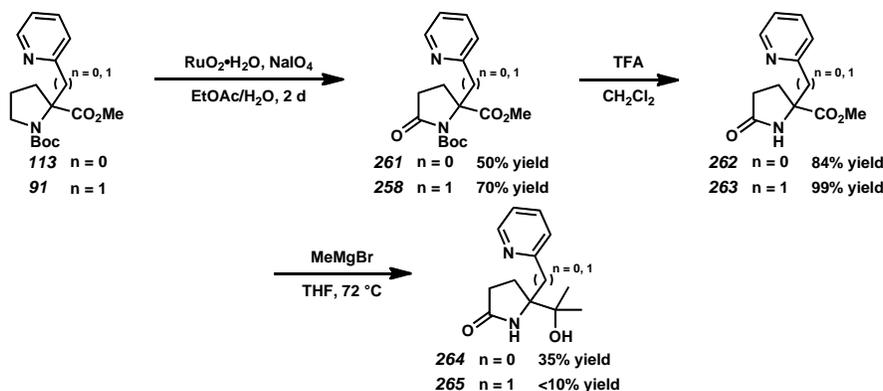
To $RuO_2 \cdot H_2O$ (4.3 mg, 0.0325 mmol) and $NaIO_4$ (4.30 mL, 10% aq) was added **113** (100 mg, 0.325 mmol) in EtOAc (1.60 mL) at 23 °C. The reaction was stirred for 2 d at 23 °C. IPA was added and stirred at 23 °C for 2 h. The black precipitate was filtered and washed with EtOAc. The organics were washed with brine (2 x 10 mL), dried over $MgSO_4$ and concentrated. The crude residue was purified via flash chromatography (3:1 to 1:1 hexanes:EtOAc) to afford **261** (59.3 mg, 57% yield, R_f = 0.25 in 1:1 hexanes:EtOAc) as a beige solid.

To **261** (59.3 mg, 0.185 mmol) in CH_2Cl_2 (0.400 mL) was added TFA (0.29 mL, 3.70 mmol). The reaction was stirred at 23 °C for 1 h. The solvent was removed under reduced pressure. The residue was neutralized with sat. aq. $NaHCO_3$. The aqueous was extracted with CH_2Cl_2 (3 x 10 mL). The organics were dried over Na_2SO_4 and concentrated to afford **262** (34.0 mg, 84% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as a beige solid.

To **262** (34.0 mg, 0.154 mmol) in THF (1.50 mL) was added MeMgBr (0.160 mL, 0.471 mmol, 3.0 M in Et_2O) at 0 °C. The reaction was warmed to 23 °C and then refluxed for 3 h. Sat. aq. $NaHCO_3$ was added upon completion. The aqueous layer was extracted with EtOAc (3 x 10 mL). The organics were washed with brine, dried over $MgSO_4$ and concentrated. The crude residue was purified via flash chromatography (40:1 to 19:1 EtOAc:MeOH) to afford **264** (11.7 mg, 35% yield, $R_f = 0.10$ in 40:1 EtOAc:MeOH) as a beige solid.

To $RuO_2 \cdot H_2O$ (4.2 mg, 0.0312 mmol) and $NaIO_4$ (4.20 mL, 10% aq) was added **113** (100 mg, 0.312 mmol) in EtOAc (1.60 mL) at 23 °C. The reaction was stirred for 2 d at 23 °C. IPA was added and stirred at 23 °C for 2 h. The black precipitate was filtered and washed with EtOAc. The organics were washed with brine (2 x 10 mL), dried over $MgSO_4$ and concentrated. The crude residue was purified via flash chromatography (3:1 to 1:1 hexanes:EtOAc) to afford **258** (104 mg, 70% yield, $R_f = 0.20$ in 1:1 hexanes:EtOAc) as a beige solid.

To **258** (72.1 mg, 0.216 mmol) in CH_2Cl_2 (0.400 mL) was added TFA (0.332 mL, 4.31 mmol). The reaction was stirred at 23 °C for 1 h. The solvent was removed under reduced pressure. The residue was neutralized with sat. aq. $NaHCO_3$. The aqueous was extracted with CH_2Cl_2 (3 x 10 mL). The organics were dried over Na_2SO_4 and concentrated to afford **263** (51.5 mg, 99% yield, $R_f = 0.0$ in 1:1 hexanes:EtOAc) as a beige solid.

To **262** (51.5 mg, 0.220 mmol) in THF (2.20 mL) was added MeMgBr (0.220 mL, 0.673 mmol, 3.0 M in Et$_2$O) at 0 °C. The reaction was warmed to 23 °C and then refluxed for 3 h. Sat. aq. NaHCO$_3$ was added upon completion. The aqueous layer was extracted with EtOAc (3 x 10 mL). The organics were washed with brine, dried over MgSO$_4$ and concentrated. The crude residue was purified via flash chromatography (40:1 to 19:1 EtOAc:MeOH) to afford **265** in less than 10% yield as a beige solid.
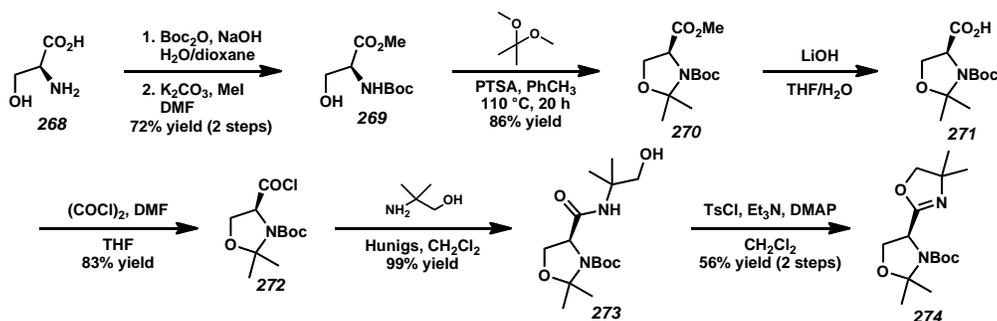
To a solution of **268** (0.500 g, 4.76 mmol) in dioxane (4.76 mL) and 1 M NaOH (9.52 mL) at 0 °C was added Boc$_2$O (1.23 g, 5.62 mmol) portionwise over 5 min. The reaction was stirred an additional 30 min at 0 °C, then 23 °C overnight. The organic solvent was removed under reduced pressure. The aqueous was acidified to pH 2 with 1 M KHSO$_4$. The aqueous was extracted with CHCl$_3$ (3 x 20 mL). The organics were washed with brine (30 mL), dried over Na$_2$SO$_4$ and concentrated to afford Boc-serine (1.11 g, 99% yield) as a thick clear oil.

To Boc-serine (4.76 mmol) in DMF (4.76 mL) at 0 °C was added K$_2$CO$_3$ (724 mg, 5.24 mmol) and stirred for 10 min at 0 °C. To this solution was added MeI (0.59 mL, 9.52 mmol) at 0 °C. The reaction was stirred at 0 °C for 30 min, the at 23 °C for 3 h. The reaction was filtered, the filtrate suspended between H$_2$O and EtOAc. The organic layer was washed with brine (2 x 15 mL), dried over Na$_2$SO$_4$ and concentrated to afford **269** (0.860 g, 72% yield, R$_f$ = 0.39 in 1:1 hexanes:EtOAc) as an amber oil.

To **269** (0.804 g, 3.67 mmol) in toluene (16.0 mL) at 23 °C was added dimethoxy propane (1.18 mL, 9.54 mmol) and PTSA (13.9 mg, 0.0734 mmol). The flask was fitted with a short path distillation apparatus, and heated to reflux until 4 mL had been collected. More dimethoxy propane (0.45 mL, 3.67 mmol) and PTSA (7.0 mg, 0.037 mmol) were added. Another 4 mL of solvent was distilled off. The reaction was then fitted with a condenser and refluxed overnight. Upon cooling to 23 °C, sat. aq. $NaHCO_3$ was added. The aqueous was extracted with EtOAc (2 x 10 mL). The organics were washed with brine, dried over Na_2SO_4 and concentrated to afford **270** (0.817 g, 86% yield, $R_f = 0.65$ in 1:1 hexanes:EtOAc) as an orange oil.

To a solution of **270** (0.817 g, 3.16 mmol) in THF (6.32 mL) and H_2O (3.16 mL) was added LiOH (75.6 mg, 3.16 mmol) and stirred overnight at 23 °C. The residue was acidified to pH 2 with 1 M $KHSO_4$. The aqueous layer was extracted with EtOAc (3 x 15 mL), the combined organics were dried over Na_2SO_4 and concentrated to afford **271** (0.809 g, 99% yield) as a clear oil.

To a solution of **271** (0.809 g, 3.30 mmol) in THF (16.5 mL) at 0 °C was added $(COCl)_2$ (1.44 mL, 16.5 mmol) and DMF (5 drops). The reaction was stirred at 0 °C for 5 min, then 23 °C for 1 h. The solvent was removed, and the residue concentrated from benzene (3 x 10 mL) to afford **272** (0.774 g, 83% yield) as a yellow solid.

To a solution of amino alcohol (1.06 mL, 11.1 mmol), DIPEA (1.93 mL, 11.1 mmol) and CH_2Cl_2 (12.0 mL) at 0 °C was added **272** (0.774 g, 2.77 mmol) in CH_2Cl_2 (13.9 mL). The reaction was stirred at 0 °C for 10 min, then 23 °C overnight. The reaction was washed with 0.5 M HCl. The aqueous was extracted with CH_2Cl_2 (2 x 15 mL). The combined organics were washed with sat. aq. $NaHCO_3$ and brine, dried over Na_2SO_4 and concentrated to afford **273** (0.756 g, 99% yield) as a yellow amorphous solid.

To a solution of **273** (0.756 g, 2.39 mmol) in CH$_2$Cl$_2$ (9.08 mL) and Et$_3$N (0.696 mL, 4.99 mL) and DMAP (27.7 mg, 0.227 mmol) was added TsCl (0.433 g, 2.27 mmol) at 23 °C. The combined organics were washed with sat. aq. NaHCO$_3$. The aqueous layer was extracted with CH$_2$Cl$_2$ (3 x 15 mL). The combined organics were dried over Na$_2$SO$_4$ and concentrated. The crude residue was prufied via flash chromatography (3:2 to 1:1 hexanes:EtOAc) to afford **274** (0.424 g, 56% yield, R$_f$ = 0.29 in 1:1 hexanes:EtOAc) as a beige solid.

To a solution of **274** (100 mg, 0.335 mmol) in MeOH was added PTSA (24.2 mg, 0.127 mmol) at 23 °C and stirred for 18 h. Sat. aq. NaHCO$_3$ was added under pH 9. MeOH was removed under reduced pressure, and the aqueous layer was extracted with EtOAc (2 x 5 mL). The organics were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (19:1 to 9:1 CH$_2$Cl$_2$:MeOH) to afford **275**.
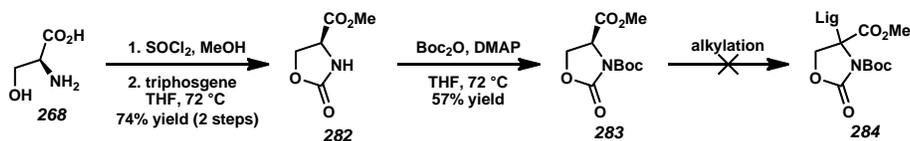
To serine methyl ester (100 mg, 0.643 mmol) in THF (5.4 mL) was added triphosgene (191 mg, 0.643 mmol) in THF (1.0 mL). The mixture was refluxed for 4 h, at which point the solvent was removed under reduced pressure. The crude mixture was purified over silica (1:1 to 3:2 hexanes:EtOAc) to afford **282** (68.8 mg, 74% yield) as a white solid.

To **282** (50 mg, 0.345 mmol) in THF (1.7 mL) was added Boc$_2$O (113 mg, 0.517 mmol) and DMAP (4.2 mg, 0.0345 mmol) at 23 °C. The mixture was refluxed for 1 h, at which time the solvent was removed. The crude residue was purified over silica (48.4 mg, 57% yield).
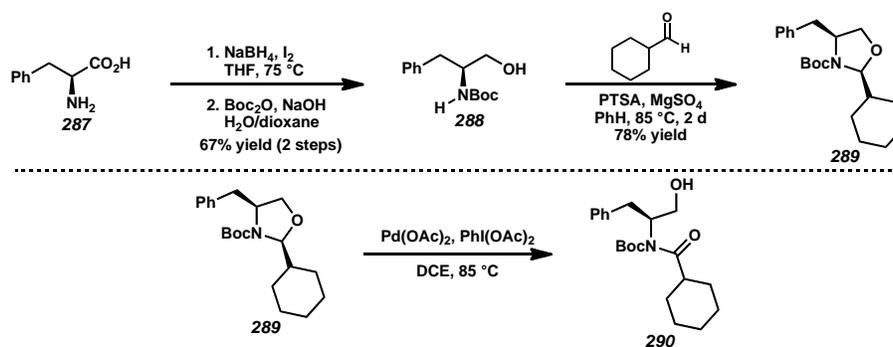
To NaBH$_4$ (275 mg, 7.27 mmol) in THF (10.1 mL) was added **287** (500 mg, 3.03 mmol) at 23 °C. I$_2$ (768 mg, 3.03 mmol) was added in THF (2.0 mL) slowly. The mixture was heated to reflux upon ceasing of bubbling for 18 h. After cooling to 23 °C, MeOH was added until the solution was clear. The solvent was removed under reduced pressure. 20% aq. KOH (8 mL) was added and the mixture stirred for 4 h. The aqueous was extracted with CH$_2$Cl$_2$ (3 x 10 mL), the organics dried over Na$_2$SO$_4$ and concentrated to afford amino alcohol (0.443 g, 97% yield).

To a solution of amino alcohol (0.443 g, 2.93 mmol) in dioxane (3.0 mL) and 1 M NaOH (5.9 mL) at 0 °C was added Boc$_2$O (0.754 g, 3.46 mmol) portionwise over 5 min. The reaction was stirred an additional 30 min at 0 °C, then 23 °C overnight. The organic solvent was removed under reduced pressure. The aqueous was acidified to pH 2 with 1 M KHSO$_4$. The aqueous was extracted with CHCl$_3$ (3 x 20 mL). The organics were washed with brine (30 mL), dried over Na$_2$SO$_4$ and concentrated to afford **288** (0.504 g, 69% yield) as a white solid.

To **288** (100 mg, 0.398 mmol) in benzene (4.0 mL) was added cyclohexanecarboxaldehyde (0.053 mL, 0.438 mmol), PTSA (1.1 mg, 5.97 μmol) and MgSO$_4$ (71.8 mg, 0.597 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The aqueous layer was extracted with EtOAc (3 x 10 mL), the organics dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (9:1 to 4:1

hexanes:EtOAc) to afford **289** (107 mg, 78% yield, R_f = 0.75 in 4:1 hexanes:EtOAc) as a white solid.
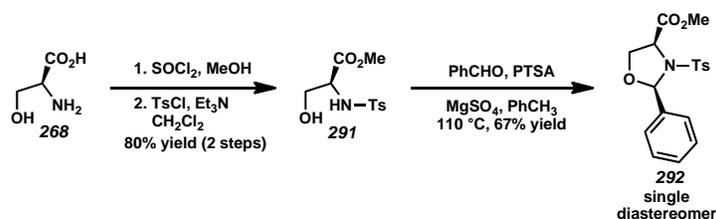
To **268** (7.5 g, 71.4 mmol) in MeOH (143 mL) at 0 °C was added $SOCl_2$ (26.0 mL, 357 mmol) slowly. Stirred an additional 5 min at 0 °C, then 23 °C overnight. The solvent was removed and the solid residue concentrated from Et_2O (3 x 50 mL) to afford the methyl ester (9.19 g, 83% yield).

To methyl ester (2.0 g, 12.9 mmol) in CH_2Cl_2 at 0 °C was added Et_3N (4.38 mL, 30.9 mmol). Stirred for 5 min at 0 °C, the TsCl (2.70 g, 14.1 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1 M $KHSO_4$, sat. aq. $NaHCO_3$ and brine sequentially. The organic layer was dried over Na_2SO_4 and concentrated to afford **291** (3.03 g, 86% yield) as a beige solid.

To **291** (1.50 g, 5.49 mmol) in toluene (27.4 mL) was added benzaldehyde (0.720 mL, 7.14 mmol), PTSA (52.2 mg, 0.274 mmol) and $MgSO_4$ (991 mg, 8.23 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. $NaHCO_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **292** (931 mg, 47% yield, R_f = 0.60 in 1:1 hexanes:EtOAc) as a white solid.

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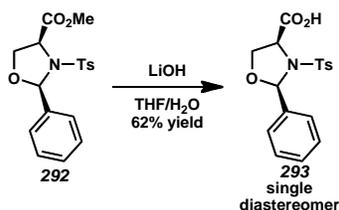
To a solution of **292** (0.856 g, 2.37 mmol) in THF (4.7 mL) and H$_2$O (2.4 mL) was added LiOH (56.7 mg, 2.37 mmol) and stirred overnight at 23 °C. The residue was acidified to pH 2 with 1 M KHSO$_4$. The aqueous layer was extracted with EtOAc (3 x 15 mL), the combined organics were dried over Na$_2$SO$_4$ and concentrated to afford **293** (0.624 g, 76% yield) as a white solid.
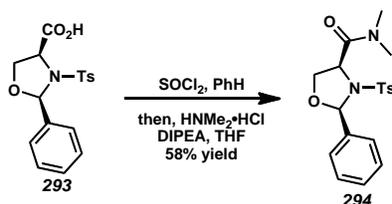
To **293** (367 mg, 1.06 mmol) in PhH (5.3 mL) was added SOCl$_2$ (0.768 mL, 10.6 mmol) at 23 °C, then heated to 60 °C for 1 h. The solvent was removed and concentrated from PhH (3 x 10 mL). To the crude residue was added THF (5.3 mL), DIPEA (0.561 mL, 3.22 mmol) and NMe$_2$H•HCl (172 mg, 2.11 mmol) at 23 °C and then stirred overnight at 23 °C. The reaction was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford **294** (211 mg, 54% yield, R$_f$ = 0.23 in 1:1 hexanes:EtOAc) as a beige amorphous solid.
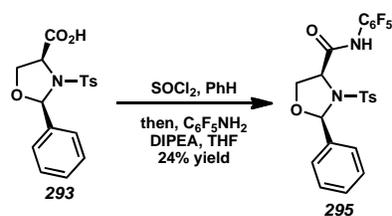
To **293** (250 mg, 0.720 mmol) in PhH (3.6 mL) was added SOCl$_2$ (0.524 mL, 7.20 mmol) at 23 °C, then heated to 60 °C for 1 h. The solvent was removed and concentrated from PhH (3 x 10 mL). To the crude residue was added THF (3.6 mL), DIPEA (0.263 mL, 1.51 mmol) and C$_6$F$_5$NH$_2$ (0.138 mL, 0.756 mmol) at 23 °C and then stirred overnight at 23 °C. The reaction was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was

dried over Na$_2$SO$_4$ and concentrated to afford **295** (89 mg, 24% yield, R$_f$ = 0.25 in 4:1 hexanes:EtOAc) as a yellow solid.
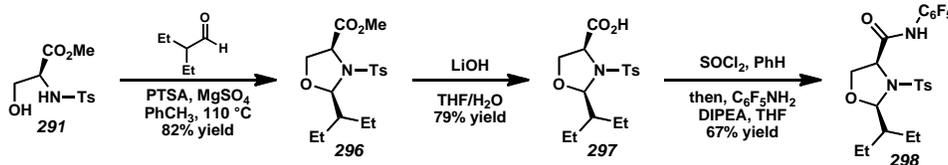
To **291** (1.00 g, 3.66 mmol) in toluene (18.3 mL) was added 2-ethylbutyraldehyde (0.809 mL, 6.59 mmol), PTSA (34.8 mg, 0.183 mmol) and MgSO$_4$ (661 mg, 5.49 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **296** (1.07 g, 82% yield, R$_f$ = 0.75 in 1:1 hexanes:EtOAc) as a white solid.

To a solution of **296** (1.06 g, 3.00 mmol) in THF (6.0 mL) and H$_2$O (3.0 mL) was added LiOH (71.8 mg, 3.00 mmol) and stirred overnight at 23 °C. The residue was acidified to pH 2 with 1 M KHSO$_4$. The aqueous layer was extracted with EtOAc (3 x 15 mL), the combined organics were dried over Na$_2$SO$_4$ and concentrated to afford **297** (0.925 g, 90% yield) as a yellow solid.

To **297** (195 mg, 0.571 mmol) in PhH (2.9 mL) was added SOCl$_2$ (0.416 mL, 5.71 mmol) at 23 °C, then heated to 60 °C for 1 h. The solvent was removed and concentrated from PhH (3 x 10 mL). To the crude residue was added THF (2.9 mL), DIPEA (0.209 mL, 1.20 mmol) and C$_6$F$_5$NH$_2$ (0.125 mL, 0.685 mmol) at 23 °C and then stirred overnight at 23 °C. The reaction was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford **298** (195 mg, 67% yield, R$_f$ = 0.34 in 4:1 hexanes:EtOAc) as an orange oil.
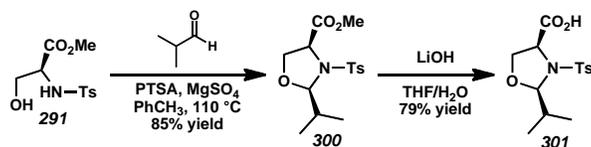
To **291** (1.58 g, 5.78 mmol) in toluene (28.9 mL) was added isobutyraldehyde (0.792 mL, 8.67 mmol), PTSA (55.0 mg, 0.289 mmol) and $MgSO_4$ (1.04 g, 8.67 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. $NaHCO_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **300** (1.17 g, 62% yield, $R_f = 0.75$ in 4:1 hexanes:EtOAc) as a white solid.

To a solution of **300** (217 mg, 0.663 mmol) in THF (1.3 mL) and H_2O (0.7 mL) was added LiOH (15.9 mg, 0.663 mmol) and stirred overnight at 23 °C. The residue was acidified to pH 2 with 1 M $KHSO_4$. The aqueous layer was extracted with EtOAc (3 x 15 mL), the combined organics were dried over Na_2SO_4 and concentrated to afford **301** (0.206 g, 99% yield) as a white solid.
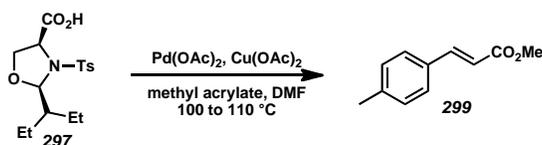
297 (50 mg, 0.146 mmol), $Pd(OAc)_2$ (3.3 mg, 0.0146 mmol) methyl acrylate (19.8 μL, 0.220 mmol) and $Cu(OAc)_2$ (53.2 mg, 0.293 mmol) were combined in DMF (0.1.5 mL) and heated to 110 °C for 8 h. Sat. aq. $NaHCO_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na_2SO_4 and concentrated.
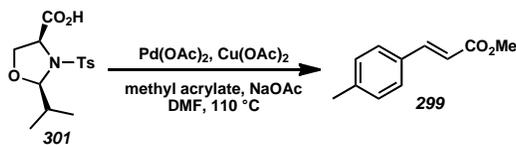
301 (20 mg, 0.0638 mmol), Pd(OAc)$_2$ (1.4 mg, 6.38 μmol), Cu(OAc)$_2$ (23.2 mg, 0.128 mmol),

methyl acrylate (8.6 μL, 0.0957 mmol) and NaOAc (10.5 mg, 0.128 mmol) were combined in

DMF (0.64 mL) and heated to 110 °C for 24 h. Sat. aq. NaHCO$_3$ was added to the crude

reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with

brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1

hexanes:EtOAc) to afford **299** (1.1 mg, 10% yield) as a clear oil.
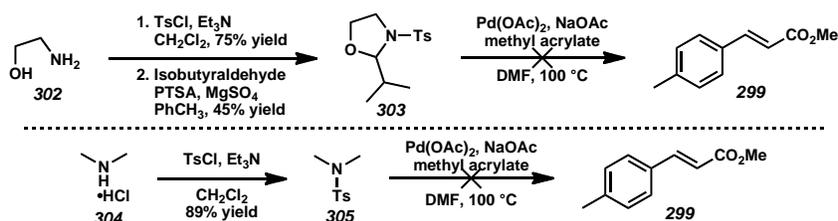
To **302** (0.250 mL, 4.14 mmol) in CH$_2$Cl$_2$ (13.8 mL) at 0 °C was added Et$_3$N (0.689 mL, 4.97

mmol). Stirred for 5 min at 0 °C, the TsCl (0.869 mg, 4.56 mmol) was added. The reaction was

stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1

M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$

and concentrated to afford sulfonamide (0.672 g, 75% yield) as a light yellow oil.

To sulfonamide (0.672 g, 3.12 mmol) in toluene (20.8 mL) was added isobutyraldehyde (0.427

mL, 4.68 mmol), PTSA (29.7 mg, 0.156 mmol) and MgSO$_4$ (0.564 g, 4.68 mmol). The

suspension was refluxed overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The

aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na$_2$SO$_4$ and

concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to

afford **303** (381 mg, 45% yield, R$_f$ = 0.75 in 4:1 hexanes:EtOAc) as a yellow oil.

To **304** (500 mg, 6.13 mmol) in CH$_2$Cl$_2$ (20.4mL) at 0 °C was added Et$_3$N (2.07 mL, 14.7

mmol). Stirred for 5 min at 0 °C, the TsCl (1.30 g, 6.74 mmol) was added. The reaction was

stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1

M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$

and concentrated to afford sulfonamide **305** (1.09 g, 89% yield) as a light yellow solid.

Pd (10 mol %)	Cu(OAc)$_2$	yield (%) 299 : 306
Pd(OAc)$_2$	2.0 equiv	10 : 4
Pd(OAc)$_2$	none	15 : 32
Pd(dba)$_2$	none	8 : 43
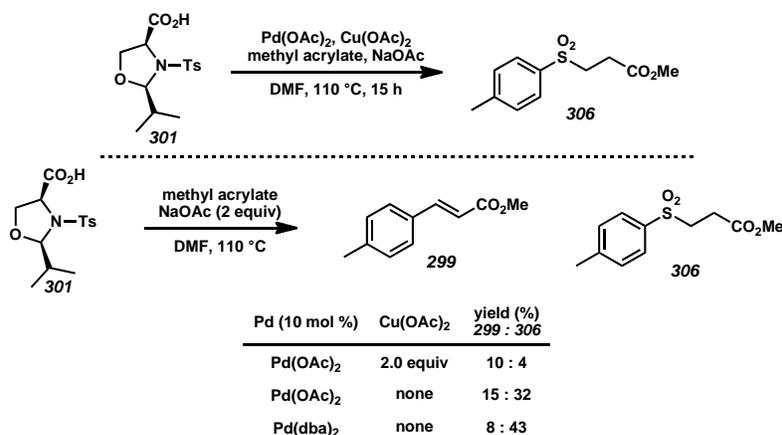
General coupling procedure: Acetal (1 equiv), palladium (10 mol %), Cu(OAc)$_2$ (2 equiv),

NaOAc (2 equiv), methyl acrylate (1.5 equiv) and DMF (0.1 M) were combined in a 2-dram vial

and capped. The reaction was heated to 110 °C for 15 h. Sat. aq. NaHCO$_3$ was added to the

crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed

with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1

hexanes:EtOAc) to afford **299** and **306** in the yields indicated.

base	solvent	yield (%) 299 : 306 : 307
NaOAc (2 equiv)	DMF	35 : 32 : 0
NaOAc (2 equiv)	t-AmOH	trace
NaOAc (2 equiv)	CH$_3$CN	none
NaOAc (2 equiv)	DCE	none
K$_2$CO$_3$ (1 equiv)	DMF	36 : 0 : 45
K$_2$CO$_3$ (1 equiv)	CH$_3$CN	15 : 15 : 46
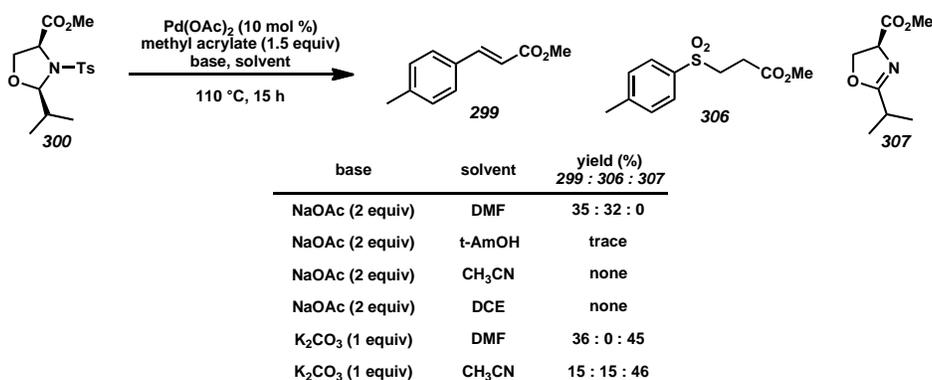
General coupling procedure: Acetal (1 equiv), Pd(OAc)$_2$ (10 mol %), base (1-2 equiv), methyl

acrylate (1.5 equiv) and solvent (0.1 M) were combined in a 2-dram vial and capped. The

reaction was heated to 110 °C for 15 h. Sat. aq. NaHCO$_3$ was added to the crude reaction

mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine,

dried over Na_2SO_4 and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299**, **306**, and **307** in the yields indicated.

Pd (10 mol %)	base	solvent	yield (%) 299 : 306 : 307
Pd(OAc)$_2$	K$_2$CO$_3$	DMSO	26 : 0 : 41
Pd(dba)$_2$	K$_2$CO$_3$	CH$_3$CN	0 : 10 : 0
Pd(dba)$_2$	K$_2$HPO$_4$	CH$_3$CN	0 : 0 : 0
Pd(dba)$_2$	Na$_2$CO$_3$	CH$_3$CN	0 : 0 : 0
PdCl$_2$	K$_2$CO$_3$	DMF	17 : 0 : 40
PdCl$_2$	K$_2$CO$_3$	DMSO	87 : 0 : 72
PdCl$_2$	Na$_2$CO$_3$	DMSO	29 : 0 : 0
PdCl$_2$	K$_2$HPO$_4$	DMSO	10 : 0 : 0
PdCl$_2$(CH$_3$CN)$_2$	K$_2$CO$_3$	DMSO	53 : 0 : 51
PdCl$_2$(PPh$_3$)$_2$	K$_2$CO$_3$	DMSO	38 : 0 : 54

General coupling procedure: Acetal (1 equiv), palladium (10 mol %), base (1 equiv), methyl acrylate (1.5 equiv) and solvent (0.1 M) were combined in a 2-dram vial and capped. The reaction was heated to 110 °C for 15 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na_2SO_4 and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299**, **306**, and **307** in the yields indicated.

base	yield (%) 299 : 306 : 307	base	yield (%) 299 : 306 : 307
KHCO$_3$ (2 equiv)	33 : 0 : 40	LiOAc (2 equiv)	59 : 0 : 52
Ba(OH)$_2$ (0.5 equiv)	none	Cs$_2$CO$_3$ (1 equiv)	complex
Ca(OH)$_2$ (0.5 equiv)	none	Li$_2$CO$_3$ (1 equiv)	none
NaOAc (4 equiv)	71 : 0 : 46	t-BuCO$_2$Na (2 equiv)	54 : 0 : 43
KOAc (2 equiv)	59 : 0 : 37	Cl$_3$CCO$_2$Na (2 equiv)	10 : 0 : 0
KOAc (1.5 equiv)	46 : 0 : 41	PhCO$_2$Na (2 equiv)	29 : 0 : 9

General coupling procedure: Acetal (1 equiv), PdCl$_2$ (10 mol %), base (0.5-4 equiv), methyl acrylate (1.5 equiv) and DMSO (0.1 M) were combined in a 2-dram vial and capped. The reaction was heated to 110 °C for 2-6 h. Sat. aq. NaHCO$_3$ was added to the crude reaction

mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na_2SO_4 and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299**, **306**, and **307** in the yields indicated.
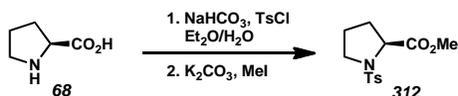
To **68** (500 mg, 4.34 mmol) in sat. aq. $NaHCO_3$/Et_2O (21.7 mL) was added TsCl (1.24 g, 6.51 mmol) at 23 °C. The reaction was stirred at 23 °C overnight. The reaction was acidified to pH 2 with 3 M HCl. The aqueous was extracted with EtOAc (3 x 15 mL). The organics were dried over Na_2SO_4 and concentrated to afford tosyl protected proline (1.07 g, 92% yield) as a white solid.

To protected proline (500 mg, 1.86 mmol) in DMF (3.71 mL) at 23 °C was added K_2CO_3 (282 mg, 2.04 mmol). After stirring for 10 min at 23 °C, MeI (0.231 mL, 3.71 mmol) was added. The reaction was stirred at 23 °C overnight. The reaction was partitioned between H_2O and EtOAc. The organic layer was washed with brine (2 x 10 mL), dried over Na_2SO_4 and concentrated to afford **312** (447 mg, 85% yield) as a white solid.
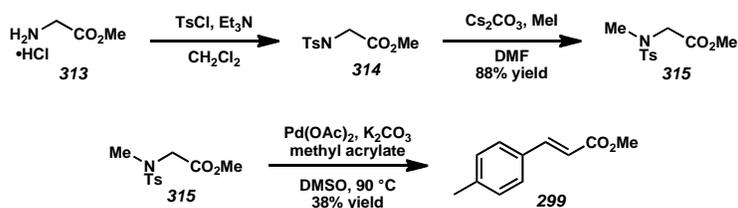
To **313** (5.0 g, 39.8 mmol) in CH_2Cl_2 (39.8 mL) at 0 °C was added Et_3N (13.4 mL, 95.6 mmol). Stirred for 5 min at 0 °C, the TsCl (8.35 g, 43.8 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1 M $KHSO_4$, sat. aq. $NaHCO_3$ and brine sequentially. The organic layer was dried over Na_2SO_4 and concentrated to afford sulfonamide **314** (9.01 g, 93% yield) as a light yellow solid.

To **314** (500 mg, 2.06 mmol) in DMF (4.1 mL) at 23 °C was added Cs$_2$CO$_3$ (737 mg, 2.26 mmol). After stirring for 10 min at 23 °C, MeI (0.192 mL, 3.08 mmol) was added. The reaction was stirred at 23 °C overnight. The reaction was partitioned between H$_2$O and EtOAc. The organic layer was washed with brine (2 x 10 mL), dried over Na$_2$SO$_4$ and concentrated to afford **315** (464 mg, 88% yield, R$_f$ = 0.54 in 1:1 hexanes:EtOAc) as a yellow oil.

Sulfonamide **315** (20 mg, 0.0777 mmol), PdCl$_2$ (1.4 mg, 7.77 μmol) K$_2$CO$_3$ (10.7 mg, 0.0777 mmol), methyl acrylate (10.5 μL, 0.117 mmol) and DMSO (0.78 mL) were combined in a 2-dram vial and capped. The reaction was heated to 110 °C for 12 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299** (5.2 mg, 38% yield).
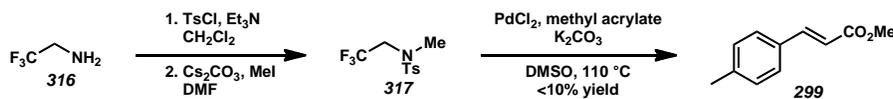
To **316** (250 mg, 1.85 mmol) in CH$_2$Cl$_2$ (9.2 mL) at 0 °C was added Et$_3$N (0.622 mL, 4.43 mmol). Stirred for 5 min at 0 °C, the TsCl (387 mg, 2.03 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford sulfonamide (412 mg, 88% yield) as a white solid.

To the sulfonamide (412 mg, 1.63 mmol) in DMF (3.3 mL) at 23 °C was added Cs$_2$CO$_3$ (583 mg, 1.79 mmol). After stirring for 10 min at 23 °C, MeI (0.152 mL, 2.44 mmol) was added. The reaction was stirred at 23 °C overnight. The reaction was partitioned between H$_2$O and EtOAc. The organic layer was washed with brine (2 x 10 mL), dried over Na$_2$SO$_4$ and concentrated to afford **317** (358 mg, 82% yield) as a white solid.

Scheme (top of page):

Top reaction:

318 (HO$_2$C–CH(Me)–NH$_2$)
1. SOCl$_2$, MeOH
2. TsCl, Et$_3$N, CH$_2$Cl$_2$
3. Cs$_2$CO$_3$, MeI, DMF
87% yield (3 steps)
→ 319 (MeO$_2$C–CH(Me)–N(Me)Ts)

PdCl$_2$, KHCO$_3$, methyl acrylate
DMSO, 90 °C, 18 h
21% yield
→ 299 (4-Me-C$_6$H$_4$–CH=CH–CO$_2$Me)

Bottom reaction:

320 (MeO$_2$C–C(Me)$_2$–NH$_3$Cl)
1. TsCl, Et$_3$N, CH$_2$Cl$_2$
2. C$_2$CO$_3$, MeI, DMF
83% yield (2 steps)
→ 321 (MeO$_2$C–C(Me)$_2$–N(Me)Ts)

PdCl$_2$, K$_2$CO$_3$, methyl acrylate
DMSO, 90 °C, 18 h
✗
→ 299 (4-Me-C$_6$H$_4$–CH=CH–CO$_2$Me)

To **318** (1.0 g, 11.2 mmol) in MeOH (44.9 mL) at 0 °C was added SOCl$_2$ (4.09 mL, 56.1 mmol) slowly. Stirred an additional 5 min at 0 °C, then 23 °C overnight. The solvent was removed and the solid residue concentrated from Et$_2$O (3 x 50 mL) to afford the methyl ester (1.59 g, 99% yield) as a white solid.

To the methyl ester (500 mg, 3.58 mmol) in CH$_2$Cl$_2$ (17.9 mL) at 0 °C was added Et$_3$N (1.21 mL, 8.60 mmol). Stirred for 5 min at 0 °C, the TsCl (751 mg, 3.94 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford sulfonamide (1.00 g, 99% yield) as a clear oil.

To the sulfonamide (3.58 mmol) in DMF (7.2 mL) at 23 °C was added Cs$_2$CO$_3$ (1.28 g, 3.94 mmol). After stirring for 10 min at 23 °C, MeI (0.446 mL, 7.16 mmol) was added. The reaction was stirred at 23 °C overnight. The reaction was partitioned between H$_2$O and EtOAc. The organic layer was washed with brine (2 x 10 mL), dried over Na$_2$SO$_4$ and concentrated to afford **319** (843 mg, 87% yield) as a yellow oil.

Sulfonamide **319** (20 mg, 0.0737 mmol), PdCl$_2$ (1.3 mg, 7.37 μmol) KHCO$_3$ (7.8 mg, 0.0737 mmol), methyl acrylate (9.9 μL, 0.111 mmol) and DMSO (0.74 mL) were combined in a 2-dram vial and capped. The reaction was heated to 110 °C for 12 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed

with brine, dried over Na_2SO_4 and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299** (2.7 mg, 21% yield).

To the methyl ester **320** (400 mg, 2.60 mmol) in CH_2Cl_2 (13.0 mL) at 0 °C was added Et_3N (0.915 mL, 6.51 mmol). Stirred for 5 min at 0 °C, the TsCl (546 mg, 2.86 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1 M $KHSO_4$, sat. aq. $NaHCO_3$ and brine sequentially. The organic layer was dried over Na_2SO_4 and concentrated to afford sulfonamide (586 mg, 83% yield) as a white solid.

To the sulfonamide (371 mg, 1.29 mmol) in DMF (2.6 mL) at 23 °C was added Cs_2CO_3 (462 mg, 1.42 mmol). After stirring for 10 min at 23 °C, MeI (0.161 mL, 2.58 mmol) was added. The reaction was stirred at 23 °C overnight. The reaction was partitioned between H_2O and EtOAc. The organic layer was washed with brine (2 x 10 mL), dried over Na_2SO_4 and concentrated to afford **321** (371 mg, 99% yield) as a yellow oil.
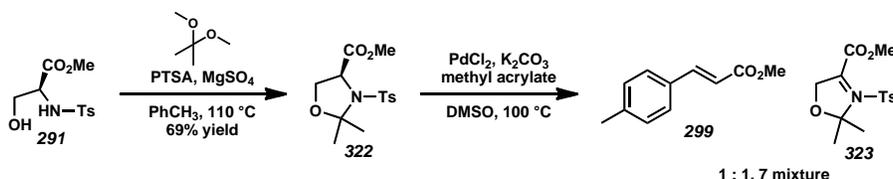
To sulfonamide **291** (328 mg, 1.20 mmol) in toluene (12.0 mL) was added dimethoxypropane (0.221 mL, 1.80 mmol), PTSA (11.4mg, 0.0600 mmol) and $MgSO_4$ (0.289 g, 2.40 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. $NaHCO_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **322** (259 mg, 69% yield, R_f = 0.72 in 1:1 hexanes:EtOAc) as a yellow oil.

Acetal **322** (20 mg, 0.0638 mmol), $PdCl_2$ (1.1 mg, 6.38 µmol) K_2CO_3 (8.8 mg, 0.0638 mmol), methyl acrylate (8.6 µL, 0.0957 mmol) and DMSO (0.64 mL) were combined in a 2-dram vial

and capped. The reaction was heated to 110 °C for 12 h. Sat. aq. $NaHCO_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na_2SO_4 and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299** and **323** in a 1 : 1.7 mixture.

base	oxidant	yield (%) 299 : 306 : 307	base	oxidant	yield (%) 299 : 306 : 307
K_2CO_3 (1 equiv)	air	18 : 5 : 54	KOAc (2 equiv)	O_2 balloon $Cu(OAc)_2$ (20 mol %)	63 : 0 : 4
KOAc (2 equiv)	air	25 : 38 : 39	KOAc (2 equiv)	O_2 balloon	67 : 0 : 29
K_2CO_3 (1 equiv)	$Cu(OAc)_2$ (2 equiv)	18 : 7 : 0	KOAc (2 equiv)	BQ (25 mol %)	36 : 0 : 35
none	$Cu(OAc)_2$ (2 equiv)	none	KOAc (2 equiv)	$NaNO_2$ (25 mol %)	27 : 0 : 7
KOAc (2 equiv)	$Cu(OAc)_2$ (2 equiv)	20 : 0 : 0	KOAc (2 equiv)	AgOAc (25 mol %)	27 : 16 : 29
K_2CO_3	BQ (1 equiv)	complex			
KOAc	BQ (1 equiv)	47 : 31 : 63			
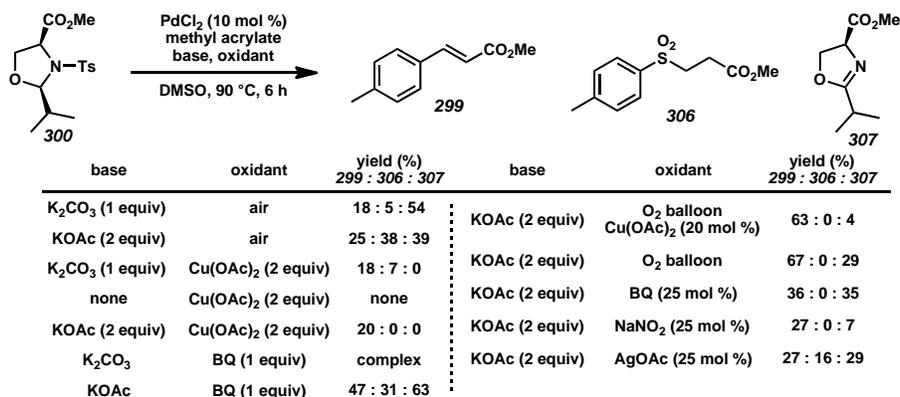
General coupling procedure: Acetal **300** (1 equiv), $PdCl_2$ (10 mol %), base (1-2 equiv), methyl acrylate (1.5 equiv), oxidant (0.25-2 equiv) and DMSO (0.1 M) were combined in a 2-dram vial and capped. Those reactions with O_2 were fitted with a septum and an O_2 balloon. The reaction was heated to 90 °C for 6 h. Sat. aq. $NaHCO_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na_2SO_4 and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299**, **306**, and **307** in the yields indicated.

base	yield (%) 299 : 306 : 307
NaOAc (3 equiv)	92 : 0 : 18
NaOAc (1.5 equiv)	52 : 0 : 0
Na_2CO_3 (1.5 equiv) AcOH (10 mol %)	70 : 0 : 7
NaOAc (3 equiv) (ground)	89: 0 : 0
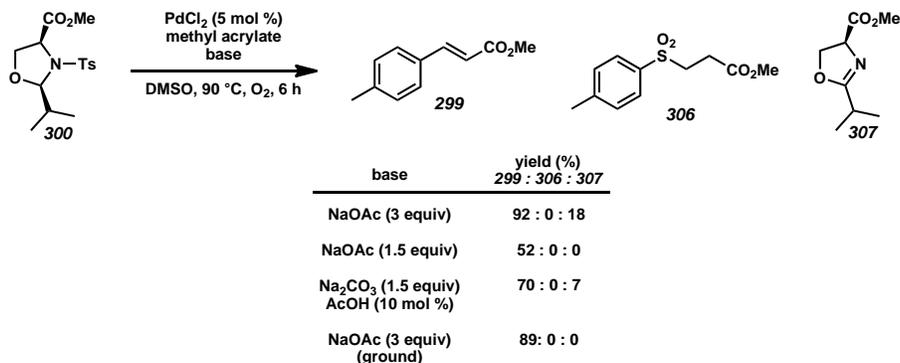
General coupling procedure: Acetal **300** (1 equiv), PdCl$_2$ (5 mol %), base (1.5-3 equiv), methyl acrylate (1.5 equiv), and DMSO (0.1 M) were combined in a 2-dram vial and were fitted with a septum and an O$_2$ balloon. The reaction was heated to 90 °C for 6 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299**, **306**, and **307** in the yields indicated.

equivalents NaOAc	yield (%) 299 : 306 : 307	equivalents NaOAc	yield (%) 299 : 306 : 307
1.0	44 : 0 : 8	3.0	18 : 22 : 12
1.0	44 : 0 : 2	3.0	65 : 0 : 12
2.0	62 : 0 : 13	4.0	54 : 5 : 0
2.0	62 : 0 : 5	4.0	76 : 0 : 13
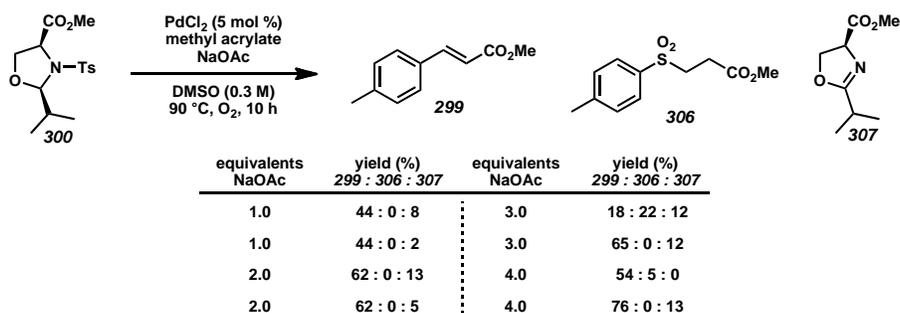
General coupling procedure: Acetal **300** (1 equiv), PdCl$_2$ (5 mol %), NaOAc (1-4 equiv), methyl acrylate (1.5 equiv), and DMSO (0.3 M) were combined in a 2-dram vial and were fitted with a septum and an O$_2$ balloon. The reaction was heated to 90 °C for 10 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **299**, **306**, and **307** in the yields indicated.
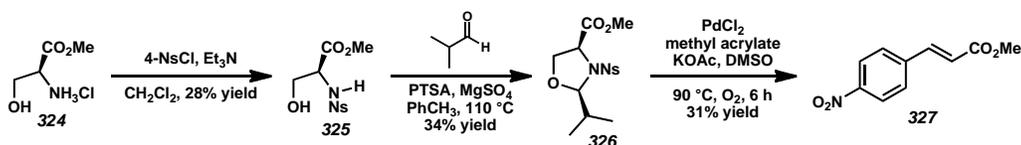
To the methyl ester **324** (500 mg, 3.21 mmol) in CH$_2$Cl$_2$ (16.1 mL) at 0 °C was added Et$_3$N (1.08 mL, 7.71 mmol). Stirred for 5 min at 0 °C, the 4-NsCl (783 mg, 3.54 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was

washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried

over Na$_2$SO$_4$ and concentrated to afford sulfonamide (274 mg, 24% yield) as a yellow solid.

To sulfonamide **325** (0.274 g, 0.90 mmol) in toluene (18.0 mL) was added isobutyraldehyde

(0.123 mL, 1.35 mmol), PTSA (8.6 mg, 0.045 mmol) and MgSO$_4$ (0.163 g, 1.35 mmol). The

suspension was refluxed overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The

aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na$_2$SO$_4$ and

concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to

afford **326** (109 mg, 35% yield, R$_f$ = 0.55 in 1:1 hexanes:EtOAc) as a yellow oil.

Acetal **326** (20 mg, 0.0558 mmol), PdCl$_2$ (1.0 mg, 5.58 μmol), K$_2$CO$_3$ (7.7 mg, 0.0558 mmol),

methyl acrylate (7.5 μmol, 0.0837 mmol), and DMSO (0.56 mL) were combined in a 2-dram vial

and capped. The reaction was heated to 100 °C for 15 h. Sat. aq. NaHCO$_3$ was added to the

crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed

with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1

hexanes:EtOAc) to afford **327** (3.6 mg, 31% yield).
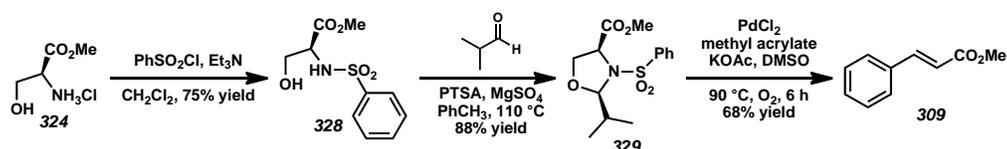
To the methyl ester **324** (750 mg, 4.82 mmol) in CH$_2$Cl$_2$ (24.1 mL) at 0 °C was added Et$_3$N (1.63

mL, 11.6 mmol). Stirred for 5 min at 0 °C, the PhSO$_2$Cl (0.679 mL, 5.30 mmol) was added.

The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture

was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was

dried over Na$_2$SO$_4$ and concentrated to afford sulfonamide **328** (932 mg, 75% yield) as a yellow

fluffy solid.

To sulfonamide **328** (0.932 g, 3.60 mmol) in toluene (23.9 mL) was added isobutyraldehyde (0.492 mL, 5.39 mmol), PTSA (34.2 mg, 0.180 mmol) and MgSO$_4$ (0.649 g, 5.39 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **329** (996 mg, 88% yield) as a beige solid.

Acetal **329** (40 mg, 0.128 mmol), PdCl$_2$ (2.3 mg, 0.0128 mmol), KOAc (25.2 mg, 0.256 mmol), methyl acrylate (17.3 μL, 0.192 mmol), and DMSO (1.28 mL) were combined in a 2-dram vial and capped. The reaction was heated to 100 °C for 15 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **309** (14.0 mg, 68% yield).
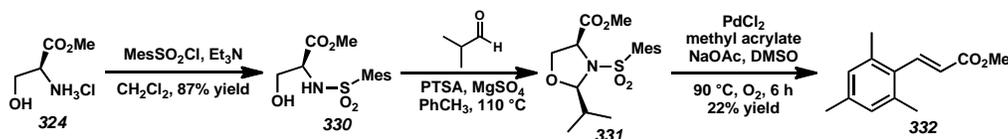
To the methyl ester **324** (500 mg, 3.21 mmol) in CH$_2$Cl$_2$ (16.1mL) at 0 °C was added Et$_3$N (1.08 mL, 7.71 mmol). Stirred for 5 min at 0 °C, the MesSO$_2$Cl (0.773 g, 3.54 mmol) was added. The reaction was stirred for an additional 1 h at 0 °C, then 23 °C overnight. The crude mixture was washed with 1 M KHSO$_4$, sat. aq. NaHCO$_3$ and brine sequentially. The organic layer was dried over Na$_2$SO$_4$ and concentrated to afford sulfonamide **330** (846 mg, 87% yield) as a white solid.

To sulfonamide **330** (0.846 g, 2.81mmol) in toluene (14.0 mL) was added isobutyraldehyde (0.384 mL, 4.21 mmol), PTSA (26.7 mg, 0.140 mmol) and MgSO$_4$ (0.507 g, 4.21 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na$_2$SO$_4$ and

concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **331** (411 mg, 41% yield) as a beige solid.

Acetal **331** (40 mg, 0.113 mmol), PdCl$_2$ (1.0 mg, 5.63 μmol), NaOAc (27.8 mg, 0.339 mmol), methyl acrylate (15.3 μL, 0.170 mmol), and DMSO (0.38 mL) were combined in a 2-dram vial and capped. The reaction was heated to 100 °C for 15 h. Sat. aq. NaHCO$_3$ was added to the crude reaction mixture. The aqueous was extracted with EtOAc (2 x 2 mL), the organics washed with brine, dried over Na$_2$SO$_4$ and concentrated. The crude residue was plugged on silica (4:1 hexanes:EtOAc) to afford **332** (5.1 mg, 22% yield).
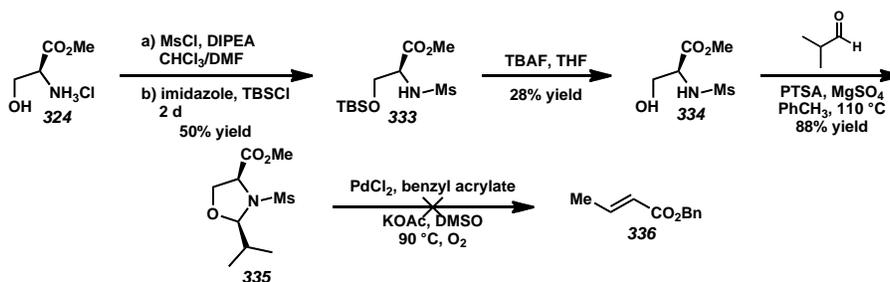
To the methyl ester **324** (500 mg, 3.21 mmol) in DMF/CHCl$_3$ (3.82 mL) at -78 °C was added DIPEA (1.40 mL, 8.04 mmol) and MsCl (0.299 mL, 3.86 mmol). The reaction was warmed to 0 °C and stirred for 2 h. Then imidazole (875 mg, 12.9 mmol) and TBSCl (581 mg, 3.86 mmol) were added at 0 °C. The reaction was stirred at 23 °C for 48 h. The reaction was quenched with 5% NaHCO$_3$, the aqueous layer extracted with Et$_2$O. The organics were washed with 5% citric acid, H$_2$O, 5% NaHCO$_3$ and H$_2$O. The organic layer was dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford sulfonamide **333** (492 mg, 49% yield) as a white solid.

To sulfonamide **333** (492 mg, 1.58 mmol) in THF (10.5 mL) was added TBAF (3.16 mL, 3.16 mmol, 1.0 M in THF) at 23 °C. The reaction was stirred at 23 °C for 2 h. The reaction was quenched with H$_2$O. The aqueous was extracted with EtOAc (3 x 20 mL), dried over Na$_2$SO$_4$

and concentrated. The crude residue was purified via flash chromatography (4:1 hexanes:EtOAc) to afford **334** (87.0 mg, 28% yield).

To sulfonamide **334** (87 mg, 0.441 mmol) in toluene (4.4 mL) was added isobutyraldehyde (60.4 µL, 0.662 mmol), PTSA (4.2 mg, 0.0221 mmol) and $MgSO_4$ (79.7 mg, 0.662 mmol). The suspension was refluxed overnight. The reaction was quenched with sat. aq. $NaHCO_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na_2SO_4 and concentrated to afford **335** (91.6 mg, 83% yield).
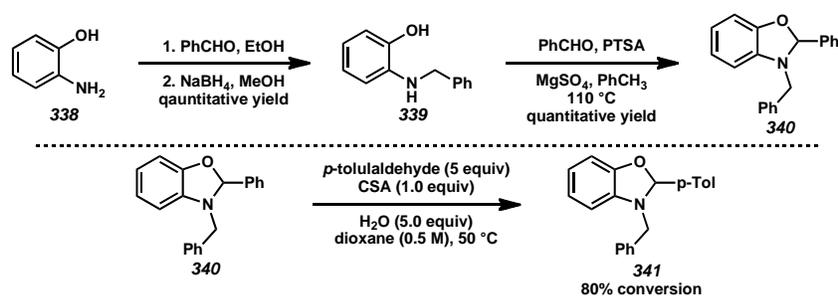
To **338** (1.0 g, 9.16 mmol) in EtOH (9.2 mL) was added PhCHO (0.94 mL, 9.26 mmol) at 23 °C. Stirred overnight at 23 °C. The solvent was removed under reduced pressure, and the crude residue concentrated from PhH (3 x 10 mL) to afford the acetal (1.84 g, 99% yield) as a beige solid.

To the acetal (1.5 g, 7.61 mmol) in MeOH (23.0 mL) was added $NaBH_4$ (575 mg, 15.2 mmol) over 10 min. After 5 min, the reaction was complete and quenched with sat. aq. NH_4Cl. Water was added to dissolve the salts. The aqueous layer was extracted with EtOAC (3 x 15 mL). The organics were washed with brine, dried over Na_2SO_4 and concentrated to afford **339** (1.60 g, 99% yield) as a brown solid.

339 (250 mg, 1.26 mmol) was mixed with benzaldehyde (0.165 mL, 1.63 mmol), PTSA (12.0 mg, 0.0628 mmol) and $MgSO_4$ (227 mg, 1.88 mmol) in toluene (12.6 mL) and refluxed

overnight. Sat. aq. NaHCO$_3$ was added to the cooled reaction. The aqueous was extracted with EtOAc (3 x 10 mL), dried over Na$_2$SO$_4$ and concentrated to afford **340** in quantitative yield.

340 (50 mg, 0.174 mmol), *p*-tolualdehyde (0.103 mL, 0.870 mmol), CSA (40.4 mg, 0.174 mmol) and H$_2$O (15.7 µL, 0.870 mmol) were combined in dioxane (0.35 mL) and heated to 50 °C overnight. The solvent was removed and analyzed by ^{1}H NMR.
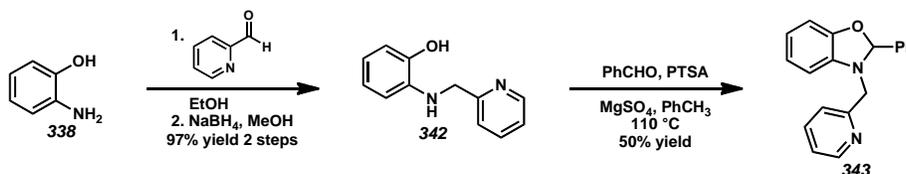
To **338** (1.0 g, 9.16 mmol) in EtOH (9.2 mL) was added 2-pyridinecarboxaldehyde (0.884 mL, 9.26 mmol) at 23 °C. Stirred overnight at 23 °C. The solvent was removed under reduced pressure, and the crude residue concentrated from PhH (3 x 10 mL) to afford the acetal in quantitative yield.

To the acetal (9.16 mmol) in MeOH (27.8 mL) was added NaBH$_4$ (693 mg, 18.3 mmol) over 10 min. After 5 min, the reaction was complete and quenched with sat. aq. NH$_4$Cl. Water was added to dissolve the salts. The aqueous layer was extracted with EtOAC (3 x 15 mL). The organics were washed with brine, dried over Na$_2$SO$_4$ and concentrated to afford **342** (1.77 g, 97% yield) as a brown solid.

339 (1.0 mg, 4.99 mmol) was mixed with benzaldehyde (0.66 mL, 6.49 mmol), PTSA (47.5 mg, 0.250 mmol) and MgSO$_4$ (902 mg, 7.49 mmol) in toluene (20.0 mL) and refluxed overnight. Sat. aq. NaHCO$_3$ was added to the cooled reaction. The aqueous was extracted with EtOAc (3 x 10 mL), dried over Na$_2$SO$_4$ and concentrated to afford **343** as a mixture of product and starting material.
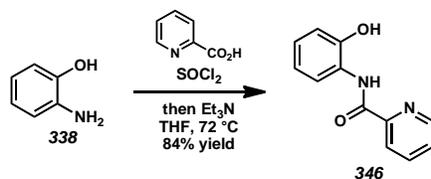
2-picolinic acid (1.13 g, 9.16 mmol) and SOCl$_2$ (3.4 mL) were combined in THF (6.5 mL) at 0 °C, then heated to 50 °C for 1 h. The solvent was removed under reduced pressure, and concentrated twice from THF (2 x 10 mL). The residue was dissolved in THF (11.2 mL), then **338** (1.0 g, 9.16 mmol) and Et$_3$N (1.93 mL, 13.7 mmol) was added at 0 °C. The reaction was refluxed overnight. The precipitate was filtered off, and the filtrate concentrated to afford **346** (1.64 g, 84% yield) as a brown solid.
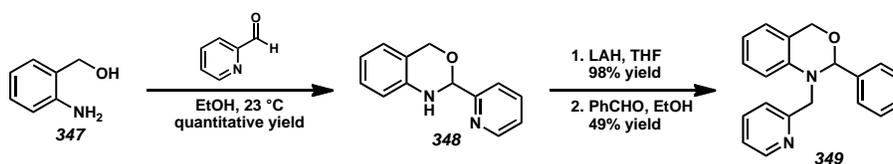
To **347** (1.0 g, 8.12 mmol) and 2-pyridinecarboxaldehyde (0.776 mL, 8.12 mmol) were combined in EtOH (8.1 mL) at 23 °C and stirred overnight. The solvent was removed under reduced pressure and the crude residue concentrated from PhH (3 x 10 mL) to afford **348** (1.80 g, 99% yield) as a red oil.

To the acetal (8.12 mmol) in THF (58.0 mL) at 0 °C was added LAH (339 mg, 8.93 mmol). The reaction was stirred at 23 °C for 2 h. The reaction was cooled to 0 °C and 0.339 mL H$_2$O, 0.339 mL 10% NaOH, and 1.02 mL H$_2$O were added sequentially. The solid was filtered off and the organic layer was dried over Na$_2$SO$_4$ and concentrated to give the amino alcohol (1.71 g, 98% yield) as a red oil.

The amino alcohol (250 mg, 1.20 mmol) and benzaldehyde (0.122 mL, 1.20 mmol) were combined in EtOH (1.2 mL) at 23 °C and stirred overnight. The solvent was removed under reduced pressure and the crude residue was concentrated from PhH (3 x 10 mL). The crude

residue was purified via flash chromatography (7:3 hexanes:EtOAc) to give **349** (177 mg, 49%
yield, $R_f = 0.60$ in 1:1 hexanes:EtOAc) as a light yellow oil.
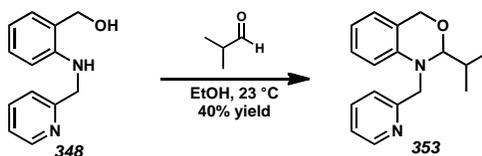
The amino alcohol (250 mg, 1.20 mmol) and isobutyraldehyde (0.160 mL, 1.75 mmol) were
combined in EtOH (1.2 mL) at 23 °C and stirred overnight. The solvent was removed under
reduced pressure and the crude residue was concentrated from PhH (3 x 10 mL). The crude
residue was purified via flash chromatography (4:1 hexanes:EtOAc) to give **349** (129 mg, 40%
yield, $R_f = 0.60$ in 1:1 hexanes:EtOAc) as a clear oil.

Appendix 1

Synthesis of Asymmetric Amino Amides

The development of asymmetric syntheses is always of interest to the synthetic community. In particula the synthesis of chiral secondary amines has been utilized for many asymmetric organocatalyzed transformations.[1] The emergence of such methods has required the ease of synthesis of these chiral catalysts.

A1.1 Development of an Asymmetric Synthesis

Our original project outline involved developing a ligand scaffold that could perform a functionalization of an unreactive C–H bond with high regio- and stereoselectivity. We envisioned utilizing a combination of an alcohol and amine as acetalization components to condense onto an aldehyde (Scheme A1.1.1). This ligand would also contain an internal ligating group that could coordinate a metal and direct a C–H acetoxylation. Upon functionalization, we envisioned hydrolysis would release the activated product and regenerate the ligand.

Scheme A1.1.1. **Revisiting the general concept**
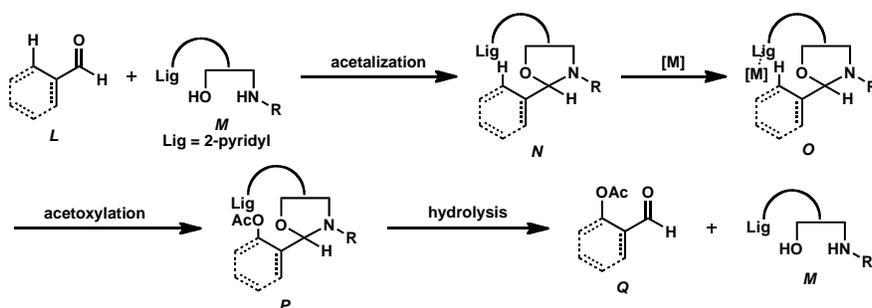
In the course of our ligand development, we uncovered an asymmetric ligand synthesis (Scheme A1.1.2). From **119** we were able to form amino amide **122** in good yield. Condensation onto isobutyraldehyde afforded *N,N*-aminal in good yield as a single diastereomer. Arylation with 2-fluoropryidine afforded ligand substrate adduct **129** in good yield as a single

diastereomer. Additionally, alkylation with 2-bromomethyl pyridine afforded **130** as a single

diastereomer. Starting from a single enantiomer (L-proline) and considering the high

diastereoselectivity, it can be assumed that the alkylated products are also single enantiomers.

Scheme A1.1.2. Synthesis of ligands 129 and 130
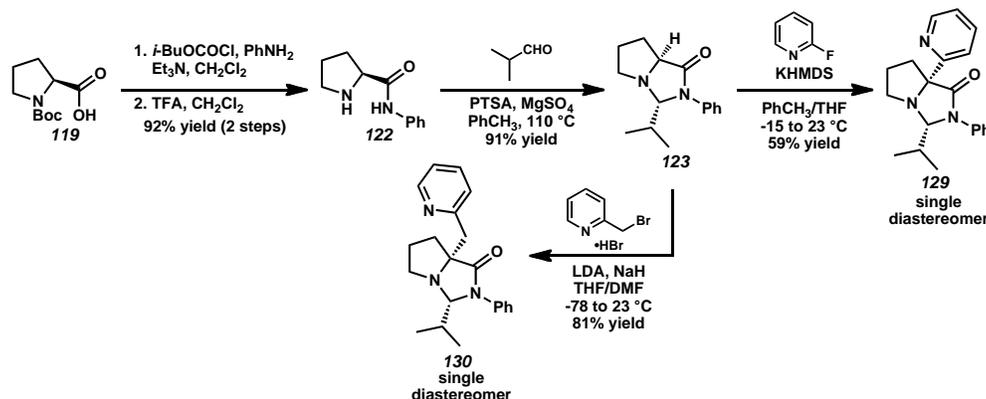
A similar example of asymmetric synthesis was employed by Seebach and coworkers in

1983 in the generation of α-substituted amino acids.[2] L-Proline was condensed onto pivaldehyde

to afford *N,O*-acetal **69** (Scheme A1.1.3). Alkylation with LDA and BnBr afforded α-substituted

N,O-acetal **70** as a single diastereomer. Cleavage of the acetal was accomplished with aqueous

acid, followed by purification with a Dowex column to afford amino acid **367** as a single

enantiomer. Although this sequence provides a short, high yielding (>80% each step) synthesis

of asymmetric amino acids, the intermediates are very unstable and difficult to work with.

Additionally, the isolation and purification of the desired amino acid is not trivial.

Scheme A1.1.3. Seebach's amino acid synthesis
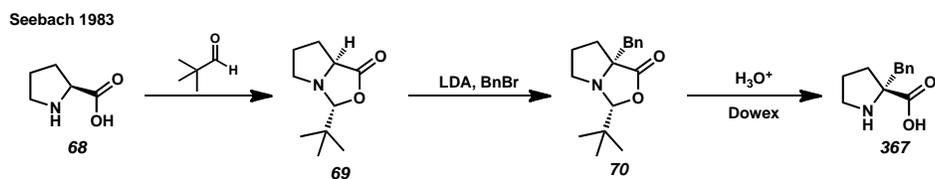
We find that our approach provides a quick straightforward synthesis of asymmetric *N,N*-

aminals without unstable intermediates or nontrivial purification steps. The only remaining

difficulty in our approach was the development of a high yielding cleavage procedure to release

the enantioenriched amino amide. To liberate our activated product, we subjected acetoxylated

181 to transamidation conditions with complete conversion to **249** (Scheme A1.1.4).[3] We

subjected our isobutyraldehyde-ligand adduct to the same conditions and observed 80%

conversion to the free ligand (**368**).[4]

Scheme A1.1.4. Hydrolysis of ligands 181 and 130
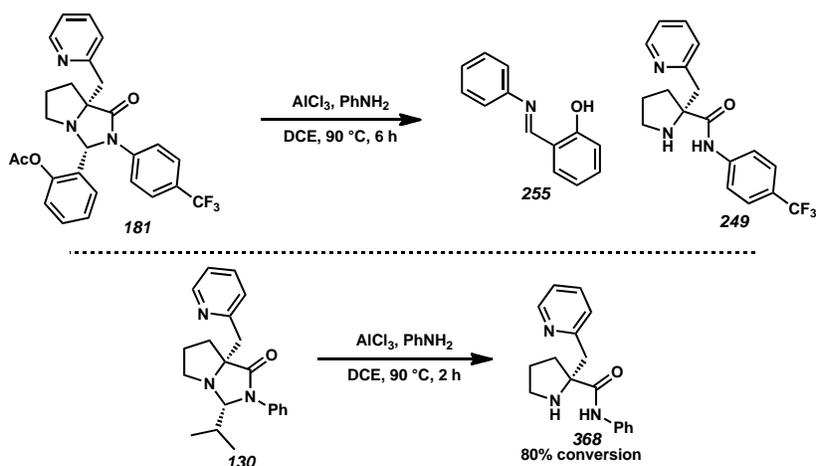
We began examining substrate scope and the feasibility of the AlCl₃ reaction conditions

for hydrolysis of the *N,N*-aminal. Alkylation of aminal **369** with MeI afforded the α-substituted

aminal in good yield (Scheme A1.1.5). Subjecting **370** to AlCl₃ and aniline in DCE for only 1 h

afforded the product in 90% isolated yield. Extending the reaction time to 3 h resulted in a

decrease of yield to 50%.

Scheme A1.1.5. Initial substrate synthesis and hydrolysis
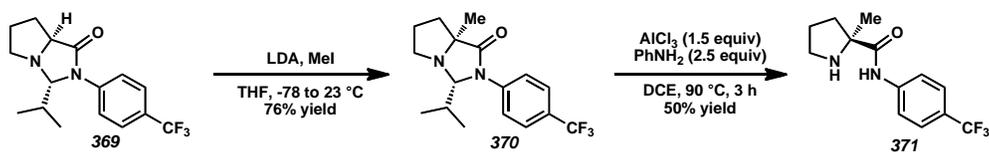
Having discovered a cleavage method providing a 90% yield of the amino amide, we

began to investigate substrate scope in terms of the amide component. Amino amide **178-OMe**

was condensed onto isobutyraldehyde to afford aminal **372**, followed by alkylation with MeI afforded α-substituted aminal **373** in good yield and as a single diastereomer (Scheme A1.1.6). Using *N,N*-aminal **123**, we installed an α-methyl group to afford aminal **374**. Additionally, we wanted to explore the tolerance of the reaction with an electron deficient 2,6-difluorophenyl amide. Formation of amino amide **190** occurred in moderate yield. Formation of aminal **375** via condensation and α-alkylation with MeI occurred in decent yield.

Scheme A1.1.6. Initial substrate scope
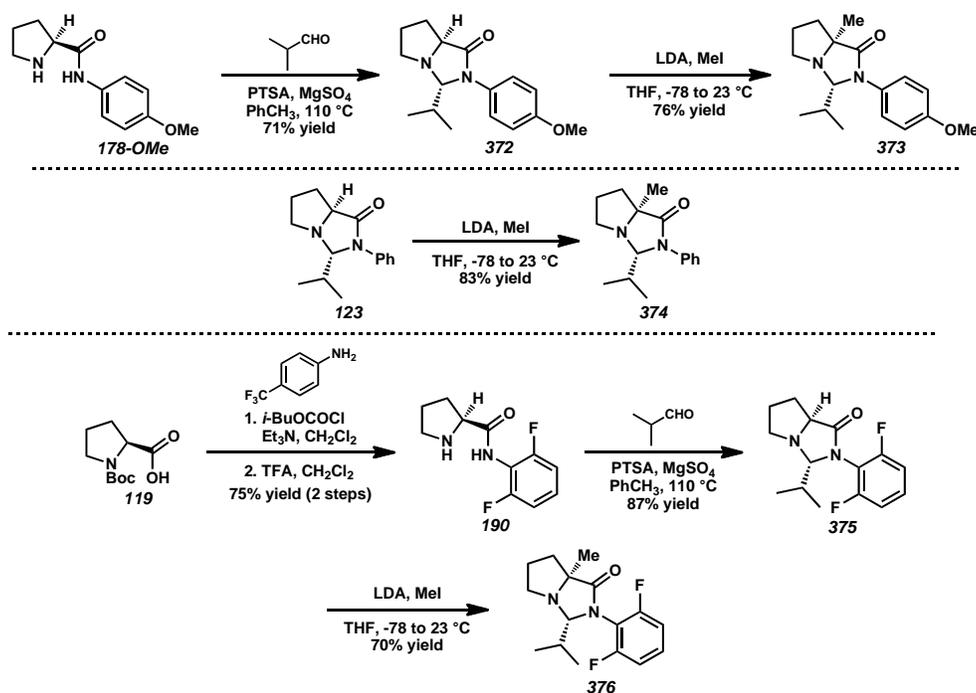
A1.2 Hydrolysis of Asymmetric *N,N*-aminals

Next we sought to examine how the electronics of the amide affected the hydrolysis of the aminal. Subjecting electron rich phenyl amide **373** to 1.1 equiv of AlCl₃ with aniline for 1 h afforded 60% conversion to amino amide **377** (Table A1.2.1). Increasing the amount of aluminum to 1.3 equiv for 16 h afforded less than 30% yield of amino amide **377**. When 1.5 equiv of AlCl₃ was employed for 3 h, a 77% isolated yield of the amino amide was observed.

We also subjected aminal **376** to hydrolysis conditions, and obtained a 38% yield of the desired amino amide, with no remaining starting material, suggesting a significant degree of decomposition.

Table A1.2.1. Examining aluminum-mediated hydrolysis

	PhNH$_2$	time (h)	conversion
1.1	2.5	1	60%
1.3	2.5	16	27% yield
1.5	2.5	3	77% yield
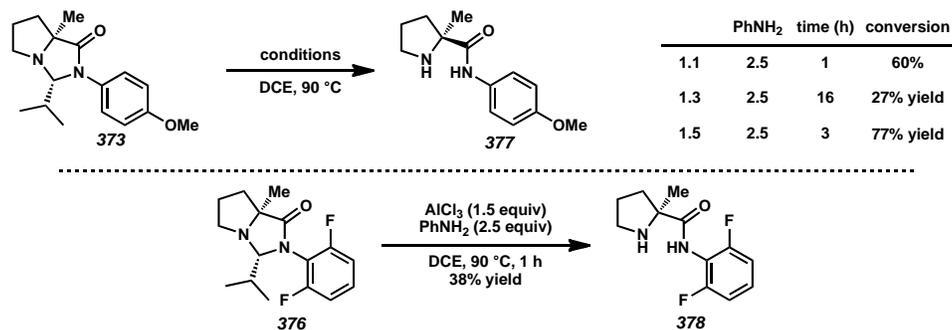
While both electron rich and electron deficient α-methyl substituted aminals had undergone hydrolysis, we next aimed to examine the electron neutral phenyl amide under hydrolysis conditions (Table A1.2.2). Subjecting **374** to typical AlCl$_3$ conditions resulted in no reaction. Employing Al(O*i*-Pr)$_3$ as the Lewis acid also gave no reaction. Utilizing AlCl$_3$ and *p*-anisidine as the amine afforded nearly a 50% yield of the hydrolysis product. We attribute this reactivity to the electron rich nature of the *p*-OMe amine. Employing FeCl$_3$ and aniline generated the amino amide in 38% yield. Other Lewis acids such as ZnCl$_2$, MgCl$_2$, and TiCl$_4$ provided less than 10% of the hydrolyzed product. When AlBr$_3$ was employed with aniline, a 40% conversion to amino amide **379** was observed.[5]

Table A1.2.2. Hydrolysis of the electron neutral aminal

conditions	result
AlCl₃ (1.5 equiv), PhNH₂ (2.5 equiv), 1 h	no reaction
Al(O-*i*-Pr)₃, PhNH₂ (2.5 equiv), 1 h	no reaction
AlCl₃ (1.5 equiv), *p*-anisidine, 16 h	48% yield
FeCl₃ (1.25 equiv), PhNH₂ (2.5 equiv), 26 h	38% yield
ZnCl₂ (1.25 equiv), PhNH₂ (2.5 equiv), 26 h	no reaction
MgCl₂ (1.25 equiv), PhNH₂ (2.5 equiv), 26 h	<10% yield
TiCl₄ (1.25 equiv), PhNH₂ (2.5 equiv), 26 h	<10% yield
AlBr₃ (1.5 equiv), PhNH₂ (2.5 equiv), 3.5 h	40% conversion
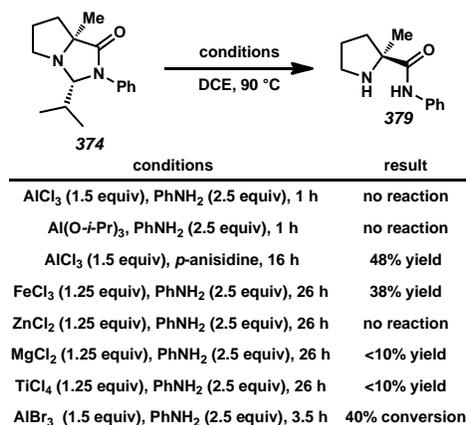
The pyridyl-substituted aminals appeared to have less reactivity towards hydrolysis than the corresponding methyl substituted aminal.[6] Subjecting pryidine **129** to AlBr₃ and aniline in DCE afforded a 14% yield of the hydrolyzed product (Scheme A1.2.1). Taking a slightly different approach we subjected **129** to CSA and aniline in IPA or MeOH affording much better conversion to product (36 and 50%, respectively).[7]

Scheme A1.2.1. Hydrolysis of aminal 129
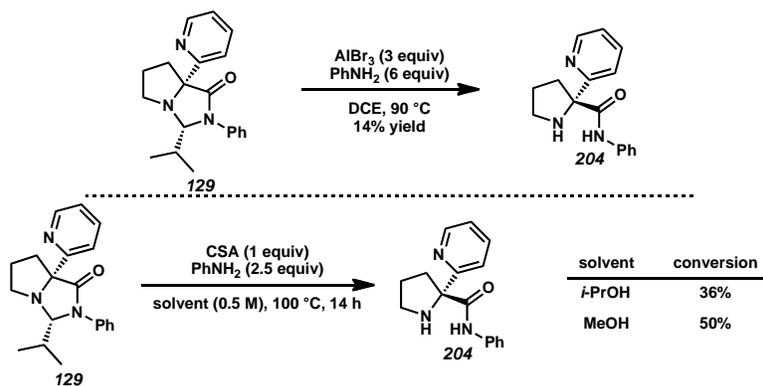
Purification of the amino amide proved difficult with the amount of aniline left in the reaction mixture.[8] We imagined using a volatile amine that could be removed upon workup. Treating **129** with CSA and butylamine resulted in no conversion to the desired amino amide (Scheme A1.2.2). When aminal **370** was subjected to the same conditions, however, almost 50% conversion to the desired product was observed.

Scheme A1.2.2. Hydrolysis of aminals with an amine catalyst
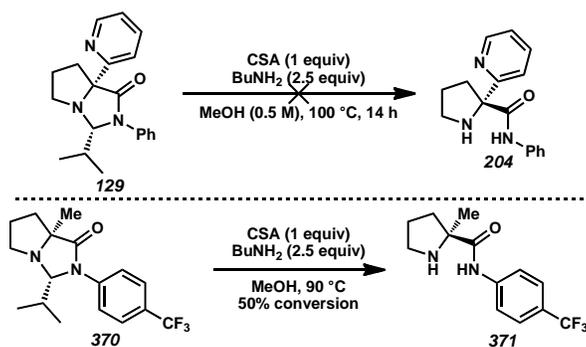
Rather than using a super-stoichiometric amount of aniline, we discovered that we could use aniline catalytically with CSA in MeOH at 110 °C, to afford 70% yield of the desired amino amide (Scheme A1.2.3). The remainder of the reaction mixture was recovered starting material.

Scheme A1.2.3. Hydrolysis of aminal 129
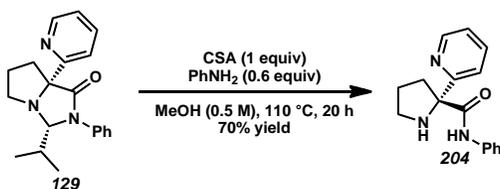
We next explored the acid mediated hydrolysis of aminal **129** in our amino amide synthesis. Treating aminal **374** with stoichiometric CSA and catalytic aniline afforded nearly 90% conversion to the product with 75% isolated yield (Table A1.2.3). Reducing the amount of CSA to 50 mol % reduced the conversion to 50% of product **379**. Utilizing 1 equiv of CSA and only 25 mol % aniline in MeOH afforded 92% conversion to the amino amide product. At 90 °C, the same conditions only afforded 50% conversion to the desired product. Realizing that 1 equiv of acid was necessary, we wanted to optimize the reaction in terms of amine catalyst loading and concentration of the reaction. The use of 1 equiv of CSA and 50 mol % aniline in 0.5 M MeOH at 100 °C afforded 76% conversion to product. The conversion improved when the amount of aniline was decreased to 25 mol % at the same concentration. Modifying the acid

to PTSA with 50 mol % aniline and a 0.5 M concentration in MeOH afforded 92% conversion to amino amide **379**. These results indicate that only 25 mol % aniline is necessary at high concentrations (0.5 M or greater).

Table A1.2.3. Acid and amine screen
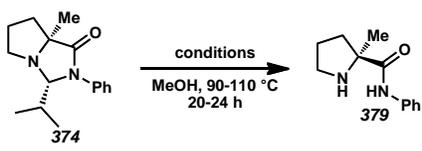
acid	PhNH$_2$		conversion (110 °C)
CSA (1 equiv)	0.6 equiv		86% (75% yield)
CSA (0.5 equiv)	0.6 equiv		50%
CSA (1 equiv)	0.25 equiv		92% (50% at 90 °C)

acid	PhNH$_2$	MeOH [M]	conversion (100 °C)
CSA (1 equiv)	0.5 equiv	0.5	76%
CSA (1 equiv)	0.25 equiv	0.5	84%
CSA (1 equiv)	0.5 equiv	0.25	77%
PTSA (1 equiv)	0.5 equiv	0.5	92%
PTSA (1 equiv)	0.5 equiv	0.25	81%
PTSA (1 equiv)	0.25 equiv	0.25	84%
PTSA (1 equiv)	0.25 equiv	0.5	91%

While we had achieved high levels of conversion, we had been unable to make the reaction proceed to completion. We imagined that the product might inhibit further hydrolysis by quenching the acid mediator. Treating aminal **374** with 1.25 equiv of PTSA and 25 mol % aniline in MeOH at 90 °C for 48 h gave complete conversion to amino amide **379** (Scheme A1.2.4). We began examining this hydrolysis reaction on electronically different amides. Treating electron rich aminal **373** with aniline and 1 equiv of acid afforded 76% conversion to the hydrolyzed product. Likewise, subjecting electron deficient aminal **370** to the same conditions afforded 88% conversion to amino amide **377**. Employing excess acid, such as 1.25 equiv of PTSA, should provide complete conversion to the hydrolyzed products.

Scheme A1.2.4. Hydrolysis substrate scope
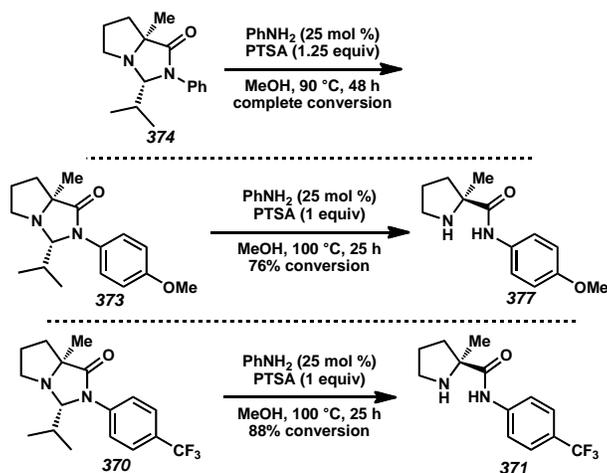
A1.3 Substrate Scope

We ultimately wanted to investigate the tolerance of α-substitution (Scheme A1.3.1). Treating aminal **123** with LDA and a variety of electrophiles allowed us to achieve a broad substrate scope in good yield. The aldehyde (**383**) could be reduced to the primary alcohol (**385**) via treatment with NaBH₄ in good yield.

Scheme A1.3.1. α-Substitution substrate scope

	R =	yield (%)
380	allyl	81
381	propargyl	63
382	CH₂CO₂Me	55
383	CHO	56
384	Et	60
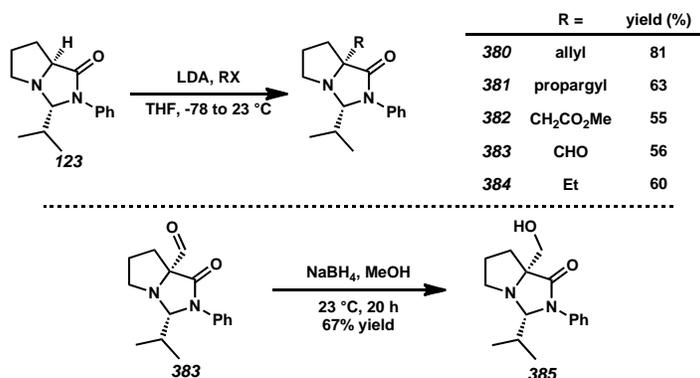
We also wanted to expand the scope of aryl amides to include alkyl amides. Formation of *t*-butyl amino amide **386** occurred in 69% yield in two steps (Scheme A1.3.2). Condensation onto isobutyraldehyde occurred in moderate yield to afford the desired aminal. Alkylation was unsuccessful, however, resulting in unreacted starting material. We attribute this lack of

reactivity to the large steric bulk of the *t*-butyl group. We were also able to synthesize the benzyl amino amide in 2 steps in good yield. Condensation onto isobutyraldehyde afforded aminal **390** in good yield. Alkylation, however, was again unsuccessful and provided a complex mixture of products.[9]

Scheme A1.3.2. Generating alkyl amides
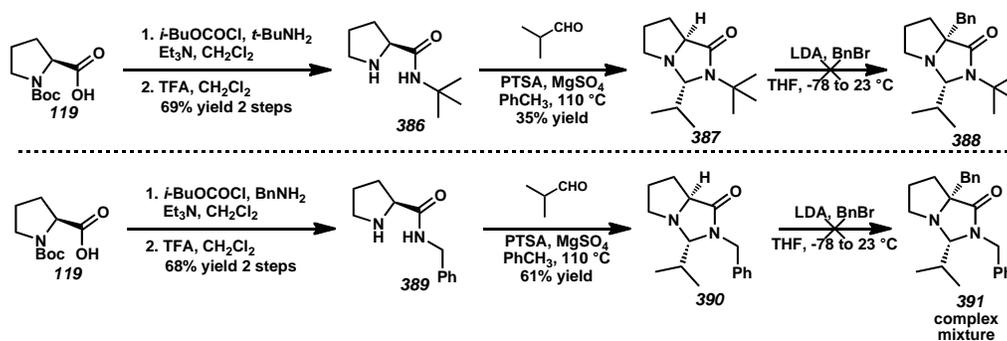
Rather than installing a *t*-butyl amide, we thought a cyclohexane amide may be less sterically hindered and function in the alkylation reaction (Scheme A1.3.3). Formation of the cyclohexane amide (**392**) followed by condensation onto isobutyraldehyde afforded aminal **393**. Likewise, we thought an *n*-butyl amide would also be significantly less sterically hindered. Formation of the butyl amide (**81**) occurred in good yield in two steps. Condensation onto isobutyraldehyde afforded aminal **53** in moderate yield. Alkylation of these two substrates with benzyl bromide would afford the desired α-substituted *N,N*-aminals.

Scheme A1.3.3. Less sterically hindered alkyl amides
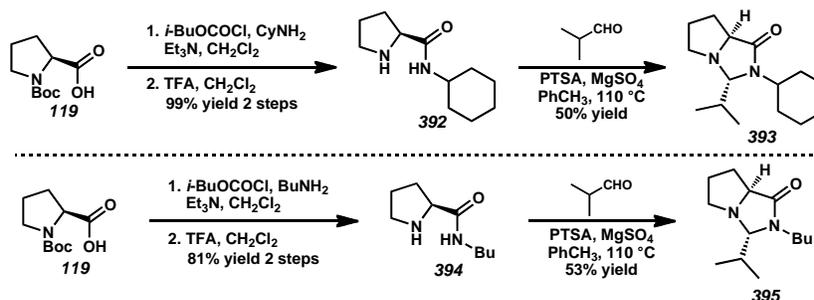
A1.4 Determination of Enantiomeric Excess

In order to determine the success of our asymmetric synthesis, we needed to examine the enantiomeric excess (*ee*) of the resultant amino amides. We have fully developed the asymmetric route to the amino amides, including hydrolysis. After developing a racemic route to the amino amides we were able to determine the *ee* by chiral HPLC. Method development on HPLC provided adequate separation between the two enantiomers. Analysis of asymmetric **379** revealed that a single enantiomer results from our synthesis.

The method we have developed for the synthesis of these asymmetric amino amides is both high yielding and decidedly stereorententive. The remaining substrates need to be hydrolyzed to reveal the asymmetric amino amides, and the corresponding racemate needs to be synthesized in order to determine the remaining *ee*'s.

A1.5 Generation of the Other Enantiomer from L-Proline

Besides retaining the stereochemistry of the L-proline starting material, we envisioned obtaining the other enantiomer via an anti-selective alkylation. Treating aminal **171** with LDA and MeI afforded a 1.5 : 1 mixture of syn to anti isomers (Scheme A1.5.1). Subjecting more electron rich aminal **398** to the same conditions afforded a 1 : 1 mixture of syn to anti isomers. It is evident that the more electron rich nature of the aminal favors more anti alkylation.

Scheme A1.5.1. Initial attempts at anti functionalization

1.5 : 1 mixture

1 : 1 mixture
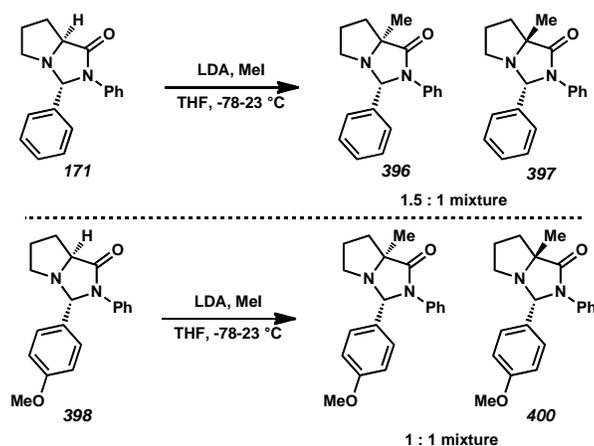
Based on this idea, we imagined employing an even more electron rich and more sterically hindered *N,N*-aminal. Condensation onto 2,4-dimethoxybenzaldehyde afforded the aminal (**401**) as a single diastereomer of unknown stereochemistry (Scheme A1.5.2). Alkylation of this aminal was unsuccessful utilizing LDA and benzyl bromide, only providing a complex mixture of products. It is likely that the highly electron rich nature of the aminal resulted in decomposition upon treatment with the alkylation conditions.

Scheme A1.5.2. Utilizing a more electron rich aromatic aldehyde
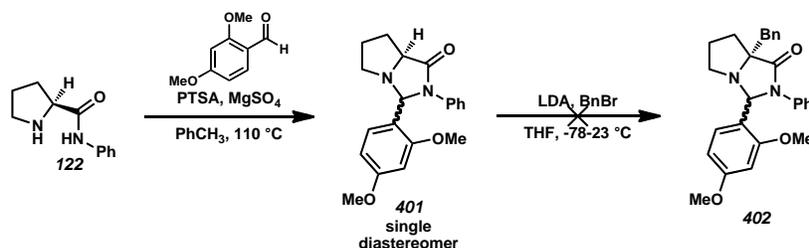
We were unable to determine if an anti alkylation would be favored under such conditions due to decomposition. The unidentified diastereomer of the condensation reaction was very intriguing, however. If the condensation reaction provided the anti diastereomer, then a syn alkylation would be more desirable to afford the other enantiomer of the amino amide. Condensation onto a sterically hindered aromatic aldehyde such as mesityl aldehyde may afford

the anti diastereomer (**403**) (Scheme A1.5.3). If we could then perform a syn alkylation, we would obtain the desired enantiomer of the amino amide.

Scheme A1.5.3. Sequence for syn alkylation
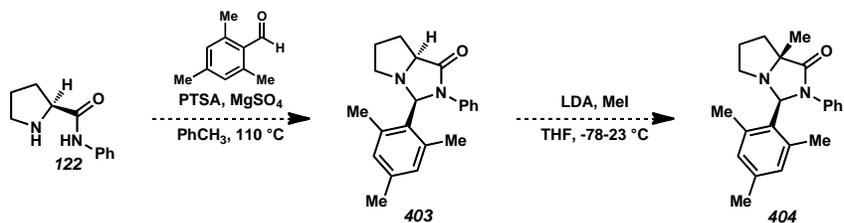
A1.6 Conclusion

We have developed a highly stereoretentive method of substituted amino amide synthesis. The synthesis of *N,N*-aminals is straightforward with simple high yielding reactions. The hydrolysis of the aminal occurs with high yield, and because only a catalytic amount of amine is needed, is easy to purify via chromatography. The amino amide we have obtained thus far has shown to be a single enantiomer by HPLC analysis. The remaining aminal substrates will need to be hydrolyzed and then analyzed by HPLC in order to determine the *ee* of the corresponding amino amides.

A1.7 References and Notes

[1] MacMillan, D. W. C. *Nature* **2008**, *455*, 304-308. (b) Erkkilä, A.; Majander, I.; Pihko, P. M. *Chem. Rev.* **2007**, *107*, 5416-5470. (c) Mukherjee, S.; Yang, J. W.; Hoffmann, S.; List, B. *Chem. Rev.* **2007**, *107* 5471-5569.

[2] (a) Boes, M.; Naef, R.; Schweizer, W. B.; Seebach, D. *J. Am. Chem. Soc.* **1983**, *105*, 5390-5398. (b) Sting, A. R.; Hoffman, M.; Seebach, D. *Angew. Chem., Int. Ed. Engl.* **1996**, *35*, 2708.

[3] Bon, E.; Bigg, D. C. H.; Bertrand, G. *J. Org. Chem.* **1994***, 59*, 4035-4036.

[4] Transamination does not occur under these conditions. Presumably $AlCl_3$ acts as a Lewis acid, activating the aminal center for attack by the stoichiometric amine. Collapse of the aminal center to form the corresponding imine occurs to provide the free amino amide.

[5] Using $Al(Oi\text{-}Pr)_3$ as a Lewis acid resulted in no reaction, suggesting that a much strong Lewis acid was needed. $AlCl_3$ provided trace amounts of reaction, so we examined $AlBr_3$ which is even stronger than $AlCl_3$.

[6] This may be due to some cooperative effect of the pyridyl nitrogen and the aluminum reagent.

[7] Li, D.; Zhang, Y.; Xia, C.; Guo, W. *Heterocycles* **2005**, *65*, 1829-1836. We abandoned the use of Lewis acids for hydrolysis due to their inconsistency across substrates and propensity to cause decomposition.

[8] Aniline could not be removed using any method other than column chromatography. Even with chromatography, the aniline generally eluted with the product.

[9] This mixture of products may be due to competitive deprotonation of a benzyl proton on the benzyl amide.

[10] Kelly, S.; Watts, J.; McKee, V.; Kelleher, F. *Tetrahedron* **2010**, *66*, 3525-3536. The authors stated that the reaction occurred in just 5 h. We found incomplete conversion, but by running the reaction for several days, near complete conversion could be obtained.

A1.8 Experimental procedures

Materials and Methods.   All reactions were performed under an argon atmosphere unless otherwise noted.   Tetrahydrofuran, *N,N*-dimethylformamide, dichloromethane, hexanes, and toluene were purified by passing through activated alumina columns.   Diisopropylamine was distilled over CaH_2. 2-Fluoropyridine was freshly distilled before use. All other reagents were used as received unless otherwise noted.   Commercially available chemicals were purchased from Alfa Aesar (Ward Hill, MA), Sigma-Aldrich (St. Louis, MO), Gelest (Morrisville, PA), Oakwood Products (West Columbia, SC), Strem (Newburport, MA), Mallinckrodt Chemicals (Phillipsburg, NJ), Spectrum (Gardena, CA) Fischer Scientific (Fair Lawn) and TCI America (Portland, OR). Qualitative TLC analysis was performed on 250 mm thick, 60 Å, glass backed, F254 silica (Silicycle, Quebec City, Canada). Visualization was accomplished with UV light and exposure to either *p*-anisaldehyde or $KMnO_4$ solution followed by heating.   Flash chromatography was performed using Silicycle silica gel (230-400 mesh). ^{1}H NMR spectra were acquired on either a Varian Mercury 300 (at 300 MHz), a Varian Inova 400 (at 400 MHz), or a Varian 400 MR (at 400 MHz) and are reported relative to $SiMe_4$ (δ 0.00). ^{13}C NMR spectra were acquired on either a Varian Inova 400 (at 100 MHz), a Varian Mercury 300 (at 75 MHz), or a Varian 400 MR (at 100 MHz) and are reported relative to $SiMe_4$ (δ 0.0).   All IR spectra were obtained on NaCl plates (film) with either a Nicolet Magna FTIR 760, a Nicolet 380 FTIR, or a Bruker Tensor 27. High resolution mass spectrometry data were acquired by the Colorado State University Central Instrument Facility on an Agilent 6210 TOF LC/MS.

General procedure for α-alkylation: To a solution of diisopropylamine (0.744 mL, 5.30 mmol) in THF (11.2 mL) at -78 °C was added *n*-BuLi (2.03 mL, 5.08 mmol, 2.5 M in hexanes). After stirring for 10 min at -78 °C, **369** (1.32 g, 4.24 mmol) was added in THF (10.0 mL) at – 78 °C. The reaction was stirred for 30 min at -78 °C. MeI (0.396 mL, 6.36 mmol) was added at -78 °C, at which time the reaction was warmed to 23 °C and stirred overnight. Water (10 mL) was added. The aqueous layer was extracted with EtOAc (3 x 20 mL). The organics were washed with brine, dried over Na_2SO_4 and concentrated. The crude residue was purified via flash chromatography (7:1 to 4:1 hexanes:EtOAc) to afford **370** (1.06 g, 76% yield, R_f = 0.32 in 4:1 hexanes:EtOAc) as a yellow solid.

General procedure for hydrolysis via $AlCl_3$: To **370** (150 mg, 0.460 mmol) in DCE (2.30 mL) was added $AlCl_3$ (91.9 mg, 0.689 mmol) and aniline (0.105 mL, 1.15 mmol) in a 2-dram vial. The reaction was heated to 90 °C for 3 h. Upon cooling, the reaction was quenched with H_2O. The aqueous layer was extracted with EtOAc (3 x 5 mL). The organic layer was dried over Na_2SO_4 and concentrated. The crude residue was purified over silica (7:3 to 1:1 hexanes:EtOAc) to afford **371** (62.3 mg, 50% yield, R_f = 0.0 in 1:1 hexanes:EtOAc) as an orange oil.
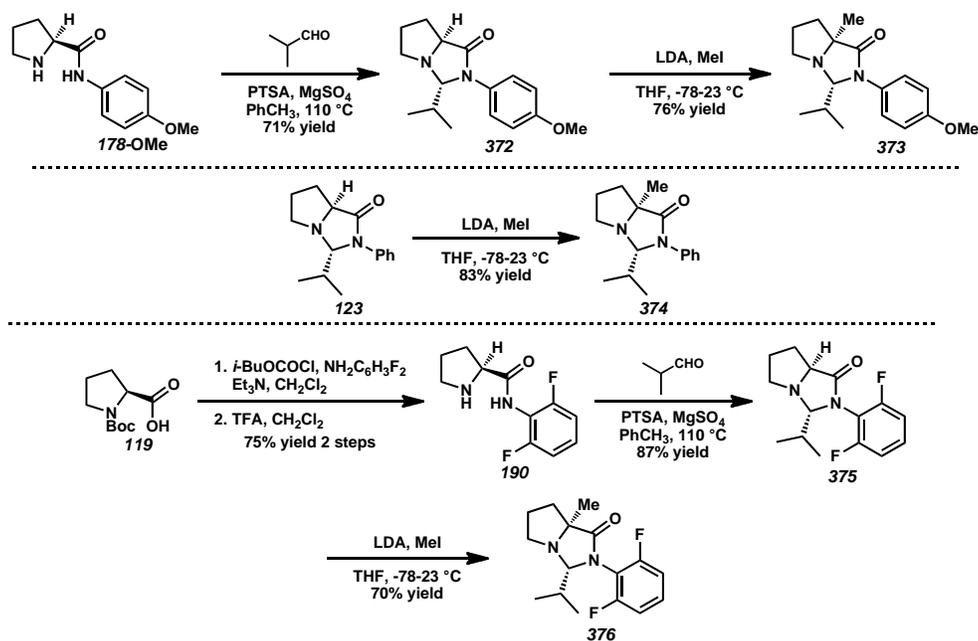
General procedure for aminal formation: **178-OMe** (826 mg, 3.75 mmol), isobutyraldehyde (0.513 mL, 5.63 mmol), PTSA (35.7 mg, 0.188 mmol) and MgSO$_4$ (677 mg, 5.63 mmol) were combined in toluene (25.0 mL) and heated to reflux overnight. The reaction was quenched with sat. aq. NaHCO$_3$. The aqueous layer was extracted with EtOAc (3 x 20 mL), the organics dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified over silica (3:1 to 7:3 hexanes:EtOAc) to afford **372** (733 mg, 71% yield, R$_f$ = 0.24 in 1:1 hexanes:EtOAc) as a yellow solid.

Aminal 373: According to the general procedure, **373** (76% yield, R$_f$ = 0.52 in 1:1 hexanes:EtOAc) as a yellow solid.

Aminal 374: According to the general procedure, **374** (83% yield, R$_f$ = 0.32 in 4:1 hexanes:EtOAc) as a yellow solid.

Aminal 375: According to the general procedure, **375** (87% yield, R$_f$ = 0.53 in 1:1 hexanes:EtOAc) as a white solid.

Aminal 376: According to the general procedure, **376** (70% yield, R_f = 0.62 in 1:1 hexanes:EtOAc) as a yellow solid.

	NH₂Ph	time (h)	conversion
1.1	2.5	1	60%
1.3	2.5	16	27% yield
1.5	2.5	3	77% yield
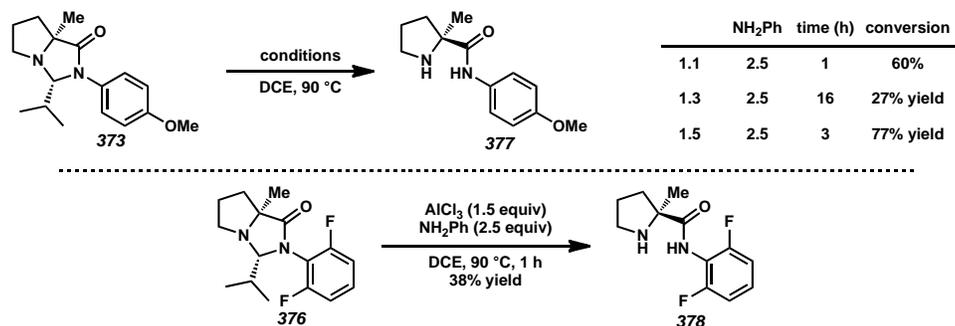
Amide 378: According to the general procedure **378** (38% yield, R_f = 0.0 in 1:1 hexanes:EtOAc) was isolated as a light brown solid.

conditions	result
AlCl₃ (1.5 equiv), NH₂Ph (2.5 equiv), 1 h	no reaction
Al(O-*i*-Pr)₃, NH₂Ph (2.5 equiv), 1 h	no reaction
AlCl₃ (1.5 equiv), *p*-anisidine, 16 h	48% yield
FeCl₃ (1.25 equiv), NH₂Ph (2.5 equiv), 26 h	38% yield
ZnCl₂ (1.25 equiv), NH₂Ph (2.5 equiv), 26 h	no reaction
MgCl₂ (1.25 equiv), NH₂Ph (2.5 equiv), 26 h	<10% yield
TiCl₄ (1.25 equiv), NH₂Ph (2.5 equiv), 26 h	<10% yield
AlBr₃ (1.5 equiv), NH₂Ph (2.5 equiv), 3.5 h	40% conversion
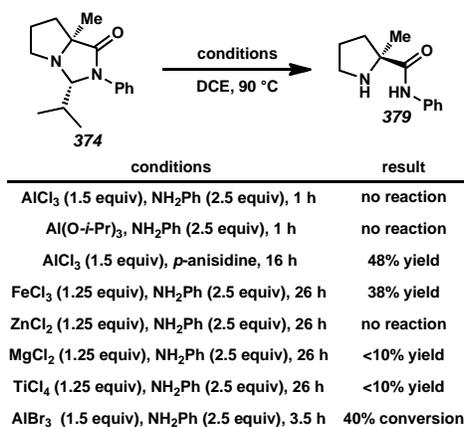
General procedure: **374** (1 equiv), Lewis acid (1.25-1.5 equiv), amine (2.5 equiv) and DCE were combined in a 2-dram vial and heated to 90 °C for the indicated time. Upon cooling, the reaction was quenched with H₂O. The aqueous layer was extracted with EtOAc (3 x 5 mL). The organic layer was dried over Na₂SO₄ and concentrated. The conversion was determined by crude ¹H NMR.
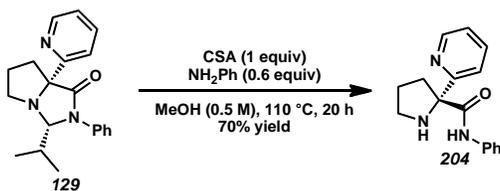
Representative procedure for acid hydrolsis: To pyridine **129** (750 mg, 2.33 mmol) in a screw cap vial with Teflon cap was added CSA (542 mg, 2.33 mmol), NH$_2$Ph (106 μL, 1.17 mmol) and MeOH (4.66 mL). The reaction was heated to 110 °C for 24 h. Upon cooling, the reaction mixture was concentrated. To the residue was added sat. aq. NaHCO$_3$ (20 mL). The mixture was extracted with EtOAc (3 x 15 mL). The combined organic layers were dried over Na$_2$SO$_4$ and concentrated. The crude residue was purified by column chromatography (3:1 to 3:2 hexanes/EtOAc eluent) to afford amino amide **204** (874 mg, 70% yield (97% yield borsm) R$_f$ = 0.05 in 1:1 hexanes/EtOAc) as a beige solid.

acid	NH$_2$Ph	conversion (110 °C)
CSA (1 equiv)	0.6 equiv	86% (75% yield)
CSA (0.5 equiv)	0.6 equiv	50%
CSA (1 equiv)	0.25 equiv	92% (50% at 90 °C)

acid	NH$_2$Ph	MeOH [M]	conversion (100 °C)
CSA (1 equiv)	0.5 equiv	0.5	76%
CSA (1 equiv)	0.25 equiv	0.5	84%
CSA (1 equiv)	0.5 equiv	0.25	77%
PTSA (1 equiv)	0.5 equiv	0.5	92%
PTSA (1 equiv)	0.5 equiv	0.25	81%
PTSA (1 equiv)	0.25 equiv	0.25	84%
PTSA (1 equiv)	0.25 equiv	0.5	91%

According to the general procedure, the conversion to **379** was determined by [1] H NMR.

	R =	yield (%)
380	allyl	81
381	propargyl	63
382	CH$_2$CO$_2$Me	55
383	CHO	xx
384	Et	xx
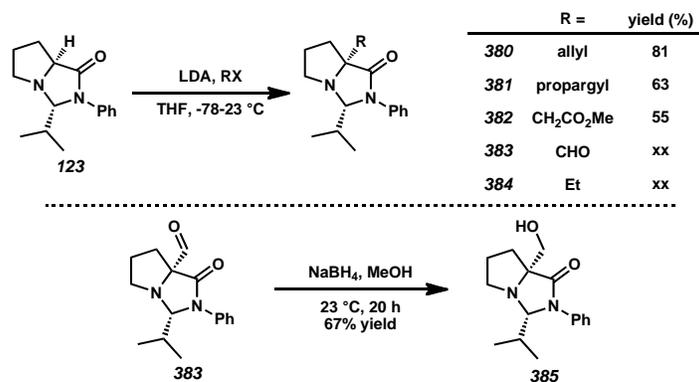
Aminal 380: According to the general procedure **380** (81% yield, R_f = 0.48 in 3:1 hexanes:EtOAc) was isolated as a light brown solid.

Aminal 381: According to the general procedure **381** (63% yield, R_f = 0.45 in 3:1 hexanes:EtOAc) was isolated as a light brown solid.

Aminal 382: According to the general procedure **382** (55% yield, R_f = 0.40 in 3:1 hexanes:EtOAc) was isolated as a light yellow oil.

Aminal 383: According to the general procedure **383** (xx% yield, R_f = 0.42 in 3:1 hexanes:EtOAc) was isolated as a light yellow solid.

Alcohol 385: According to the general procedure **385** (67% yield, R_f = 0.25 in 3:1 hexanes:EtOAc) was isolated as a light yellow solid.
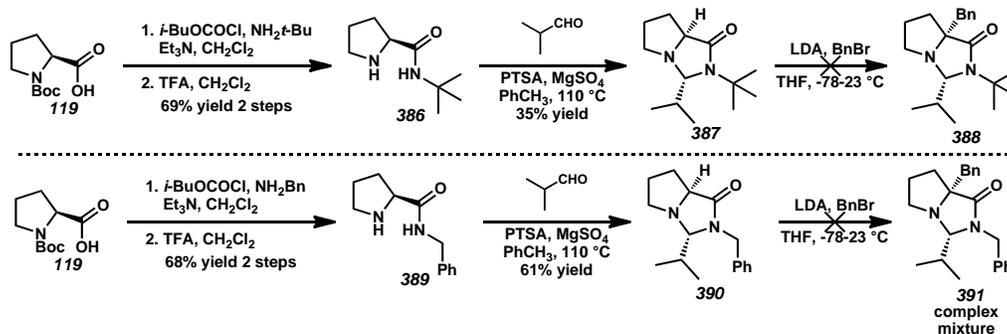
Representative procedure for amino amide formation: To a solution of (*S*)-*N*-Boc proline (750 mg, 3.48 mmol) in CH_2Cl_2 (17.4 mL) at 0 °C was added isobutyl chloroformate (0.501 mL, 3.83 mmol) and triethylamine (0.539 mL, 3.83 mmol). After stirring for 20 minutes at 0 °C, $BnNH_2$ (0.419 mL, 3.83 mmol) was added, and the reaction was warmed to 23 °C and stirred overnight. The reaction was washed sequentially with aq. $KHSO_4$ (1 M, 20 mL), sat. aq. $NaHCO_3$ (20 mL) and brine (20 mL). The organic layer was dried over Na_2SO_4 and concentrated to afford a pale brown solid. To a solution of crude amide in CH_2Cl_2 (6.97 mL) at 23 °C was added TFA (5.37 mL, 69.7 mmol). The solution was stirred at 23 °C for 1 h, and the solvent was removed under

reduced pressure. The residue was taken up in CH_2Cl_2 (20 mL) and neutralized with solid Na_2CO_3 until pH ~9. Water (10 mL) was added and the aqueous layer was extracted with CH_2Cl_2 (3 x 25 mL). The combined organic layers were dried over Na_2SO_4 and concentrated to afford amino amide **389** (482 mg, 68% yield, R_f = 0.00 in 1:1 hexanes/EtOAc) as a light brown solid, which was sufficiently pure to be taken on to the next step.

Aminal 390: According to the general procedure **390** (61% yield, R_f = 0.25 in 1:1 hexanes:EtOAc) was isolated as a light yellow solid.

Amide 386: According to the general procedure **386** (69% yield, R_f = 0.0 in 1:1 hexanes:EtOAc) was isolated as a white solid.

Aminal 387: According to the general procedure **387** (35% yield, R_f = 0.20 in 1:1 hexanes:EtOAc) was isolated as a light yellow solid.
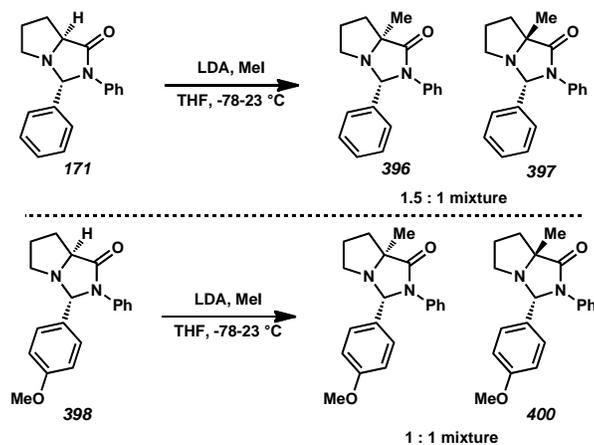
According to the general procedure. The product ratios were determined by crude ^{1}H NMR.

Appendix B: Spectra Associated with Chapter 2

146

173

174

179-OMe

179-CF₃

180a-OMe

180b-OMe

180a-CF₃

180b-CF₃

181-OMe

181-CF₃

203

206

207b

210

213

214

214b

217
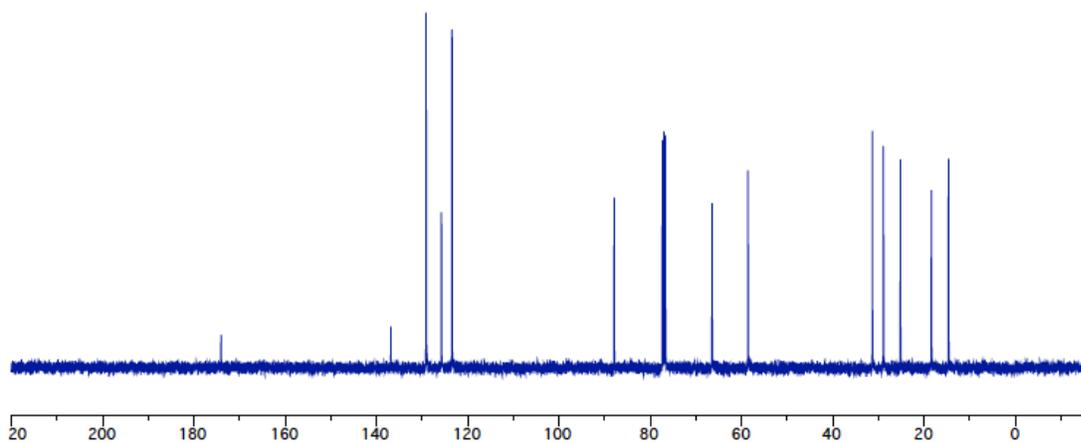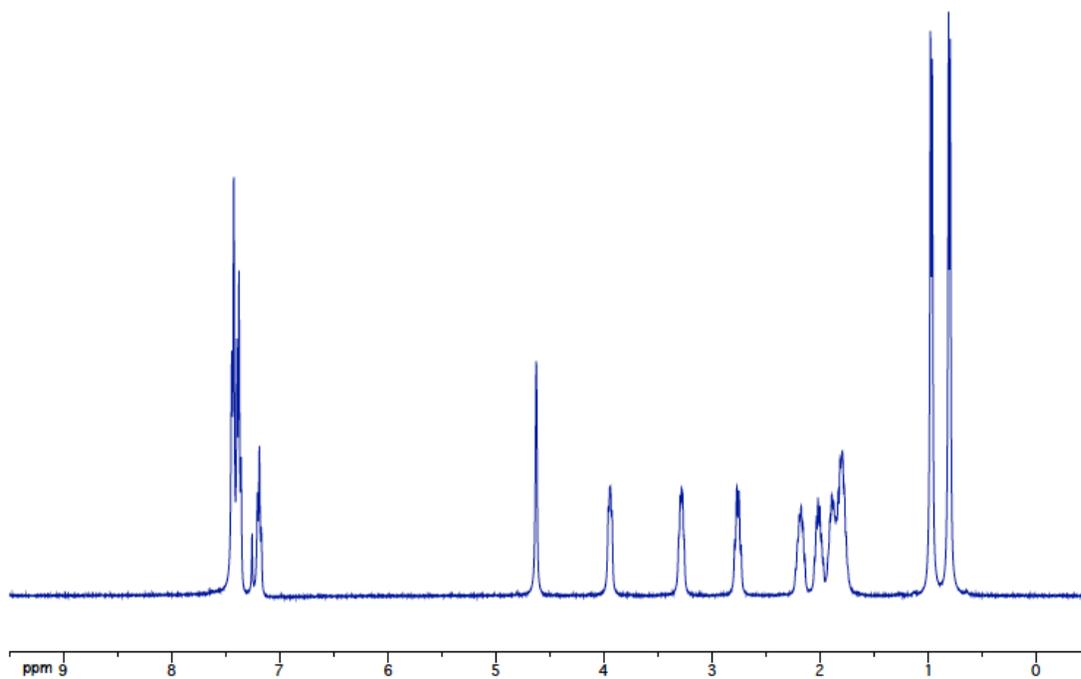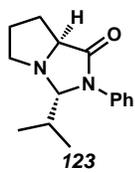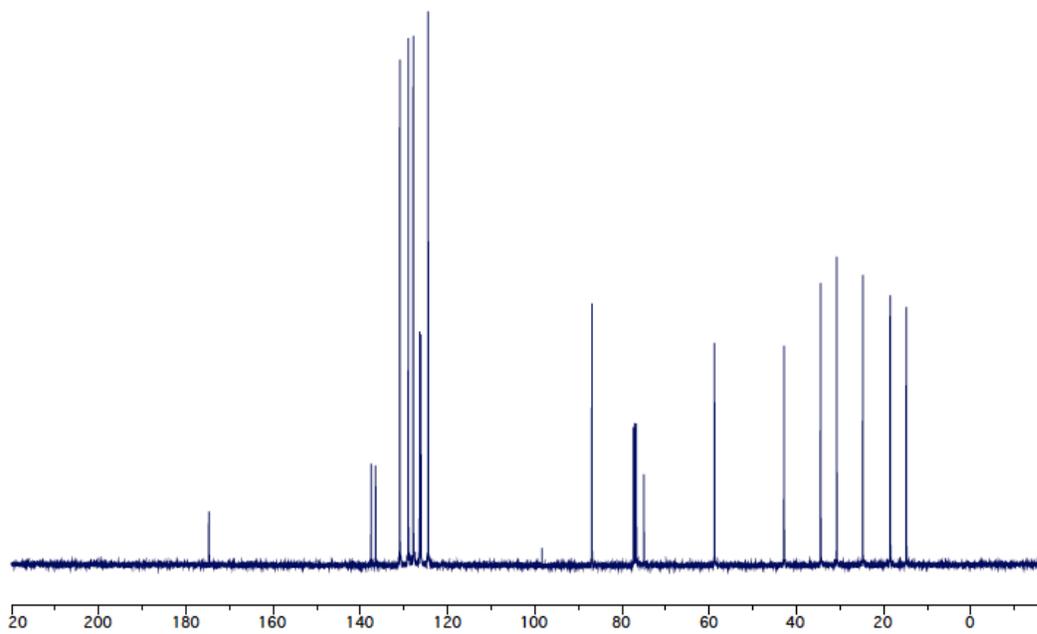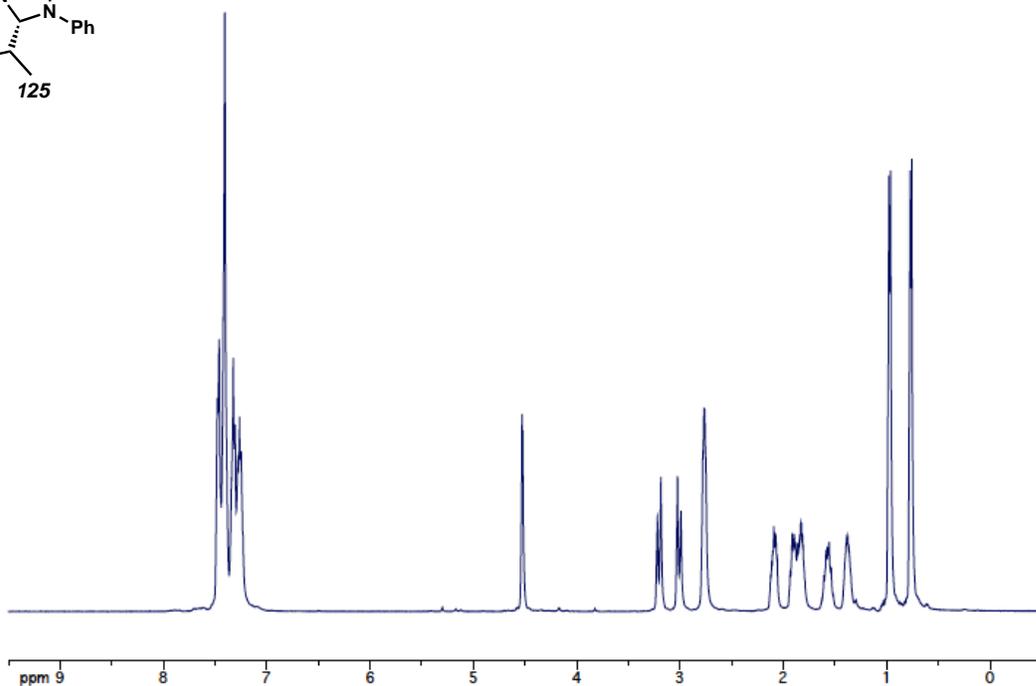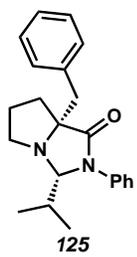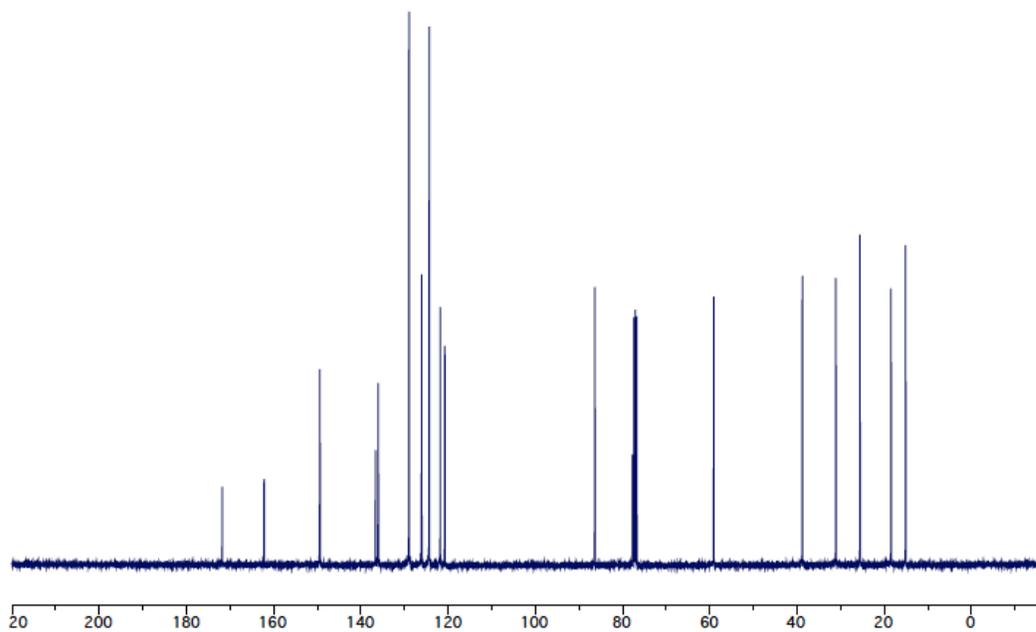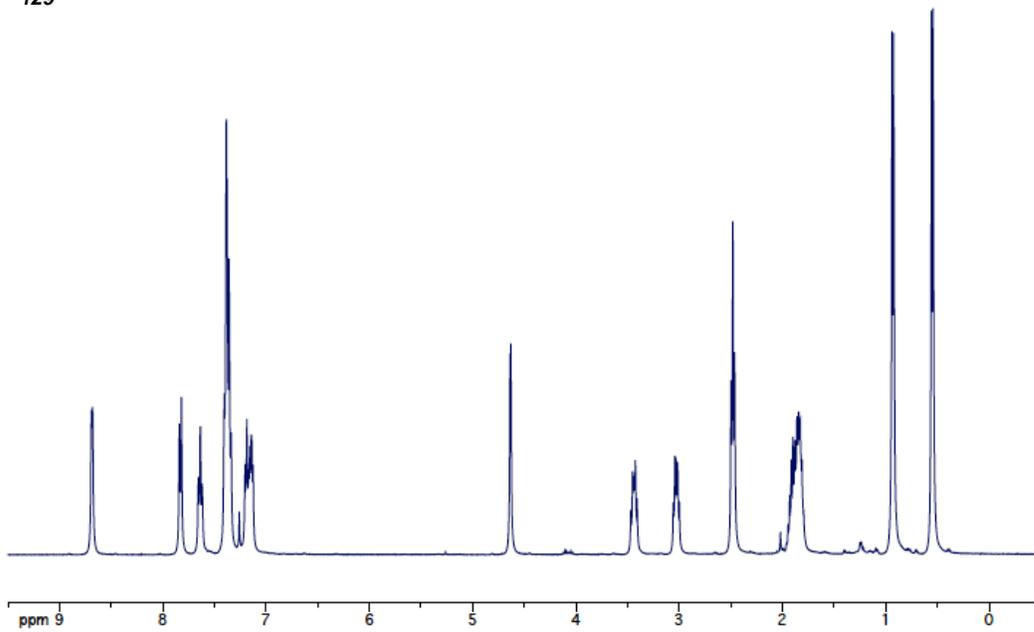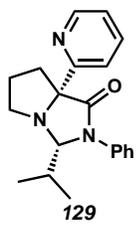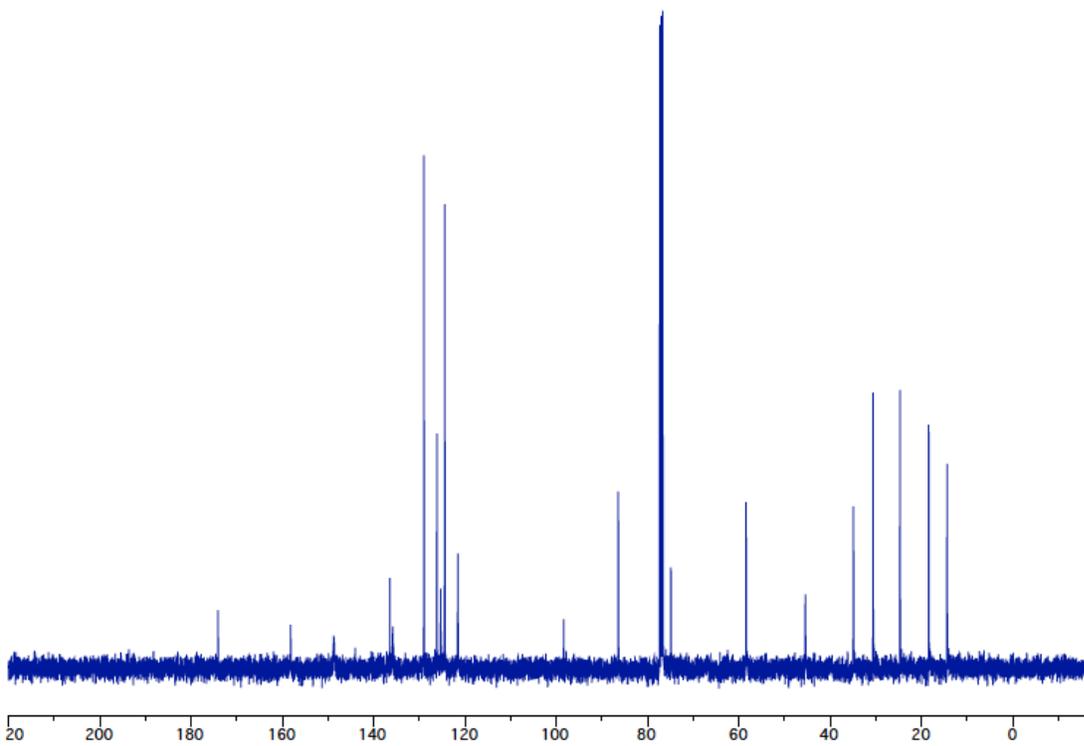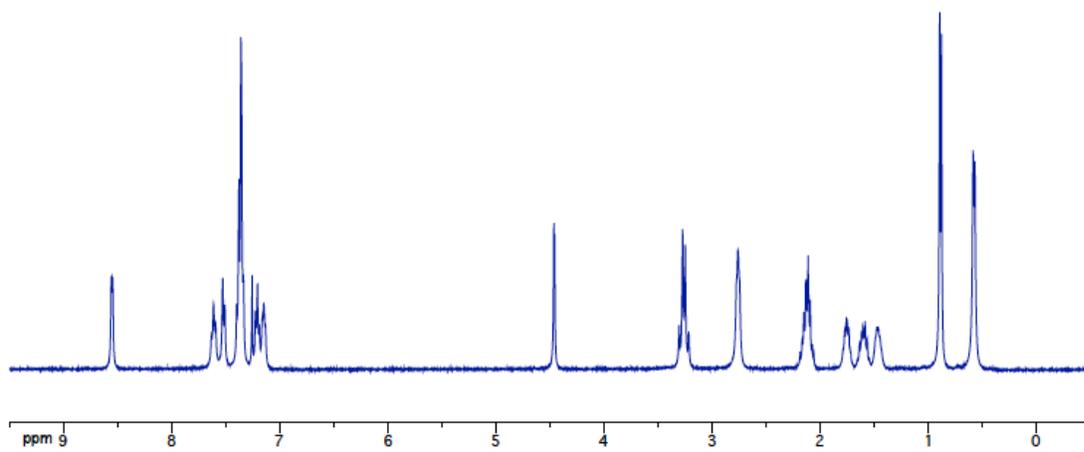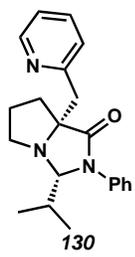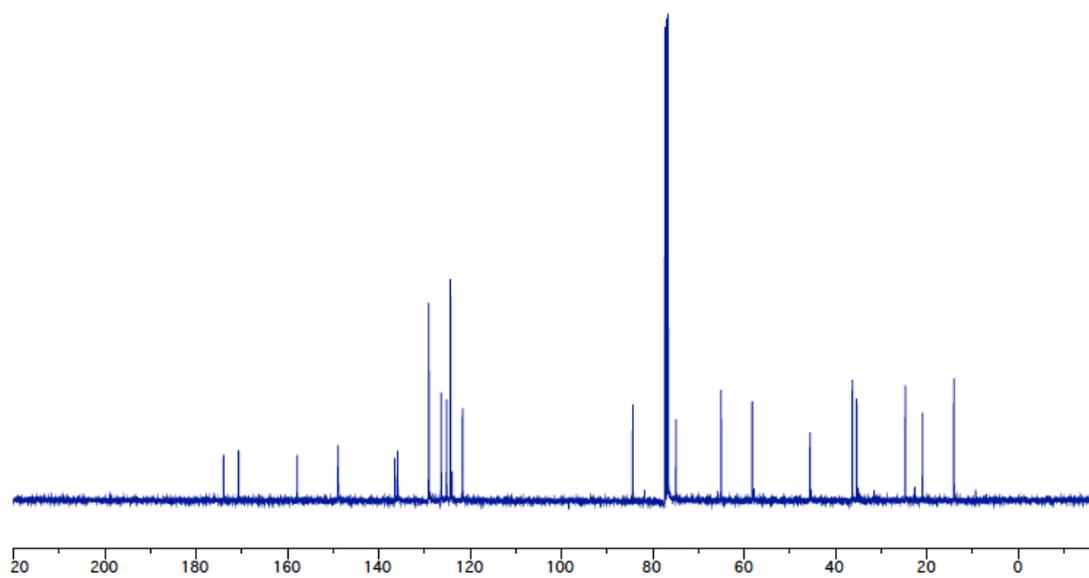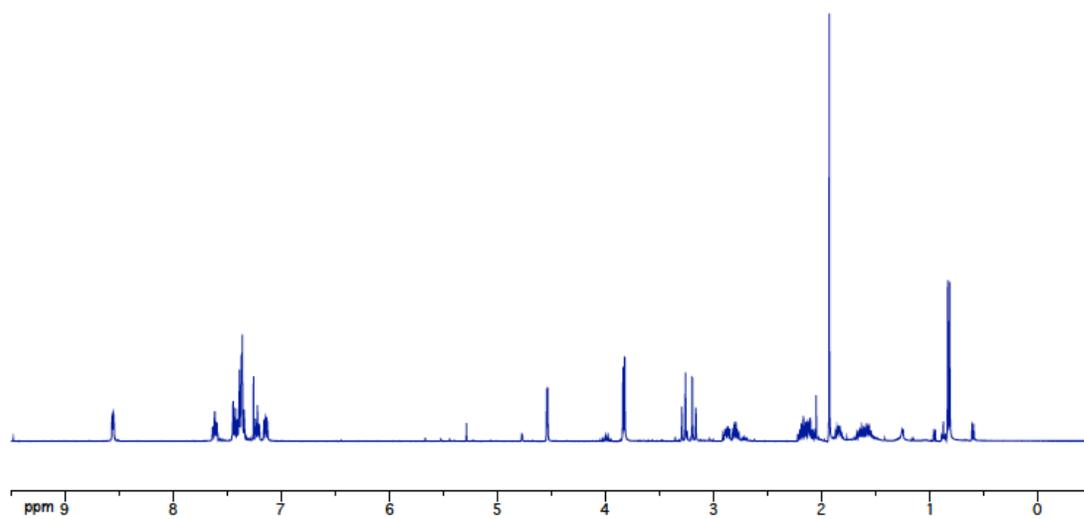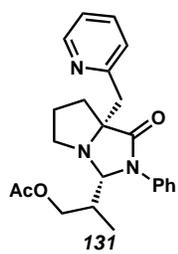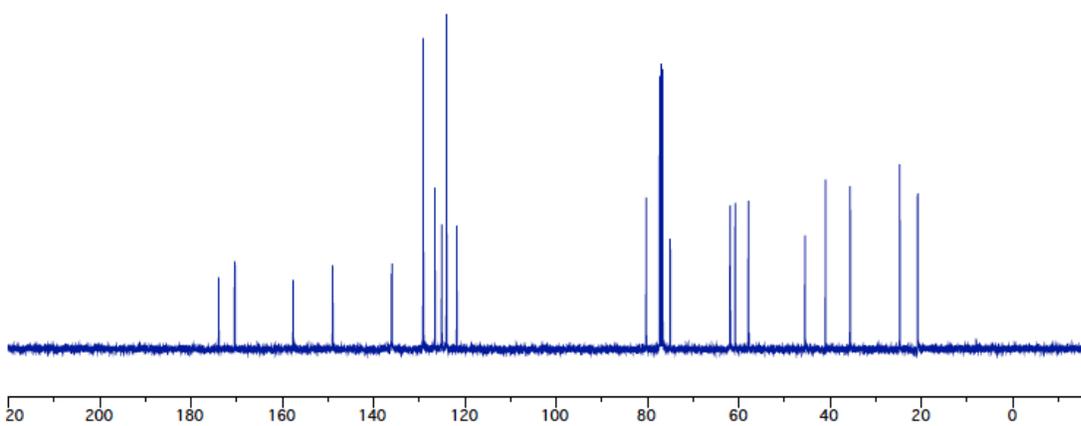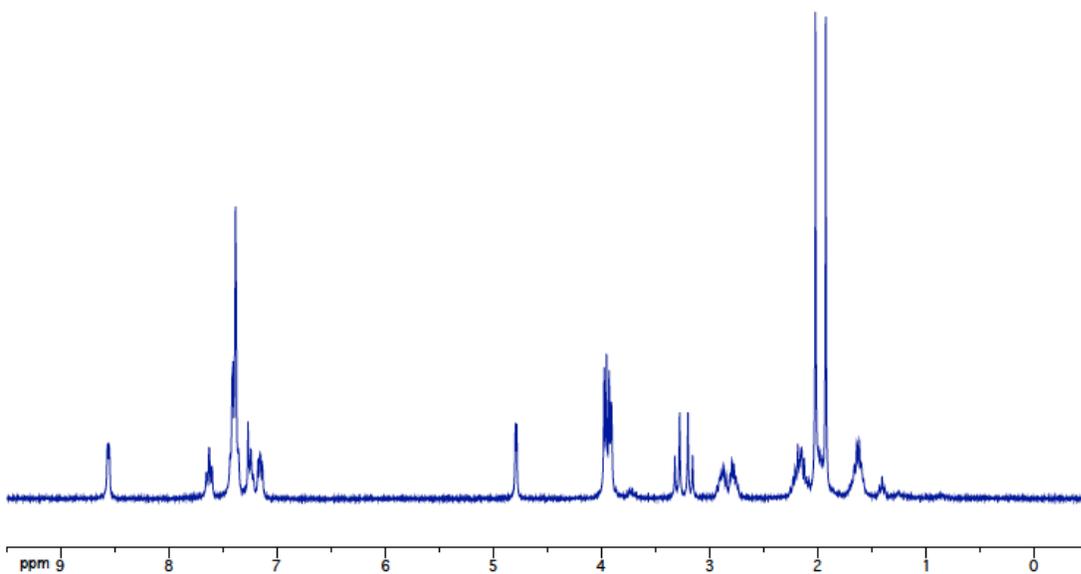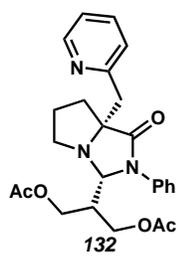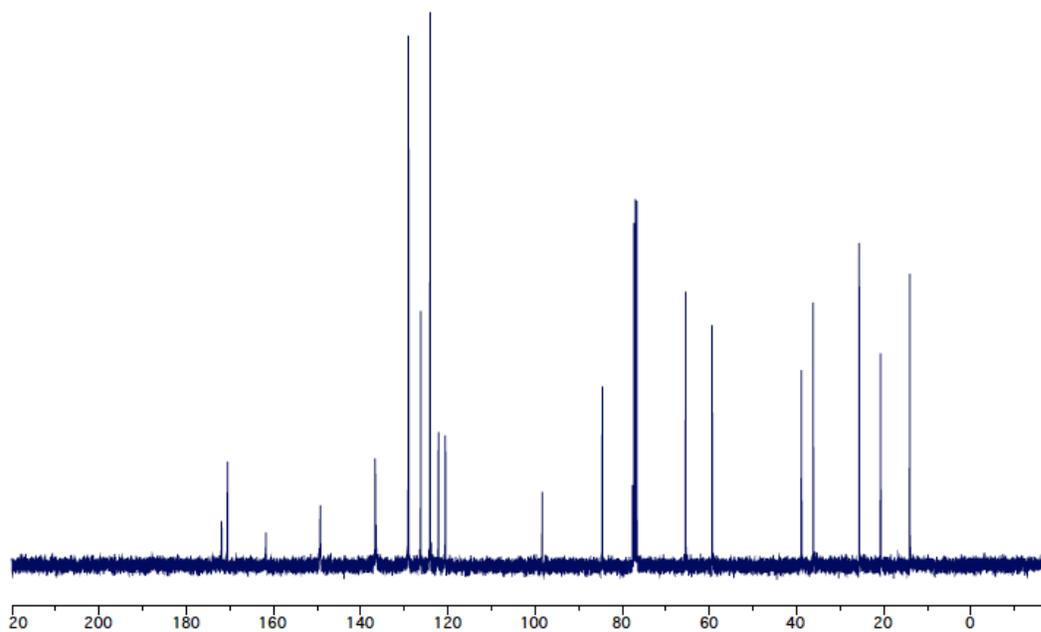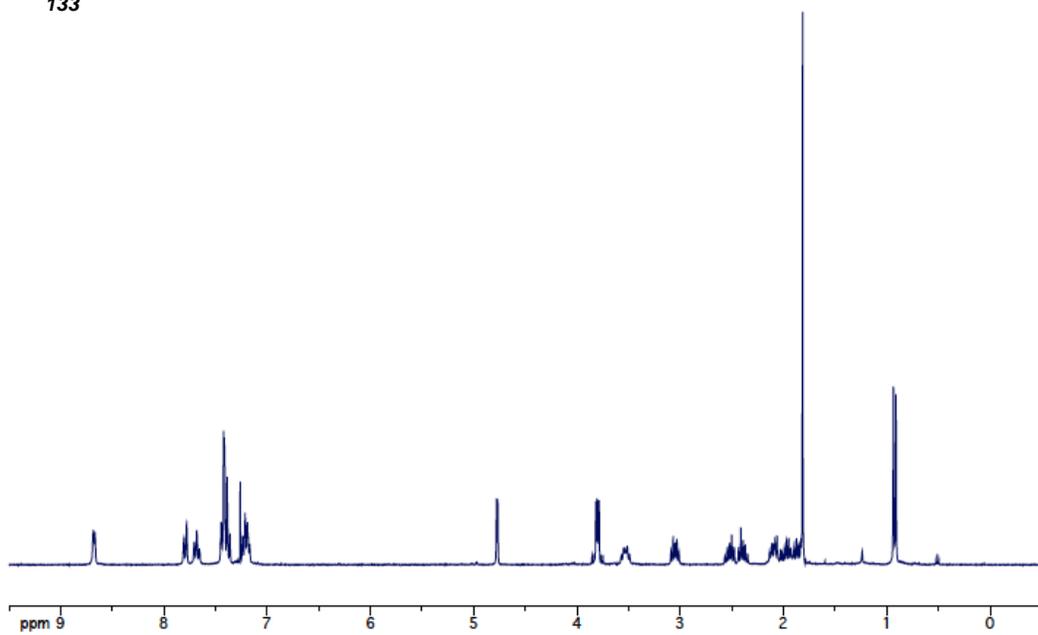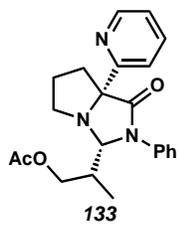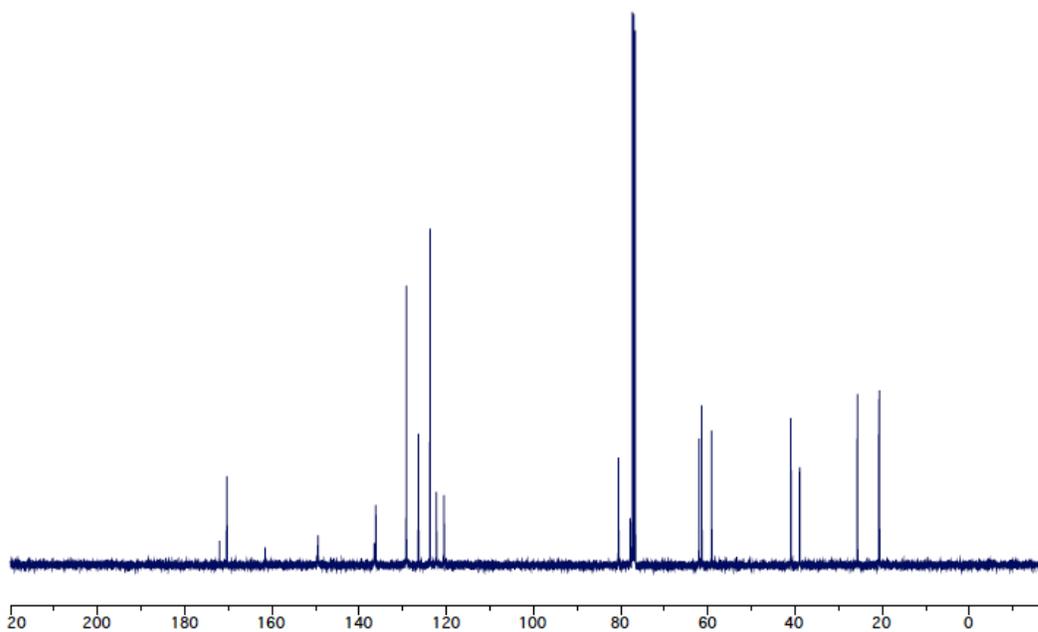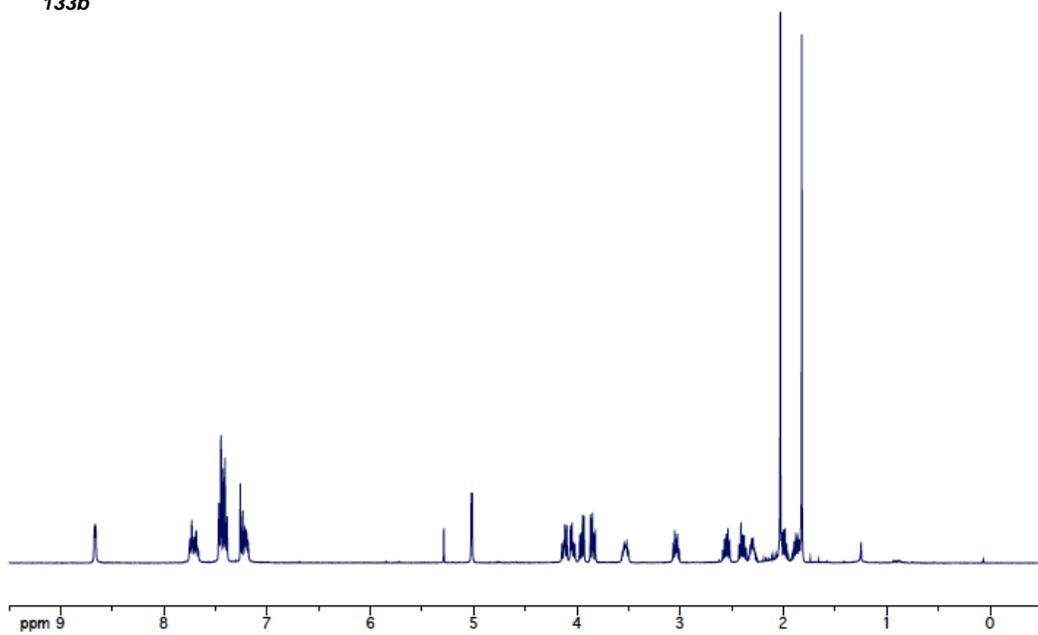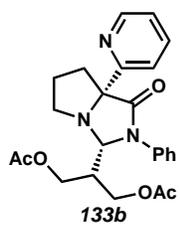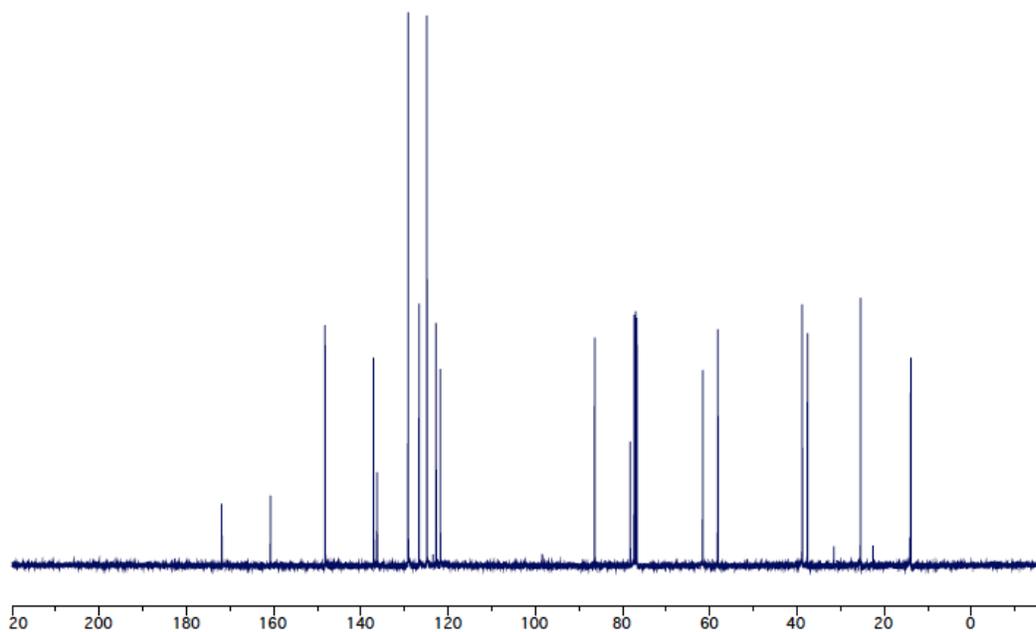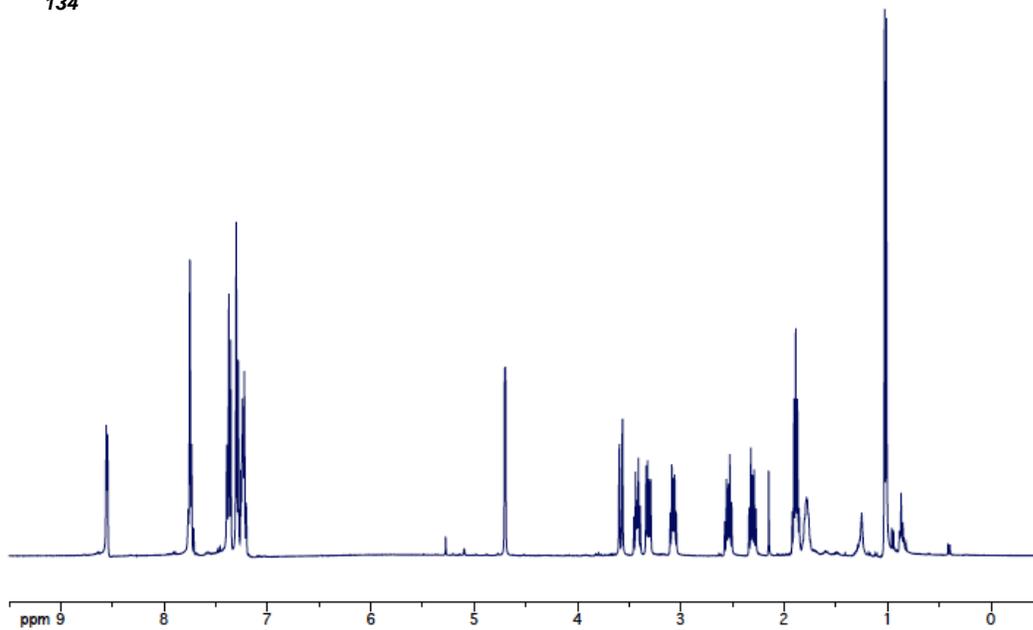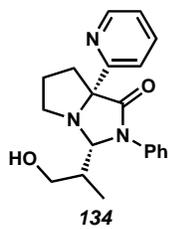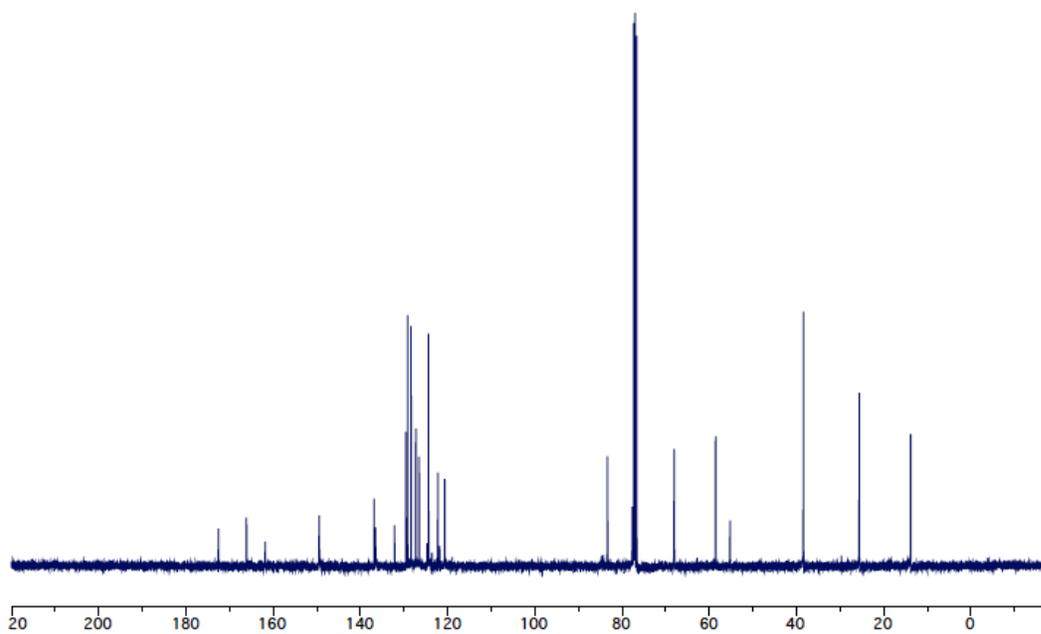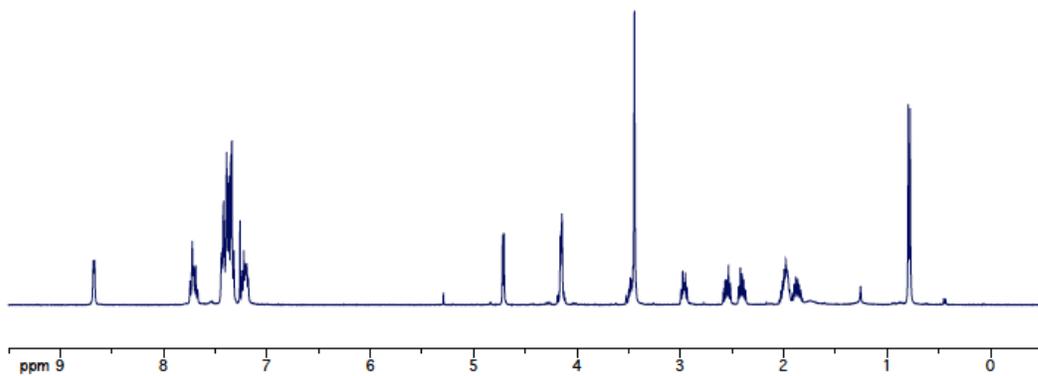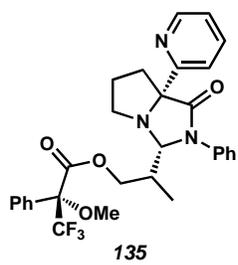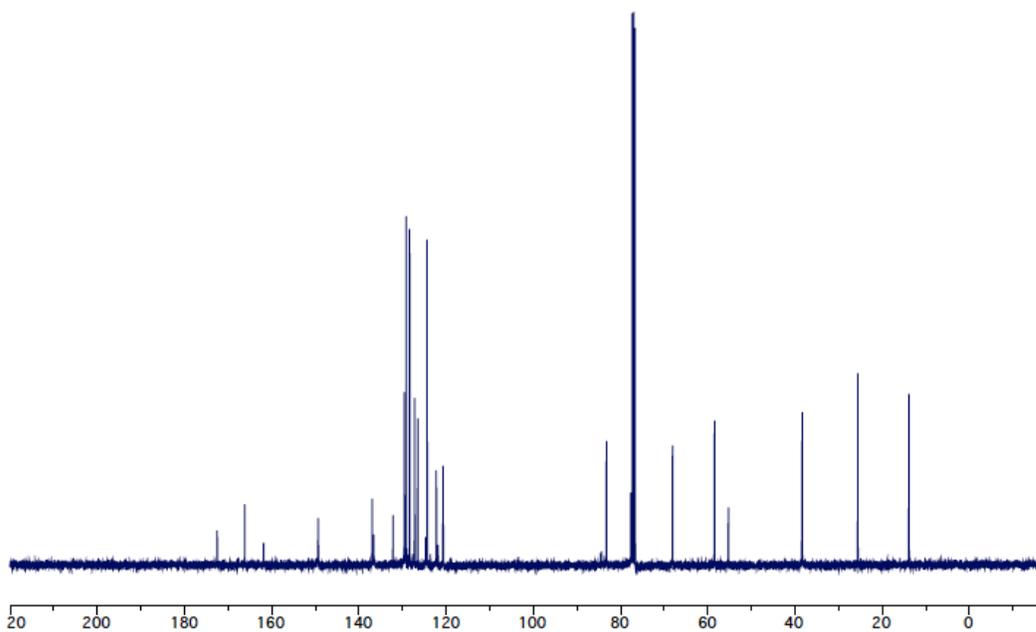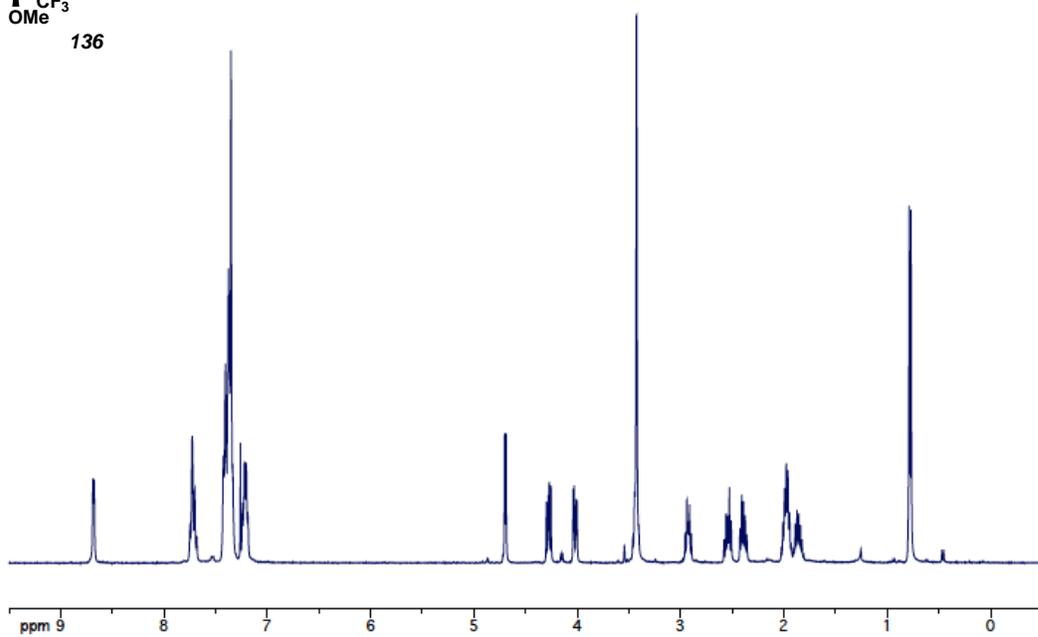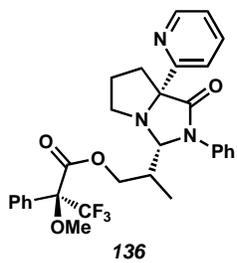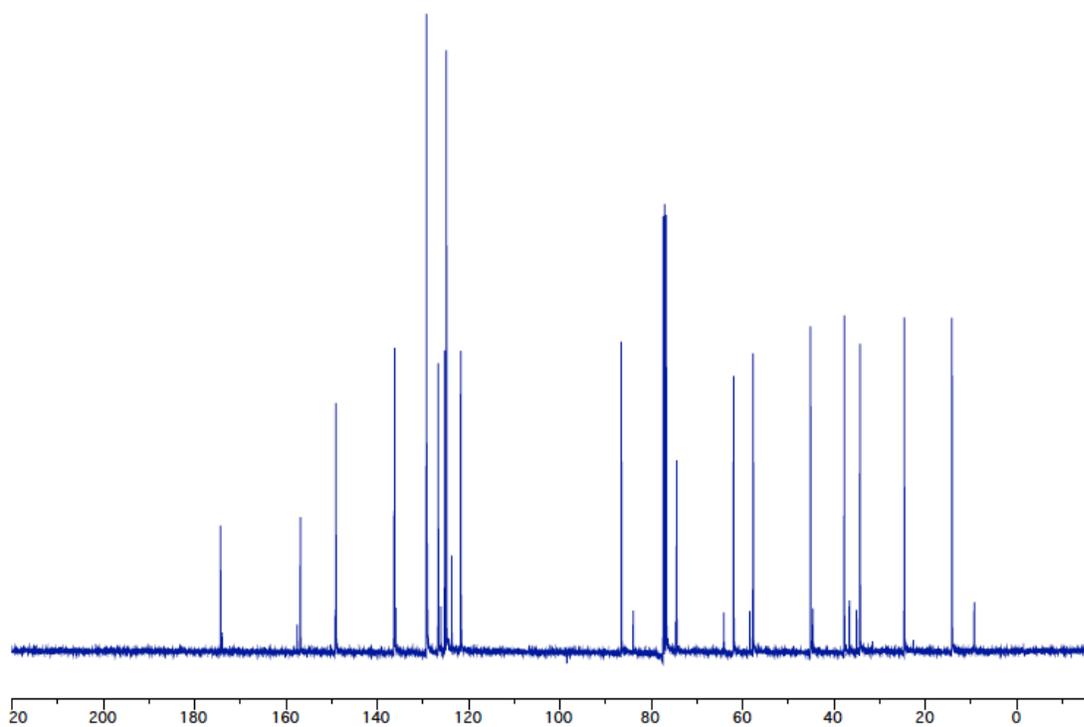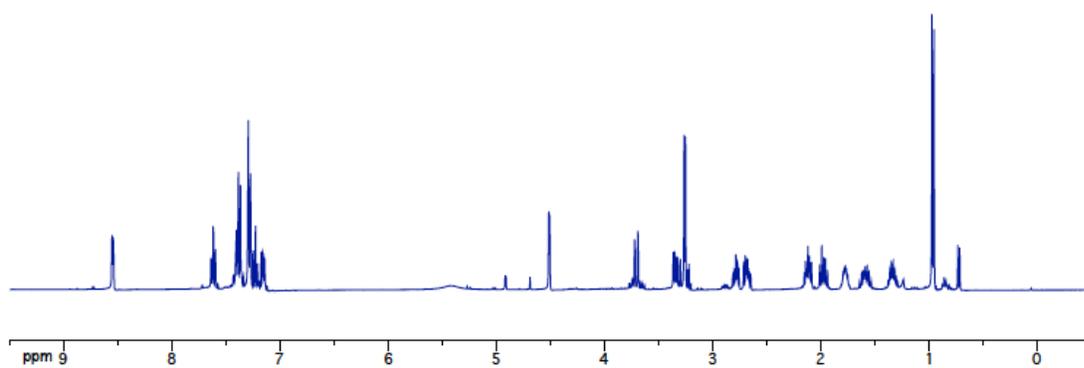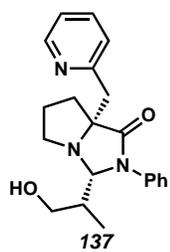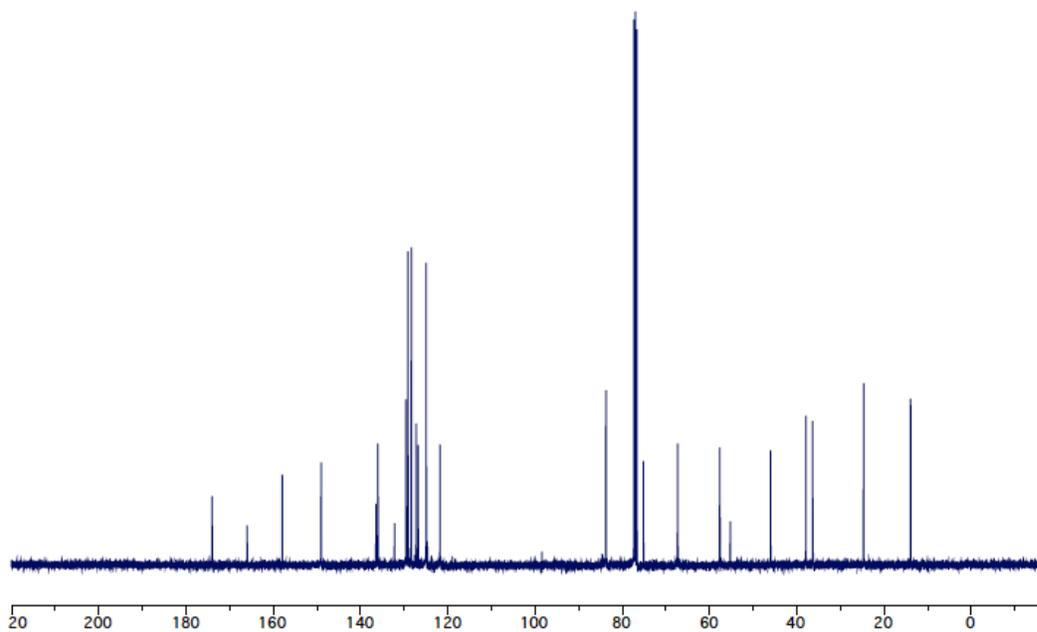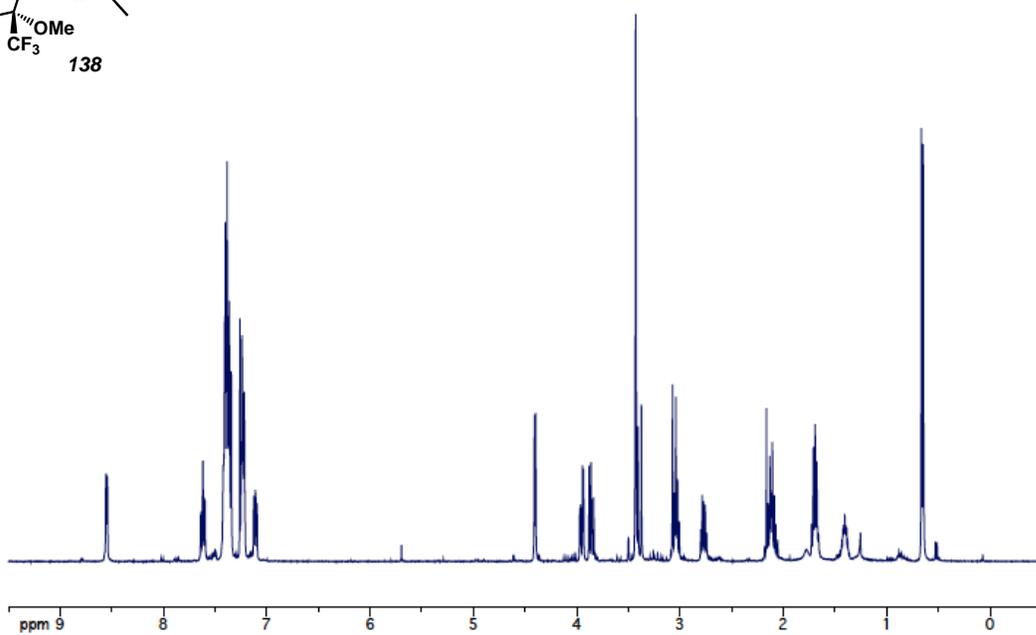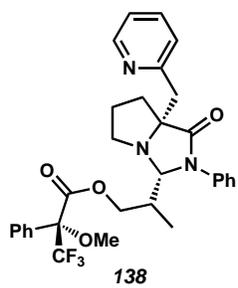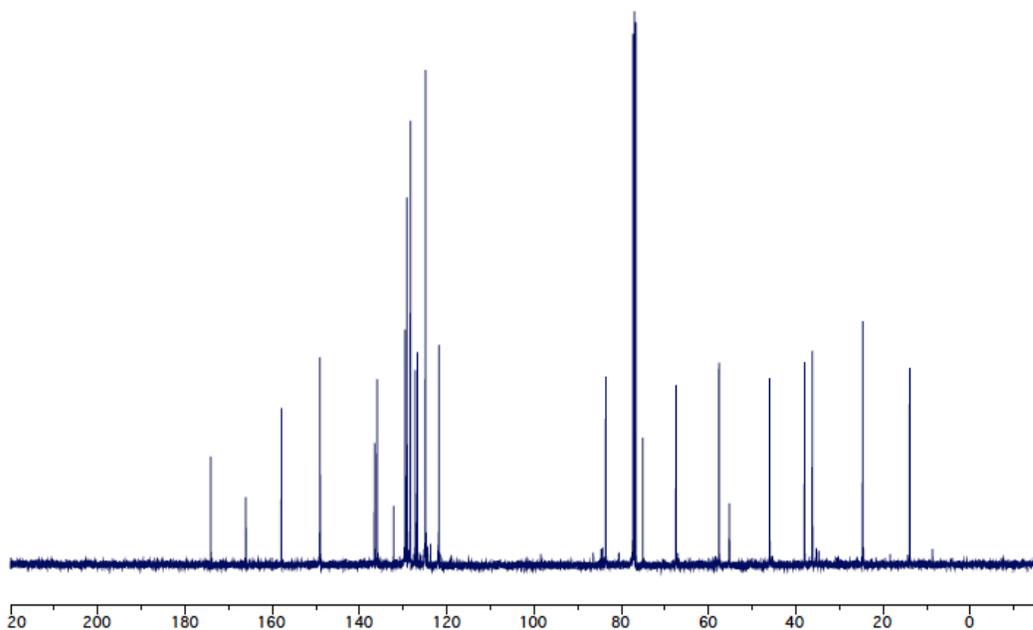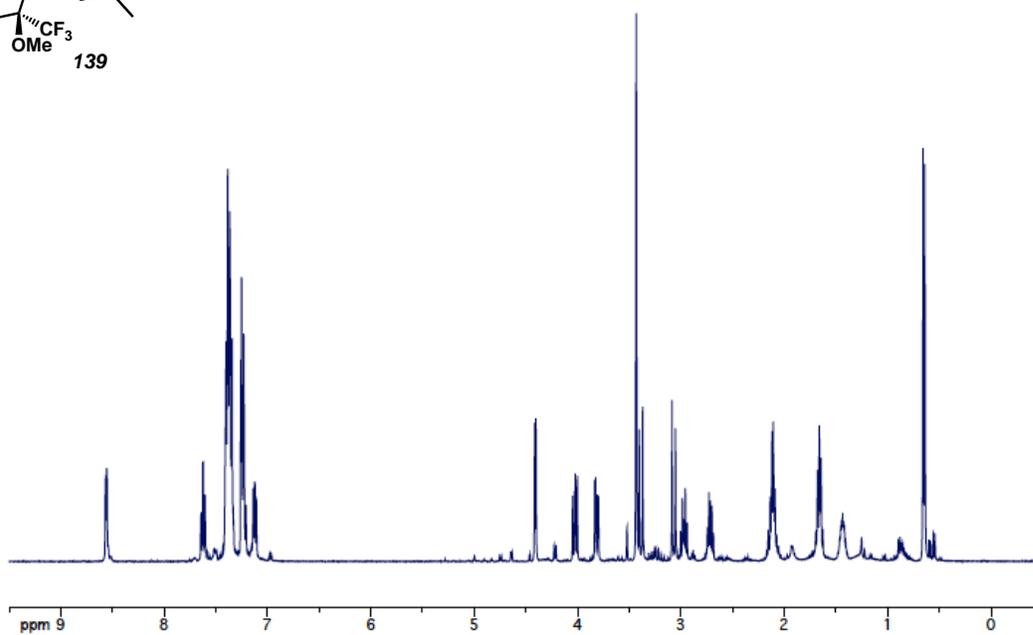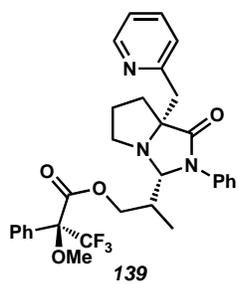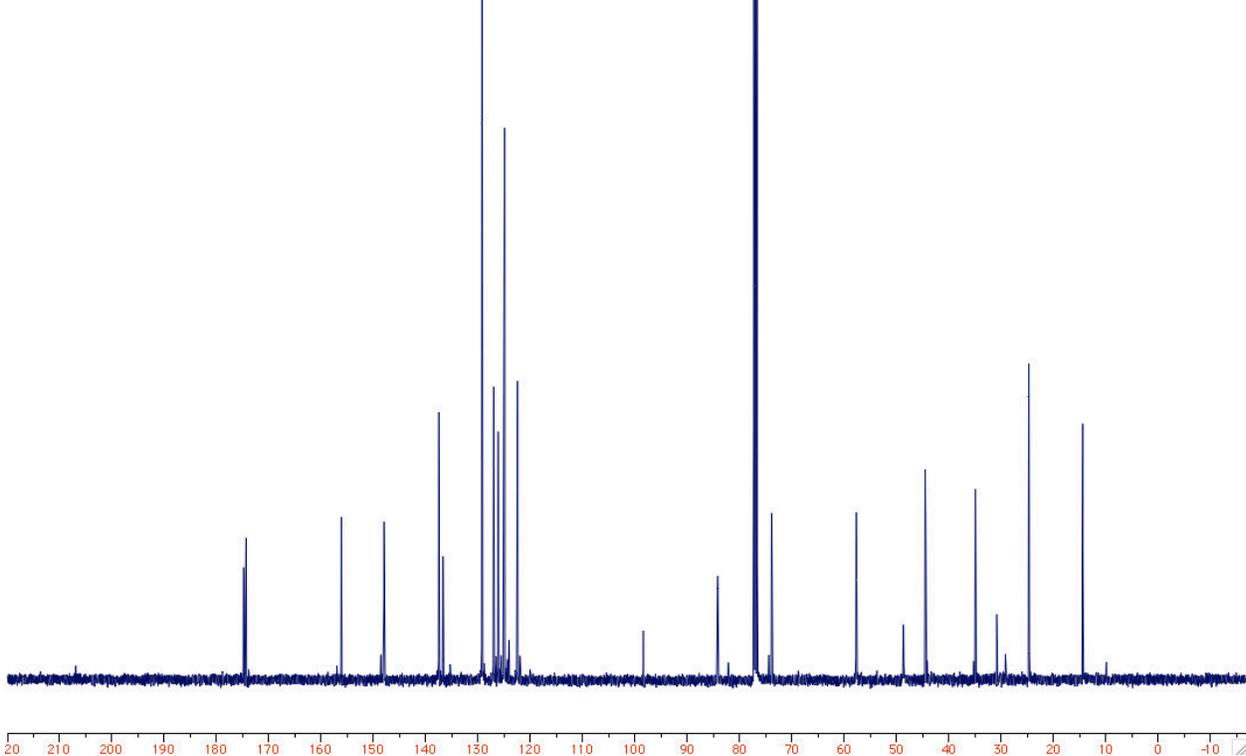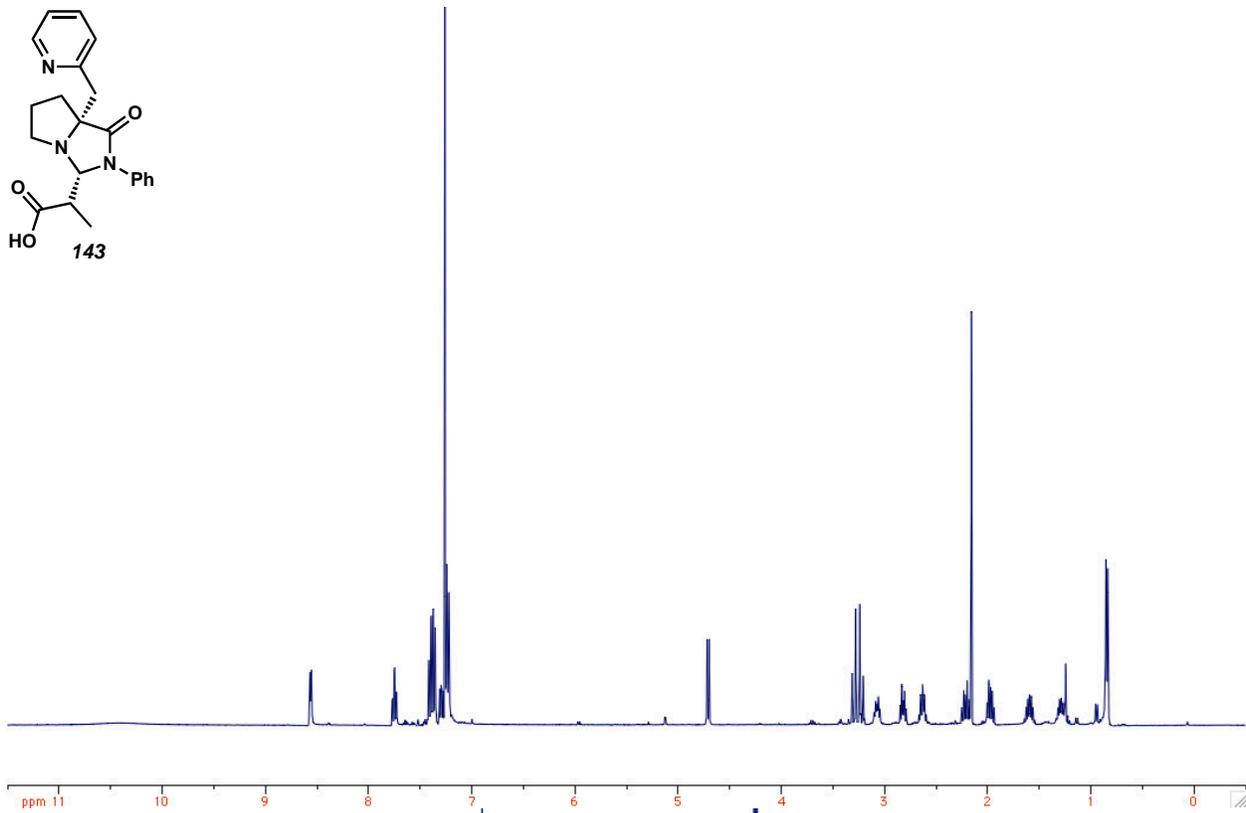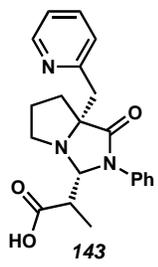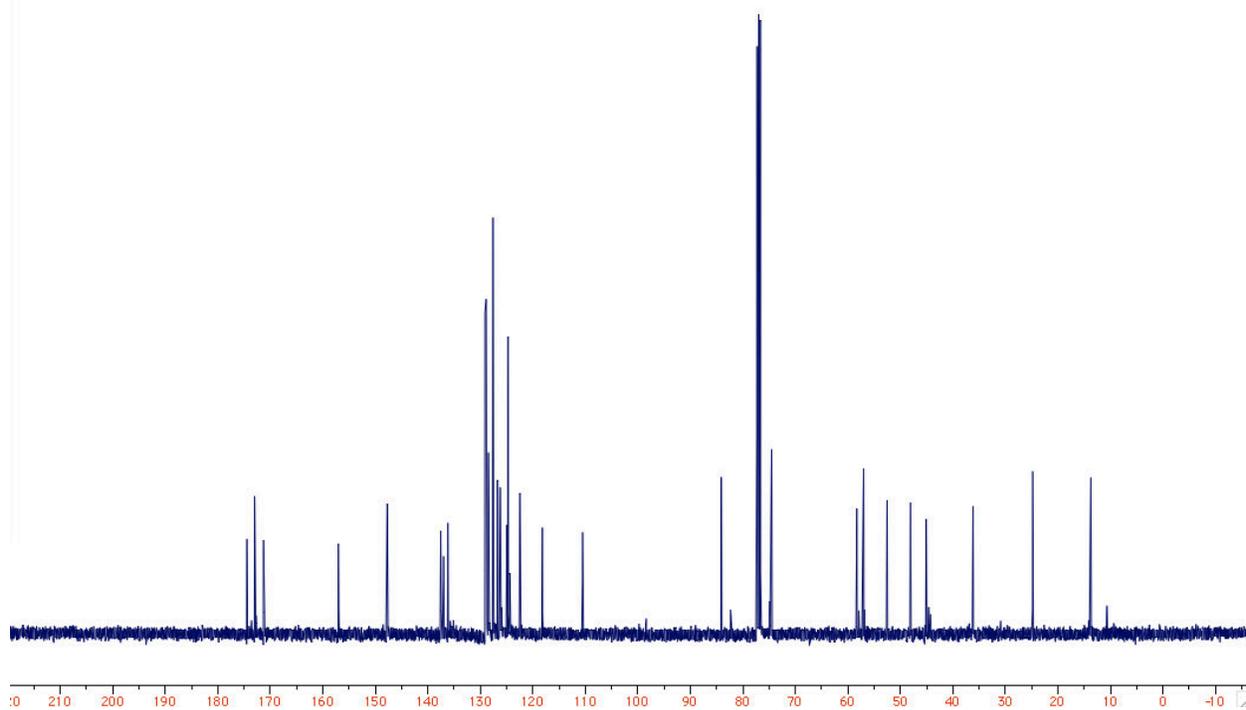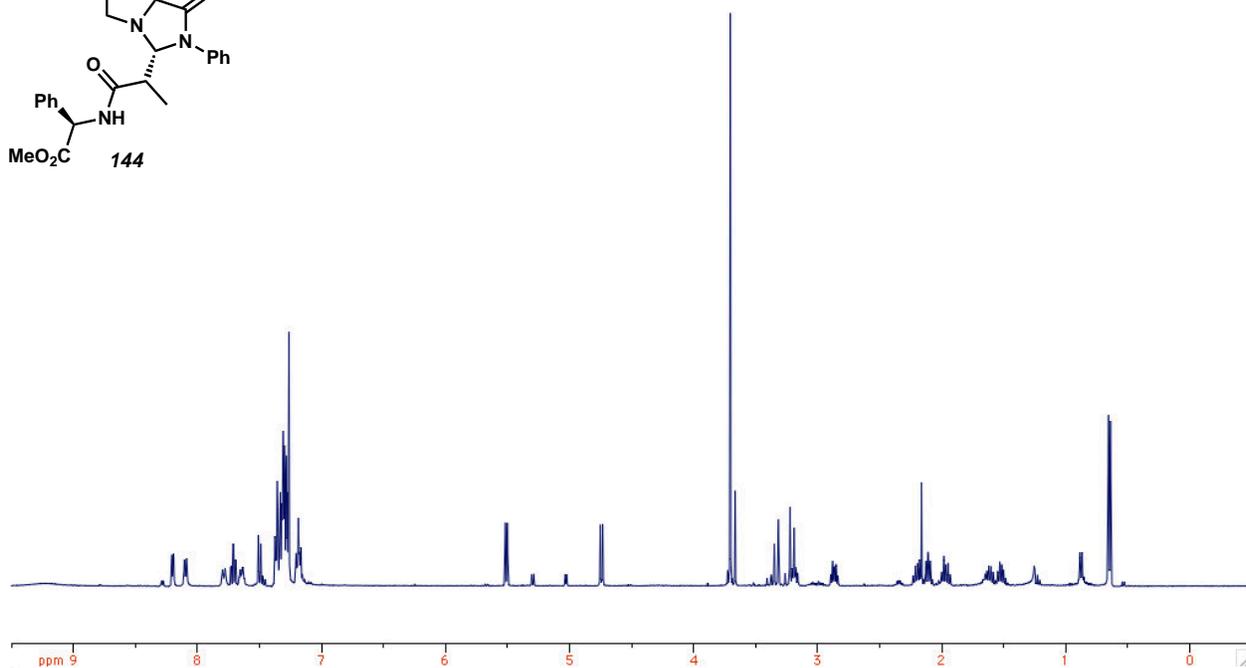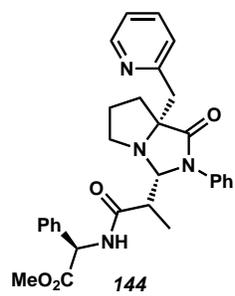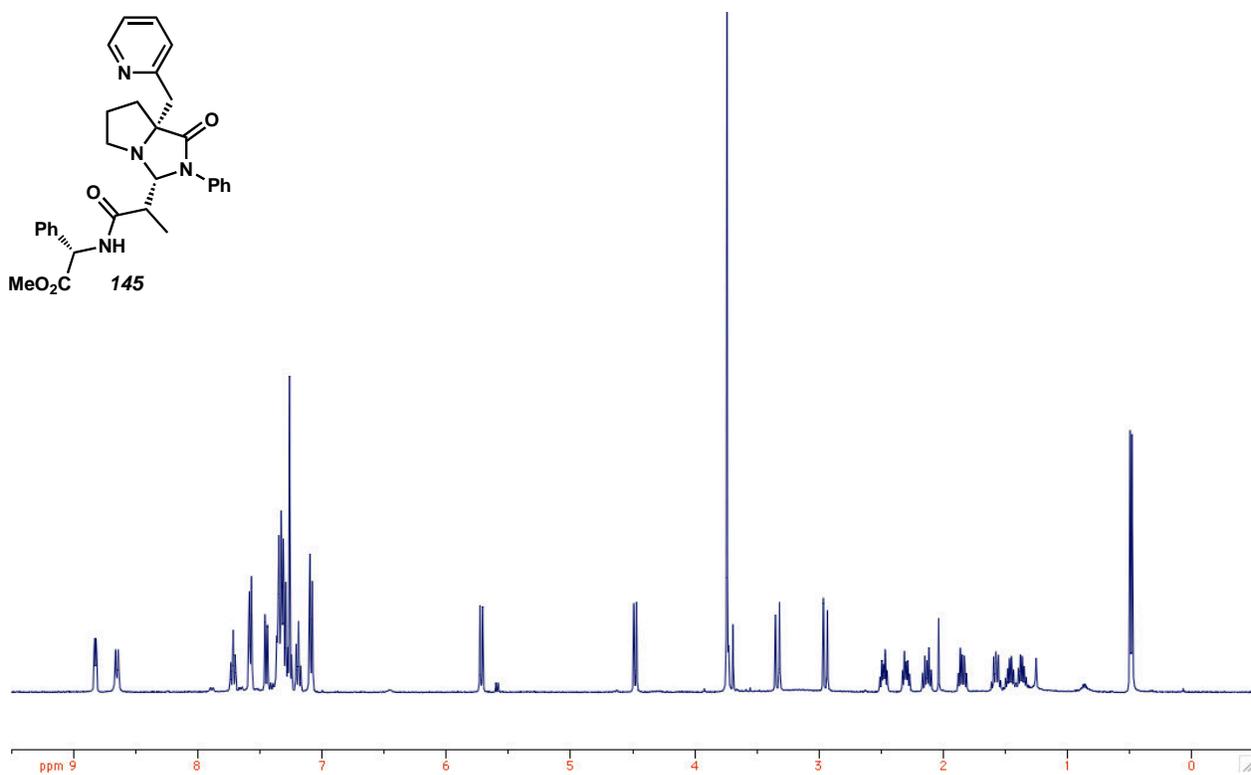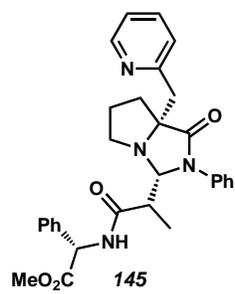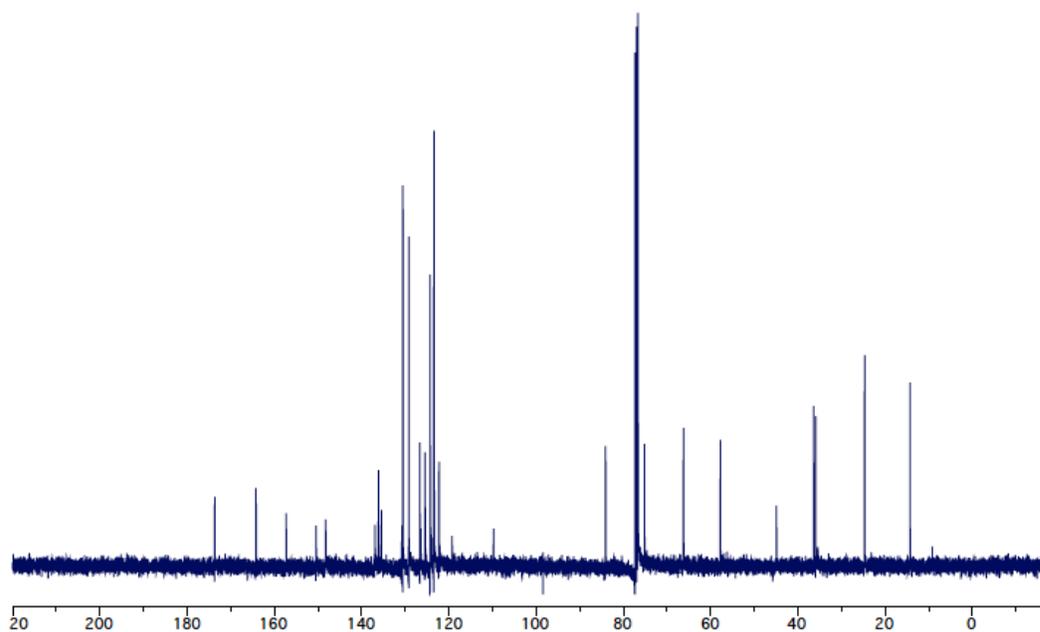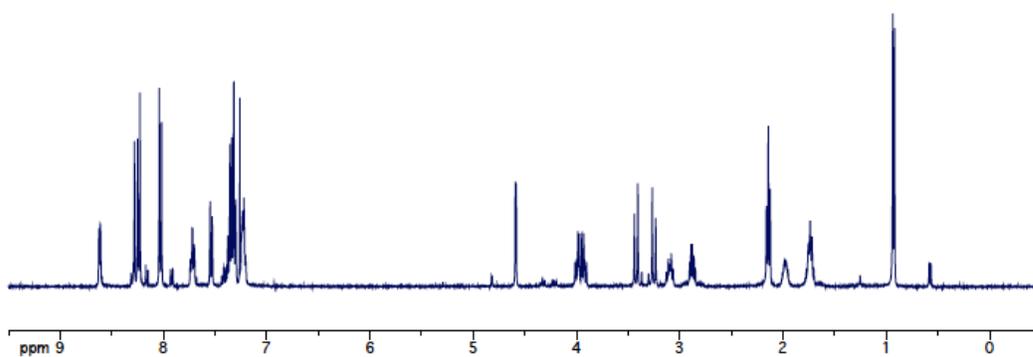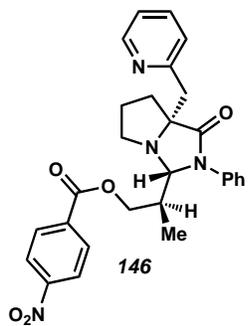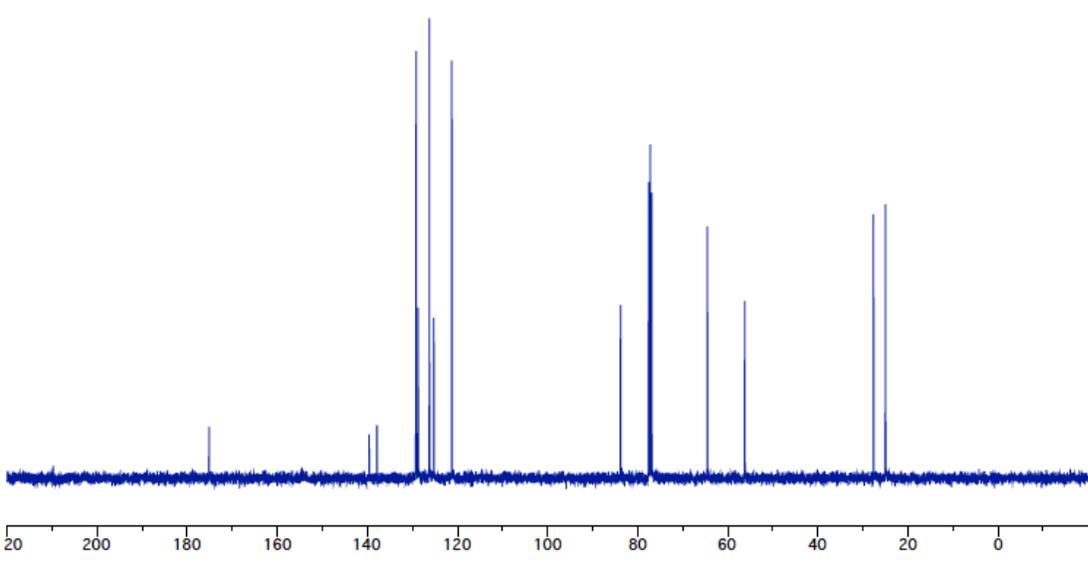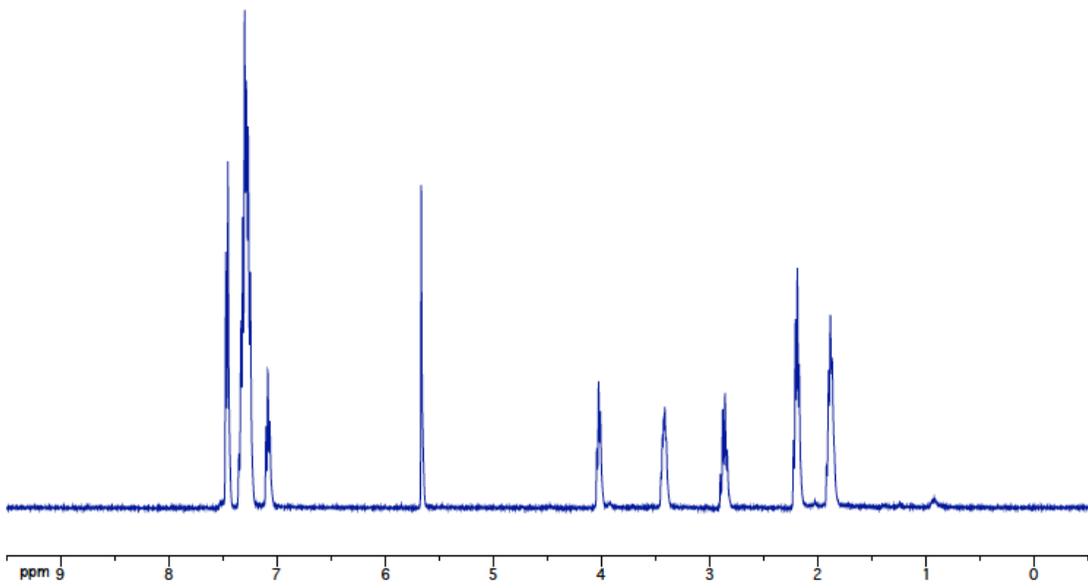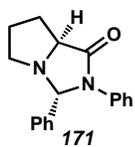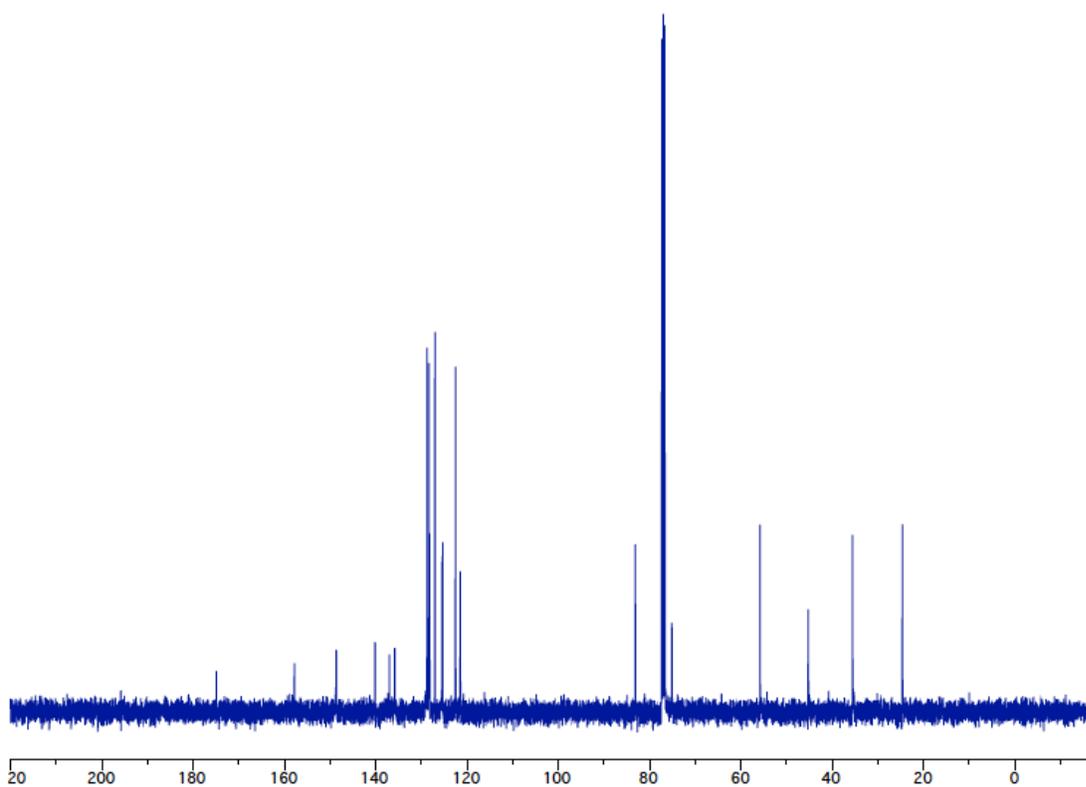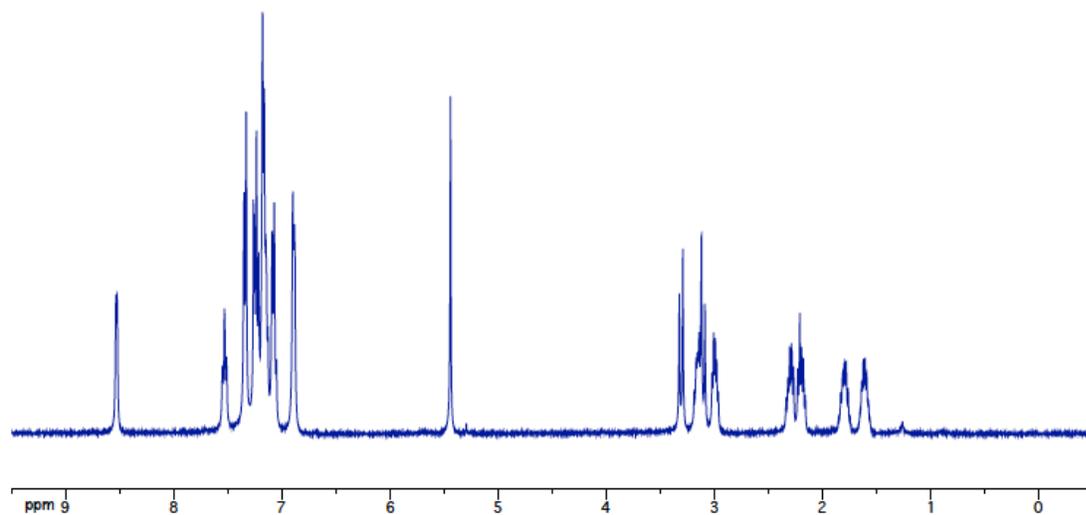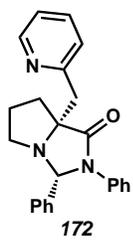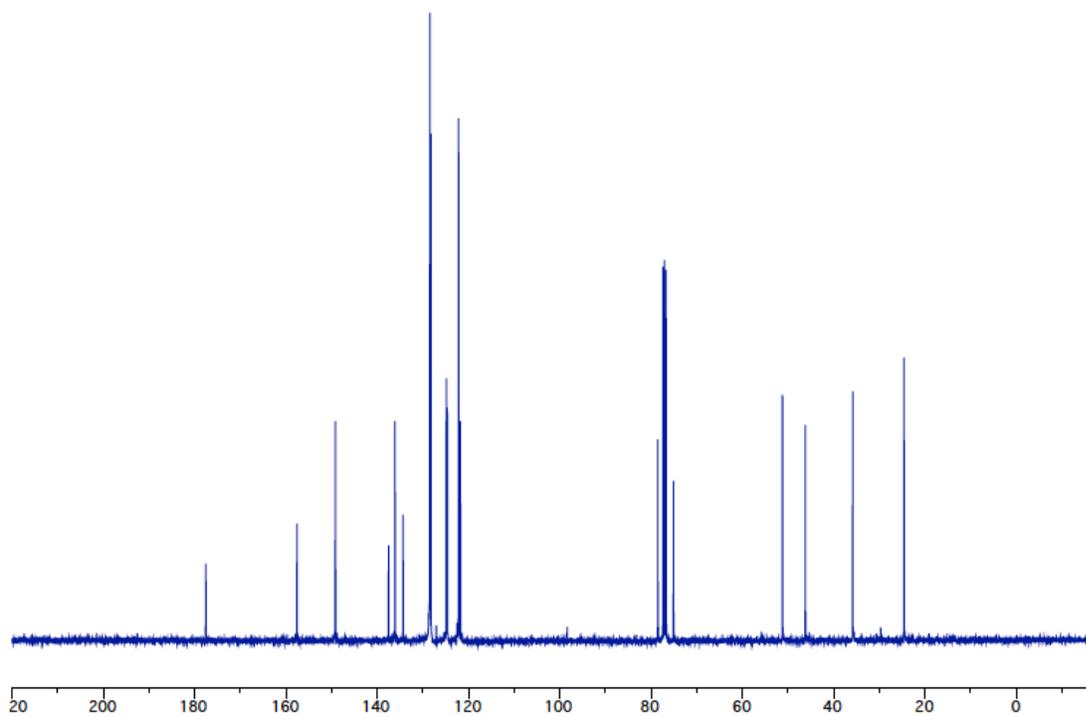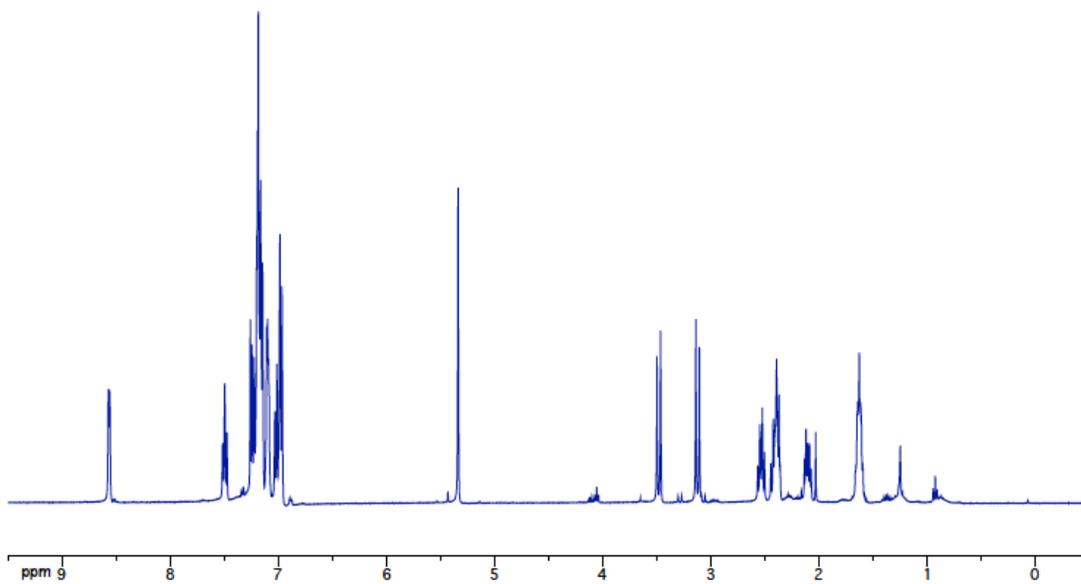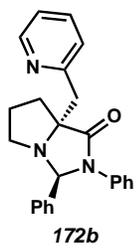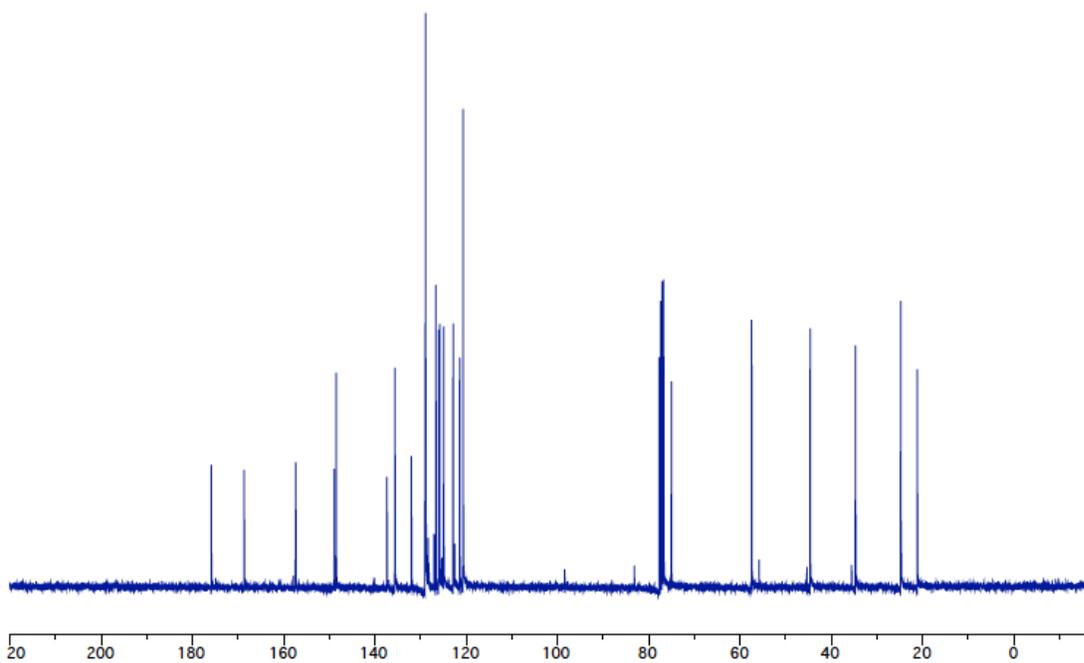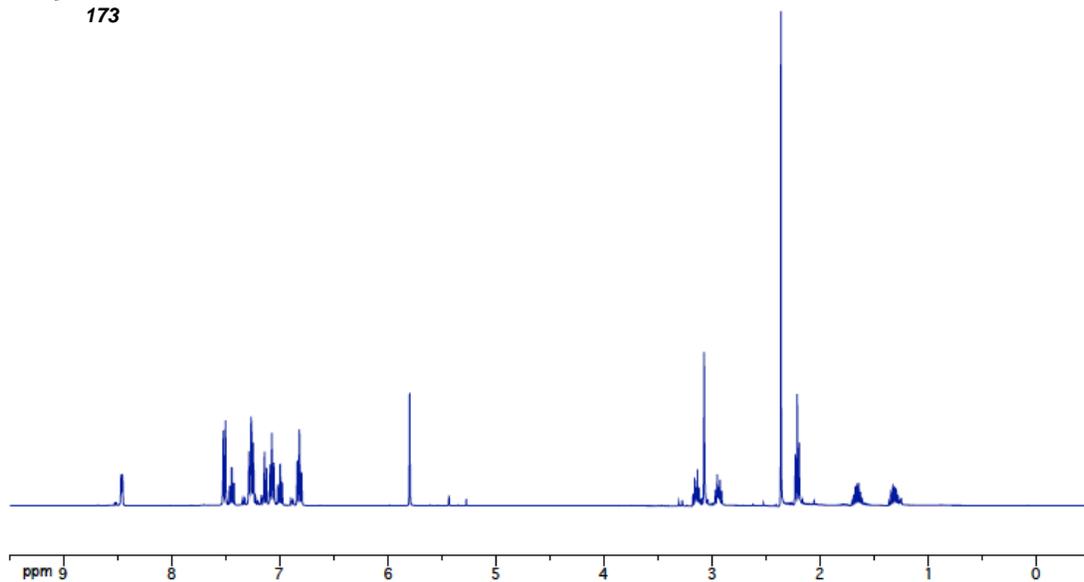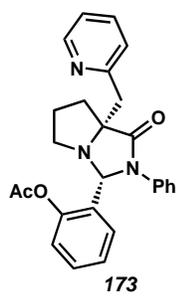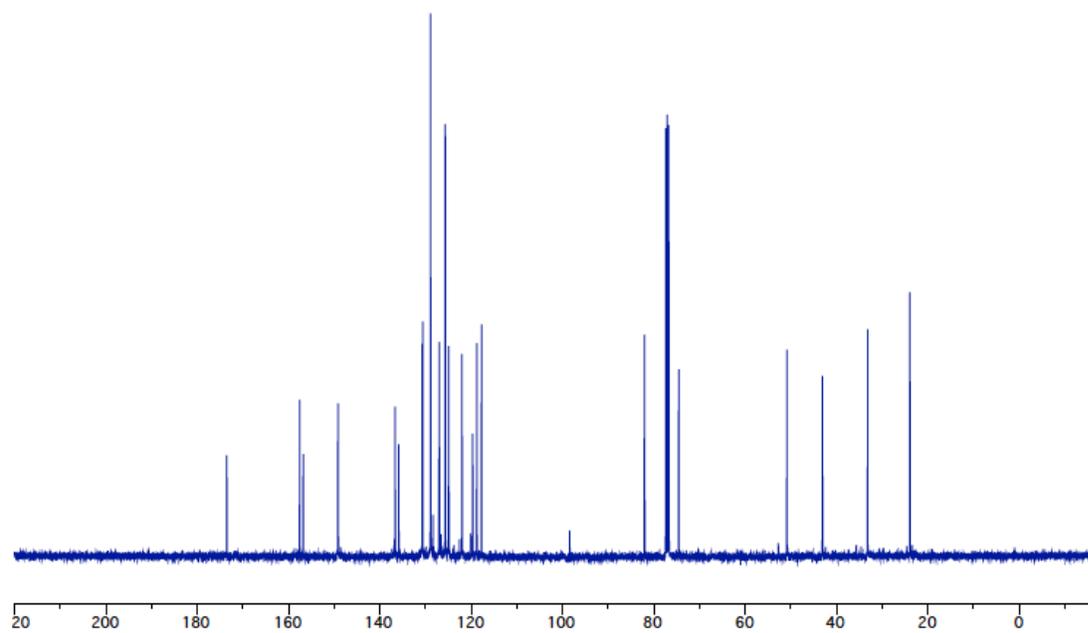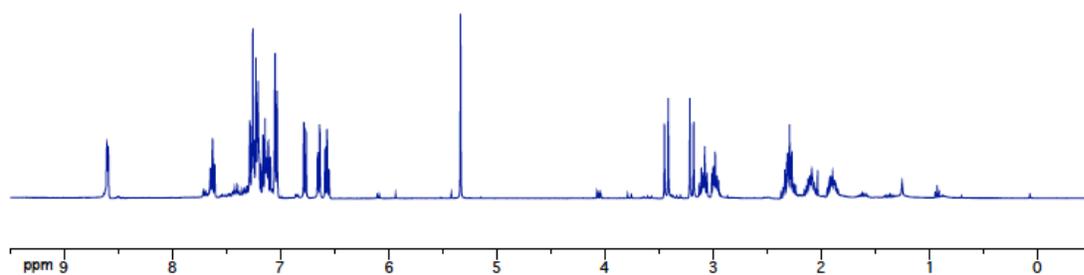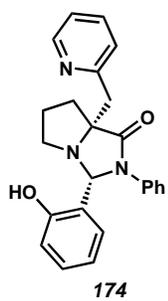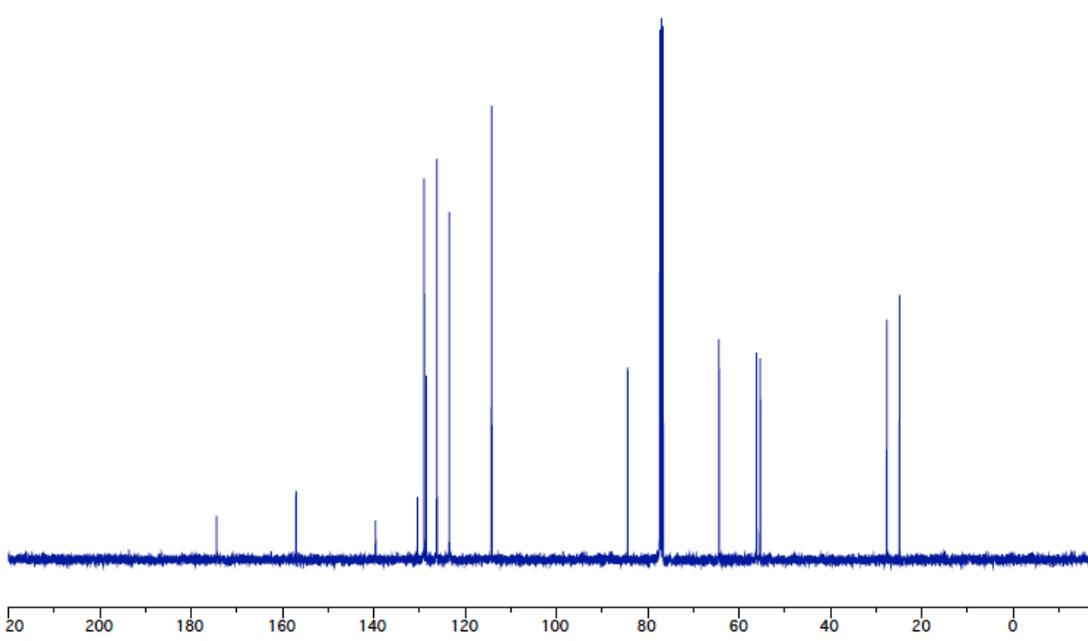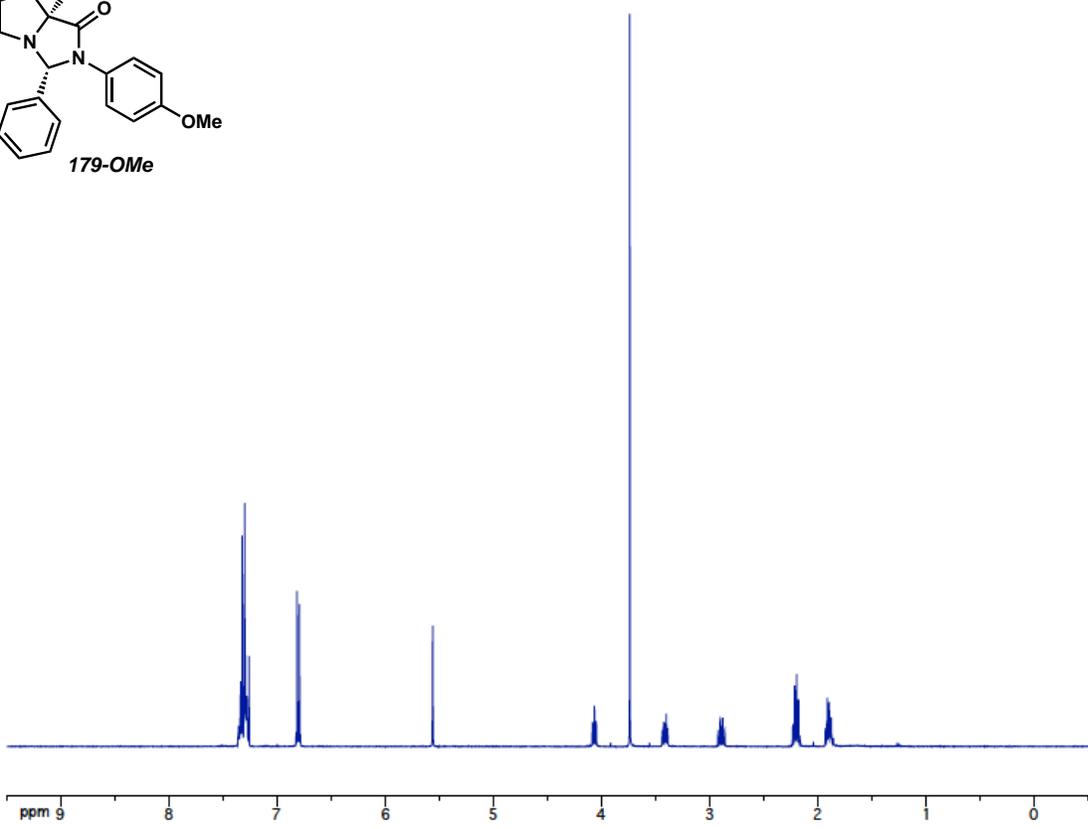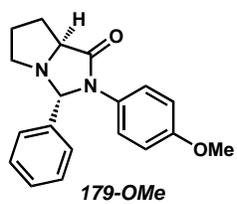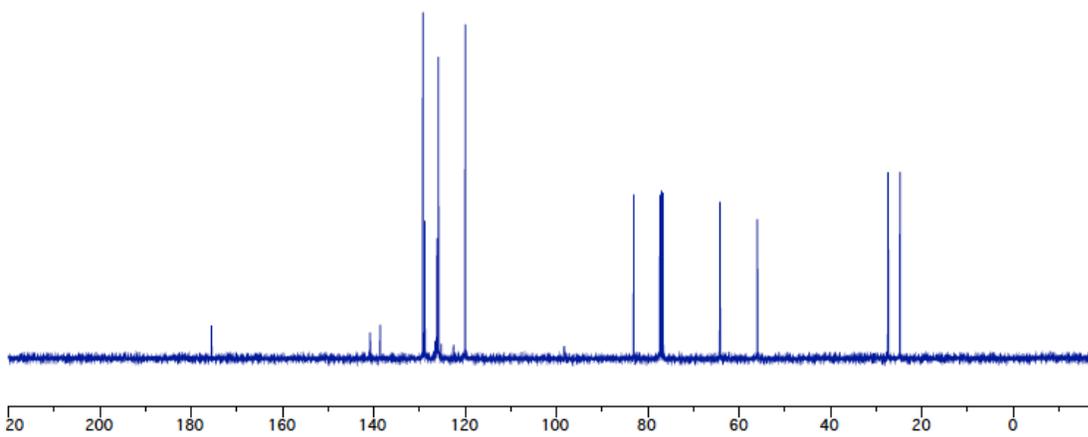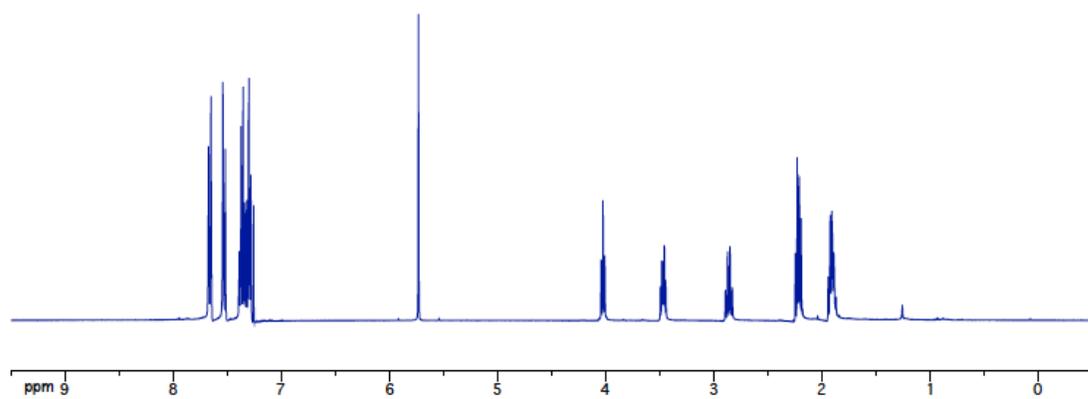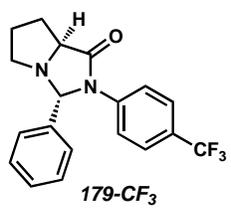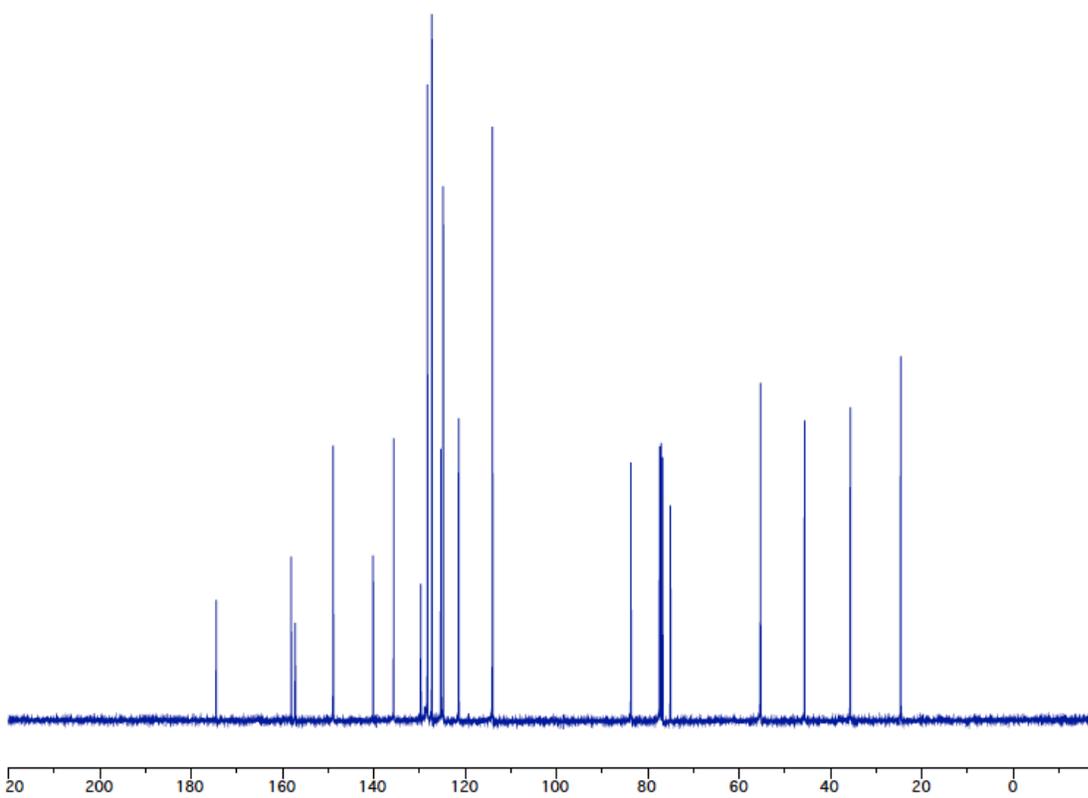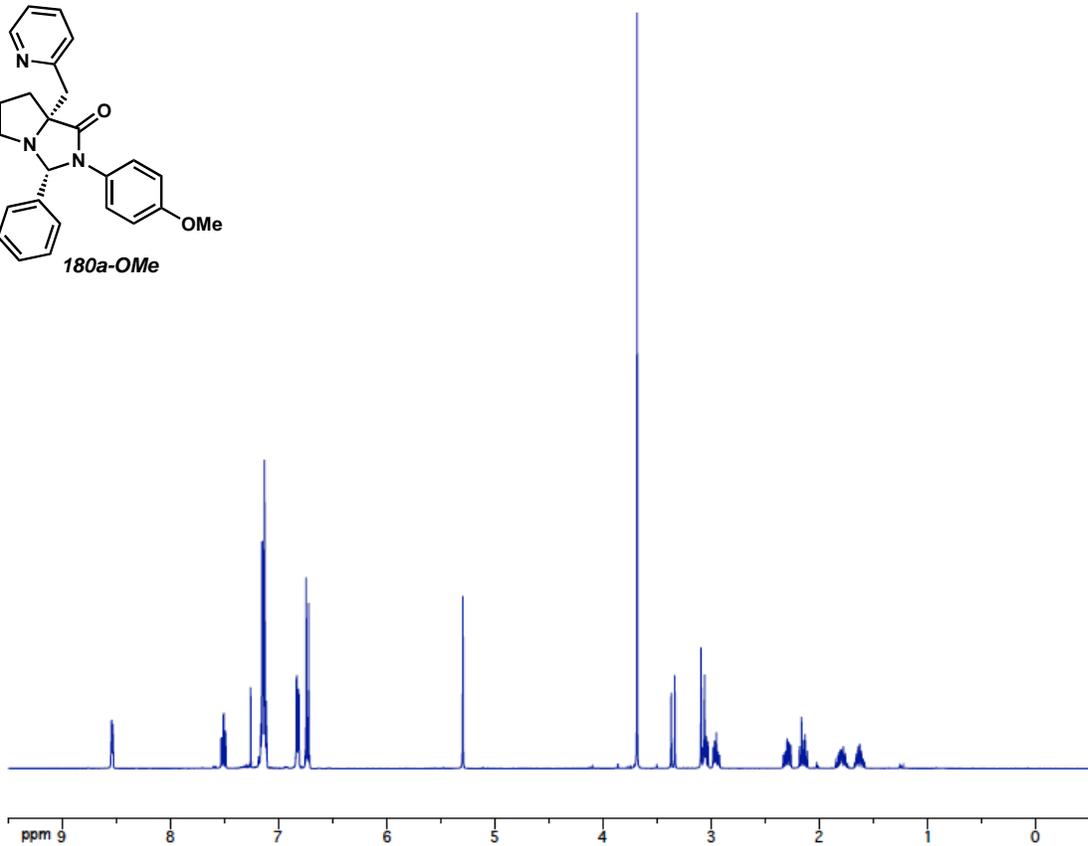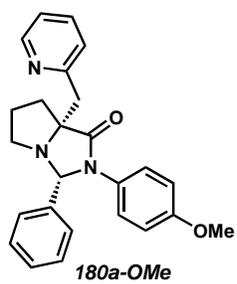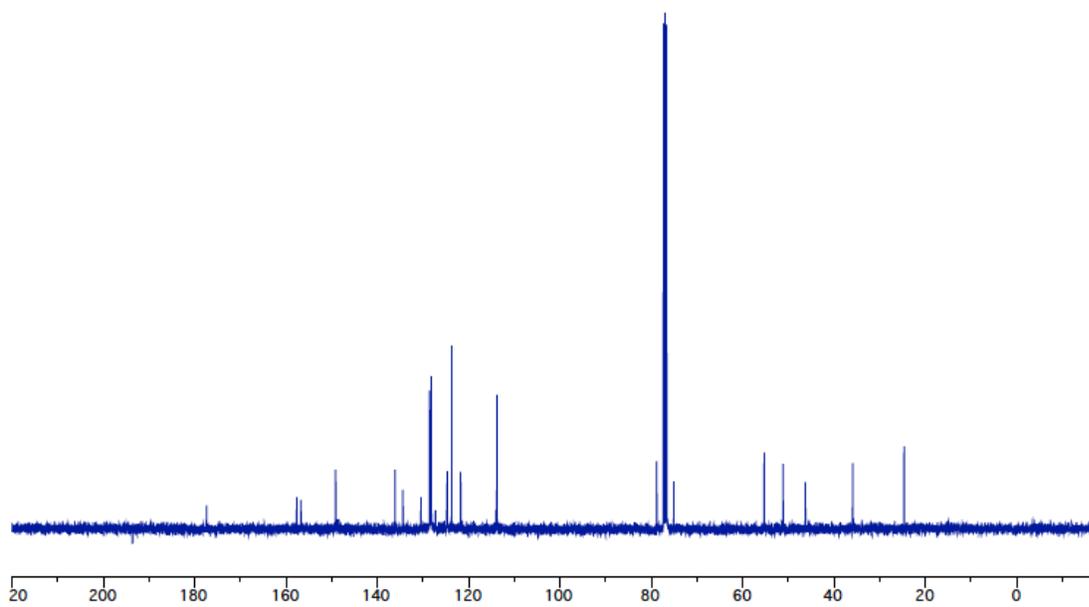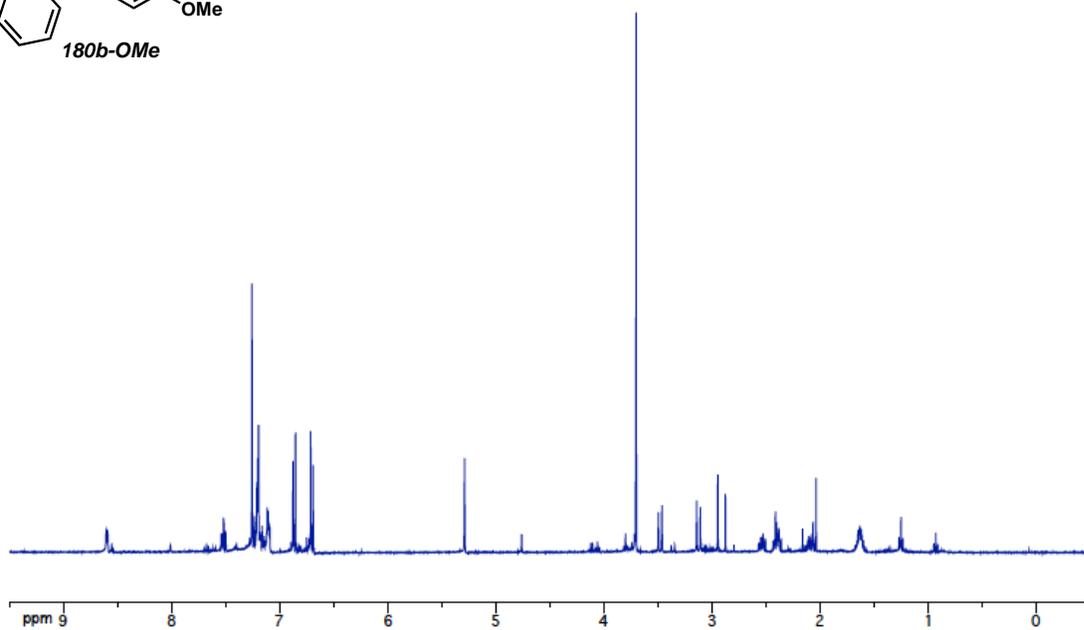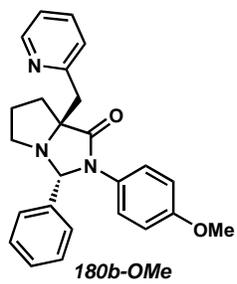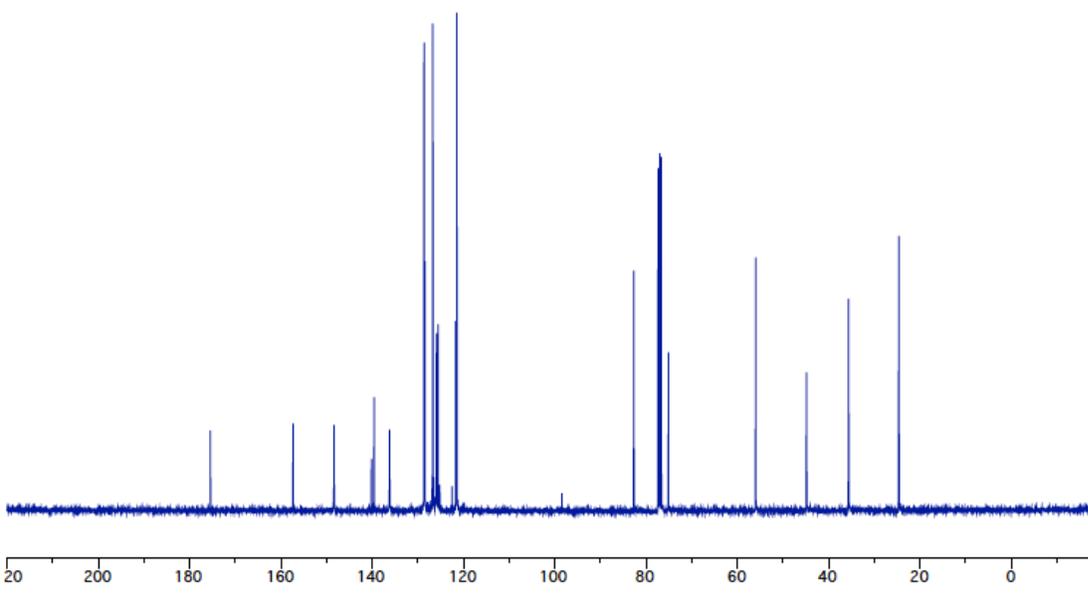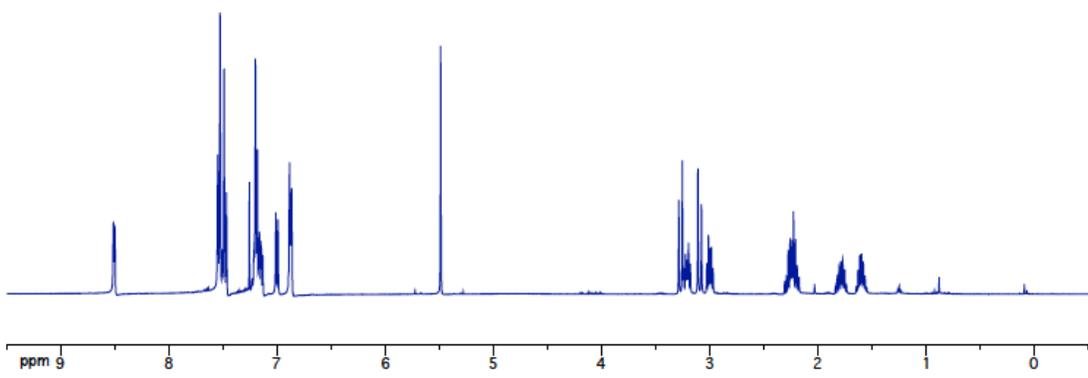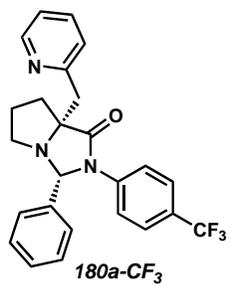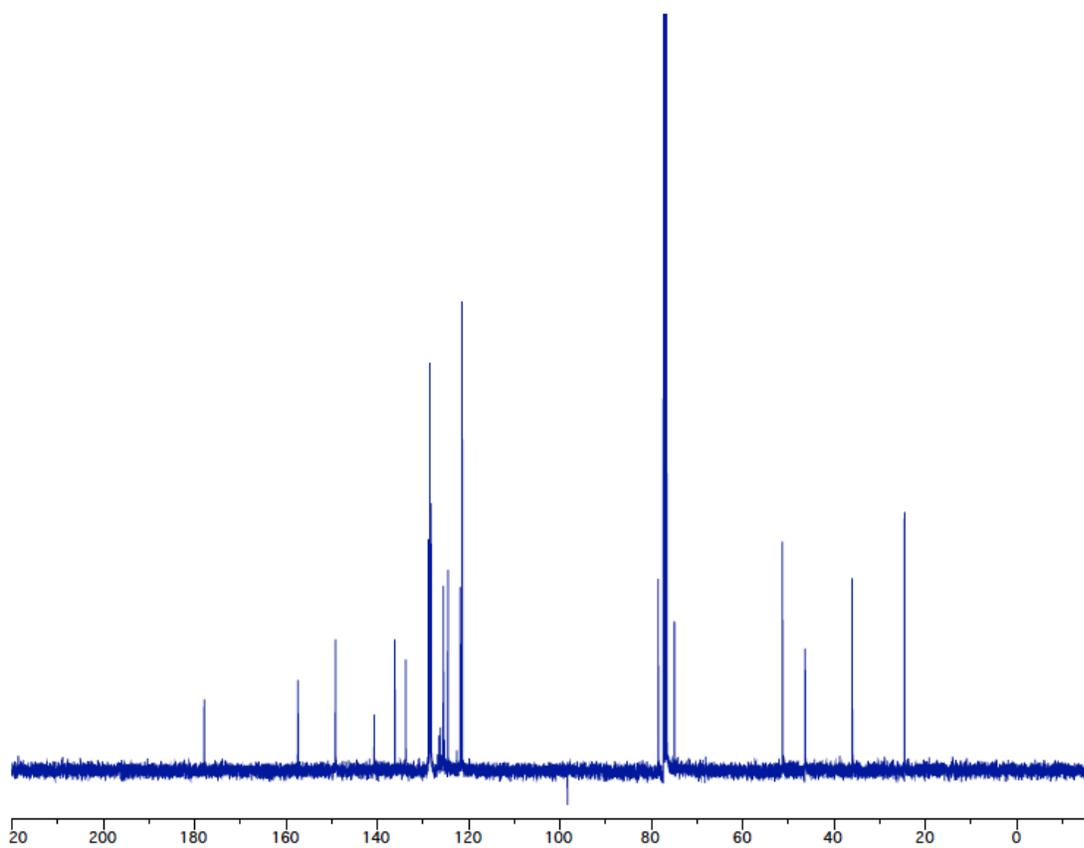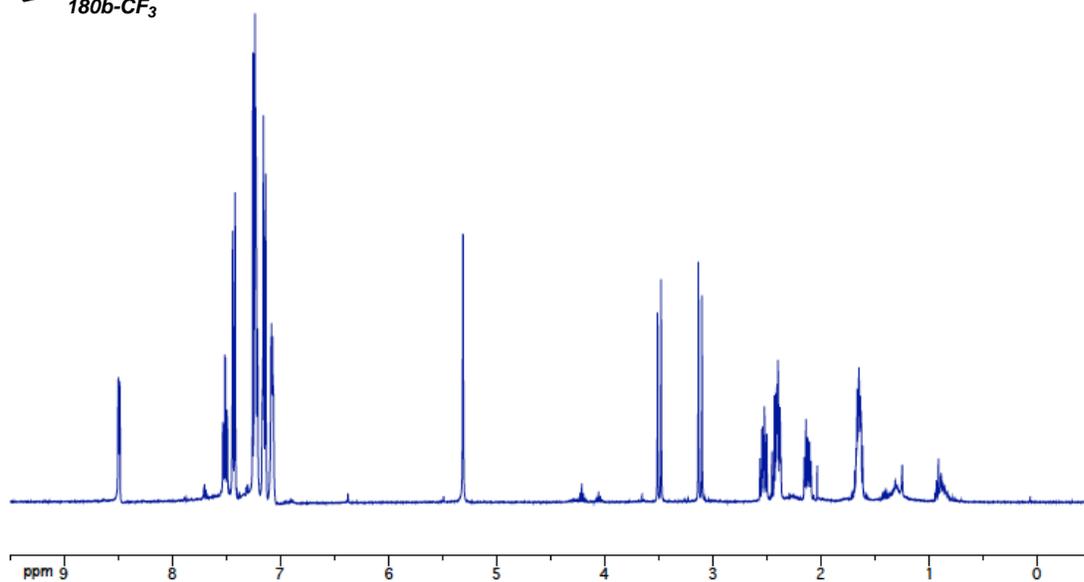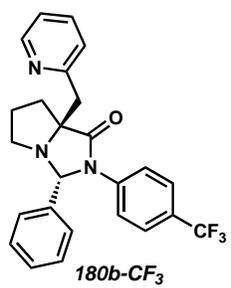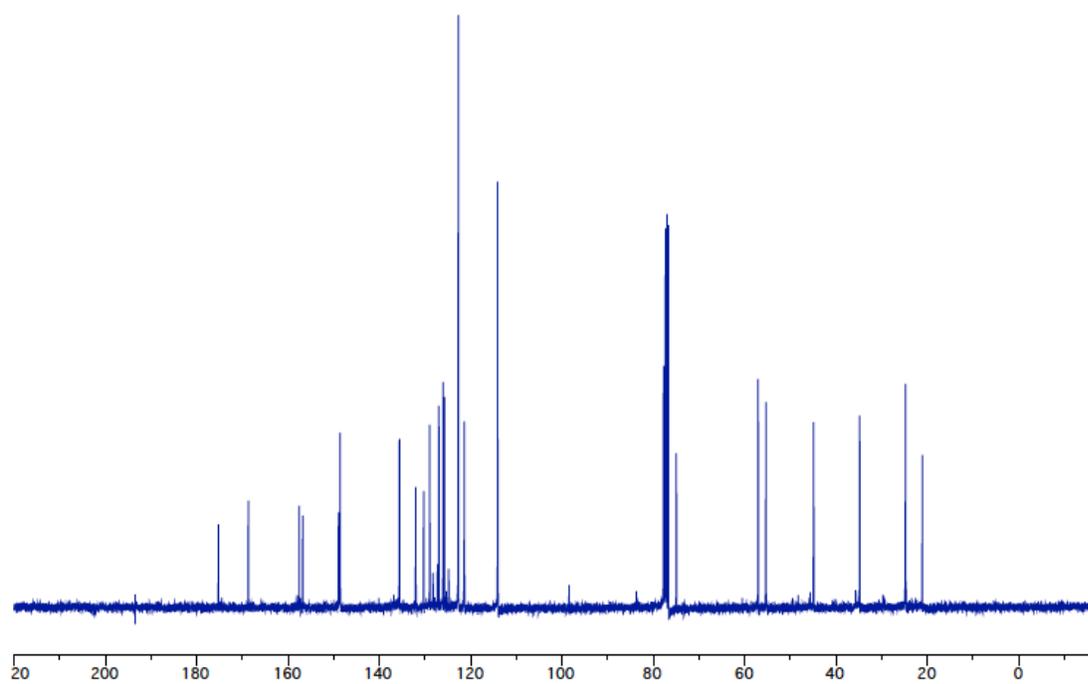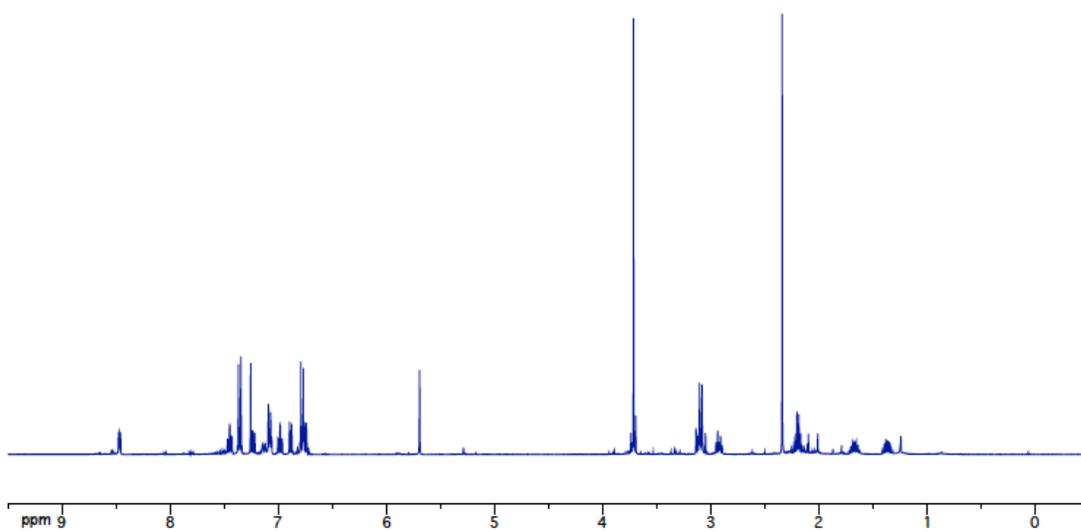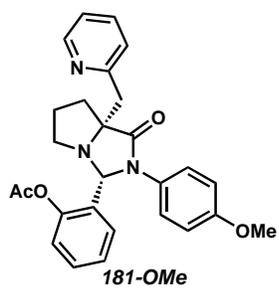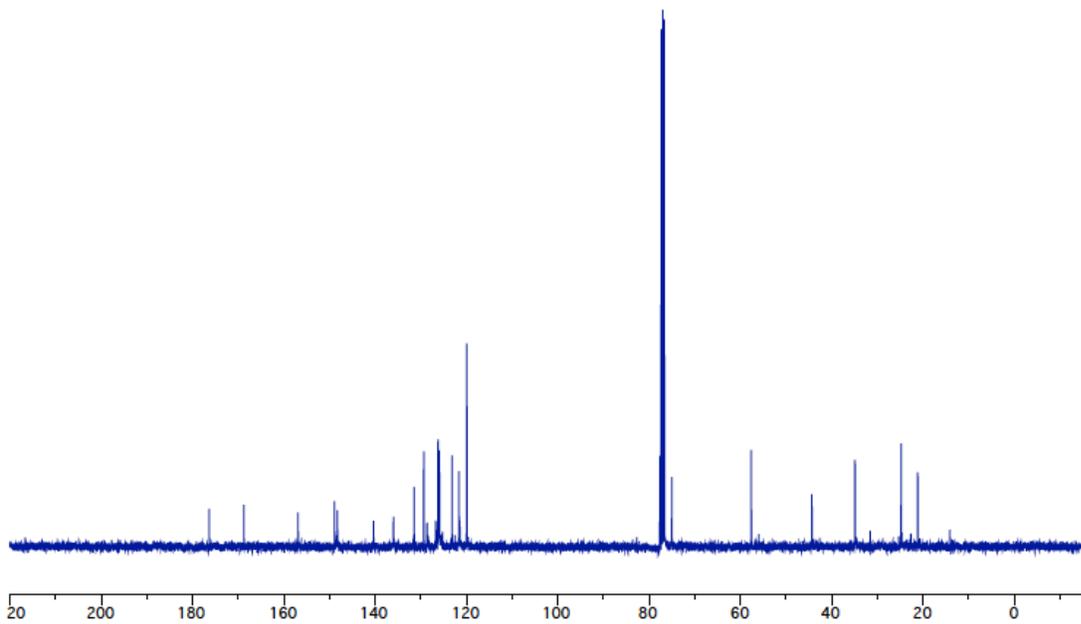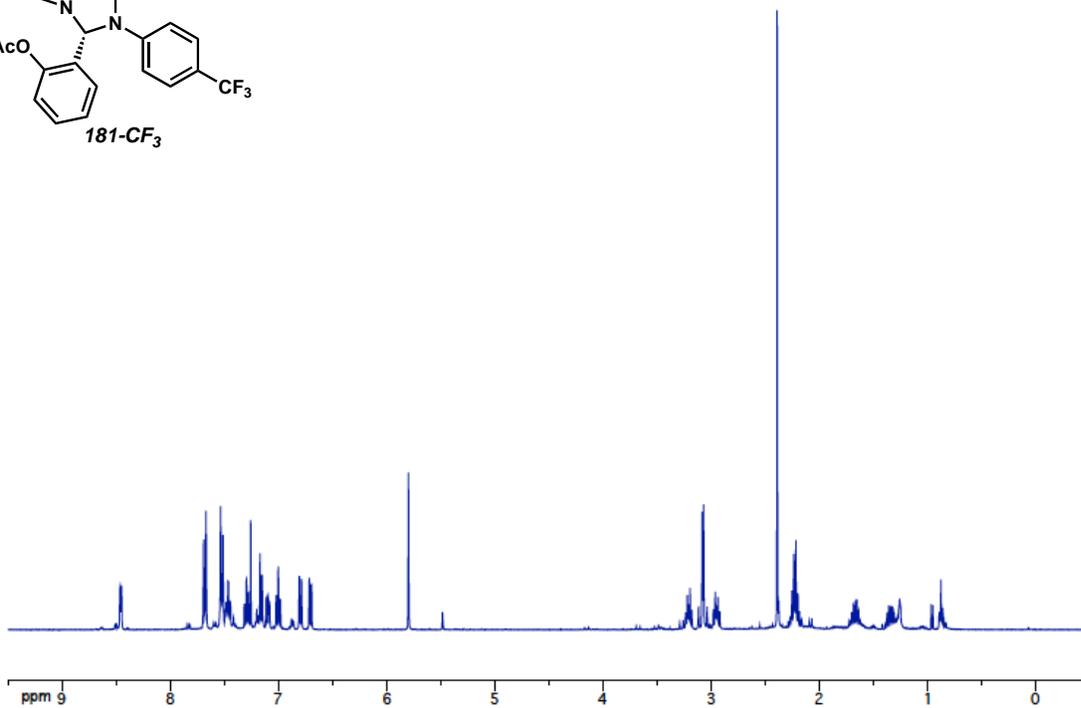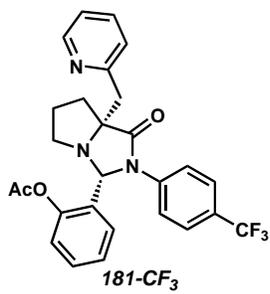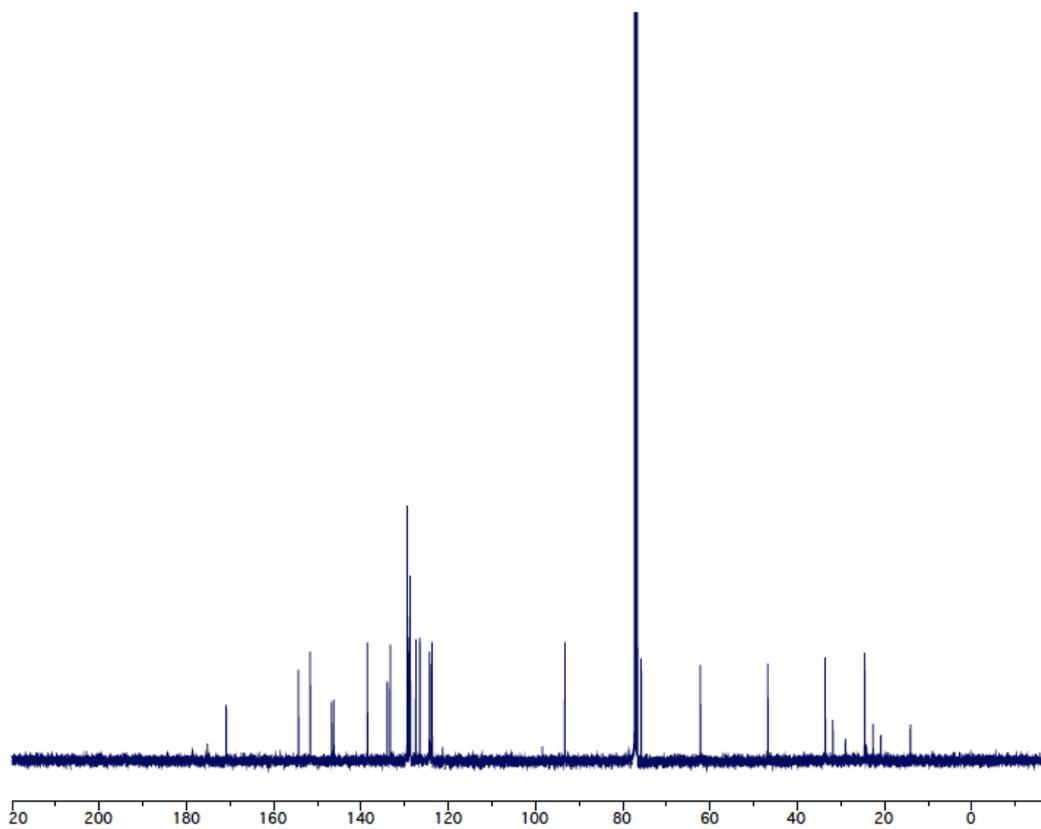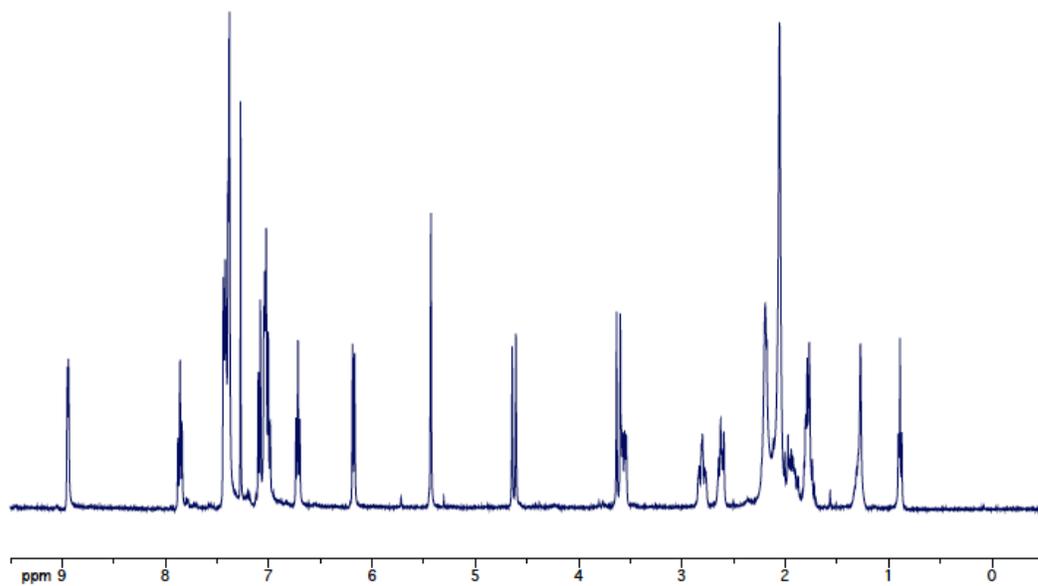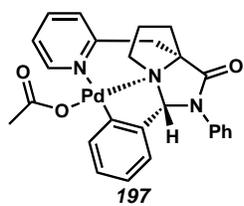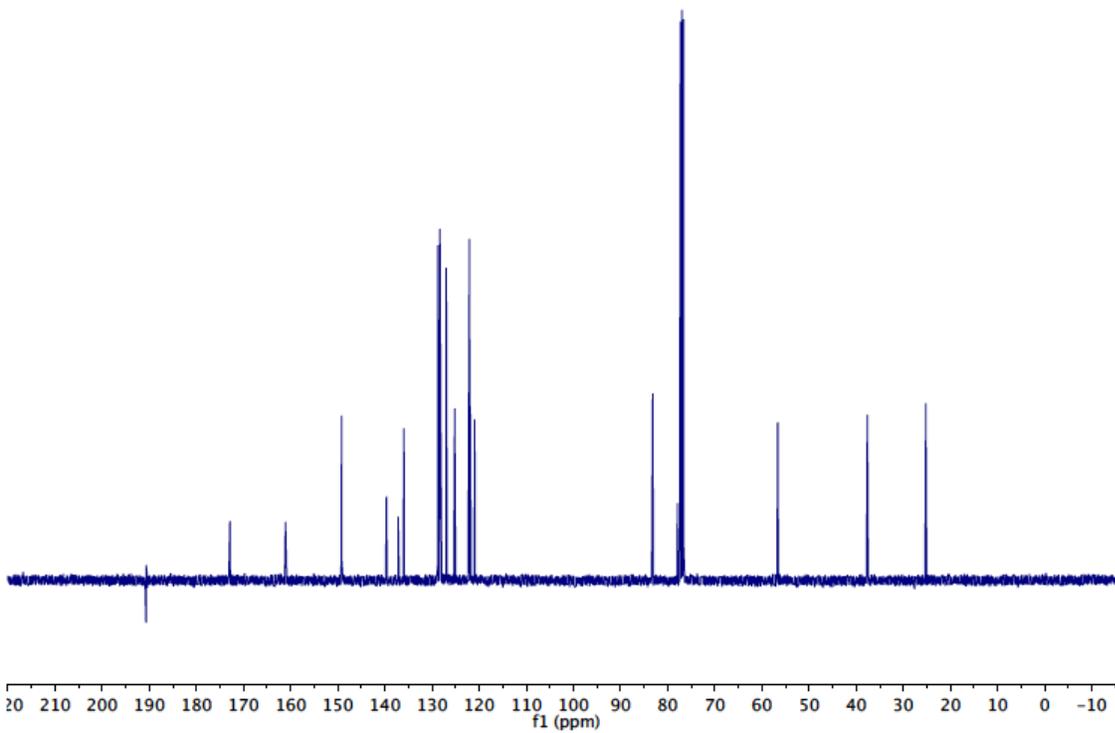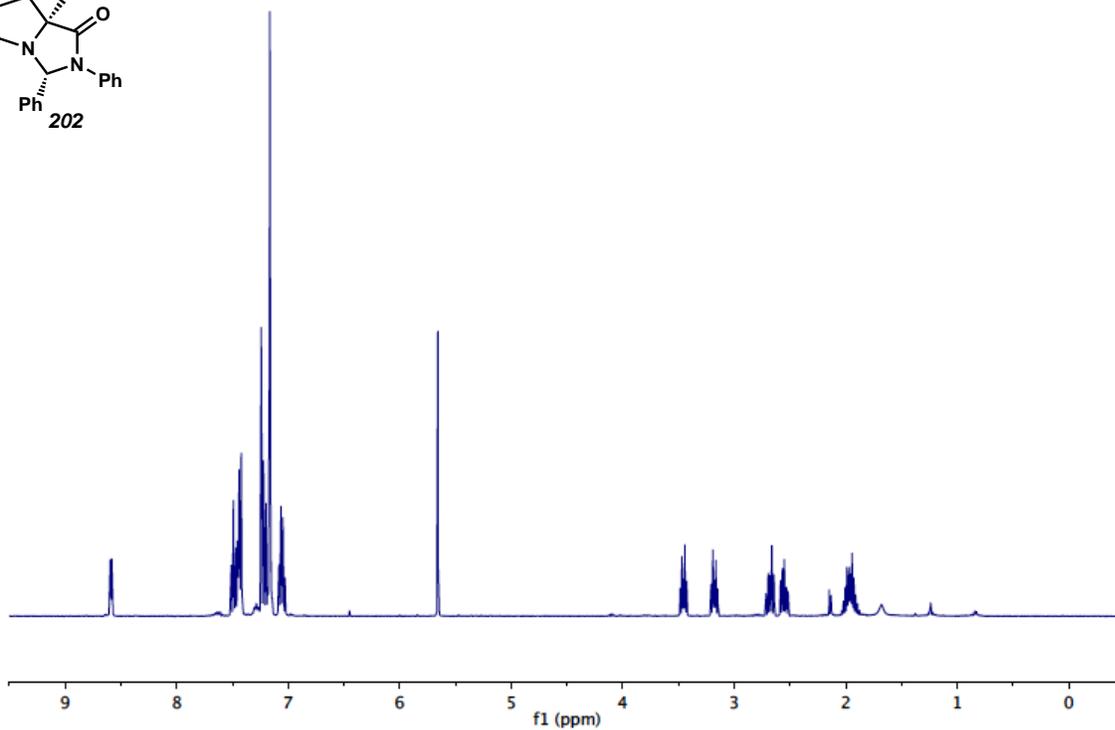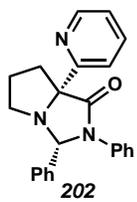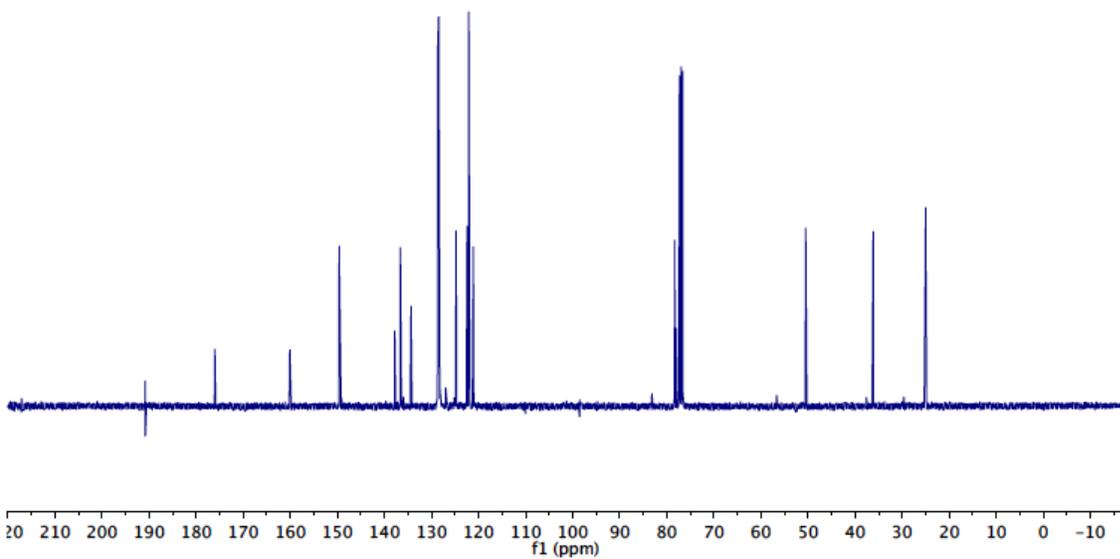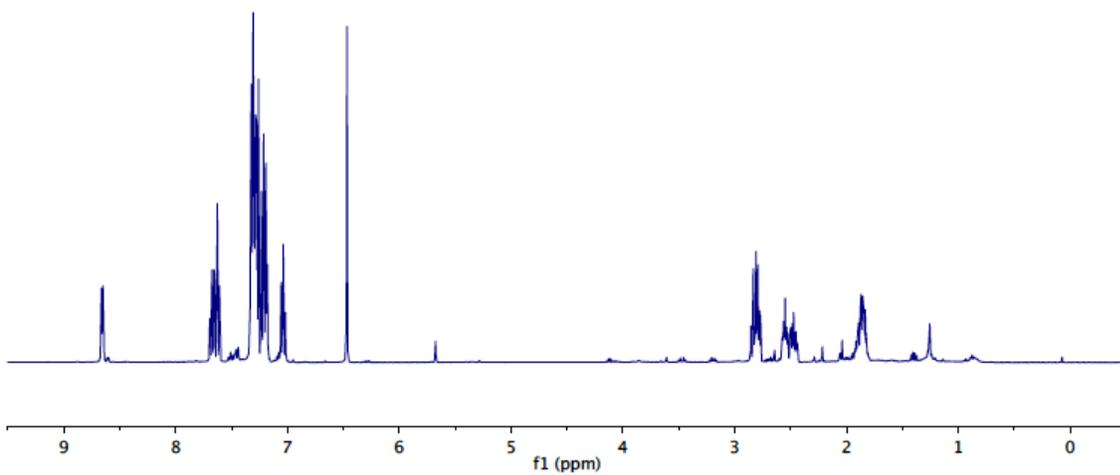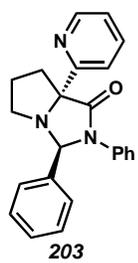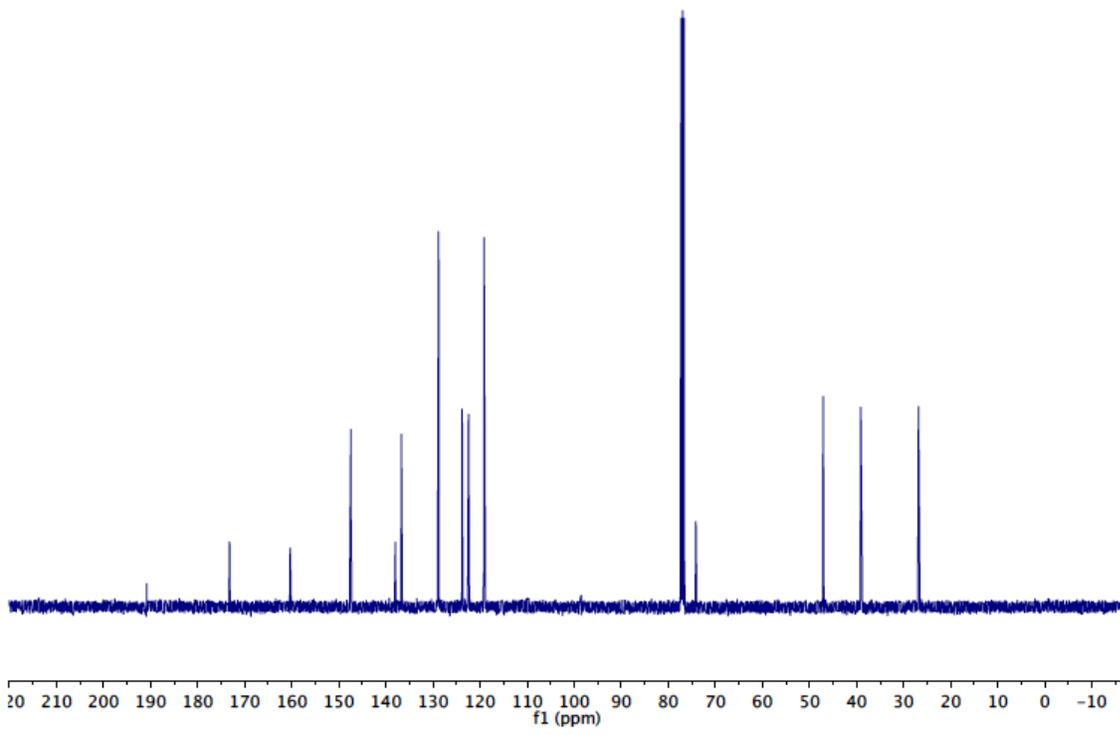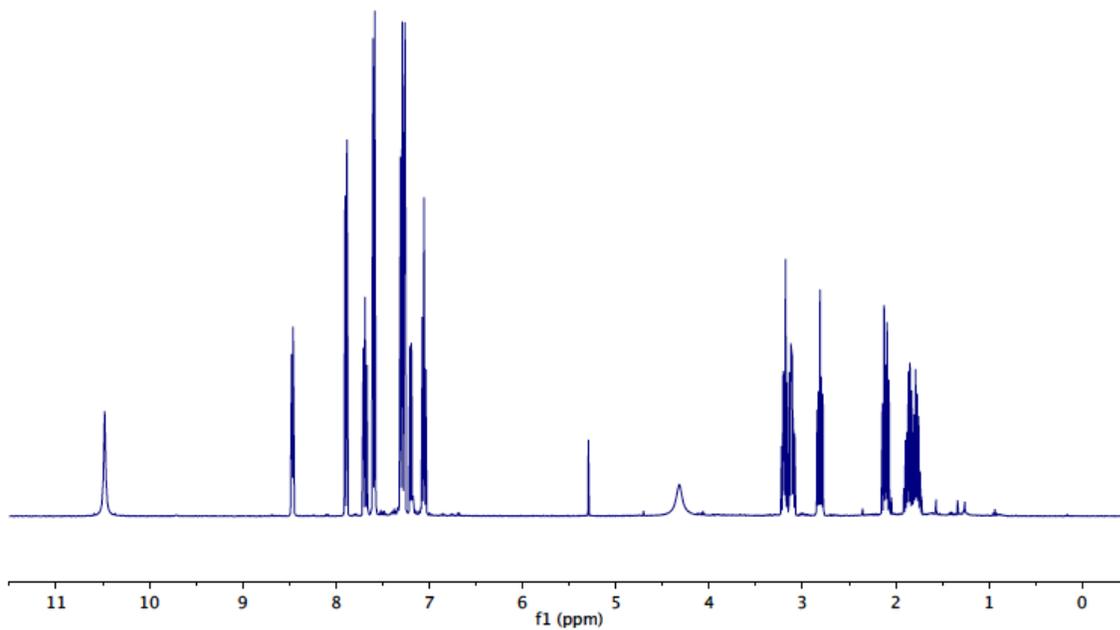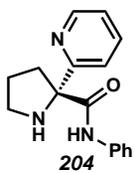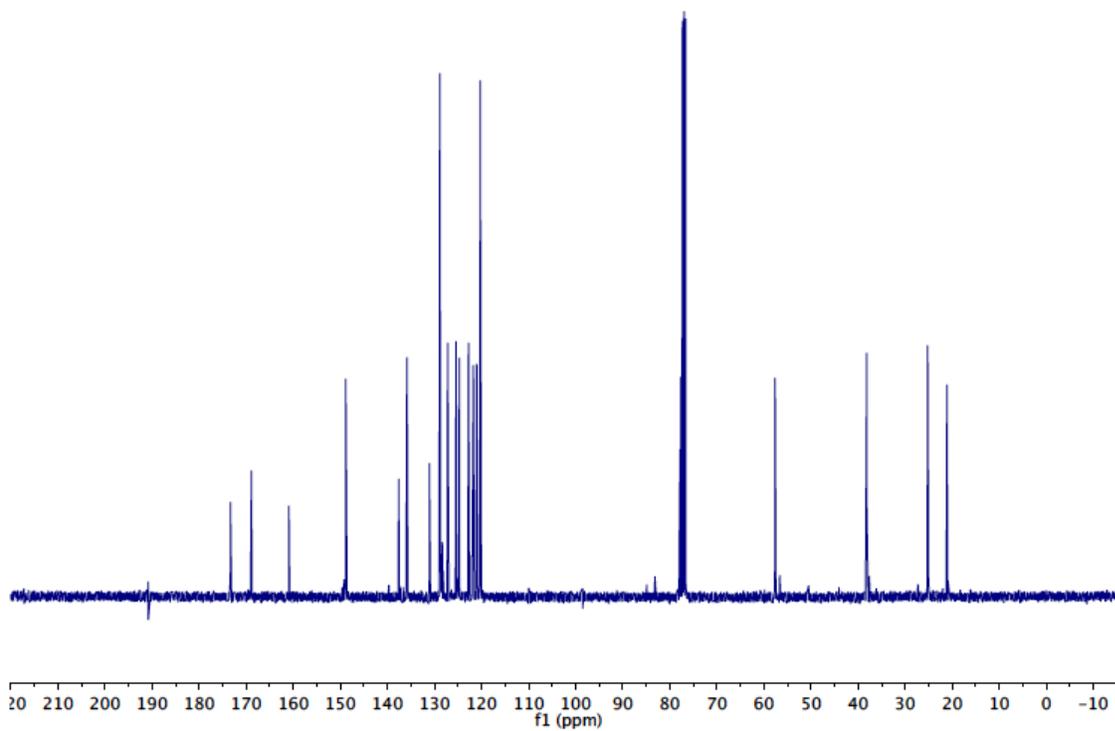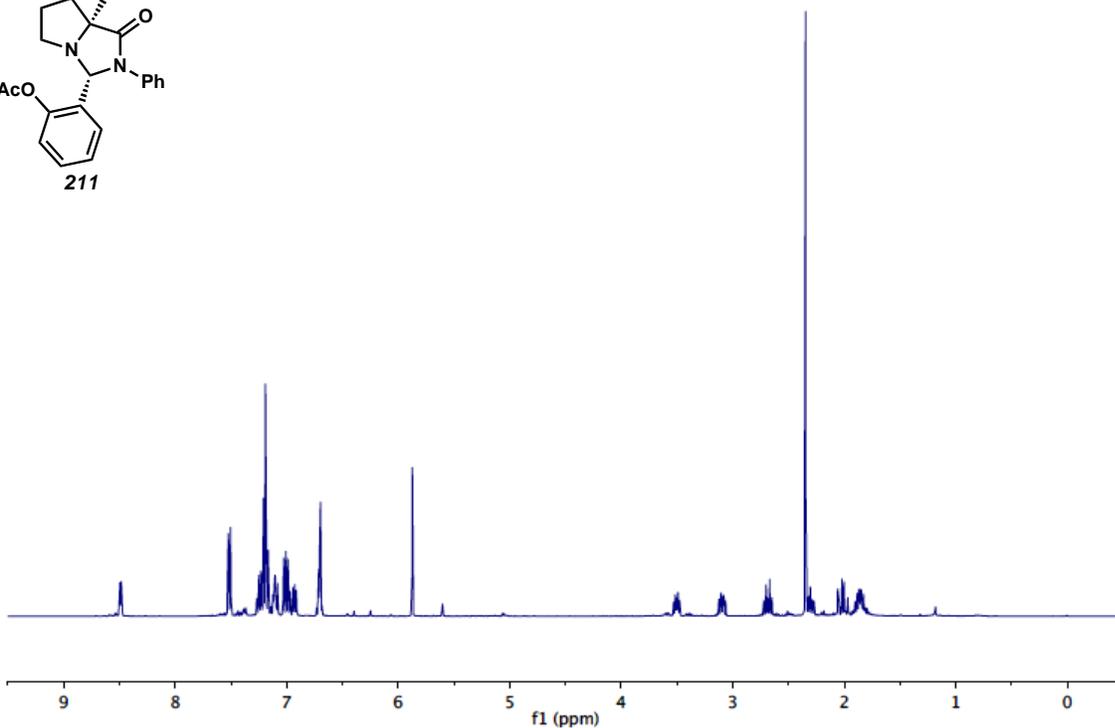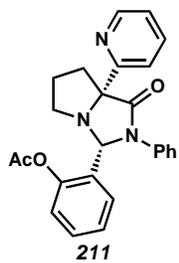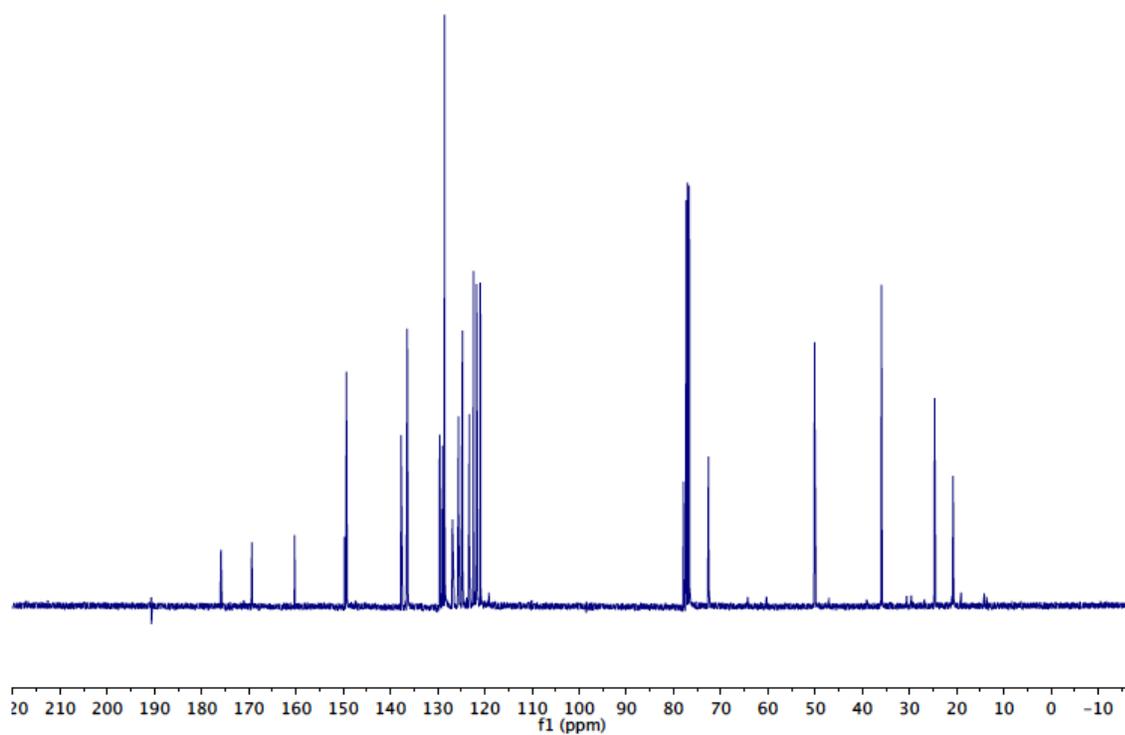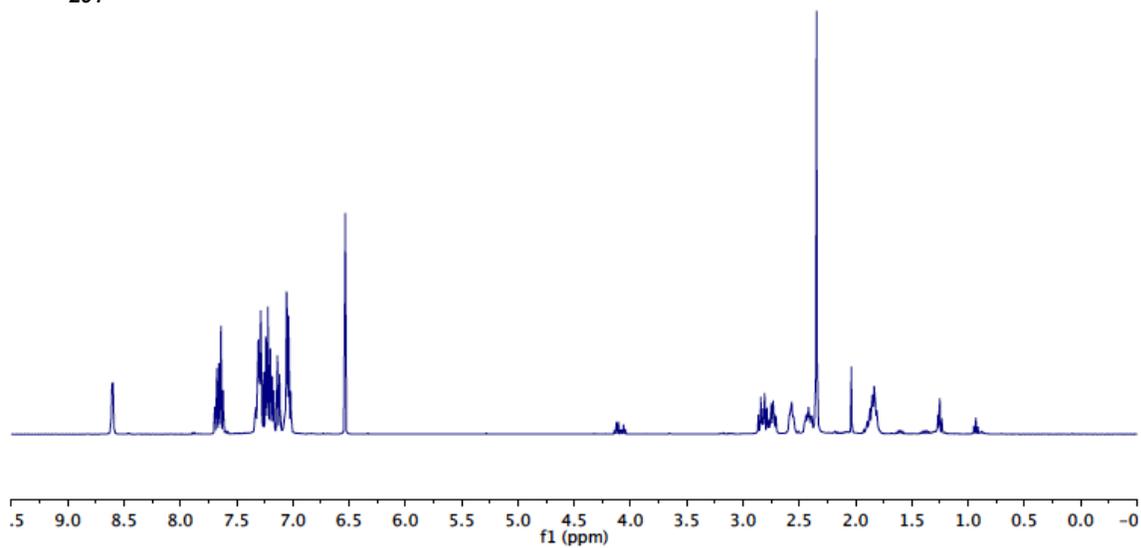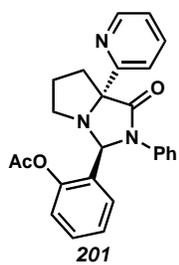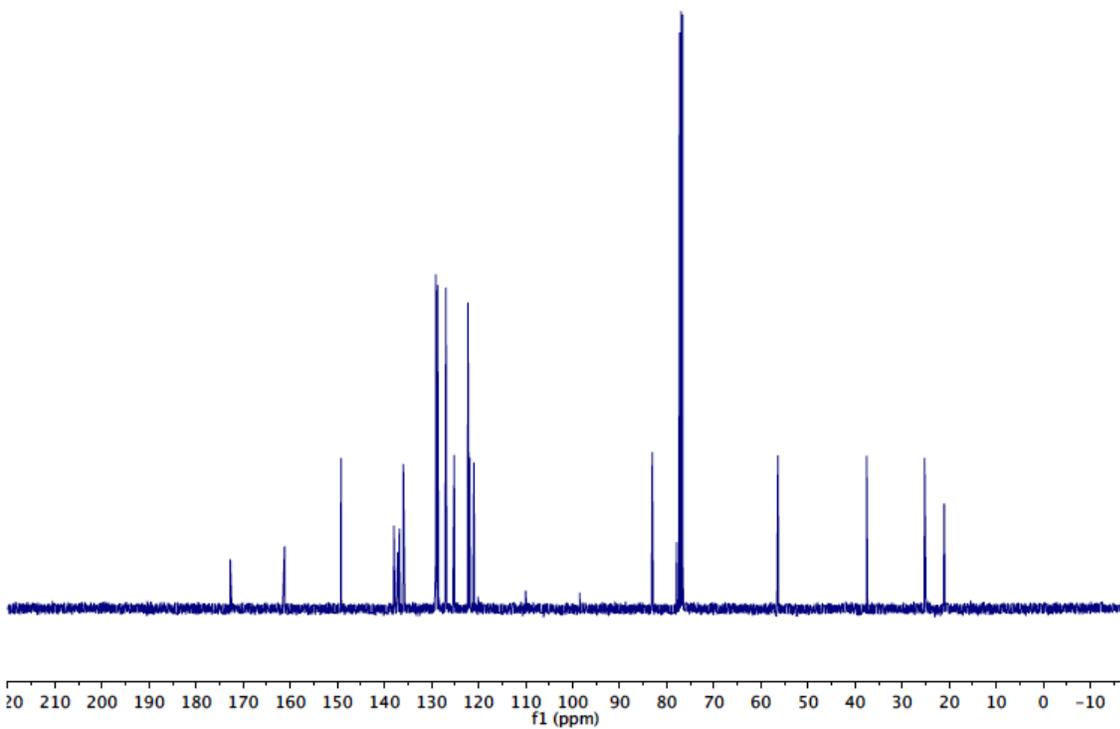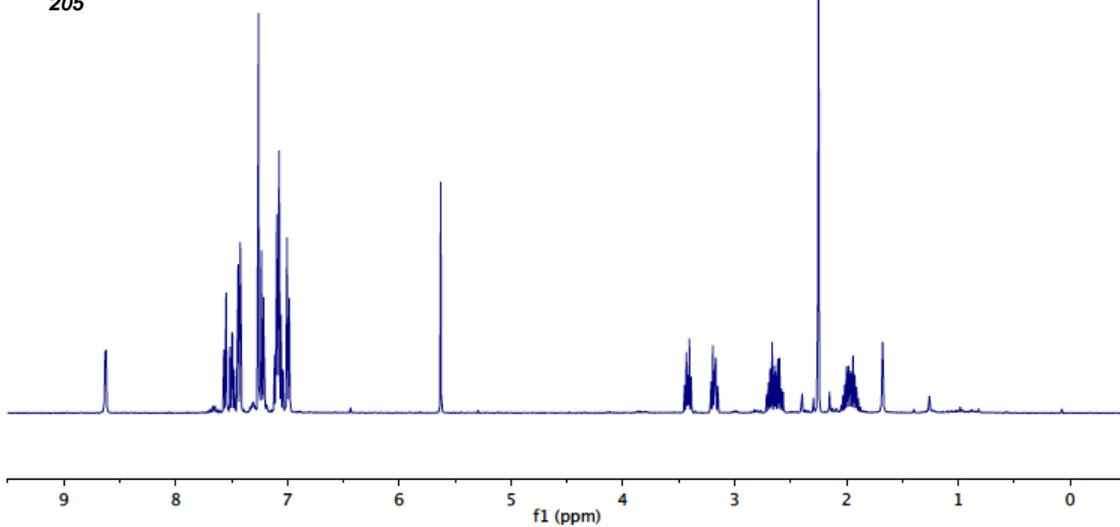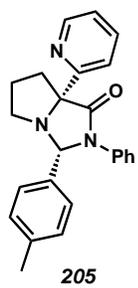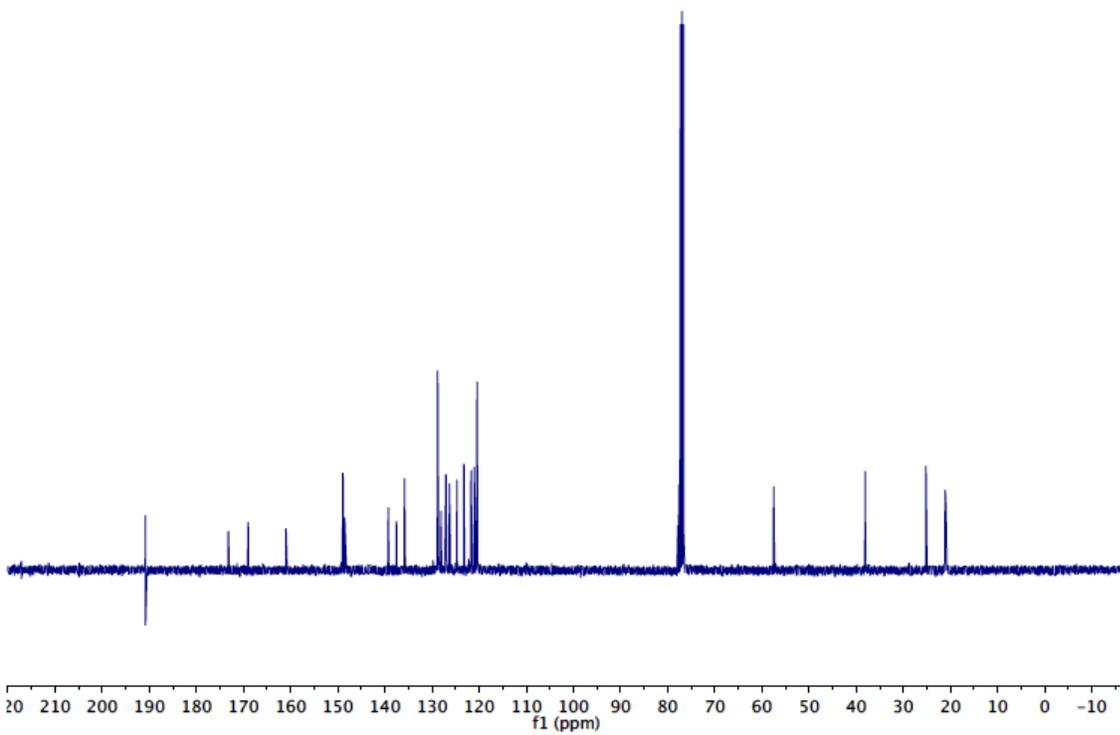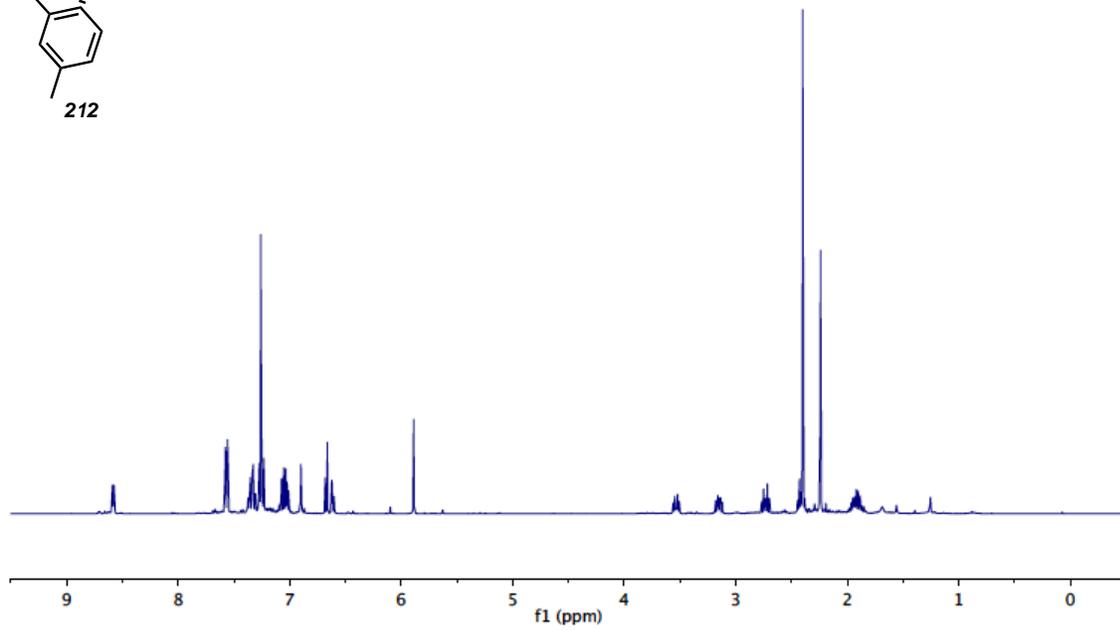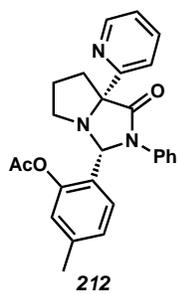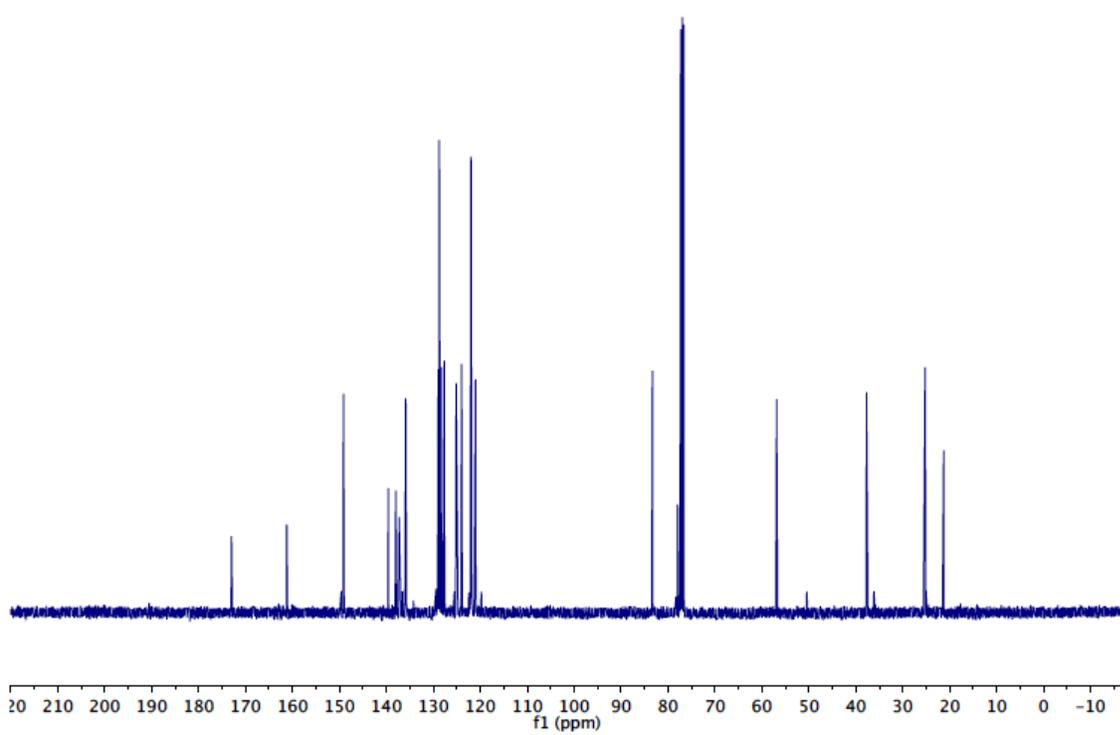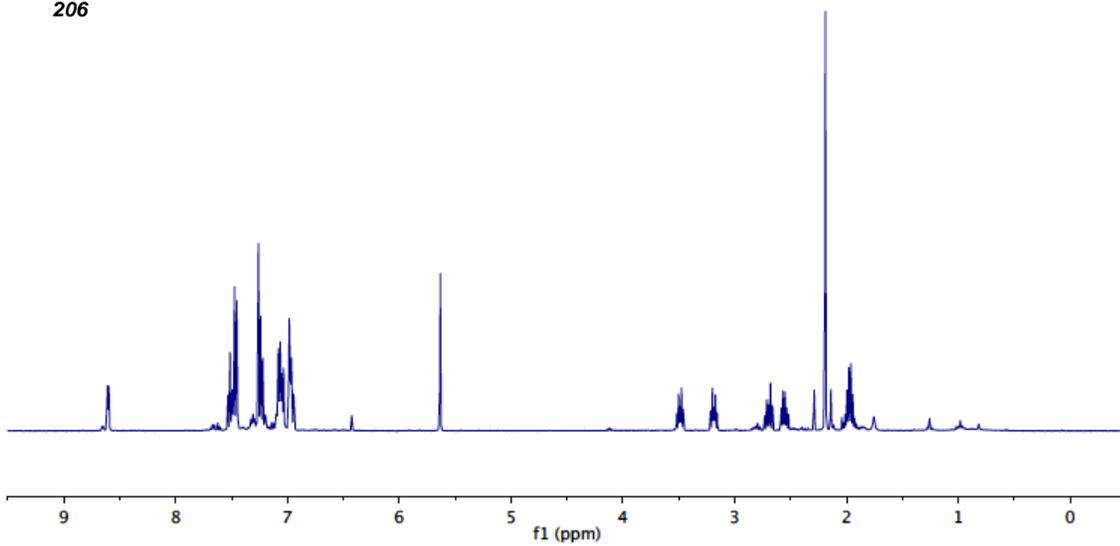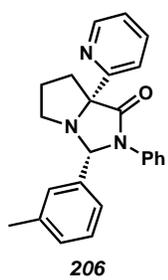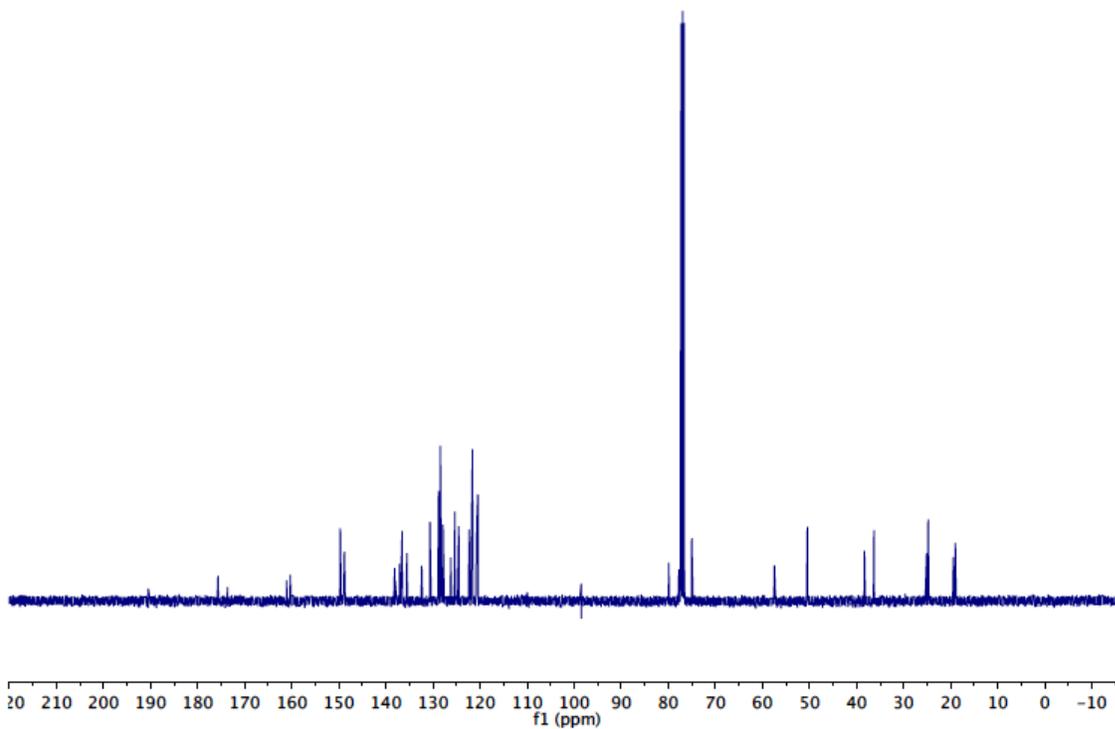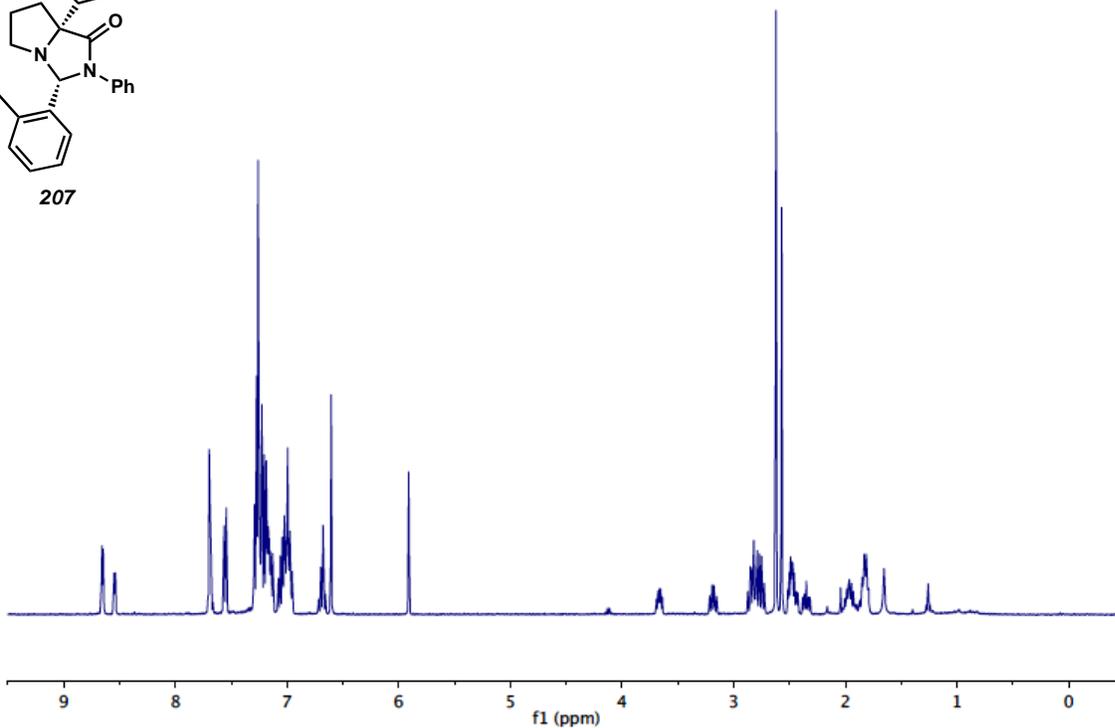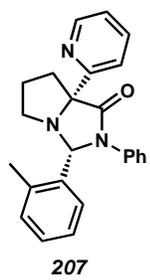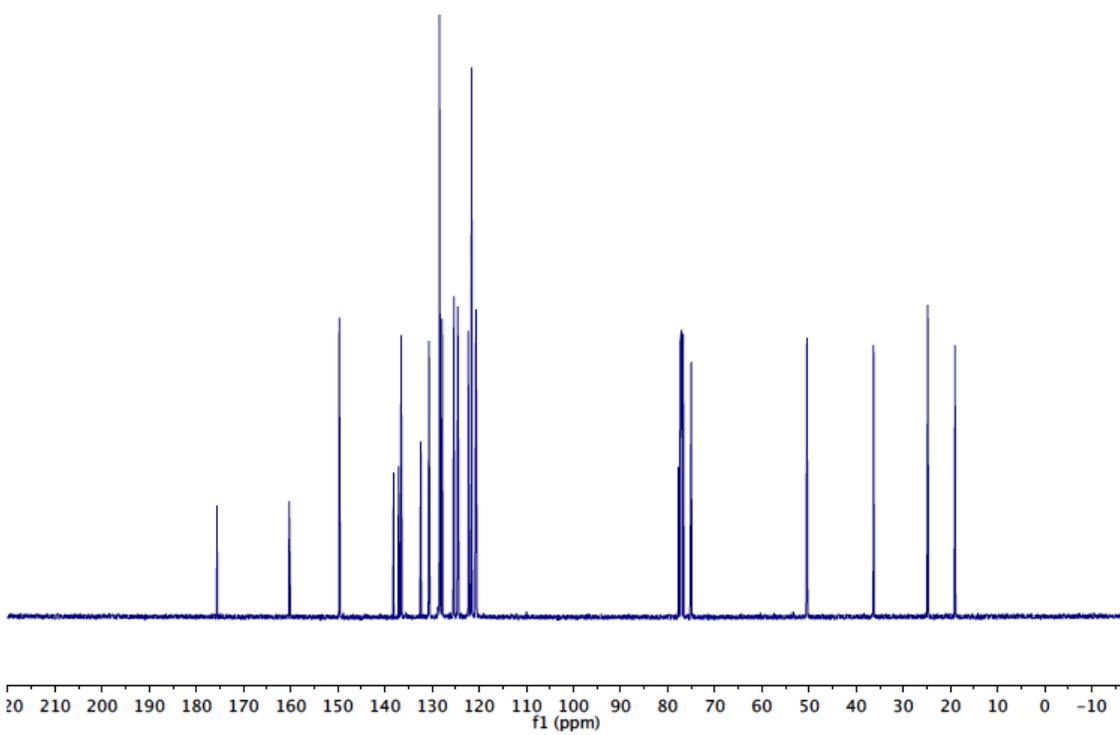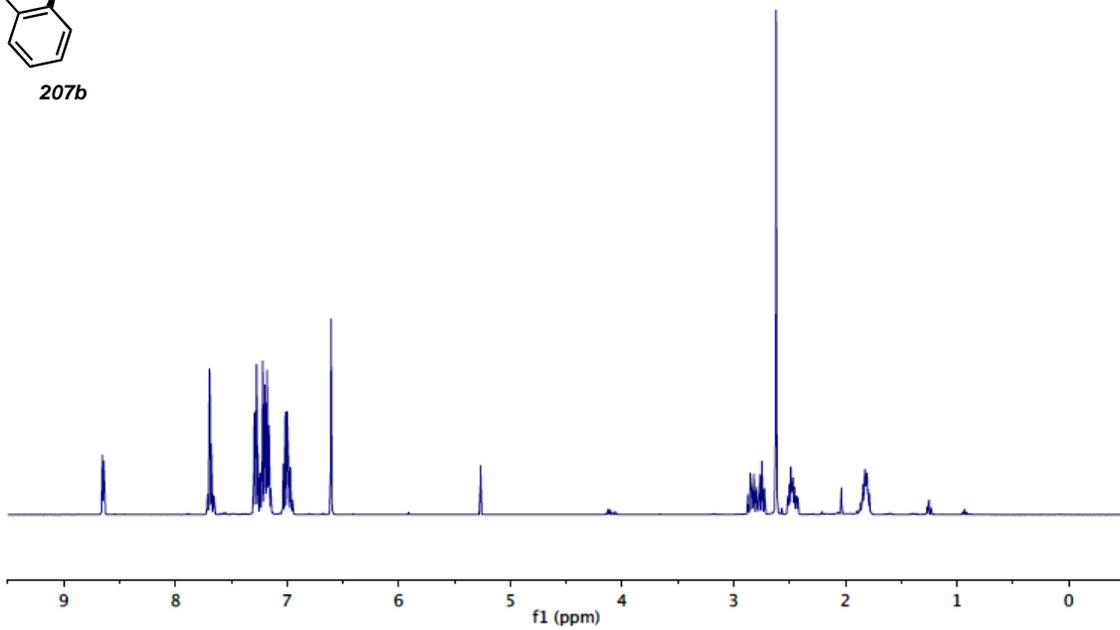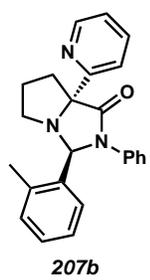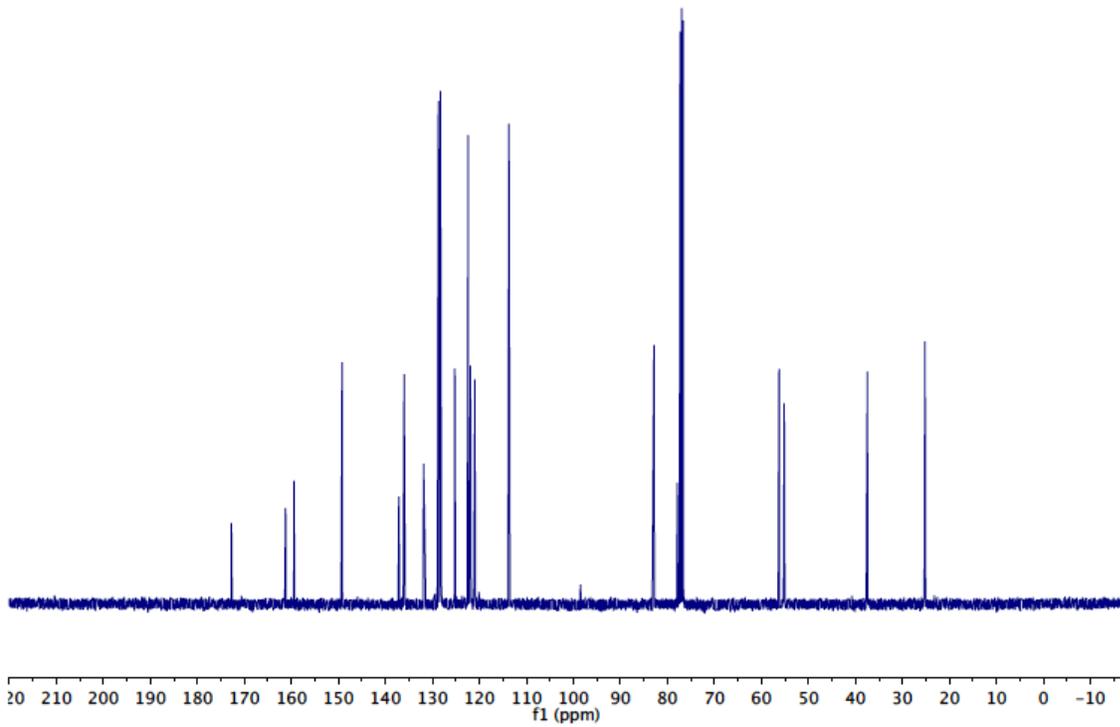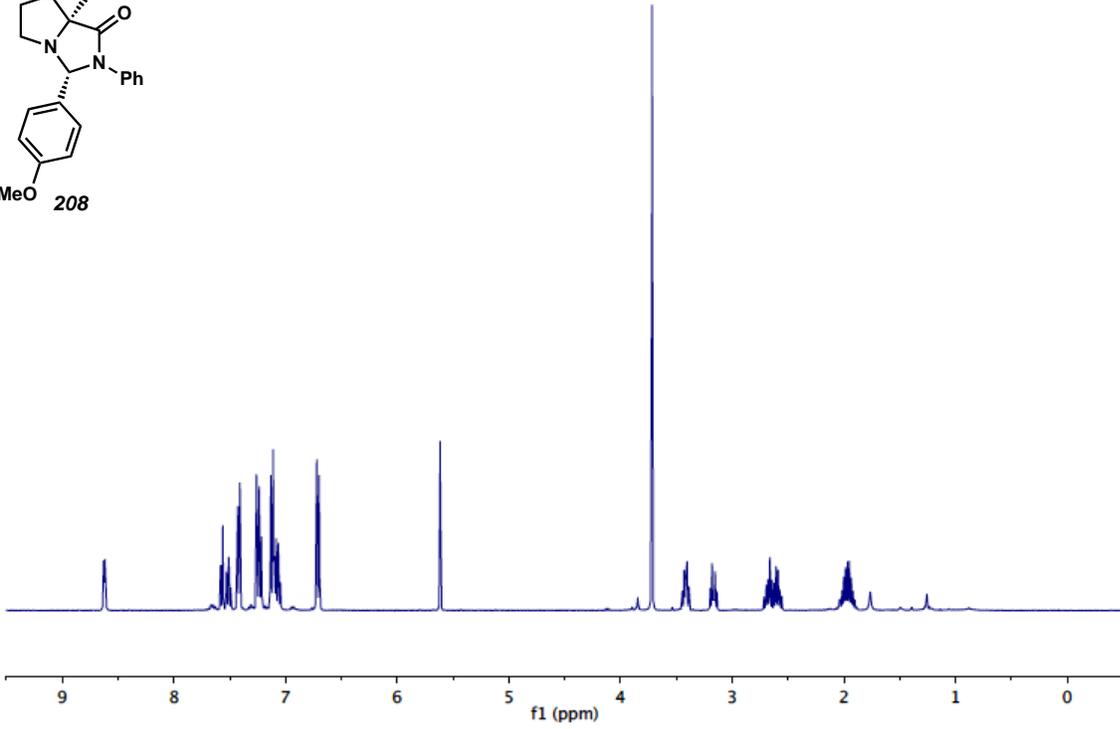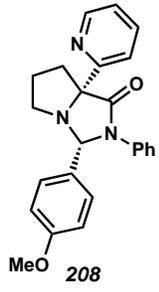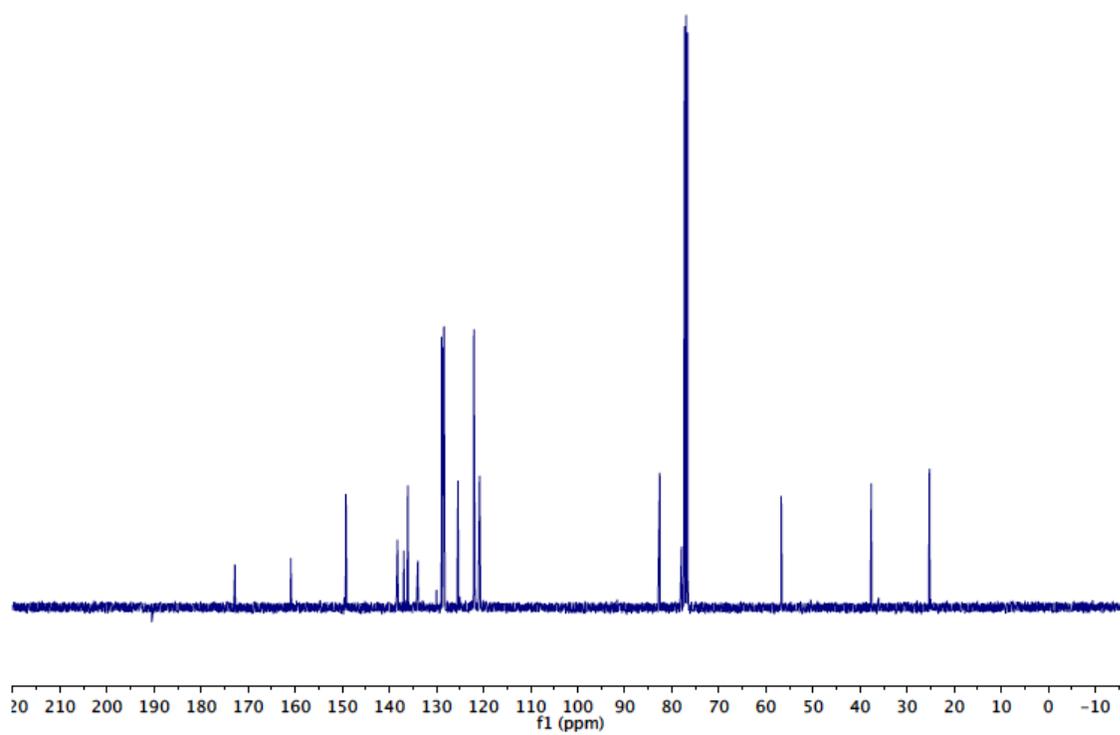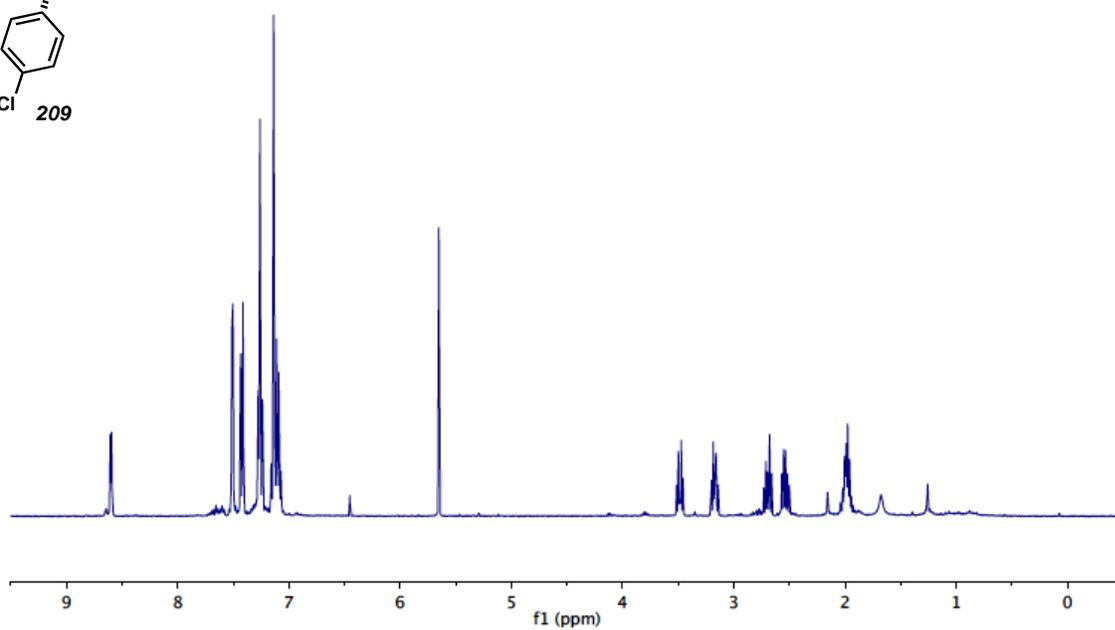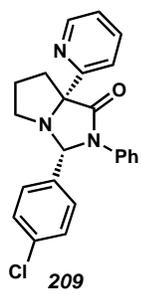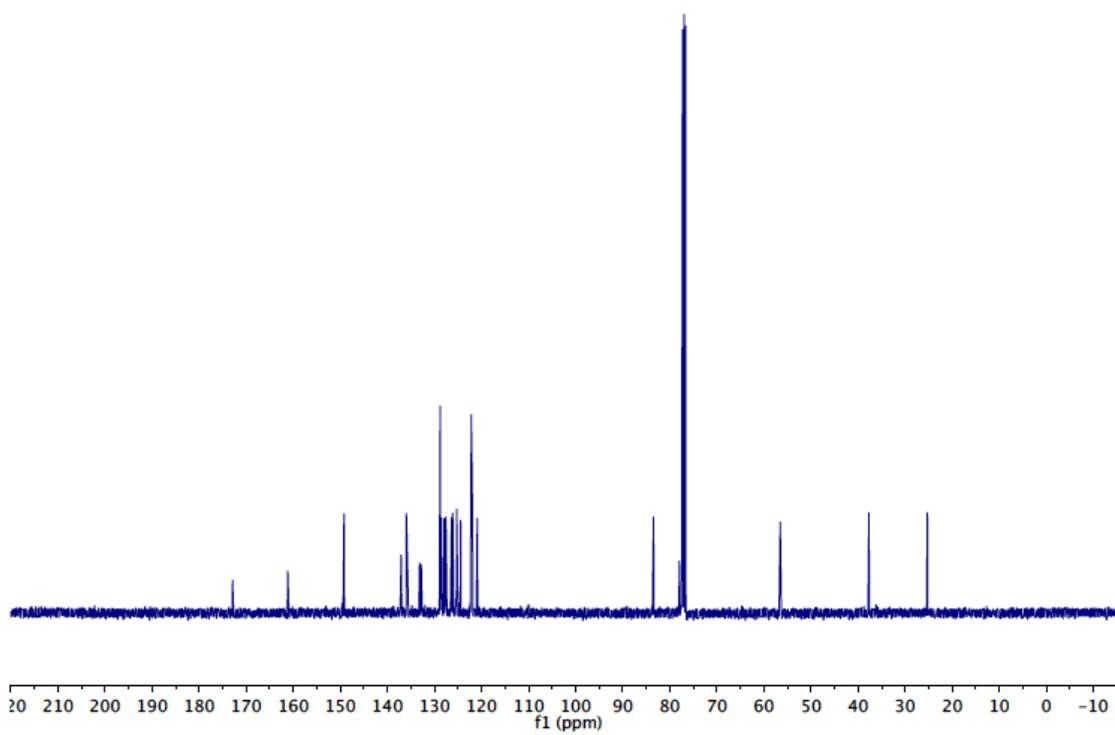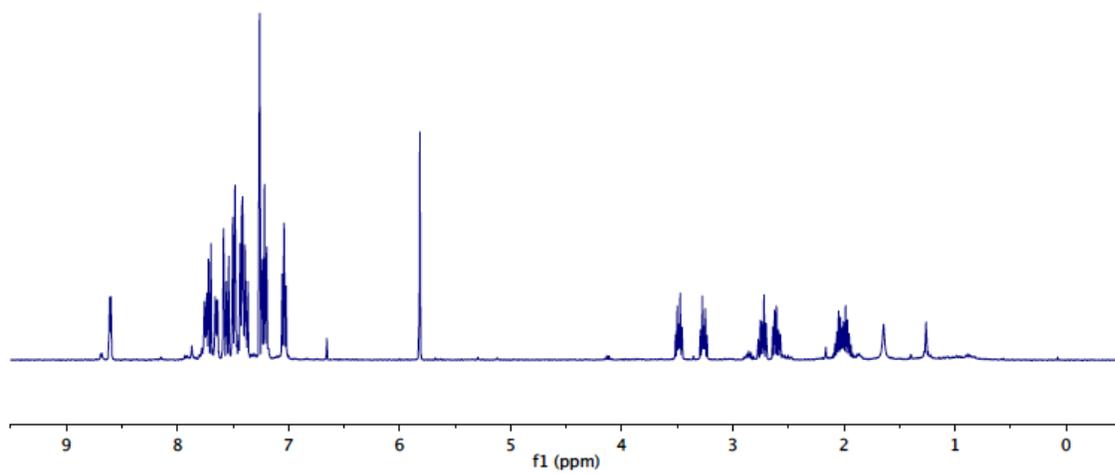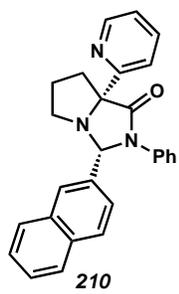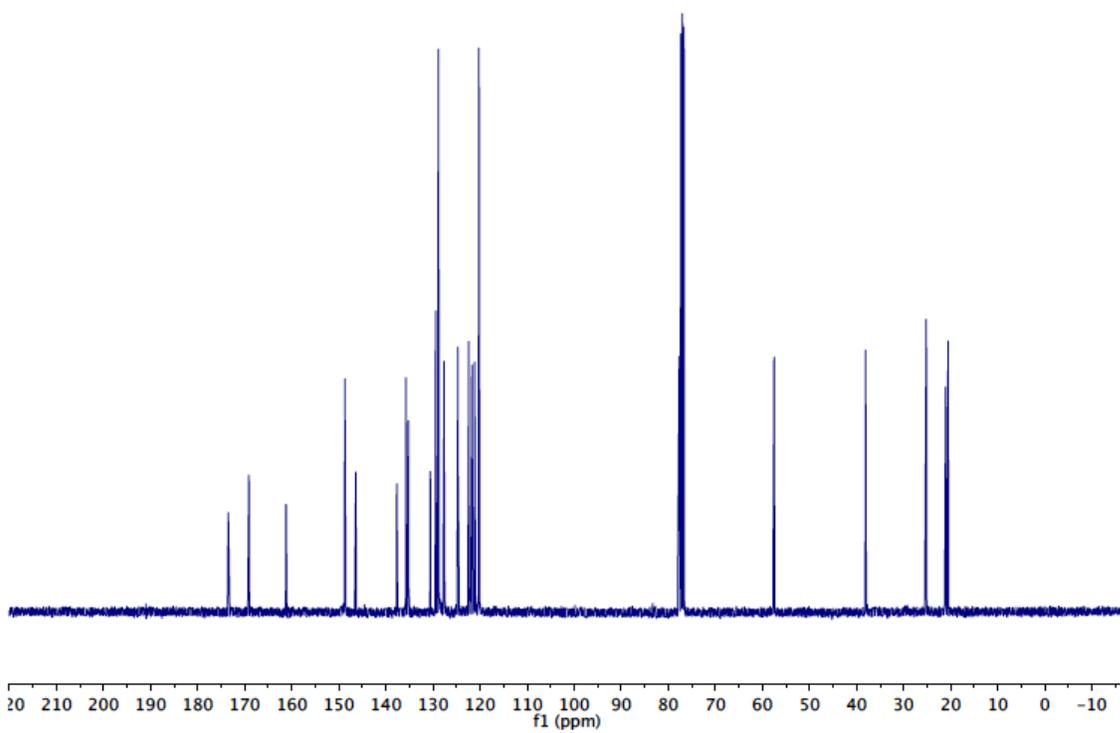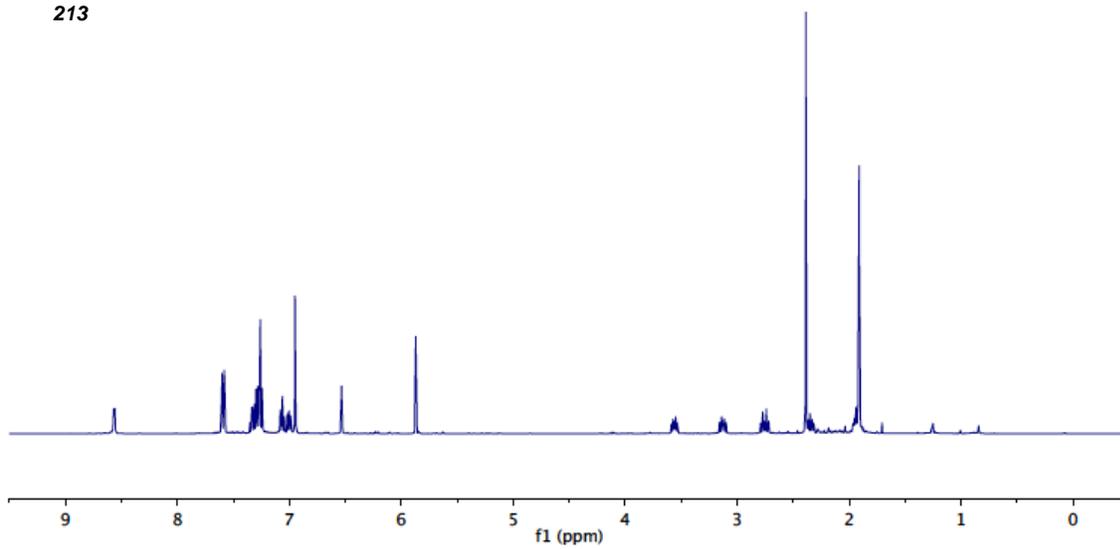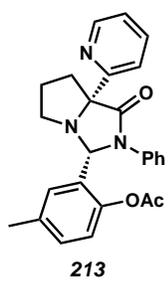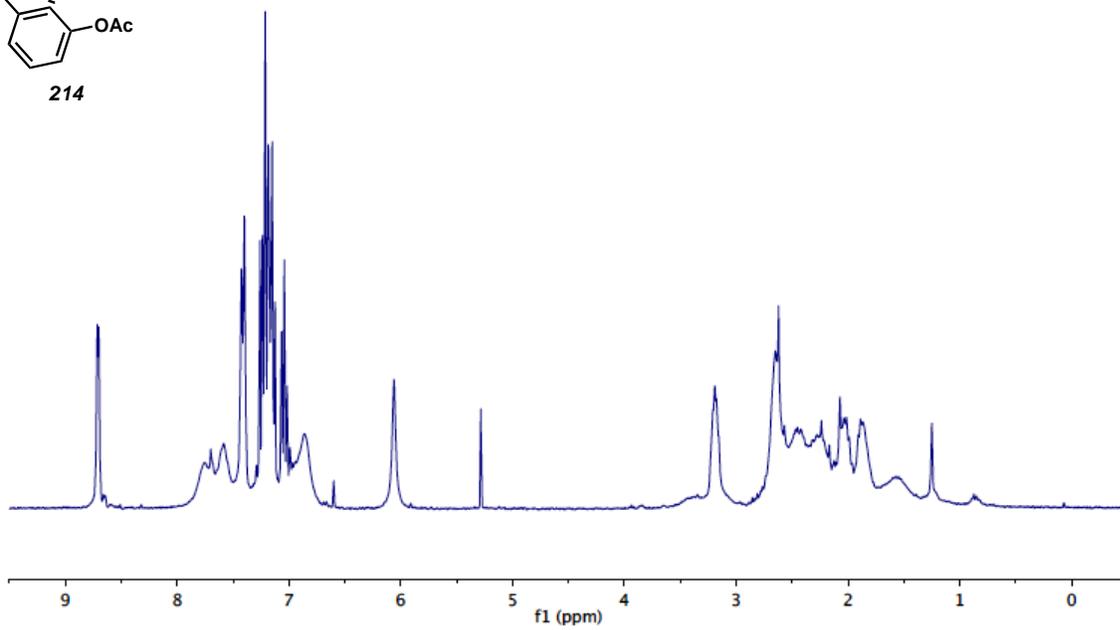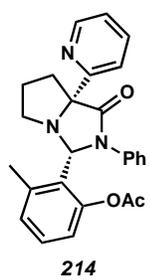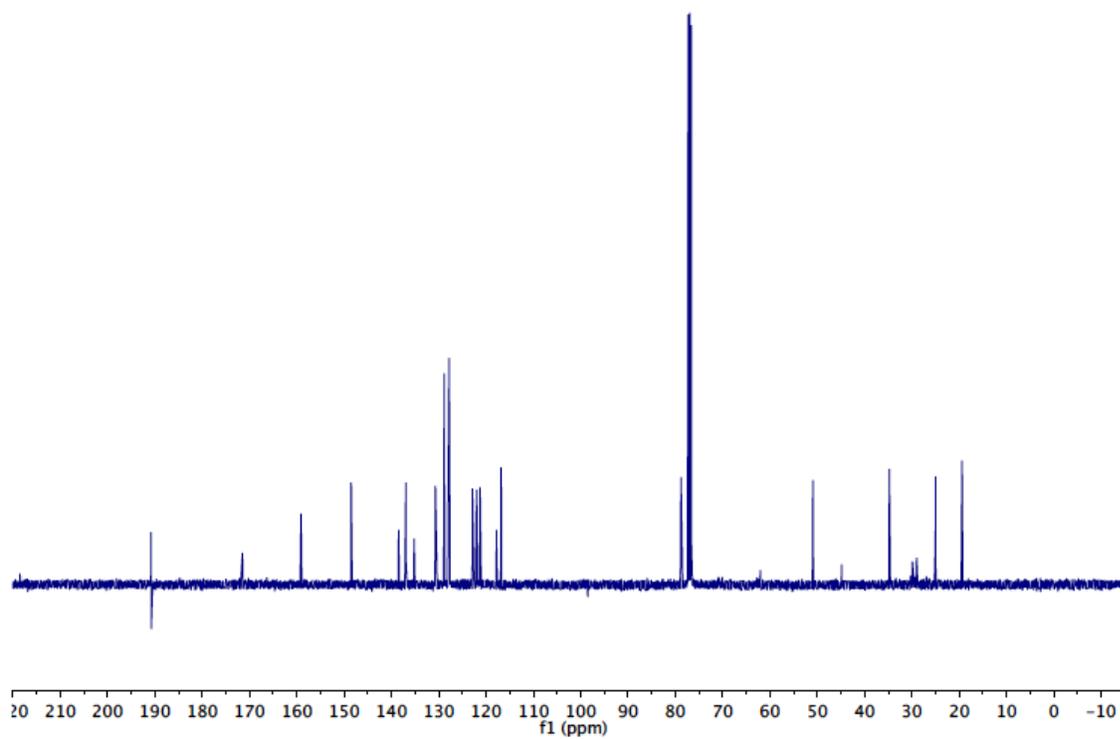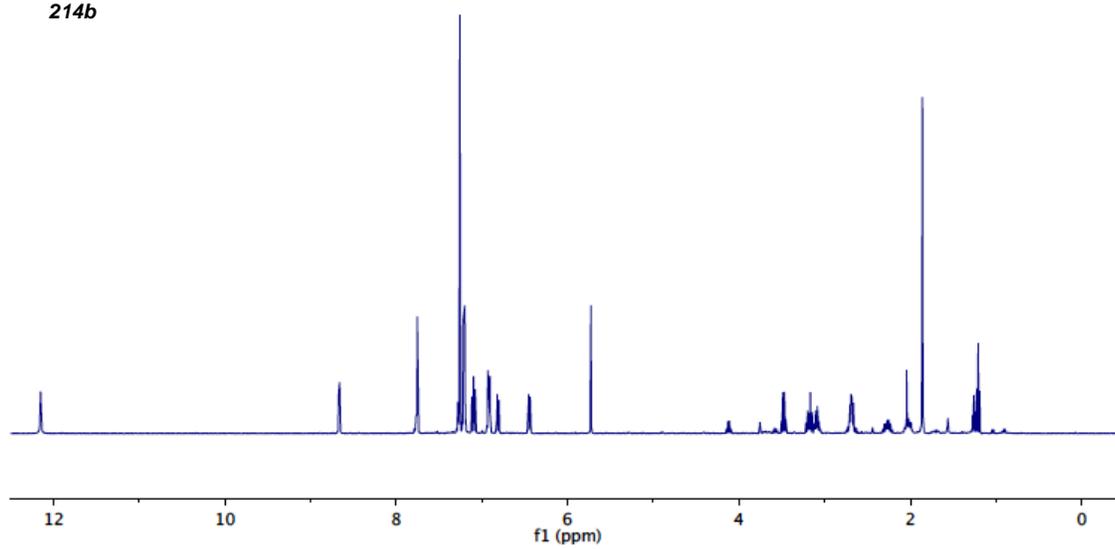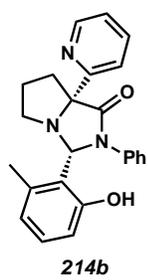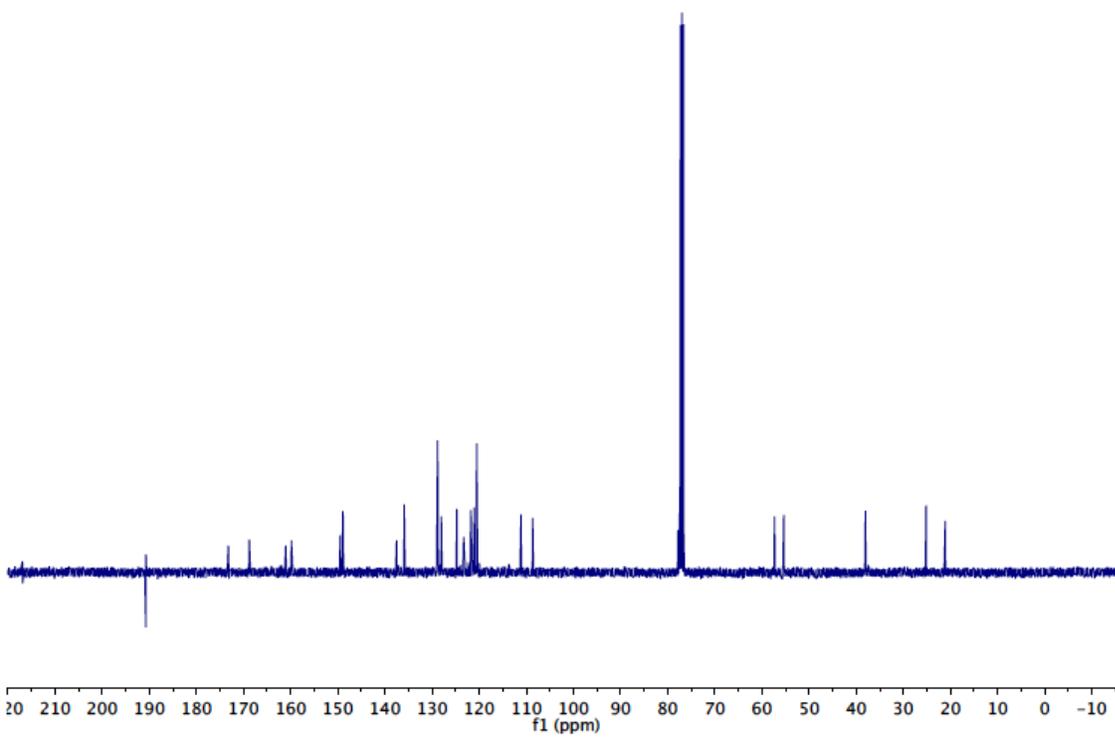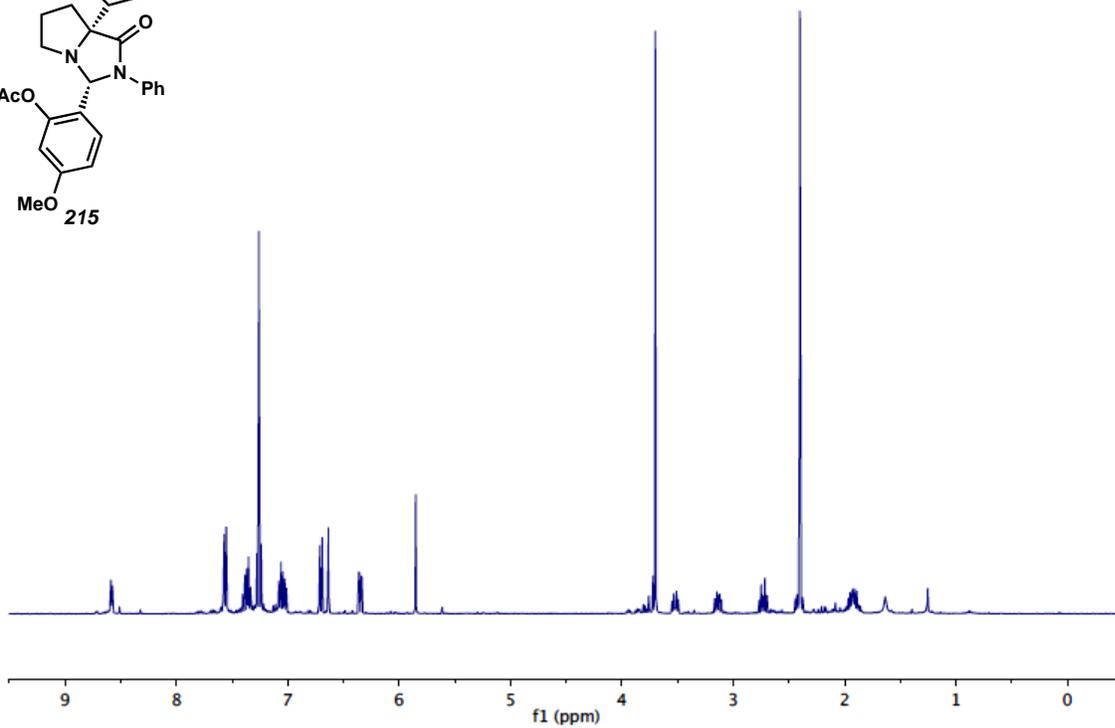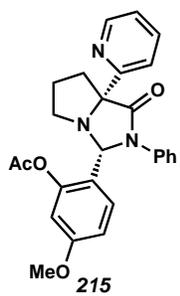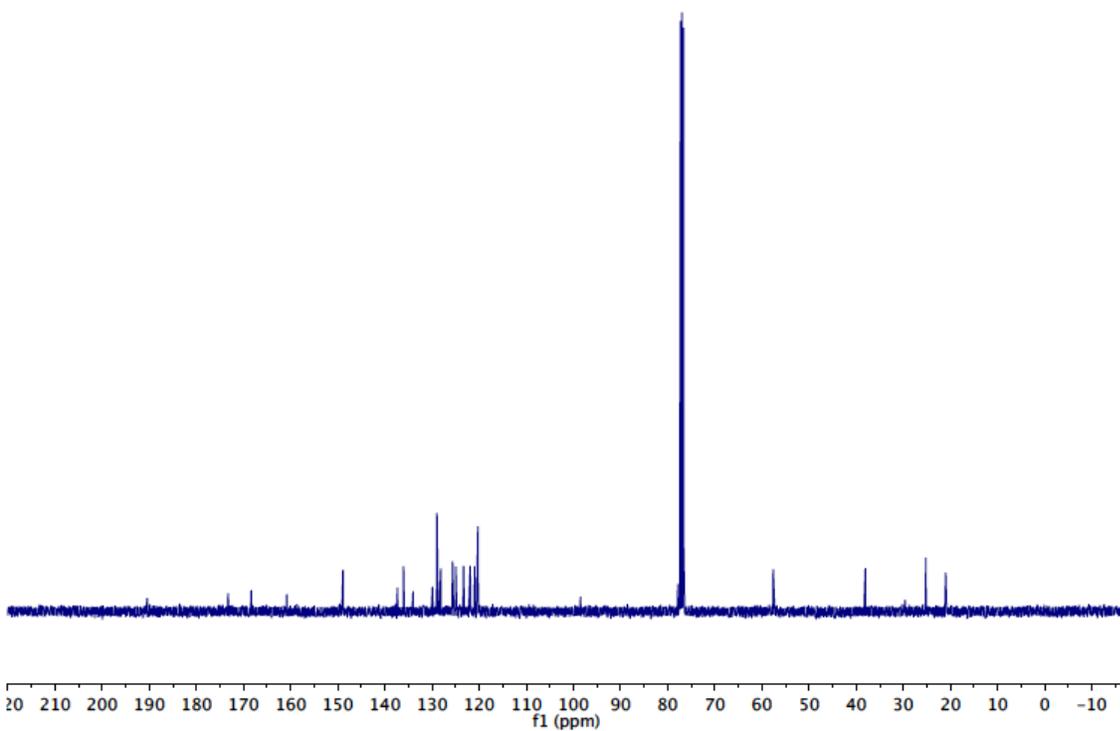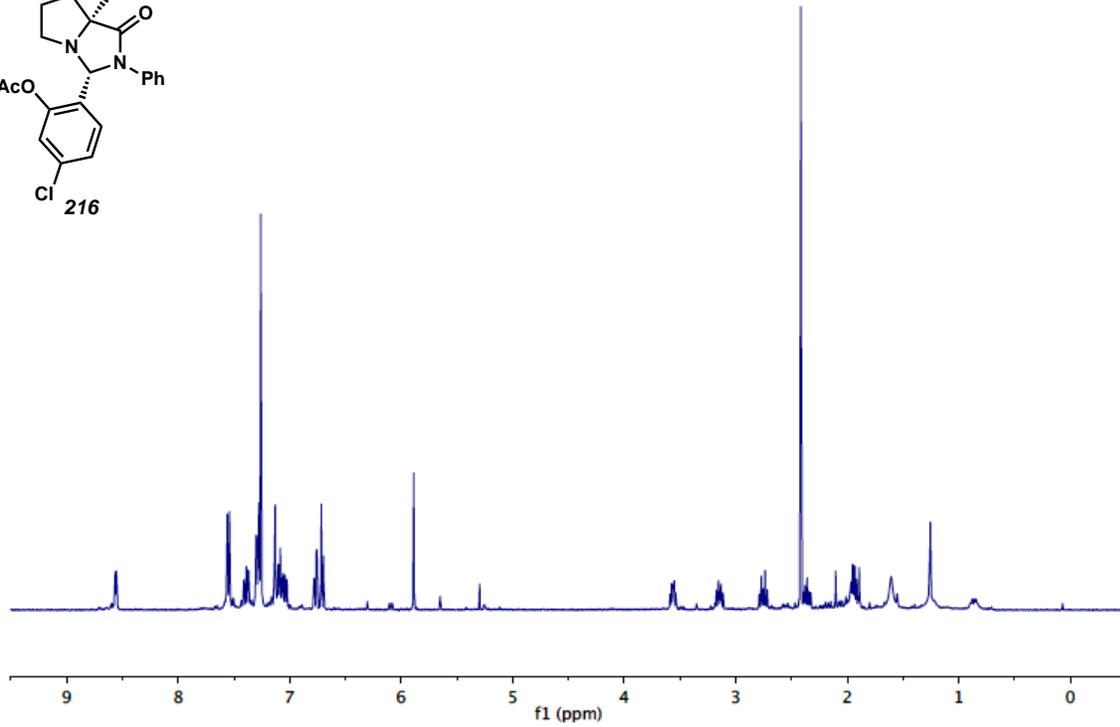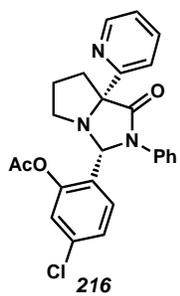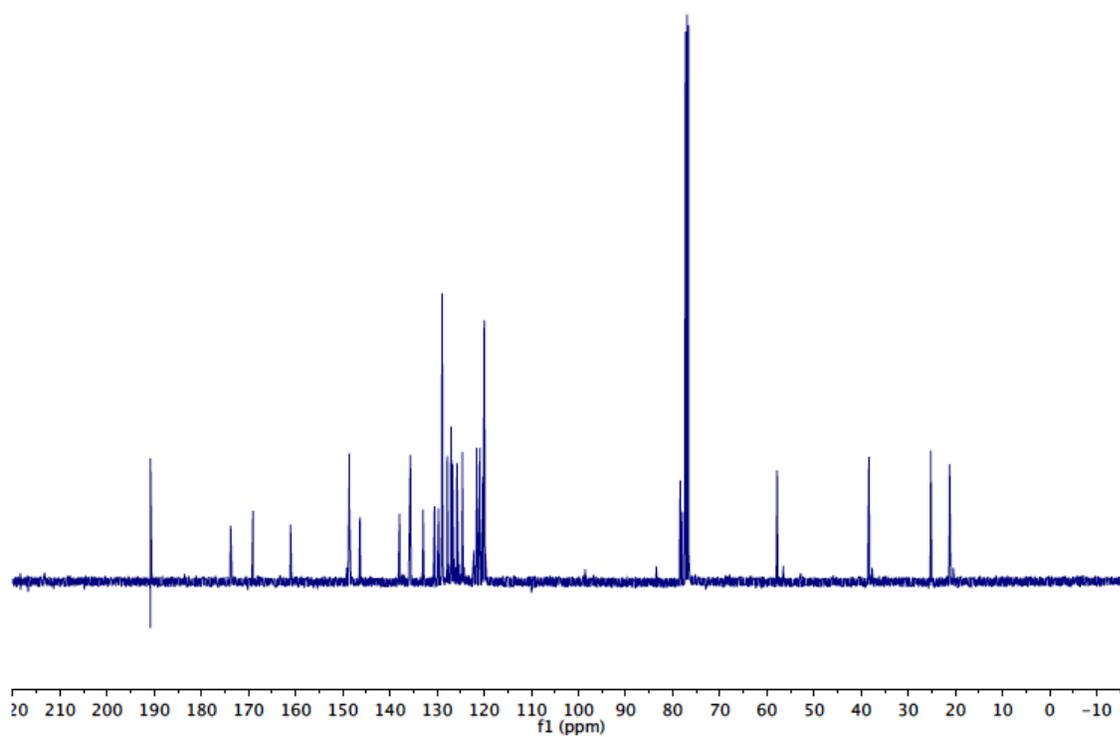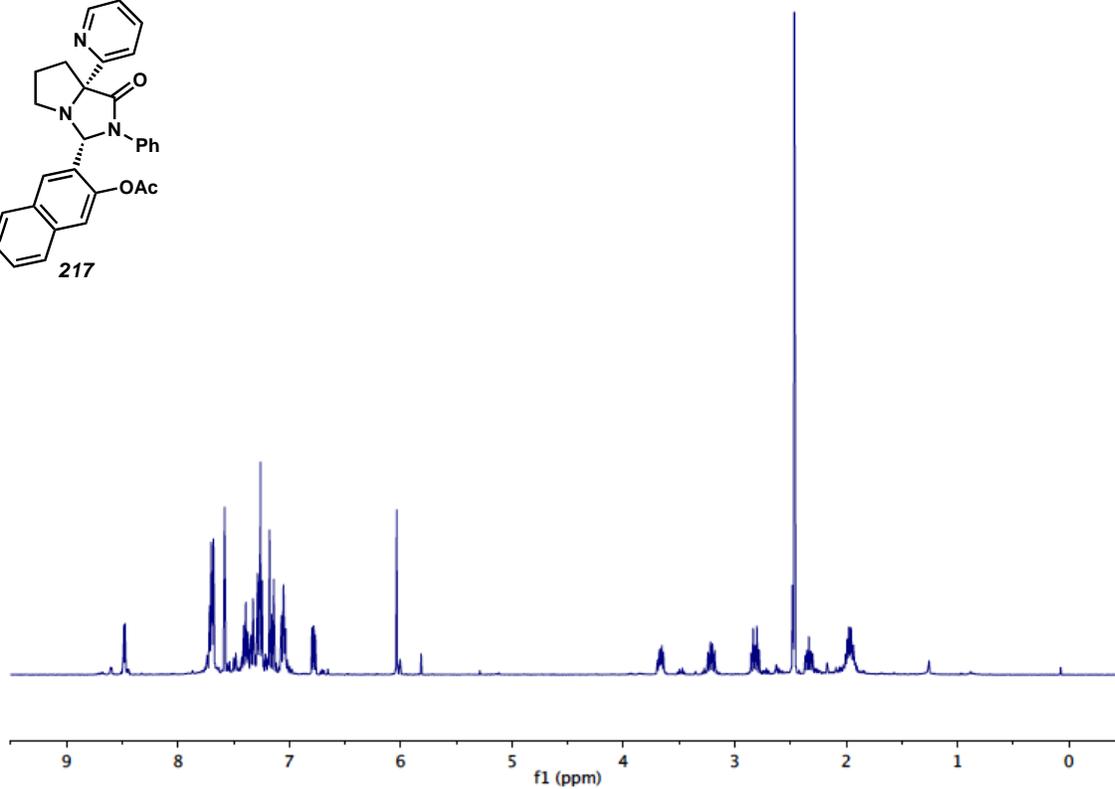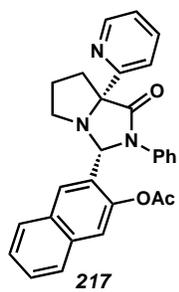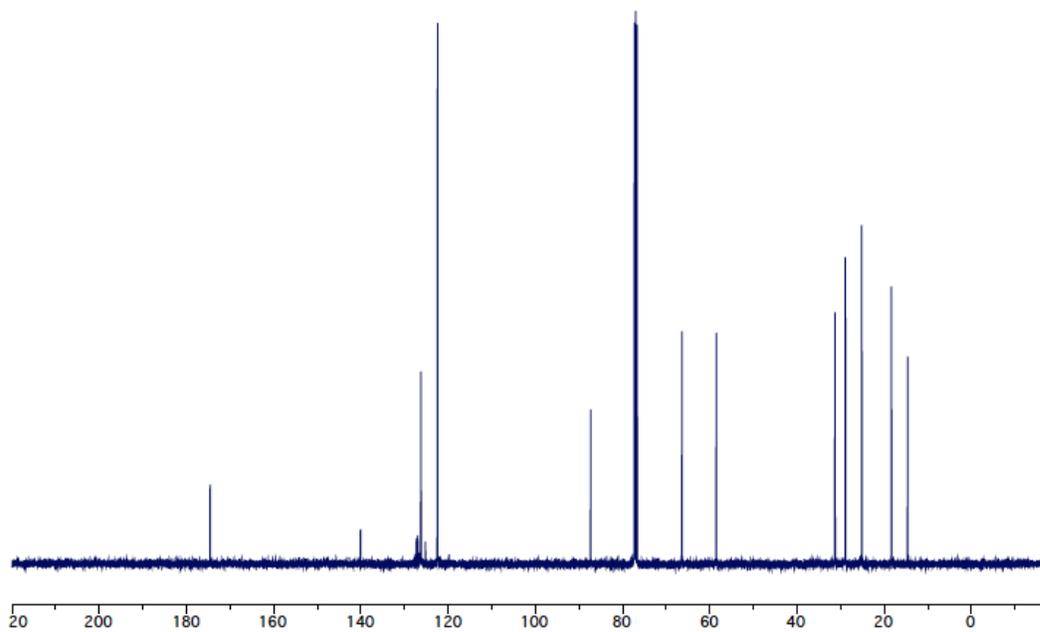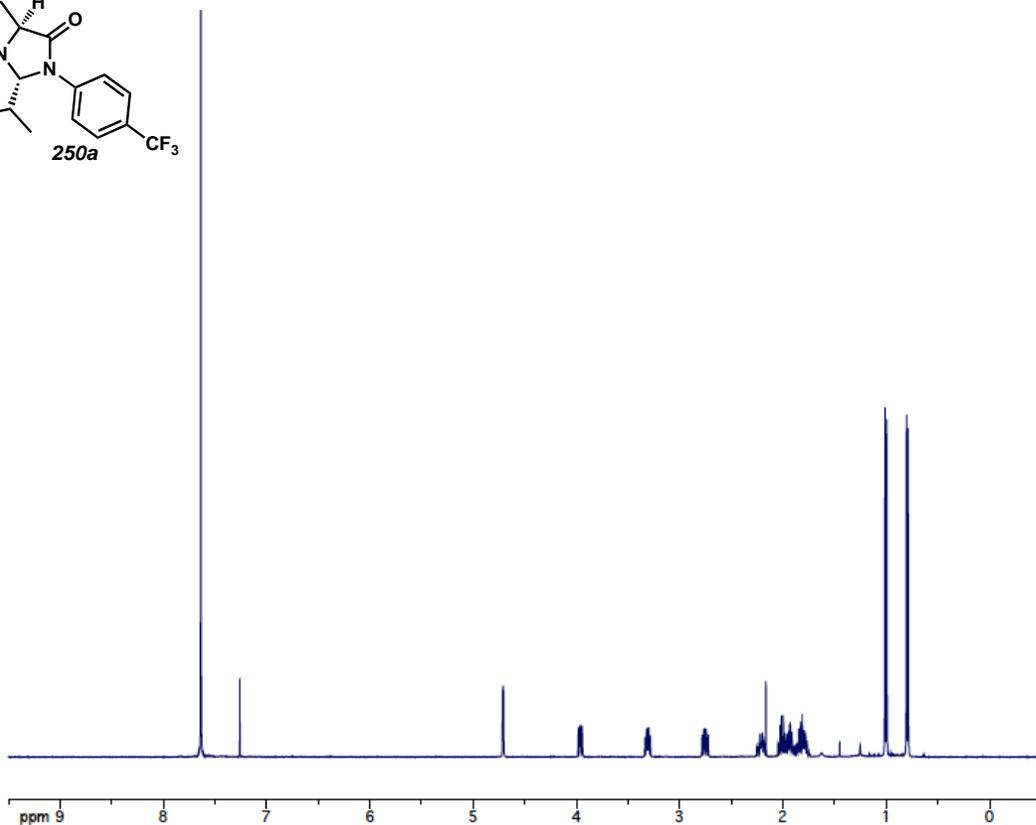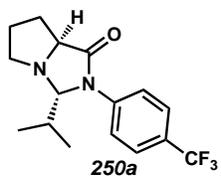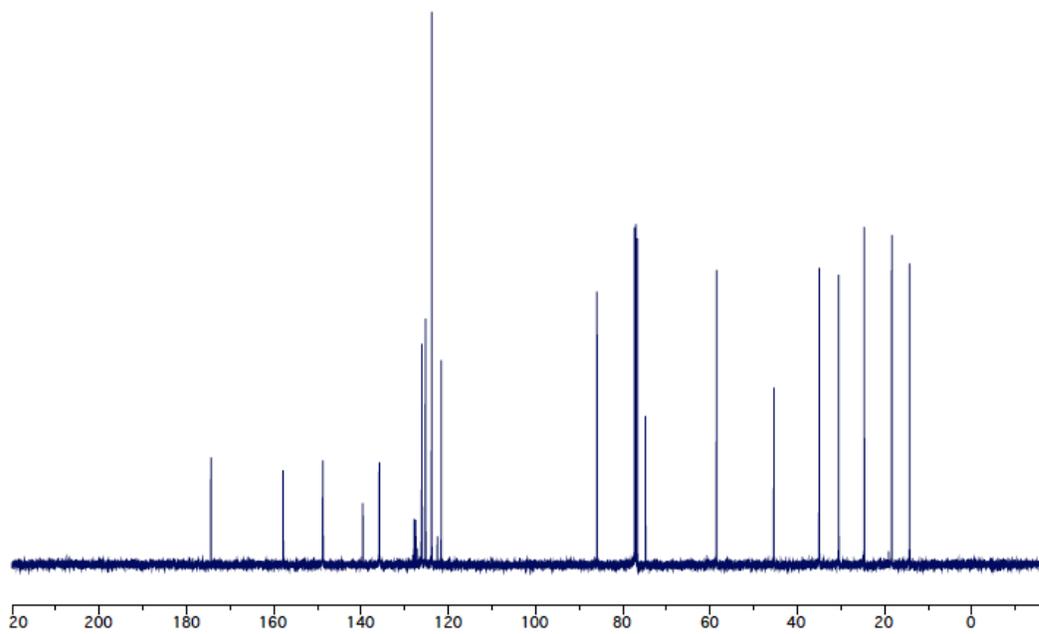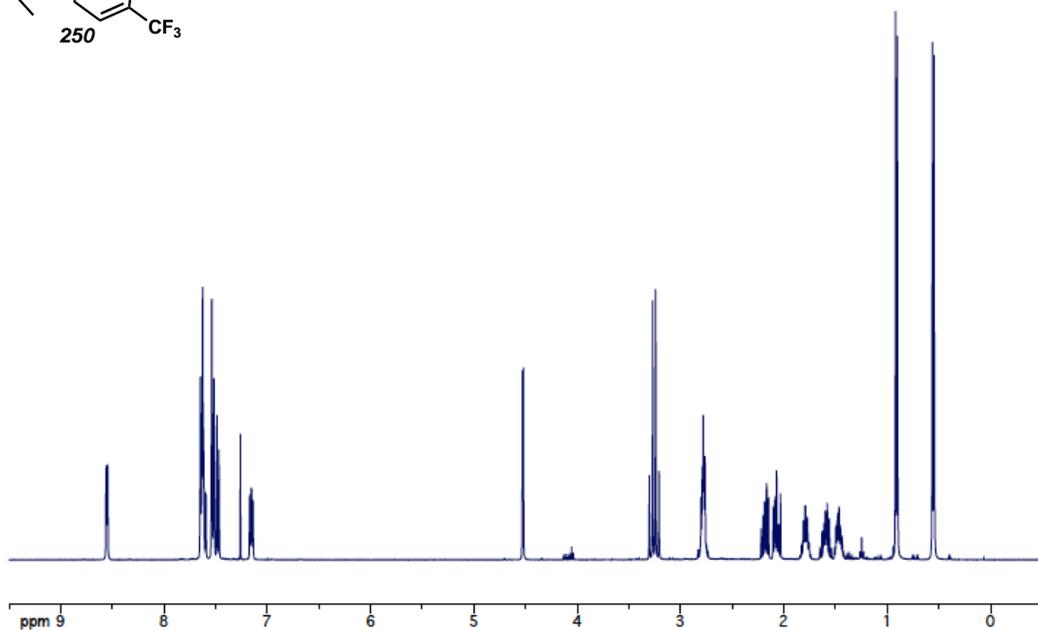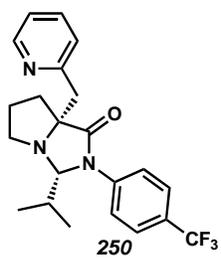
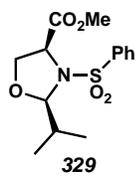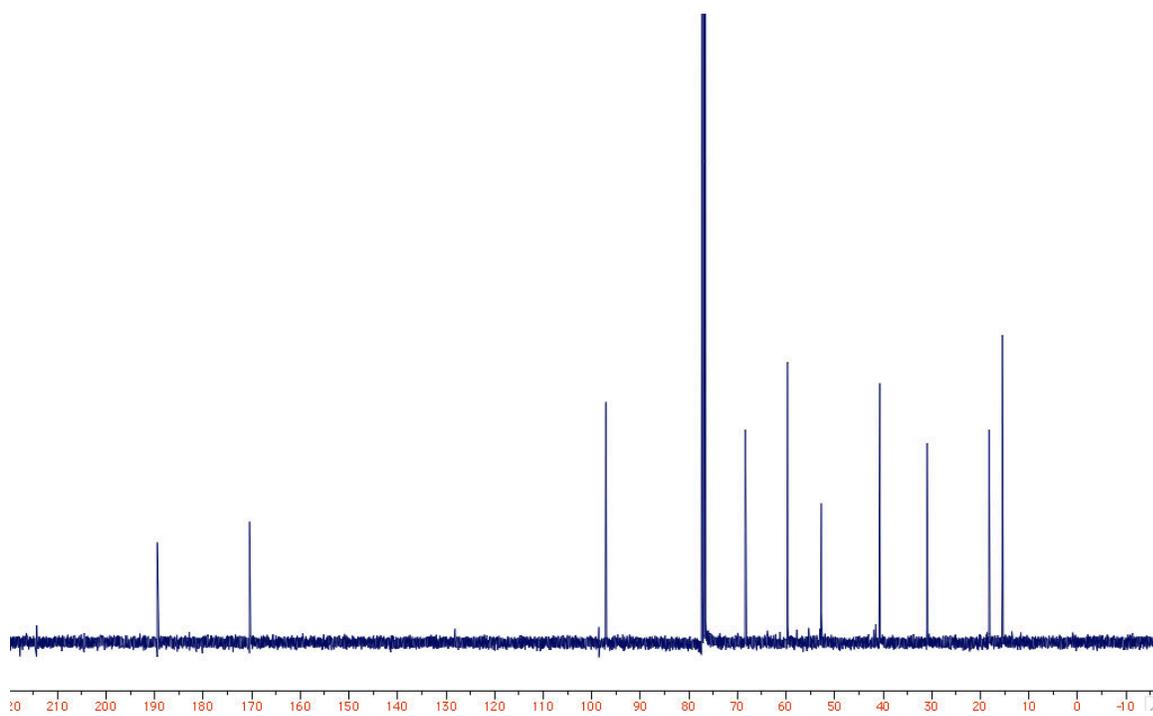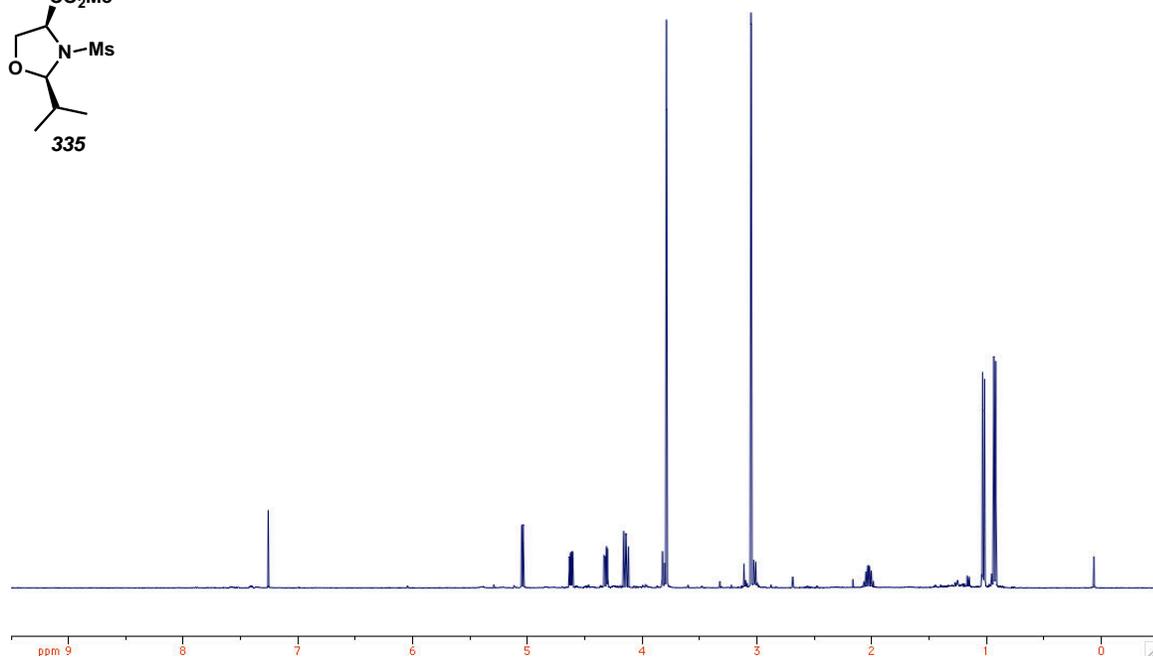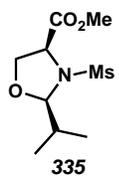# Appendix 3: Spectra Associated with Chapter 3

262

300

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Ts

315

326
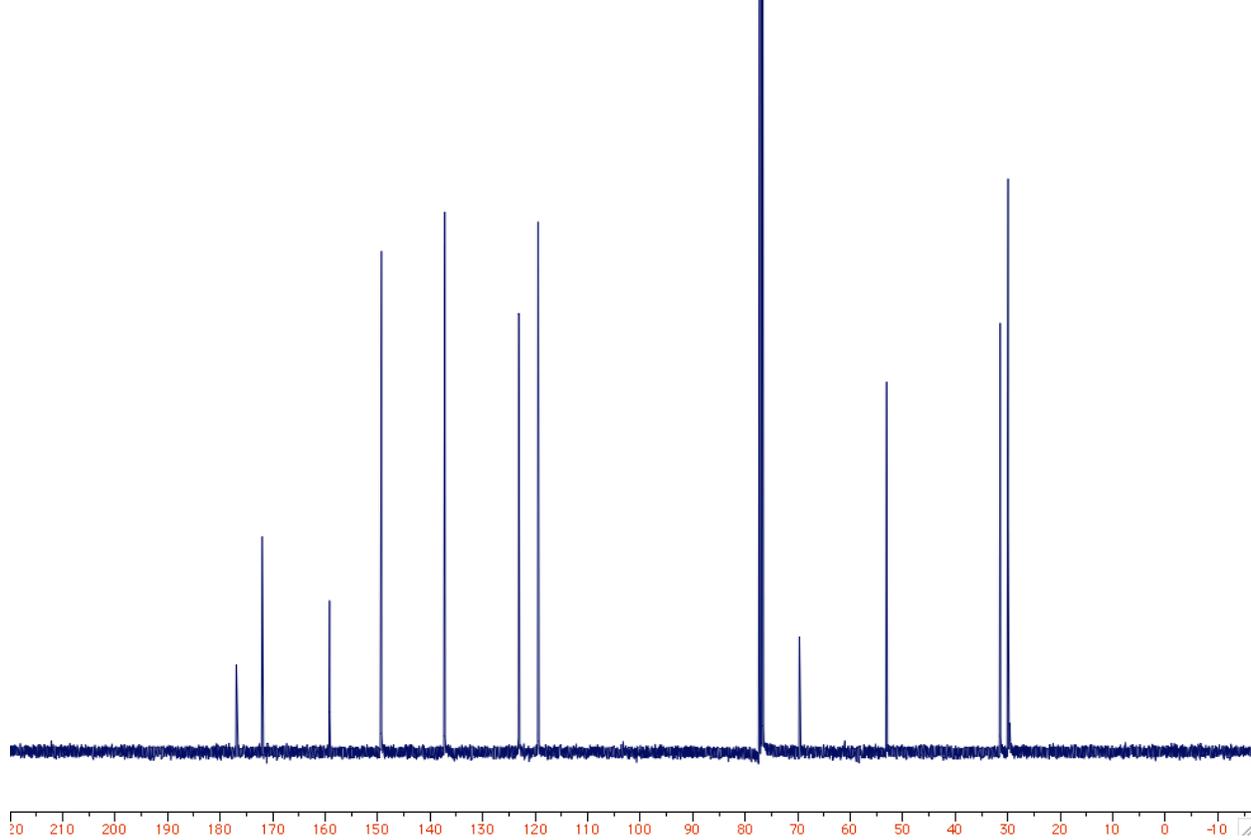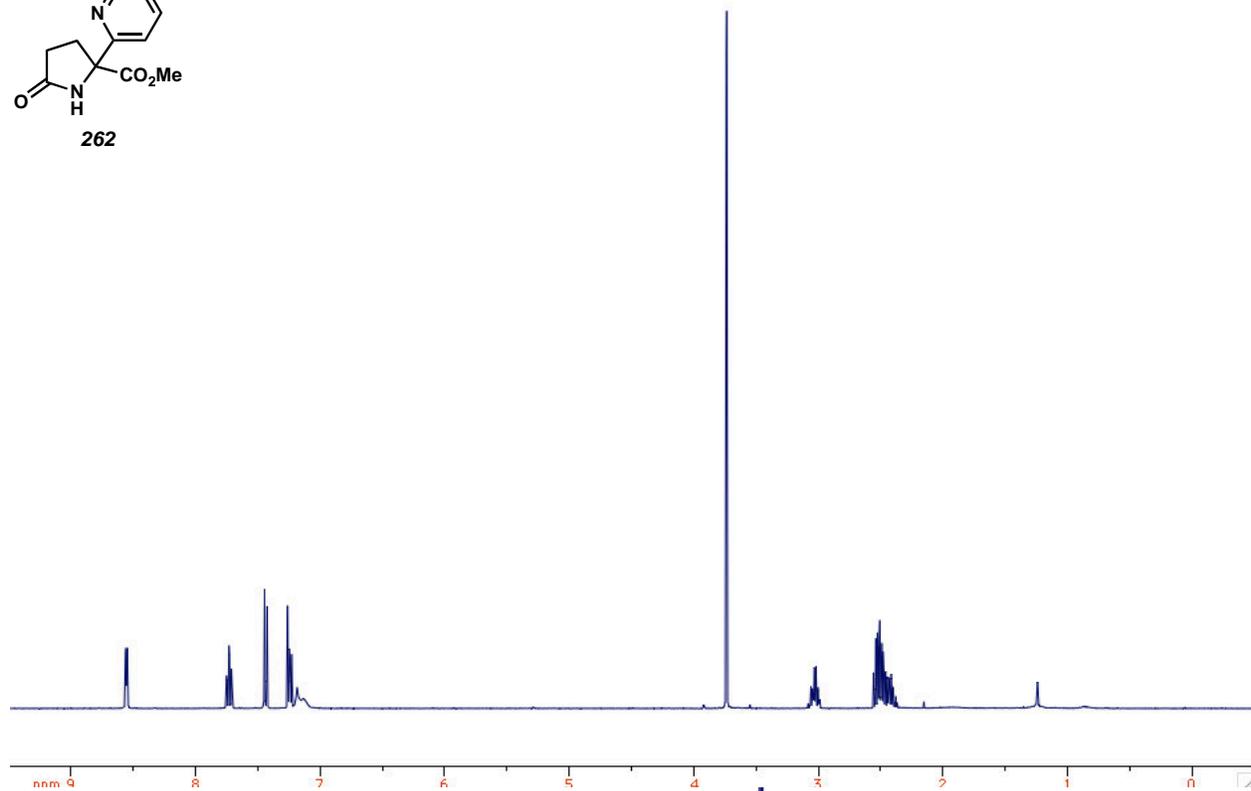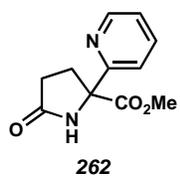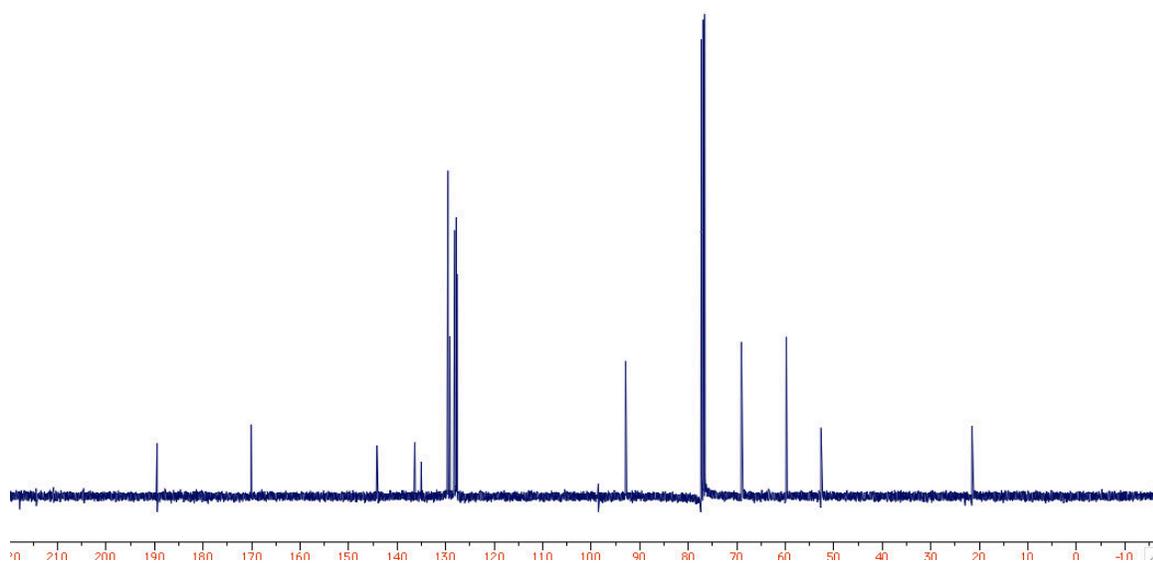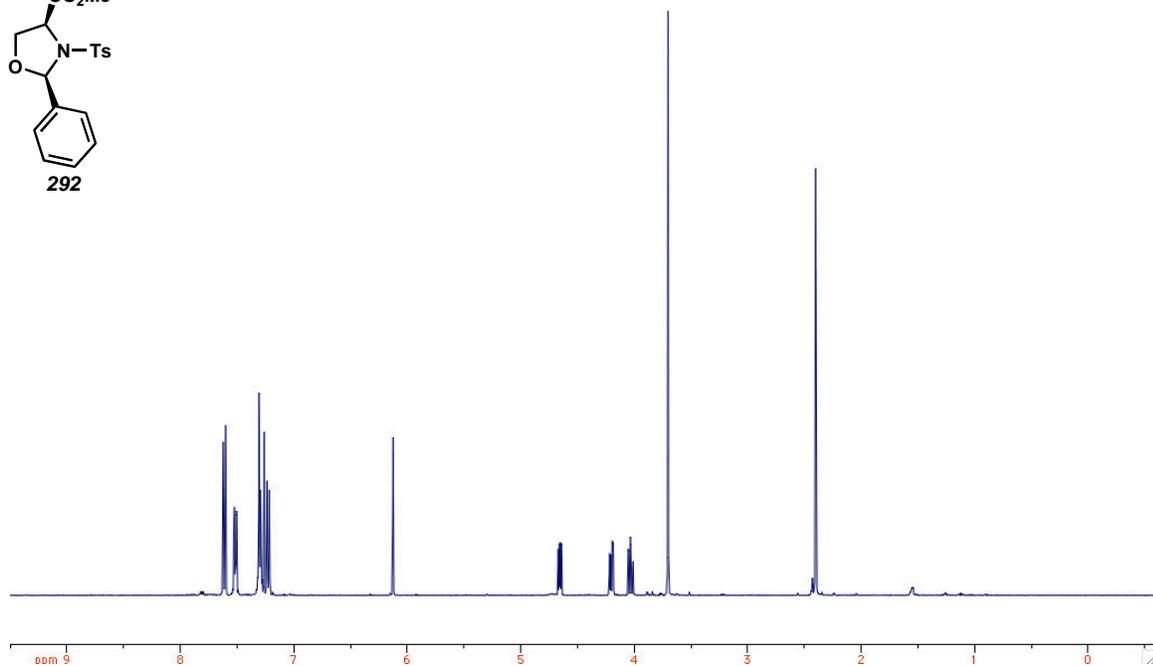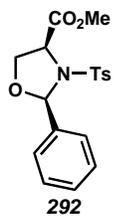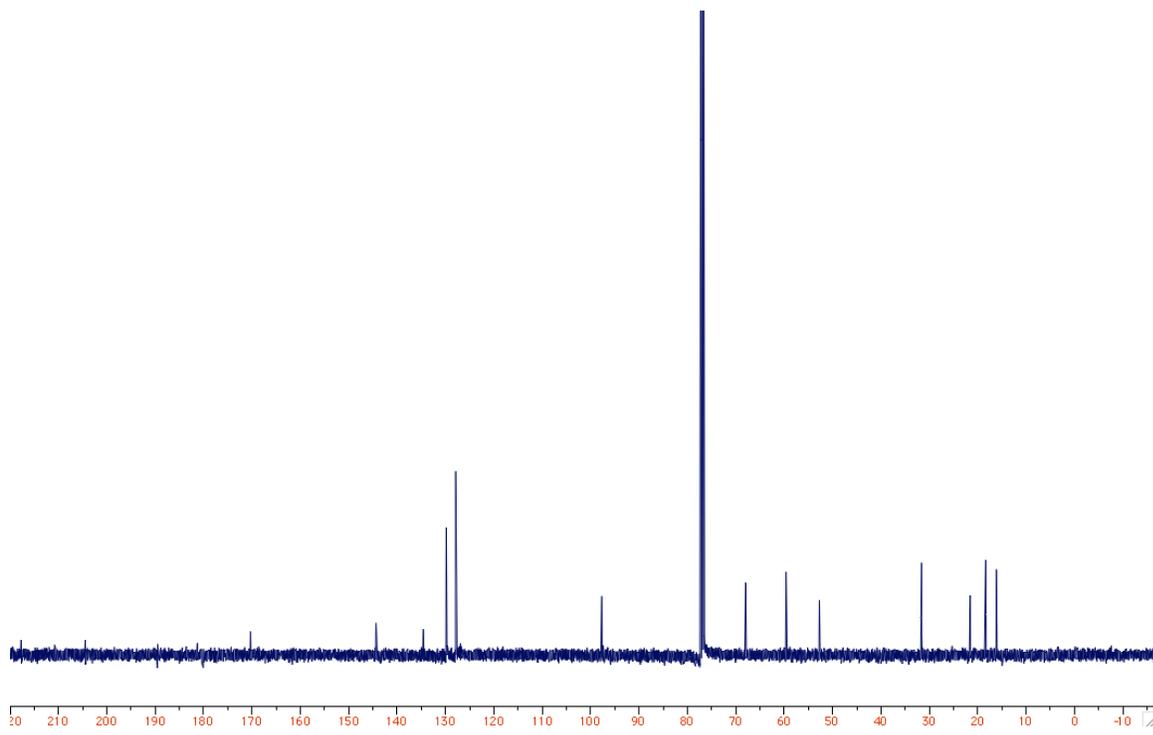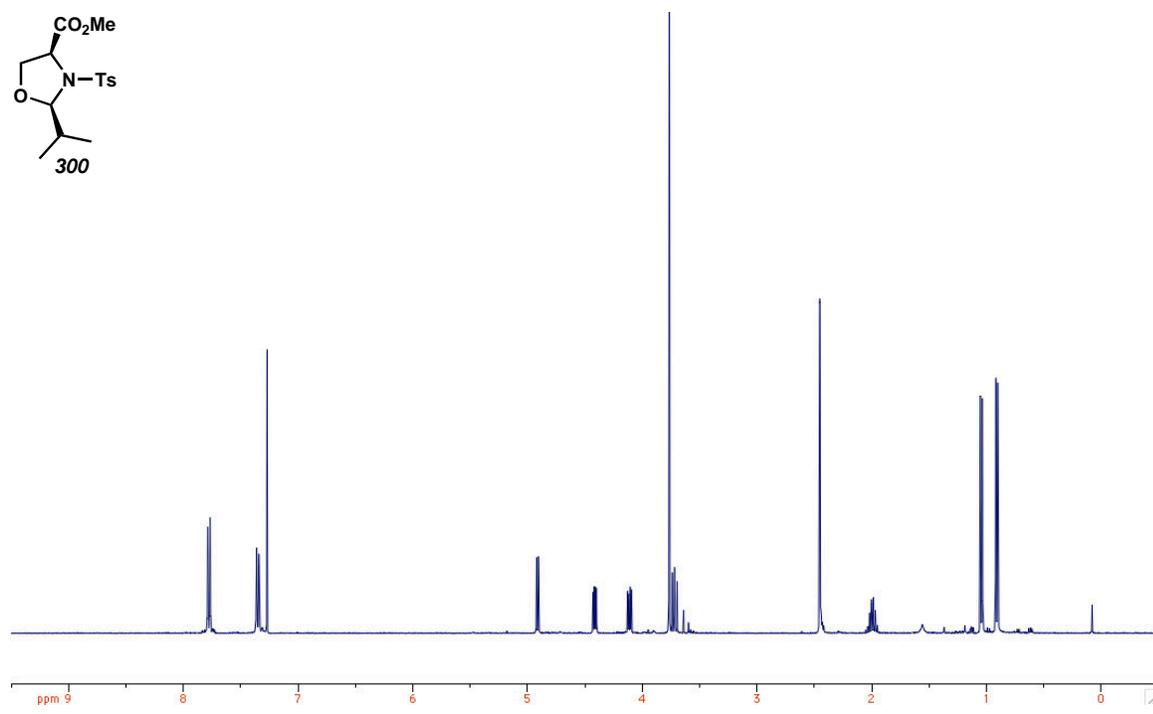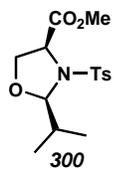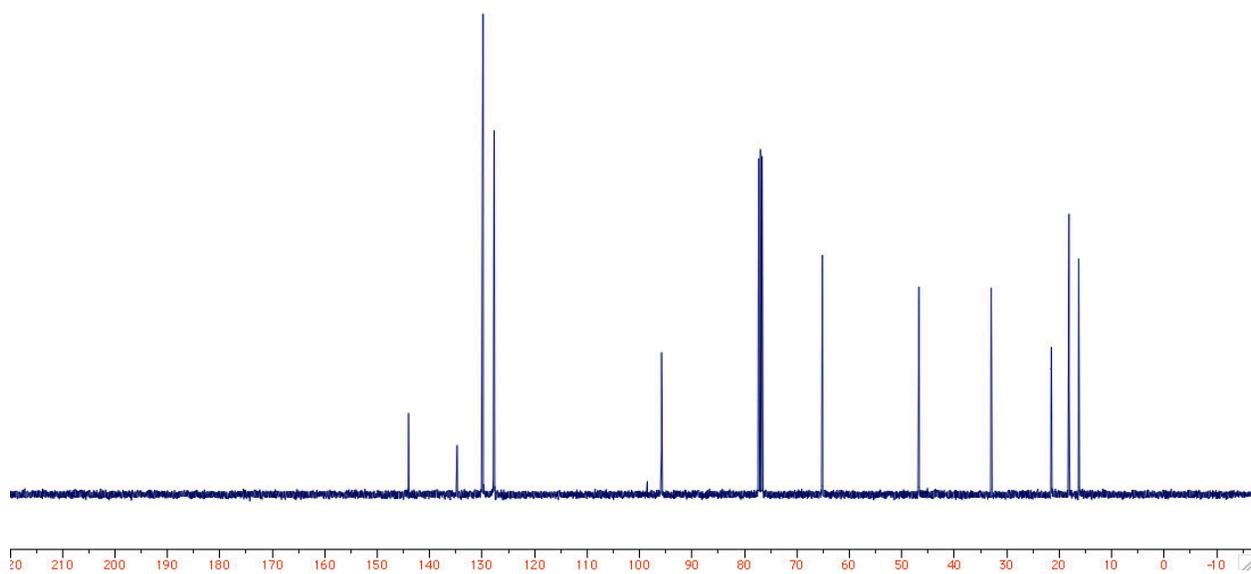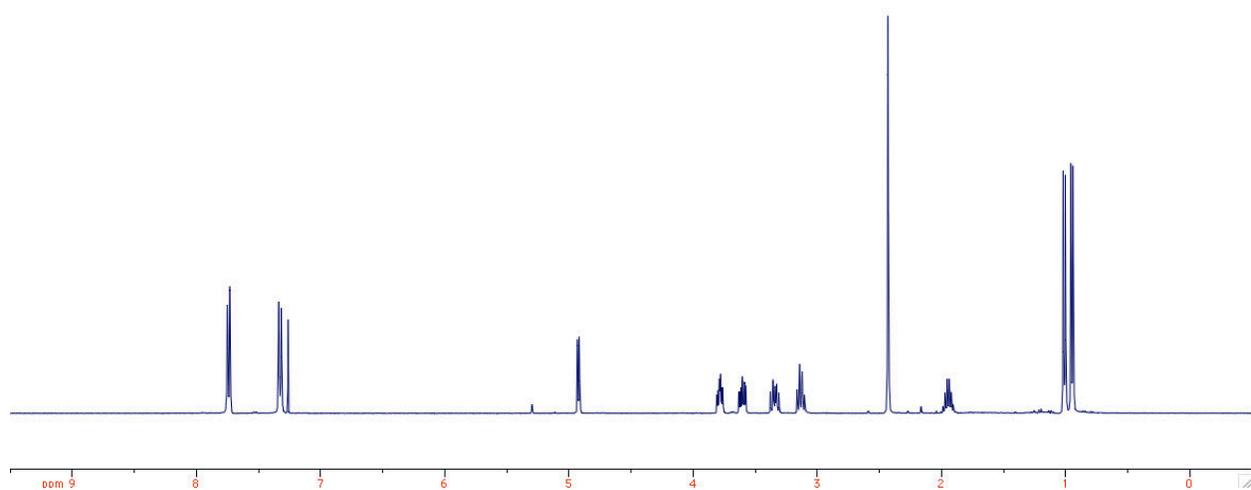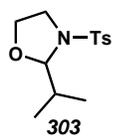
335
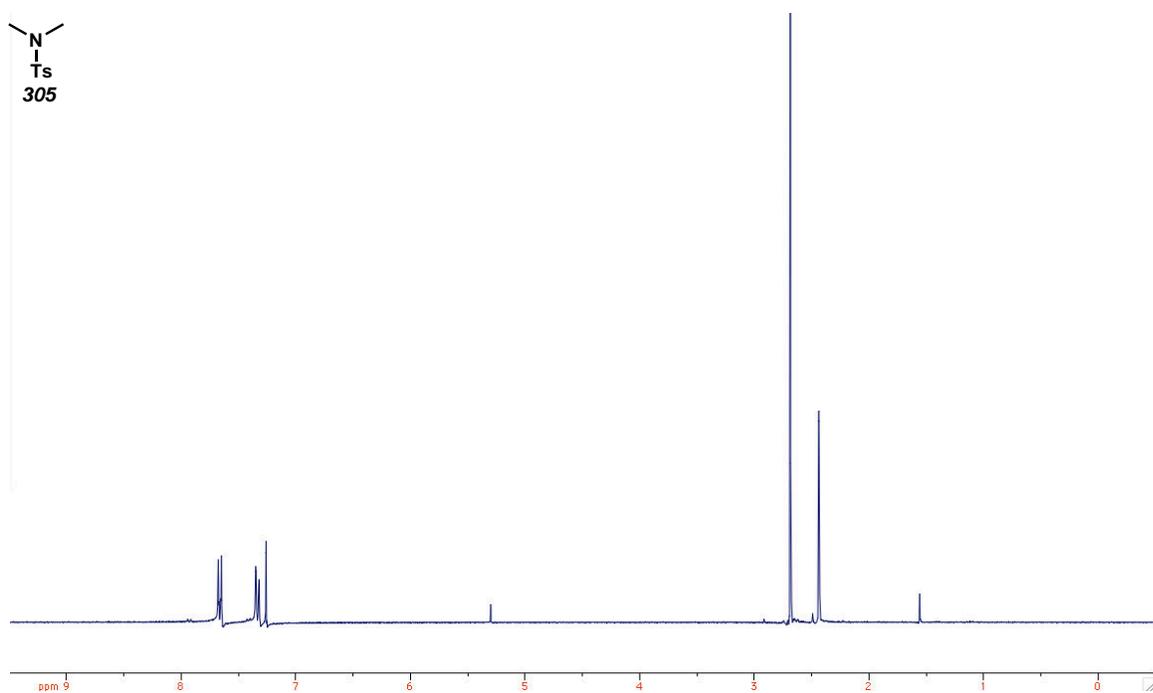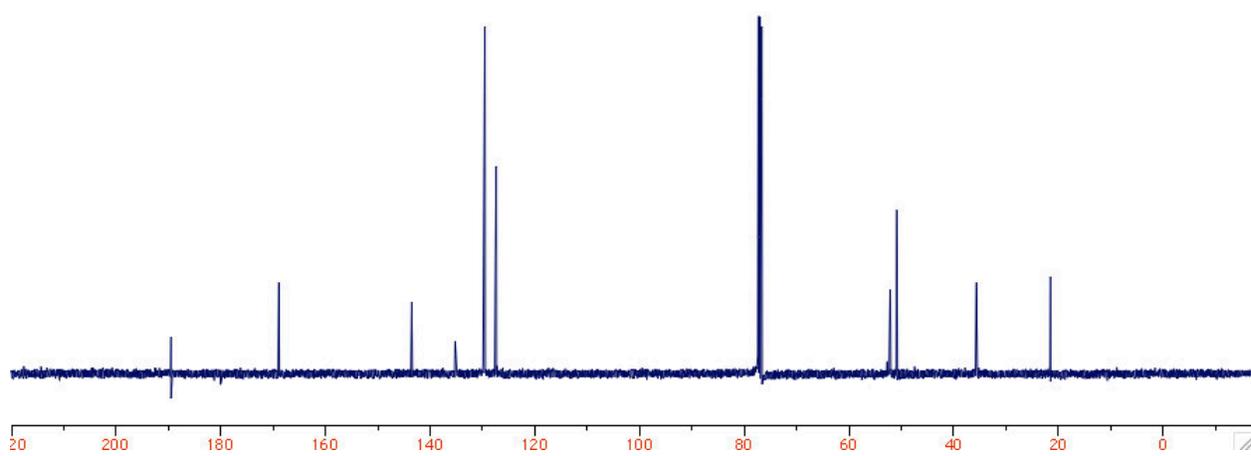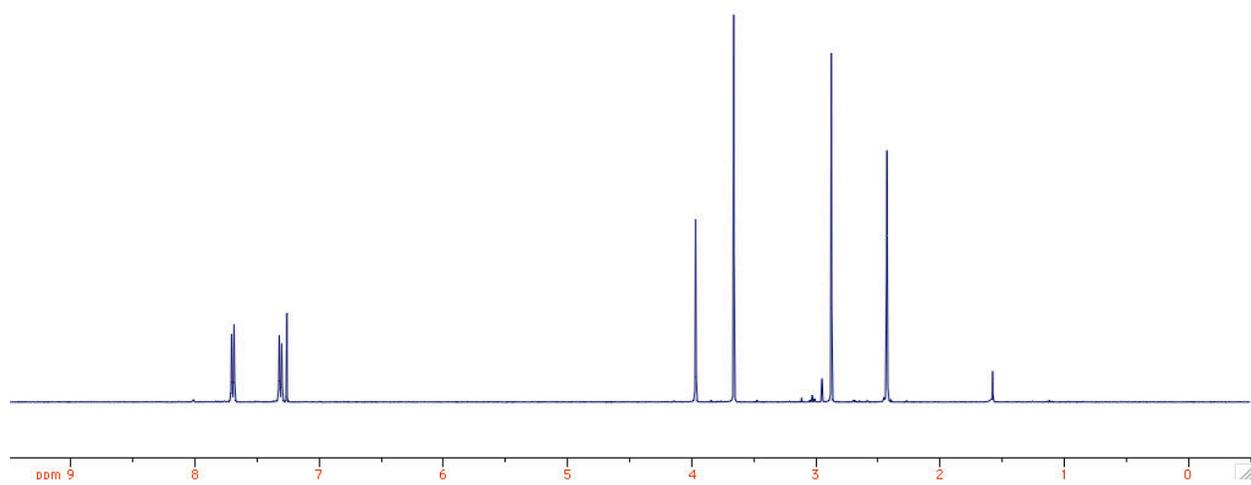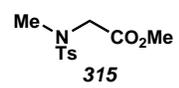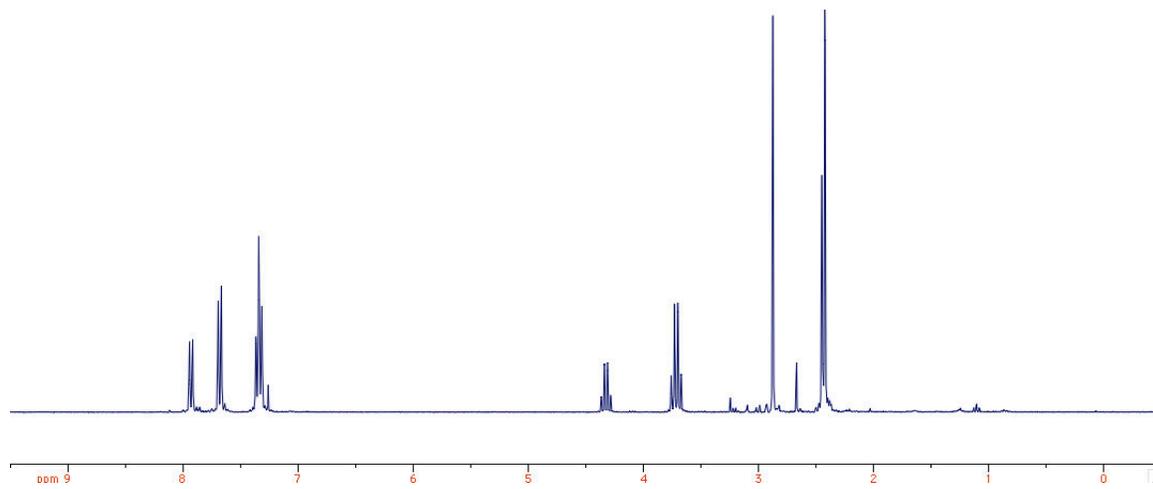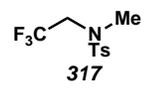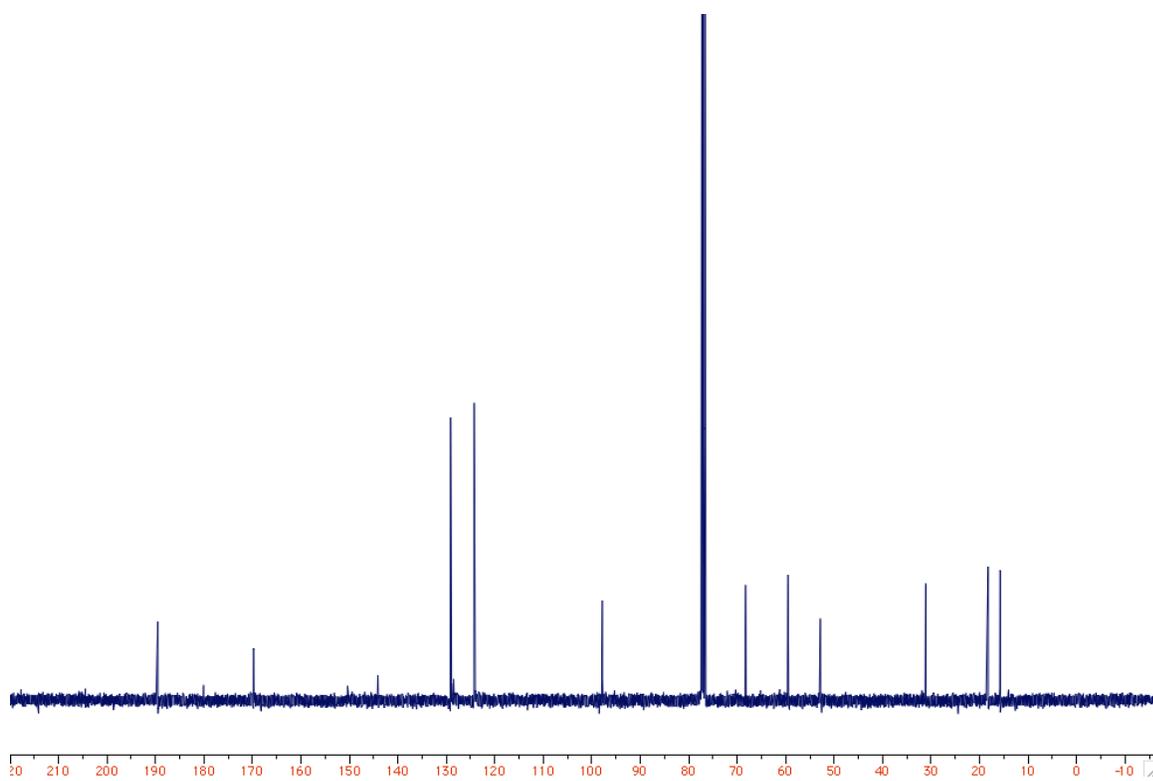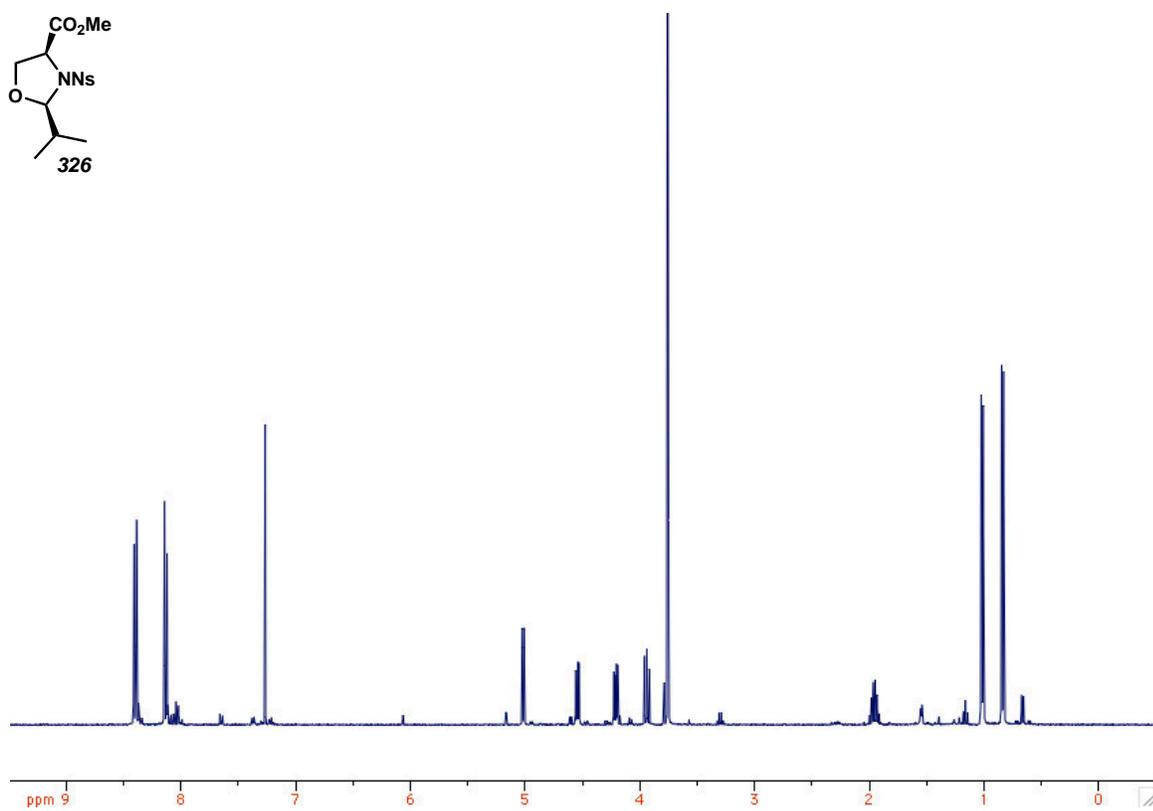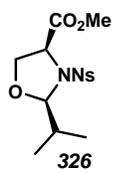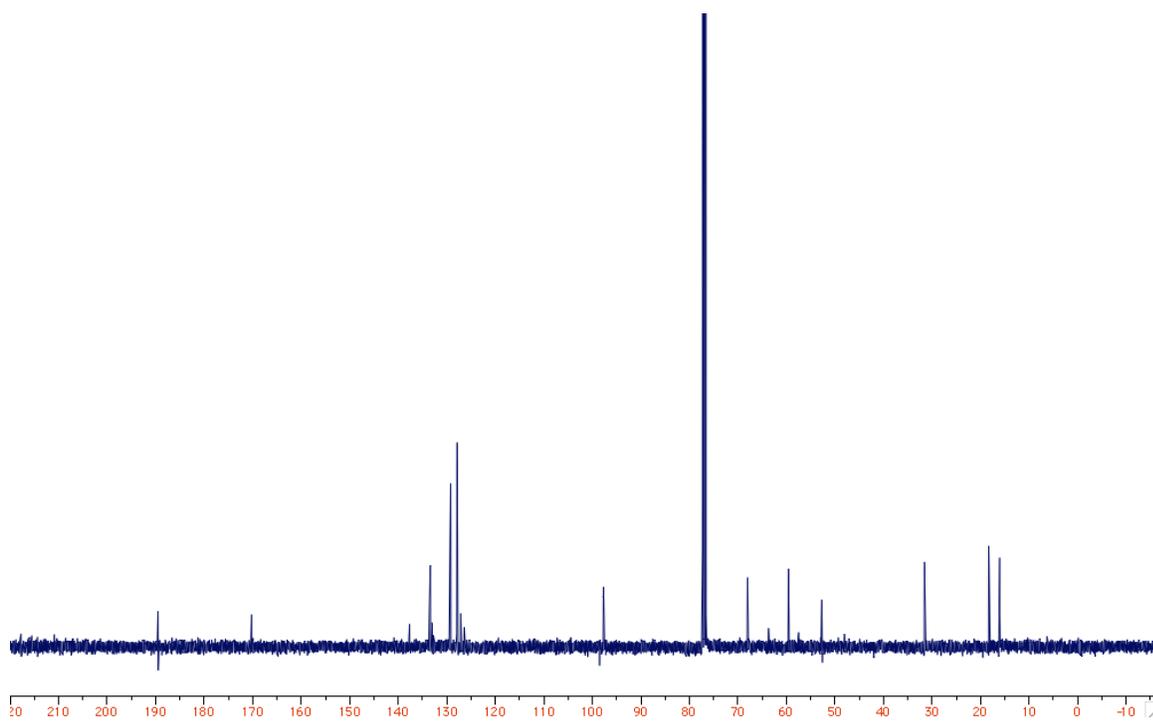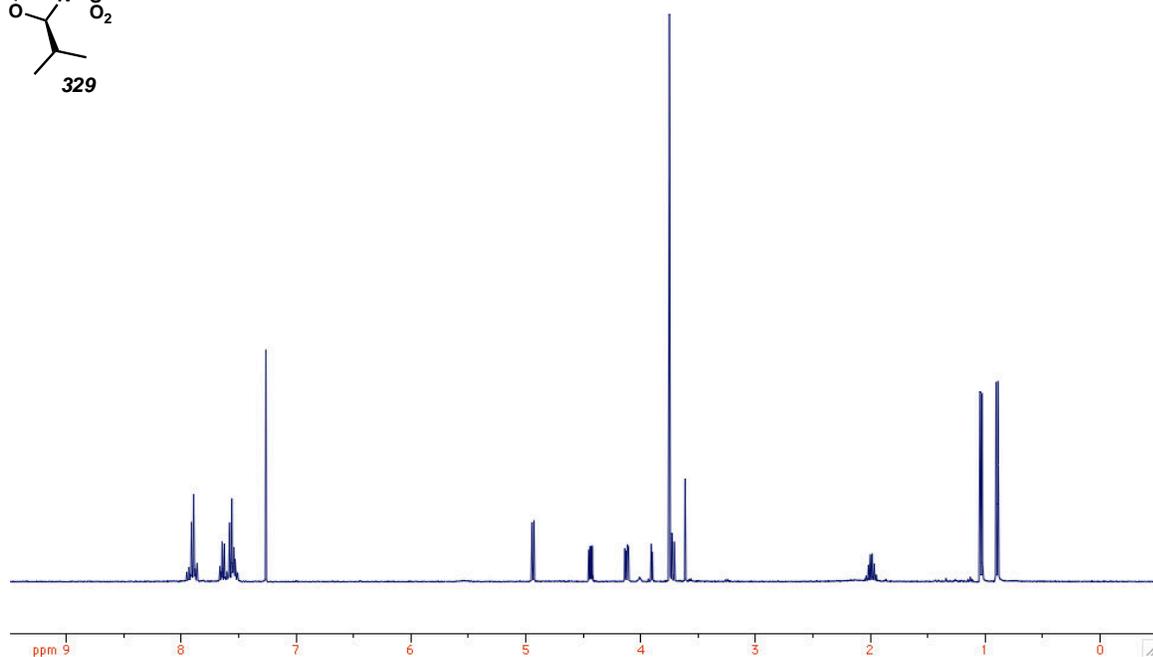
Appendix 4: Spectra for Appendix 1

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