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PARALLEL ALGORITHMS FOR TIME DEPENDENT DENSITY FUNCTIONAL THEORY IN REAL-SPACE AND REAL-TIME

A Dissertation Presented

by

JAMES KESTYN

Submitted to the Graduate School of the University of Massachusetts Amherst in partial fulfillment of the requirements for the degree of

DOCTOR OF PHILOSOPHY

SEPTEMBER 2018

Electrical and Computer Engineering

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PARALLEL ALGORITHMS FOR TIME DEPENDENT DENSITY FUNCTIONAL THEORY IN REAL-SPACE AND REAL-TIME

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JAMES KESTYN

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ABSTRACT

PARALLEL ALGORITHMS FOR TIME DEPENDENT DENSITY FUNCTIONAL THEORY IN REAL-SPACE AND REAL-TIME

SEPTEMBER 2018

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Directed by: Professor Eric Polizzi

Density functional theory (DFT) and time dependent density functional theory (TDDFT) have had great success solving for ground state and excited states properties of molecules, solids and nanostructures. However, these problems are particularly hard to scale. Both the size of the discrete system and the number of needed eigenstates increase with the number of electrons. A complete parallel framework for DFT and TDDFT calculations applied to molecules and nanostructures is presented in this dissertation. This includes the development of custom numerical algorithms for eigenvalue problems and linear systems. New functionality in the FEAST eigenvalue solver presents an additional level of distributed memory parallelism and is used for the ground state DFT calculation, allowing larger molecules to be simulated. A parallel domain decomposition linear system solver has also been implemented. This approach uses a Schur complement technique and a combination of direct sparse solvers to outperform black box distributed memory solvers in both performance and scalability. All other aspects of the code have been rewritten to operate in the domain decomposition framework and have been parallelized using both MPI and OpenMP. Numerical experiments demonstrate that our all-electron code can be applied to systems containing up to a few thousand atoms.

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CHAPTER 1 INTRODUCTION

1.1 Motivation

The ability to control materials and understand their properties has been a driving force for technological breakthroughs over the course of human history. As understanding has progressed so have the tools used to facilitate scientific discovery. First principle calculations offer a unique approach to study materials starting directly from the mathematical equations that describe physical laws and do not require any empirical parameters aside from fundamental constants. They are known as electronic structure calculations when applied to the configuration of electrons in a molecule or solid which determine most of the physical properties of matter through chemical bonding. Fundamentals laws governing the physics have been known since the beginning of the 20^{th} century with the development of quantum mechanics. However, the difficulty to solve these problems scales exponentially with the number of electrons. Recent progress to improve calculations of the many-electron wave functions can be broadly separated into three categories: (i) physical modeling to reduce the complexity of the full solution while keeping much of the important physics, (ii) discretization and mathematical modeling that transform physical equations into the language of linear algebra, and (iii) computing and numerical algorithms to solve the resulting problems.

Physical Models: Exact solutions to the many-body Schrödinger equation can only be computed for the simplest molecular systems. This has lead to different treatments and approximations to reduce complexity of the full many-body problem. These include Hartree-Fock [186, 187, 89, 72] and other post Hartree-Fock approaches (notably coupled-cluster [29] and configure interaction [183]) and density functional theory [96, 120]. Quantum Monte Carlo [73] has also been used extensively. All of these methods are approximate, with varying degrees of accuracy. This presents trade offs between accuracy and computational complexity; the most accurate of the approaches above can only be applied to small atomic systems containing a few atoms.

Although density functional theory (DFT) is not as accurate as traditional quantum chemistry techniques that calculate the many-electron wave function as a Slater determinant expansion of single particle states, it is the method of choice when dealing with moderate sized systems containing more than a handful of atoms [217, 85]. The work in this thesis will focus exclusively on DFT, which has been widely applied with great success in the solid state physics [119], material science, chemistry [117], and biology [57]. DFT uses the electron density as the fundamental variable, which in principle determines all ground state properties of a system. Many physical quantities can be determined using DFT and one can calculate, for example, total energies, band structures [80], molecular geometries [28, 148], static polarization, vibrational frequencies [192], and potential energy surfaces. DFT is also used to calculate input parameters for more advanced excited states models and for molecular dynamics simulations (e.g. see 2013 Nobel Prize in Chemistry), which have become vital in drug development. Hybrid molecular dynamics approaches using DFT for electronic degrees of freedom and classical dynamics to treat the motion of the nuclei have also been developed [42].

Starting in the 1980s, the formulation of time-dependent DFT (TDDFT) by Runge and Gross [173] has allowed for excited-states calculations and the study of nonequilibrium physics of atoms, molecules and nanostructures [38, 71, 185]. Similar to DFT, where observables of the ground state system can be calculated using knowledge of the electron density, TDDFT uses the time-dependent electron density as the fundamental variable. It is closely related to the field of spectroscopy and can be used to accurately model optical absorption and emission spectrum in the X-ray, ultraviolet and infrared frequency ranges [102, 101, 142]. There has been interest in using TDDFT to study and develop terahertz electronic devices [43, 206] because of its ability to describe electron dynamics at the attosecond to picosecond time scales. Additionally, TDDFT has been applied to calculate other physical observable of the time-dependent Hamiltonian, such as excitation energies [160] and complex permittivities, bond breaking and time-dependent modulations [39], as well as other non-linear phenomena such as ionization due to laser pulses [204, 205, 207]. Multiple approaches exist and many quantum chemistry codes incorporated linear response TDDFT through the evaluation of the Casida eigenvalue problem [44] in frequency space. However, it is also possible to use density functional perturbation theory (Sternheimer equation) [193, 83, 82, 19] or a real-time propagation of the groundstate DFT wave functions [216, 146].

Discretization and Mathematical Models: In order to numerically compute the solution to partial differential equations, such as the Schrödinger equation, it is necessary to discretize the problem by expanding the solution as a linear combination of basis functions. This will result in a matrix equation that can be solved for the coefficients of the expansion. Most commercial or academic electronic structure codes use either a plane wave [11, 8, 1] or a linear combination of Gaussian-type orbitals (LCGTO) [3, 9] basis set. Others [5, 4] even offer both. A comprehensive list can be found in [12]. Plane waves have traditionally been used within the solid-state physics community because of their natural periodicity that can be easily applied to crystal structures. However, this can be cumbersome when dealing with finite systems where the computational domain must be made much larger than the molecular size to ensure interactions due to neighboring unit cells are negligible. Additionally, they must use pseudopotentials to mimic the effects of core electrons, which do not directly participate in chemical bonding and would require a very large number of plane waves to capture their high-frequency variations. On the other hand, LCGTO uses an expansion of local functions around each atom and is the standard method in quantum chemistry for finite-systems. These codes can include all-electron effects (i.e. directly including effects of core electrons), but require a fine-tuned basis set to produce accurate results. A major drawback to both plane wave and LCGTO is the resulting system matrix. Plane waves produce a fully dense Hamiltonian. LCGTO can be formulated as a linear scaling method by explicitly setting the atomic orbitals to zero at a certain cutoff radius. This will neglect the overlap of basis functions between certain atoms and cause many of the matrix elements to be zero. However, a limitation of both LCGTO and plane wave approaches are the dense system matrices which can be too large to fit on disk. This necessitate the use of less robust iterative techniques requiring only matrix-vector multiplication. Furthermore, with a plane wave basis each matrix vector product must Fourier transforms between the frequency domain and real-space, where the potential is diagonal. Overall, both of LCGTO and plane waves are limited in their ability to scale problem sizes and simulate large scale molecules containing many atoms.

The other possibility is a real-space method using wavelets [2], finite differences [6, 7, 36], or finite elements [194]. The local-in-space basis functions have a non-zero overlap with a small fraction of neighbors and produce large sparse system matrices. This is a major advantage that enables the use of domain decomposition approaches. It is ideal for simulating large-scale molecules on modern computing architectures because of the additionally level of distributed-memory parallelism non-existent in plane wave and LCGTO codes. Real-space methods have traditionally been used with pseudopotentials. However, the finite element method allows for mesh refinement at arbitrary locations (i.e. around the atomic nucleus) and can be used for all-electron calculations [129].

Additional mathematical modeling questions are specific to the physical problem at hand. For example, the non-linear eigenvalue problem of DFT can either be computed directly [78] or, more commonly, using self-consistent-field iterations combined with a density mixing scheme. In fact, solution to the eigenvalue problem can be avoided entirely by evaluating select diagonal elements of the density matrix [136, 124]; although, this may be associated with a loss of accuracy since error in the numerical integration translates to the electron density and the SCF iterations converge to a perturbed result. Further modeling questions arise for solving the time dependent partial differential equation of TDDFT. Real-time TDDFT, where the ground state wave functions are propagated directly in time, must use an approximation to the propagation operator. The most common approach in TDDFT is to use the Crank Nicolson propagation operator [60]. An integral method based on diagonalizing the time-dependent Hamiltonian using FEAST has been previously investigated [52] and allows for larger time steps with more capacity for parallelization.

Computing and Numerical Algorithms: Much of the progress in this field is directly tied to advancements allowing larger and more complex systems to be simulated. Molecules with many atoms and electrons can only be treated by addressing the efficiency and scalability of algorithms or, more generally, with increasing processing power. The latter has consistently improved over the last 60 years and is summarized by the famous Moore's Law. However, new algorithms are necessary to take advantage of modern peta- and exascale architectures. These supercomputers are comprised of hundreds of thousands of processors and contain multiple hierarchical levels of parallelism. Multi-core CPUs, where each core has access to a common shared-memory, have allowed for near optimal scaling of many numerical linear algebra operations though threading and OpenMP [61] directives. Performance of each core has also increased with the advent of SIMD and AVX registers that can operate on multiple data elements within a single clock cycle. Traditional multicore CPUs contain at most a few tens of cores, but the emergence of GPUs and Intel's Xeon Phi accelerator has pushed this number into the hundreds; although each individual core is less powerful and the coding difficulty is drastically increased. However, to fully utilize the available processing power of modern machines one must move to distributed-memory, where many thousands of CPUs (or accelerators) are linked together via a high-speed interconnect and communicate through a standard message passing interface such as MPI [84].

Attaining high efficiencies for distributed memory codes can be challenging due to the cost of communication compared to computation. The core elements are numerical linear algebra routines that operate on large matrices and vectors. Solution to linear systems and eigenvalue problems represent the most demanding of these operations. Development of scalable algorithms for distributed memory is essential to performance. Efficient libraries exist for small to medium sized dense matrices and have become standardized in BLAS [126], LAPACK [17] and ScaLAPACK [33]. However, no such standard has emerged for large sparse matrices, where the matrix structure can vary drastically between applications. Packages such as PETSc [27] and SLEPc [91], Trillinos [92], and MATLAB have assembled many different algorithms, but lack flexibility for application specific problems and require software to be implemented in their specific framework. To develop scalable methods adapted to our own application it is necessary to develop custom numerical algorithms.

Domain decomposition methods have been well studied [115, 201, 188] and are a natural framework for applications with large sparse matrices. They can be applied at the discretization level, where separate meshes are generated for different regions of space, or directly to the matrix, generally with the use of a graph partitioner (e.g. see [109, 110]) to divide the elements into non-intersecting subsets. The partitioning depends on the structure of the mesh or matrix and is specific to the application. Distributed memory numerical algorithms must take into account the data distribution and are also application specific.

1.2 Objectives

The major goal of this work is to develop numerical algorithms and simulation software for electronic structure that scale system sizes up to a thousand atoms, without resorting to additional approximations beyond traditional DFT and TDDFT. The target architecture is a computing cluster (including high speed interconnect) with multiple shared memory nodes each containing 1 or 2 many core CPU. Our approach addresses the three categories outlined in Section 1.1, which are all interrelated and must be taken into account in order to develop a practical and efficient electronic structure code. The techniques will harness the processing capabilities of modern machines using domain decomposition and other advantages of a realspace finite element basis. Specific objectives can be broken down into three main categories. First, the development of parallel *numerical algorithms* to solve linear algebra problems that arise in DFT/TDDFT. Next, a parallel framework for DFT and TDDFT simulations built on top of the numerical algorithms, where multiple levels of MPI and OpenMP parallelism are used for all mesh and matrix operations. Finally, the execution of the new code to perform simulations of large scale molecules and push the current limits of all-electron DFT/TDDFT calculations.

Objectives:

- Numerical Algorithms: Develop novel numerical algorithms for major computational operations within DFT and TDDFT. This will include numerical methods for eigenvalue problems and solving linear-systems.
- **Parallel Framework for DFT and TDDFT Simulations:** Create a fully parallel and distributed code, where subroutines for DFT, TDDFT and auxiliary operations are performed without directly constructing the full wave function solution.
- Simulations of Large Scale Molecules: Perform linear-response real-time TDDFT simulations of organic molecules containing up to 1000 atoms. This will be achieved by optimizing the strong scalability of the code and addressing the weak scaling of the solver, where the simulation of larger molecules is made possible through the use of additional computing resources.

1.3 Contributions

Specific design decisions have evolved over many years starting prior to my involvement and continuing with my contributions. Previous developments to the NESSIE electronic structure code have been on the mesh/discretization [220, 218, 219, 132, 131], real-time TDDFT simulations [52, 54, 103, 168], and other aspects of the modeling process [78]. Much of this previous work is used, but the software has been upgraded to run in a fully parallel environment. This has been achieved rewriting the code to operate within a domain decomposition framework. In this case both the wave functions and Hamiltonian are distributed and all numerical algorithms and mesh operations must act directly upon the distributed vectors and matrices. Furthermore, custom numerical algorithms to compute eigenvalues and linear system solutions have been developed. All other aspects of the code have been parallelized using MPI and OpenMP.

The major bottleneck in DFT is the computation and storage of the DFT wave functions. These are dense $n \times m$ matrices with memory requirements that depend on the number of electrons m as $O(m^2)$ - both the number of discretization points and the number of needed eigenvectors increase with the number of atoms in the molecule. Even with tractable first-principle approach such as DFT, it is practically difficult to scale electronic structure problems. It becomes necessary to compute subsets of the eigenspectrum independently, reducing the number of vectors stored in the wave function matrix. The FEAST algorithm [164] offers this ability. NESSIE is built on top of the FEAST Eigenvalue solver [166], which is released as open source software and is included in Intel's Math Kernel Library [212]. A major part of this work is focused on both the FEAST algorithm and solver. A new development is PFEAST [113], a fully parallel extension of FEAST that can be linked with a distributed memory linear system solver, which will released in FEAST v4.0. The ability to distribute the eigenvector matrix by row across compute nodes presents an additional level of parallelism for strong or weak scaling. This brand new extension has been integrated in NESSIE and plays a key role in the distributed version of the code since it can be easily interfaced with a domain decomposition linear solver.

A Schur complement linear system solver, a standard approach in domain decomposition, has been implemented within the NESSIE muffin-tin decomposition framework. This is a parallel method where different domains can be solved independently by forming and solving a smaller global problem that accounts for the interactions between them (called the Schur complement). PFEAST, which must compute the multiple complex (not Hermitian) linear systems with many right-hand-sides, has been linked with the linear solver and integrated within NESSIE. NESSIE is only using local exchange and correlation functionals and the other significant computation of our DFT approach is the Poisson equation, which amounts to solving a real symmetric-positive-definite linear system with one right-hand-side. Because the Poisson equations has only a single right-hand-side, it is much less expensive than the linear solves of the eigenvalue computation and the same domain decomposition approach is used. In fact, the computation of the boundary condition for Poisson is actually more time consuming, but is very scalable. For time dependent calculations using the Crank Nicolson scheme [60], a complex linear systems must be solved at each time step to propagate the wave functions in time and the same exact solver linked to PFEAST is used.

Other routines that act on the finite element mesh must also be performed in parallel and be customized to the domain decomposition. These operations can become the bottlenecks if run on a single thread or if not properly optimized. The most computationally demanding of these routines are to (1) add potential to the Hamiltonian using numerical integration and the finite element basis, (2) reconstruct the electron density from ground state wave functions and (3) perform matrix multiplication. All of these operations mainly involve loops over the mesh/matrix indices and have been parallelized using both MPI and OpenMP.

Further developments in FEAST include the creation of a robust non-Hermitian eigenvalue algorithm and software extension [114], which can be much more difficult than the Hermitian case. Although the non-Hermitian routine is not currently used within NESSIE, it is still of importance for real-space DFT and TDDFT simulations since it can be used with complex symmetric matrices resulting from absorbing boundary conditions. Non-Hermitian FEAST can be used in TDDFT with absorbing boundary conditions for simulating charge transfer and ionization. Both transport calculations and the identification of resonant states can also produce non-Hermitian matrices [47]. Importantly, it also generalizes the FEAST algorithm to all linear cases.

Details on the modeling framework can be found in Chapter 2, which provides background on the physical model (DFT and TDDFT), the discretization (finite element combined with a muffin-tin domain decomposition) and numerical algorithms for solving linear algebra problems (mainly for eigenvalues and linear systems). Chapter 3 provides a step-by-step example for running the code starting with the mush generation and then discussing details for DFT and TDDFT calculations. Results are then presented in Chapter 4 for a collection of small molecules as well as a more detailed investigation of (3,3) carbon nano-tubes up to forty unit cells. Further discussion on the performance and scalability of the code can be found in Chapter 5.

CHAPTER 2

MODELING FRAMEWORK

2.1 Overview

Our electronic structure code uses a real-space finite element discretization and domain decomposition to perform all-electron ground-state DFT and excited-states TDDFT calculations in parallel over many distributed memory compute nodes. Custom numerical algorithms have been developed for the eigenvalue problems and linear systems, which are the major linear algebra operations within the software. The following summarizes the topics that will be discussed in this Chapter.

• Discretization: A real-space finite element discretization is associated with many benefits outlined in Section 1.1. It is well suited for all-electron calculations that require more accuracy around the atomic nucleus since the mesh can be arbitrarily refined. A muffin-tin domain decomposition is used to separate the computational domain into two sets: the atomic regions that directly surround each atom and the interstitial connecting mesh. The domain decomposition allows for each atomic mesh/matrix to be operated on in parallel. The full interstitial mesh is known on each MPI process. However, the interstitial matrix is not local to a single MPI process as with a single atom. The matrix is instead distributed across available computing resources and requires communication to perform linear algebra operations.

• Ground state calculation: With DFT, many eigenvalue problems must be solved in a self consistent loop. Self consistency is reached once the input electron density is equal to the output density, which can be recalculated from the eigenvector solutions (wave functions). At each self consistent field (SCF) iteration a new input electron density for the next iteration must be computed. The simplest approach is a simple mixing scheme where a linear combination of the current density and the previous density is used. We have used the Anderson mixing scheme [16], which takes a linear combination of past densities. As the density begins to converge the previous eigenvector solutions become a very good initial guess for the current SCF iteration. Since the eigenvalue convergence criteria is set to only slightly exceed the current SCF convergence, PFEAST must only perform a single subspace iteration on average and, with parallelism, solve a single linear system per diagonalization.

• Excited states calculation: A real-time TDDFT approach is used to time propagate ground state wave functions. This requires, as a first step, a ground state calculation. The propagation uses a Crank Nicolson scheme [60], where a complex linear system with many right-hand-sides must be solved at each time step. This proceedure can require thousands of time steps, but is more scalable than frequency space methods.

2.2 Physical models

Since the exact solution to the Schrodinger equation can not be computed exactly, a physical model must be used to reduce the complexity. Many techniques, each with benefits and drawbacks, have been developed. In this work density functional theory (DFT) and time-dependent density functional theory (TDDFT) are considered.

2.2.1 Density functional theory

The development of density functional theory, which won Walter Kohn the Nobel prize in Chemistry in 1998 [118], was a major advance in the attempt to solve the many body Schrödinger equation. It is an exact theory, although in practice approximations must be made to include many body effects. The general idea is to use the electron density, instead of the position of each electron, as the fundamental parameter. The many body Schrödinger equation with the many electron wave function $\Psi(r_1, r_2, ..., r_{N_e})$ is intractable for all but the smallest molecular systems. Here, r_i represents the spacial coordinates for the i^{th} electron and all other quantum numbers; although in DFT, spin is not necessarily included. DFT does not solve the many body problem or compute the many-electron wave function. The problem is reformulated so that "fictitious" wave functions $\psi(r)$ of the single particle Schrödinger equation are computed instead. This drastically reduces the degrees-of-freedom for the problem allowing larger molecular systems to be studied.

2.2.1.1 Hohenberg Kohn Theorem

The basis of DFT is the Hohenberg Kohn formalism [96], which provides a proof of one-to-one correspondence between the electron density and external potential. Essentially this shows that the Hamiltonian can be written solely as a functional of the electron density. The result is a variational problem that can be solved by a minimization of the total energy.

The Hamiltonian for a set of electrons moving in an external potential is given by

$$\hat{H} = \hat{T} + \hat{U} + \hat{V}, \qquad (2.1)$$

where \hat{T} represents the kinetic energy of the electrons and \hat{U} the Coulomb interaction between the electrons (we use ' $\hat{}$ ' since they are operators). The third term \hat{V} is the external potential and can include, for example, the Coulomb interaction between the negatively charged electrons and positively charged nuclei.

The Hohenberg Kohn Theorem [96] shows that there is a one-to-one correspondence between the electron density n(r) and the external potential \hat{V} . The groundstate electron density can be generated by at most one potential. Likewise this potential will generate only a single unique electron density. For the many body Hamiltonian, the solution to the stationary Schrödinger Equation

$$\hat{H} |\Psi_0\rangle = E_0 |\Psi_0\rangle \tag{2.2}$$

will give the ground state wave function $|\Psi_0\rangle$ and the energy eigenstate E_0 . The many body Hamiltonian is given by

$$\hat{H} = \left(-\frac{\hbar^2}{2m}\sum_{j=1}^{N_e}\sum_{j'=1}^{N_e}\nabla_j \cdot \nabla_{j'}\right) + \left(\frac{1}{2}\sum_{\substack{j=1\\j'\neq j}}^{N_e}\sum_{\substack{j'=1\\j'\neq j}}^{N_e}\frac{e^2}{|r_j - r_{j'}|}\right) - \left(\sum_{j=1}^{N_e}\sum_{l=1}^{N_e}\frac{Z_le^2}{|r_j - R_l|}\right), \quad (2.3)$$

with fundamental constants \hbar , m and e. The summations are over the number of electrons N_e and number of atoms N_{at} . Since the Born-Oppenheimer approximation is used, which assumes the atomic nuclei are fixed in space and obey classical mechanics, the kinetic energy term only sums over the electrons and there is no nucleus-nucleus interaction. The electron density n(r) can be calculated from the diagonal elements of the density matrix $|\Psi_0\rangle \langle \Psi_0|$ or by applying the density operator,

$$\hat{\rho} = \sum_{i=1}^{N_e} \delta(r - r_i), \qquad (2.4)$$

to the wave function:

$$n(r) = \langle \Psi_0 | \hat{\rho} | \Psi_0 \rangle = N_e \int |\Psi_0(r, r_2, ..., r_{N_e})|^2 dr_2 ... dr_{N_e}.$$
(2.5)

Here, $\delta(r - r_i)$ is the Dirac delta function and the many body wave functions $|\Psi_0\rangle$ depends on the coordinates $\{r_i\}$ of all N_e electrons in the system. The final result in

(2.5) can be written without a summation since electrons are fermions and exchanging any two arguments of $|\Psi_0\rangle$ will change its sign,

$$\Psi_0(r_1, r_2) = -\Psi_0(r_2, r_1), \tag{2.6}$$

but not its magnitude. The electron density therefore depends also on the external potential \hat{V} that defined \hat{H} though its solution $|\Psi_0\rangle$.

The reverse must also be true for a one-to-one correspondence between n(r) and \hat{V} : the external potential \hat{V} must be a unique functional of the electron density n(r). This is proved via a contradiction, which is reached when it is first assumed that there exist two potentials \hat{V}^A and \hat{V}^B which both give rise to the same electron density n(r). Two different potentials will result in two different Hamiltonians \hat{H}^A and \hat{H}^B and two separate ground state wave functions $|\Psi_0^A\rangle$ and $|\Psi_0^B\rangle$. We must have the relation

$$E_0^A = \left\langle \Psi_0^A \right| \hat{H}^A \left| \Psi_0^A \right\rangle < \left\langle \Psi_0^B \right| \hat{H}^A \left| \Psi_0^B \right\rangle \tag{2.7}$$

because of the Rayleigh-Ritz principle and since $|\Psi_0^A\rangle$ represents the ground state of the system \hat{H}^A . The strict inequality is used since equality is only possible for $|\Psi_0^A\rangle = |\Psi_0^B\rangle$. With $H^A = H^B + V^A - V^B$ this becomes

$$E_0^A < E_0^B + \left\langle \Psi_0^B \right| \hat{V}^A - \hat{V}^B \left| \Psi_0^B \right\rangle = E_0^B + \int \left[v^A(r) - v^B(r) \right] n(r) dr.$$
(2.8)

The exact same approach can be used for the opposite case where A^{A} , and B^{B} , arguments are of (2.7) are swapped:

$$E_0^B < E_0^A + \int \left[v^B(r) - v^A(r) \right] n(r) dr.$$
(2.9)

Now, adding (2.8) and (2.9) results in a contradiction $E^A + E^B < E^B + E^A$. Therefore we arrive at the conclusion that $\hat{V}^A = \hat{V}^B$. Additionally, it must be shown that the ground state density n(r) results in the minimum energy for the system. This is a variational principle where the energy E[n(r)] must reach a minimum relative to deviations n'(r) away from the ground state density (with the constraint that n(r) integrates to the total number of electrons):

$$E[n(r)] < E[n'(r)] \quad \forall n'(r) \neq n(r).$$

$$(2.10)$$

This relation follows directly from the fact that the ground state density is constructed from the ground state wave functions for which there exists a variational principle [133].

2.2.1.2 Kohn Sham Equations

Kohn Sham DFT [120] uses a set of self consistent equations, evaluated iteratively, to find the ground state electron density. This evolved from attempts by Thomas [199], Fermi [69, 70], Hartree [89], and Slater [187] to develop an approximate single particle Schrödinger equation. In theory, the Kohn Sham equations present an exact method to find the ground state density by explicitly incorporating many body effects into the single particle Hamiltonian. However, in practice DFT is only able to approximately reproduce the true ground state electron density since the exact form of many body effects is unknown.

The idea is to consider each electron as independent and moving in an effective (fictitious) potential. This is not the potential of the real physical system. Furthermore, solutions will not produce the correct many body wave function. However, the Kohn Sham single particle wave functions should construct the true ground state electron density of the physical system. The Kohn Sham wave functions $\psi_i(r)$ and energy eigenstates E_i can then be calculated from the single particle Schrödinger equation,

$$H[n(r)]\psi_i(r) = E_i\psi_i(r), \qquad (2.11)$$
by re-writing the expression for the total energy functional,

$$E = \hat{E}[\hat{H}] = \hat{E}[\hat{T}] + \hat{E}[\hat{U}] + \hat{E}[\hat{V}], \qquad (2.12)$$

in terms of a non-interacting set of electrons. The electron-electron interaction (i.e. \hat{U} and \hat{T}) can be replaced by their non-interacting counterparts \hat{U}_s and \hat{T}_s plus an additional "exchange and correlation" term that represents the interaction. This results in the relation,

$$E = \hat{E}[\hat{T}_s] + \hat{E}[\hat{U}_s] + \hat{E}[\hat{V}] + E_{xc}[n(r)], \qquad (2.13)$$

for the total energy, where the first three terms can be calculated exactly, but the last term,

$$E_{xc}[n(r)] = \int n(r)\epsilon_{xc}[n(r)]dr, \qquad (2.14)$$

is unknown and must be approximated.

This then defines a set of self consistent equations seen in Figure 2.1 that must be solved iterative starting with an initial guess for the electron density. The Hamiltonian can be formed directly from the electron density, where the Hartree potential,

$$v_H(r) = \int \frac{n(r')}{|r - r'|} dr',$$
(2.15)

is the classical electrostatic potential of a charge distribution (i.e. non-interacting electron density) and the external potential will include the electrostatic potential from the nuclei,

$$v_{ext}(r) = \sum_{l=1}^{N_{at}} \frac{Z_l e}{|r - R_l|},$$
(2.16)

but could also account for additional sources. The exchange and correlation potential,

$$v_{xc}(r) = \frac{dE_{xc}}{dn},\tag{2.17}$$

Kohn Sham Equations:

1.
$$\hat{H} = -\frac{\hbar^2}{2m} \nabla^2 + v_H(r) + v_{ext}(r) + v_{xc}(r)$$

2. $\hat{H}\psi_i(r) = E_i\psi_i(r)$
3. $n(r) = 2\sum_i |\psi_i(r)|^2$

Figure 2.1: Self consistent set of Kohn Sham equations that are solved in DFT calculations. Starting with an initial guess of the electron density n(r) allows for the construction of the Kohn Sham Hamiltonian operator. Solving the resulting Schrödinger equation results in a set of single particle wave functions that can be used to reconstruct n(r). The factor of 2 corresponds to the spin, which we do not explicitly take into account.

approximates many body effects. After building the single particle Hamiltonian, the Kohn Sham wave functions and energy eigenstates can be determined from a single particle Schrödinger equation. These wave functions do not have any physical meaning beyond reproducing the true ground state electron density. Besides the largest occupied energy eigenstate, which corresponds to the ionization energy, the Kohn Sham eigenvalues also lack a physical interpretation.

2.2.1.3 Exchange and correlation functionals

The exchange and correlation potential must account for the many body effects not included in the single particle Hamiltonian: (i) *exchange:* switching the position of any two electrons in the many body wave function should change its sign (antisymmetric property for fermions); and (ii) *correlation:* to account for the fact that an electron does not interact with itself, which is lost in the density representation. Since the first three terms if the Kohn Sham Hamiltonian can be computed exactly, any improvements to the physical model must be made by addressing the exchangecorrelation potential. Although there is no systematic approach to improve this approximation, many sophisticated techniques are available. No known universal functional exists to recovers the true many-body potential, but different exchange and correlation functionals are available for specific use-cases. A large collection of research works exist that have produced hundreds of functionals. These range from the simplest case, the local density approximation (LDA) that considers only the local electron density [156, 30, 157, 127], to the generalized gradient approximation (GGA) [156, 30, 157, 127] and meta-GGA [31, 58, 198] which consider higher order derivatives of the electron density, and to even more sophisticated approaches. It must be noted, again, that these beyond-LDA approaches do not systematically improve the Kohn Sham potential toward the true many body potential and, in some cases, can actually produce worse results.

This is the major limitation of DFT, but has also been touted a feature since complicated, and computationally more expensive, approaches must only be used when the situation warrants. Many functionals are specific to a certain observable quantity, for example, to improve band gap calculations which DFT notoriously underestimates when using LDA. Others functionals should be employed for specific types of materials: notably transition elements and strongly correlated materials. In particular, hybrid functionals [32, 192], which blend in a portion of the exact exchange into v_x calculated from LDA or GGA, have been widely adopted by the quantum chemistry communities, which require "chemical accuracy" (around 0.04 eV per atom).

In this work, and in NESSIE, only the LDA and GGA family of functionals are considered. The focus of this dissertation is mainly on parallelism and simulating large molecules consisting of a few thousand electrons. Other exchange and correlation functionals may require drastically different parallelization techniques and are not considered here.

The local density approximation is the most basic approximation to $v_{xc}(r)$. With LDA, the local density at a given point is used to calculate the exchange and correla-

tion potentials at that same point using the solution of a uniform electron gas, which can be computed [120]. The exchange and correlation energy,

$$E_{xc} = E_x + E_c, \tag{2.18}$$

can be decomposed into separate terms. The total exchange and correlation energy of the system E_{xc} is often written as,

$$E_{xc} = \int_{\Omega} n(r)\epsilon_{xc}d\Omega, \qquad (2.19)$$

where ϵ_{xc} is the energy per electron per unit volume. The potential can be found by taking the functional derivative of (2.19):

$$v_{xc}(r) = \epsilon_{xc}[n(r)] + n(r)\frac{de\epsilon_{xc}[n(r)]}{dn(r)}.$$
(2.20)

As was the case for the energy in (2.18), it is often written as two separate terms:

$$v_{xc}(r) = v_x(r) + v_c(r).$$
 (2.21)

For the homogeneous electron gas of density n_0 the exchange energy per unit volume is known analytically. In the absence of spin the expression for the exchange potential can be written:

$$v_x[n_0] = -\left(\frac{3}{\pi}\right)^{1/3} n_0^{1/3}.$$
(2.22)

The expression for the correlation energy (per unit volume), however, is unknown even for the homogeneous electron gas. It is has been calculated to a high degree using Monte Carlo simulations for which a closed form expression can be derived [210, 158, 155]. The Perdew Zunger correlation functional [159] is used in NESSIE. The potential for the homogeneous electron gas can be computed as,

$$v_c[n_0] = \epsilon_c - \frac{r_s}{3} \frac{d\epsilon_c}{dr_s},\tag{2.23}$$

where,

$$\epsilon_c = \begin{cases} -0.0480 + 0.031ln(r_s) - 0.0116r_s + 0.002r_sln(r_s) & r_s < 1\\ -0.1423/(1 + 1.0529\sqrt{r_s} + 0.3334r_s) & r_s > 1 \end{cases},$$
(2.24)

and,

$$r_s = \left(\frac{3}{4\pi n_0}\right)^{1/3},$$
 (2.25)

is the average spherical radius containing one electron. The exchange and correlation potentials at each point r are then computed from equations (2.22) and (2.23) using the electron density at that point:

$$v_x(r) = v_x[n(r)]$$
 and $v_c(r) = v_c[n(r)].$ (2.26)

Also considered are the generalized gradient family of functionals. These attempt to improve the approximation for the exchange and correlation term by including the gradient of the electron density in the calculation. In this case, the exchange and correlation energy is written as a functional of both the electron density and its gradient. These can be much more difficult to compute and write down and only the general approach is described here. In GGA, as with LDA, the potential at a given spacial coordinate is computed from the local value electron density. However, unlike LDA, GGA includes a correction from the gradient vector of the electron density. The exchange and correlation terms are usually generated separately and the following discussion only considers a generic potential that could be either. The GGA exchange and correlation potential can be decomposed into two terms which correspond to the potential dependent only on the electron density and a potential dependent on the density and its gradient:

$$v[n(r), \nabla n(r)] = v_n[n(r)] + v_g[n(r), \nabla n(r)], \qquad (2.27)$$

Here,

$$v_n[n] = \epsilon + n \frac{d\epsilon}{dn},\tag{2.28}$$

has a form to similar to LDA, but the gradient dependent term,

$$v_g[n, \nabla n] = -\nabla \cdot \left[n \frac{dE}{d\nabla n} \right], \qquad (2.29)$$

is the divergence of the vector,

$$n\frac{dE}{d\nabla n} = \left[n\frac{dE}{d\nabla_x n}, n\frac{dE}{d\nabla_y n}, n\frac{dE}{d\nabla_z n}\right].$$
(2.30)

To perform GGA calculations an external library LIBXC [143] has been used. This offers a large variety of different functionals to choose from. LIBXC takes as input the electron density n(r) and, instead of the vector field $\nabla n(r)$, the scalar function $\sigma(r) = \nabla n(r) \cdot \nabla n(r)$. The library then returns two scalar functions, where the first corresponds to $v_n(r)$ and is associated with the electron density. The seconds is the scalar function $nd\epsilon/d\sigma$ from which, with the help of the chain rule, the scalar potential can be formed:

$$v_g(r) = -\nabla \cdot \left[n \frac{d\epsilon}{d\nabla n} \right] = -\nabla \cdot \left[\left(n(r) \frac{d\epsilon}{d\sigma} \times 2\nabla n(r) \right) \right].$$
(2.31)

Unfortunately, $v_g(r)$ in (2.31) will result in numerical instability due to the higher order derivative [144]. The point-wise GGA potential can not be included in the Kohn Sham Hamiltonian in the same way as the LDA potentials. Instead, the divergence in (2.29) must be applied at the level of the basis. After defining the vector,

$$\mathbf{V}_{\mathbf{g}}(r) = \left[n \frac{dE}{d\nabla n} \right] = \left[\left(n(r) \frac{d\epsilon}{d\sigma} \times 2\nabla n(r) \right) \right], \qquad (2.32)$$

the potentials can be included in the Hamiltonian by computing each matrix element corresponding to the overlap of basis functions ω_i and ω_j :

$$\langle \omega_j | v_n(r) + \mathbf{V}_{\mathbf{g}}(r) | w_i \rangle = \int \left(v_n w_j^* w_i \right) + \left(w_j^* \mathbf{V}_{\mathbf{g}} \cdot \nabla w_i + w_i \mathbf{V}_{\mathbf{g}} \cdot \nabla w_j^* \right).$$
(2.33)

This requires only slight modification to the standard approach with LDA, which would contain only the $v_n(r)$ term.

2.2.1.4 All-electron approach

The all-electron approach employed in NESSIE makes no additional simplifications beyond traditional approximations for the exchange and correlation. This is opposed to using a pseudopotential method to reduce the computational complexity of the full Kohn Sham DFT problem. Outer-shell valence electrons directly participate in chemical bonding. Underlying core electrons that interact indirectly, can be removed from the calculations by using a non-local function known as a pseudopotental [90]. This mimics the combined effects of core electrons and nucleus to account for a screened electrostatic potential away from the 1/r nuclear potential. The solutions near each atom are also taken to be "pseudo wave functions". Norm-conserving pseudopotentials [88] require the wave function to match the all-electron wave function outside of some cutoff radius surrounding the atom. Ultra-soft pseudopotentials [209] relax this requirement in order to further reduce computational complexity, but result in a generalized eigenvalue problem.

The number of basis functions needed to capture the solution is reduced by only considering valence electrons. Their associated single particle wave functions are much more delocalized and slow varying than for the inner core electrons. Fewer basis functions are needed to expand the solution to a desired accuracy without the high-frequency variation of core states near the atomic nucleus. Additionally, with pseudopotentials only wave functions of valence electrons are computed, reducing the number of eigenvectors calculated and stored in memory. However, this reduction in computational complexity comes with a cost.

Pseudopotentials approaches are less accurate than full all-electron calculations that explicitly consider core electrons. Although pseudopotentials are used to model molecules and crystal structures composed of many different atoms, they are fitted specifically for a single atom (i.e. for each chemical element). A similar, but more accurate, approach using augmented basis function [214] is also widely used, but is non-linear and still represent an linearized approximation. Furthermore, it is not clear what type of effect pseudopotentials have in TDDFT since they are fitted to the ground state. Explicitly including the effects of core electrons offers a more intuitive method since no additional approximation is needed. The all-electron calculations can be performed in NESSIE with little or no additional cost if executed in a parallel environment.

In the finite-element approach used in NESSIE the difficulty becomes the creation of a sufficiently fine and structured mesh to capture the 1/r potential at and around the atomic nucleus. The singularity can instead be treated numerically and is included into the matrix system by integrating over the element using a quadrature rule, as will be described in a later section. The potential must then be evaluated only at quadrature nodes and never directly on the singularity.

2.2.2 Time dependent density functional theory

Time dependent DFT (TDDFT) for the time dependent Schrödinger equation,

$$i\hbar\frac{\partial}{\partial t}\Psi(r_1,...,r_N,t) = \left(-\frac{\hbar^2}{2m}\Delta + v(r,t)\right)\Psi(r_1,...,r_N,t),$$
(2.34)

is similar to ground state DFT where the explicit dependence on the position of each electron is replaced the electron density and a non-interacting set of wave functions moving in an effective potential:

$$i\hbar\frac{\partial}{\partial t}\psi_i(r,t) = \left(-\frac{\hbar^2}{2m}\Delta + v_{KS}[n(r,t)]\right)\psi_i(r,t).$$
(2.35)

In principle, this method can be used to calculate any time dependent observable as a functional of the electron density. However, the functional dependence may not be known. TDDFT has had considerable success modeling the interaction of electromagnetic fields with matter [141]; particularly absorption and emission spectrum, which can be easily calculated from the time dependent electron density.

2.2.2.1 Important TDDFT theorems

There exist two particularly important theorems in TDDFT. Although time dependent calculations using the Kohn Sham system were initially performed in the late 1970s [18], the beginning of TDDFT is associated with the paper of Runge and Gross [173] in 1984. This put TDDFT on a solid theoretical foundation by providing an analogous proof of the Hohenberg Kohn theorem for time dependent systems. However, the Runge Gross theorem is not enough to justify a time dependent Kohn Sham scheme. It was not until 1999 that van Leeuwen [208] showed the true time dependent many body potential can replaced with an effective potential of a noninteracting system, and that this effective potential will generate the same electron density. The two theorems are transcribed below, but the formal proofs are involved and have been omitted.

The Runge Gross Theorem:

For every single particle potential v(r,t) which can be expanded into a Taylor series with respect to the time coordinate around $t = t_0$, a map G: $v(r,t) \rightarrow n(r,t)$ is defined by solving the time dependent Schrödinger equation with a fixed initial state $\psi(t_p) = \psi_0$ and calculating the corresponding densities n(r, t). This map can be inverted up to an additive merely time dependent function in the potential.

The van Leeuwen Theorem:

A time dependent particle density n(r, t) obtained from a given many particle system can, under mild restrictions on the initial state, always be reproduced by an external potential v(r, t) in a many particle system with different two-particle interactions and given the initial state of this other many particle system, the potential v(r, t) is unique up to a purely time dependent function.

In essence the Runge Gross theorem shows that there is a one-to-one correspondent between the time dependent potential v(r,t) and the time dependent electron density n(r,t). Two time dependent potentials, $v_1(r,t)$ and $v_2(r,t)$, that differ by more than a purely time dependent function $(v_1(r,t) - v_2(r,t) \neq c(t))$ can not produce the same time dependent density n(r,t) and two time dependent densities $n_1(r,t)$ and $n_2(r,t)$ can not generate the same potential v(r,t). This implies that all observables of the time dependent Hamiltonian are functionals of the electron density; although, they may not be easily expressed in this form. The van Leeuwen theorem then allows for a time dependent Kohn Sham Hamiltonian written in terms of the non-interacting electron density plus an exchange and correlation potential $v_{xc}(r,t)$ accounting for many body effects.

2.2.2.2 Real-time approach

In principle TDDFT allows for the calculation of true electron density by replacing the interacting system with a non-interacting system plus an effective potential. And just like with DFT, properties of the system can be extracted from directly from the density, which is now time dependent. Unlike DFT, the effective potential in TDDFT is not only a functional of the electron density, but also the initial many body state Ψ_0 and initial non-interacting state ψ_0 . However, this functional dependence on Ψ_0 and ψ_0 vanishes if the initial state ψ_0 is given from the solution of the grounds state Kohn Sham system since, according to the Hohenberg Kohn theorem, both Ψ_0 and ψ_0 are functionals of the ground state electron density alone. In this case the time dependent effective potential becomes a functional of the electron density alone. Because of this, the first step in TDDFT is always to compute the DFT ground-state.

The most straight-forward way to compute the time dependent electron density is to propagate the ground state Kohn Sham wave functions directly in time. The electron density can be recomputed at each time step from the time dependent Kohn Sham wave functions:

$$n(r,t) = 2\sum_{i} |\psi_i(r,t)|^2.$$
(2.36)

The time dependent Kohn Sham potentials are also similar to their DFT counterparts. The Hartree term,

$$v_H(r,t) = \int \frac{n(r',t)}{|r-r'|} dr',$$
(2.37)

inherits time dependence from the density. The external potential,

$$v_{ext}(r,t) = \tilde{v}_{ext}(r,t) + \sum_{l=1}^{N_{at}} \frac{Z_l e}{|r - R_l|},$$
(2.38)

now includes an additional term accounting for the perturbing potential $\tilde{v}_{ext}(r, t)$, which is responsible for exciting the system away from its ground state. Just as before, the exchange and correlation potential includes all other many body effects:

$$v_{xc}(r,t) = v(r,t) - v_H(r,t) - v_{ext}(r,t), \qquad (2.39)$$

where v(r,t) here is the true many body potential. However, the situation is even worse than in DFT: $v_{xc}(r,t)$ is not only non-local in space, but also in time. In principle $v_{xc}(r,t)$ depends on the electron density at all previous instances of time. Determination of the exchange and correlation potential is a difficult and open questions in DFT and TDDFT. Since the focus of this work is to scale TDDFT calculations using a distributed memory environment, the simplest approximation, where the exchange and correlation potential is assumed to be local in space and time, is used. These are widely used and known as the adiabatic approximation and can be applied to both the local density approximation ALDA or the generalized gradient approximation AGGA. Practically speaking, this is the same approximation used in ground state DFT:

$$v_{xc}(r,t) = v_{xc}[n_0](r)_{n_0 \to n(r,t)}.$$
(2.40)

The exchange and correlation potential is updated at each time step using the local density and instantaneous electron density calculated from (2.36). ALDA, in our experience, has worked relatively well for organic molecules and nanostructures. The code also has the capability to used GGA functionals for ground-state calculations and, within the adiabatic approximation, for time dependent calculations as well.

2.2.2.3 Time-dependent observables

As is the case for DFT, all observables must be expressed in terms of the electron density, which is the only physical quantity. The time-dependent electron density itself is the most easily computed observable in real-time, since this quantity is trivially known. To investigate the electron density corresponding to a specific resonance, a sinusoidal stimulus at that frequency can be applied. The spatial electron density should be visualized at the maximum and minimum of the dipole. It can also be viewed, as an animation, changing from the maximum to minimum in time.

The functional form is not known for many observables and TDDFT may not be applicable. However, a very useful case of TDDFT is related to spectroscopy, where the absorption and emission spectrum corresponding to electronic excitations can be computed. In particular, we would like to compute a measure of the photoabsorption. The photoabsorption cross section gives the probability that a photon passing through an atom or molecule absorbed and the oscillator strength is a measure of how strong the interaction is.

Before going further, the dynamic polarizability must be defined. Polaizability α relates the induced dipole moment d to an applied electric field E. In general, E and d are vectors quantities with \hat{x} , \hat{y} and \hat{z} components and α is a tensor. The following will only consider an isotropic medium, which will have a diagonal polarizability, and an electric field applied along one coordinate axis. In this case only scalar quantities must be considered. The polarizability can be defined in the time domain,

$$d(t) = \int_{-\infty}^{t} \alpha(t - t') E(t') dt',$$
(2.41)

where the dipole depends on the applied electric field at all times t' < t. However, this form is not very useful. Instead, the relation is generally considered in frequency space,

$$d(\omega) = \alpha(\omega)E(\omega), \qquad (2.42)$$

which can be calculated as the Fourier transform of (2.41). The induced dipole $d(\omega)$ depends on the medium through $\alpha(\omega)$, but also on the frequency and strength of the applied electric field $E(\omega)$.

The imaginary part of the dynamic polarizability is proportional to the photoabsorption [23]. The oscillator strength can be computed from the polarizability [95, 53]:

$$\sigma(\omega) = \frac{4\pi\omega}{c} \operatorname{Im}(\alpha(\omega)).$$
(2.43)

With real-time TDDFT, direct access to the electron density is available. Also available is the time-dependent induced dipole moment:

$$d_{\mu}(t) = \int_{\Omega} (\mu - \mu_0) \left(\rho(t, r) - \rho_0(r)\right) dr, \qquad (2.44)$$

where μ takes on x, y or z depending on the direction considered. The molecular center of mass, μ_0 , is calculated as,

$$\mu_0 = \frac{1}{qn_e} \sum_{i=1}^{n_{at}} R_i \times Z_i, \qquad (2.45)$$

where $n_a t$ is the number of atoms in the molcule, R_i is the atomic coordinates, Z_i is the nuclear charge, and where qn_e is the total elecronic charge (i.e. the same as the total nuclear charge). In this case the dynamic polarizability could be calculated by inverting (2.42),

$$\alpha(\omega) = \frac{d(\omega)}{E(\omega)},\tag{2.46}$$

assuming the Fourier transform of the electric field is known. This is the case for an impulse excitation $\delta(t)$ and step potential u(t), for which the equations are given in [215] and presented in Table 2.1. The oscillator strength can be computed according to (2.43).

Table 2.1: Polarizability for excitations in the form of an impulse potential and a step potential.

$$\begin{array}{c|c|c} E(t) & \alpha(\omega) \\ \hline \\ \hline \\ E_0 \times \delta(t) & -\frac{1}{E_0} \times d(\omega) \\ \hline \\ E_0 \times u(t) & -\frac{i\omega}{E_0} \times d(\omega) \end{array}$$

In practice, it is necessay to introduce artificial damping signal into the computed dipole moment before taking the Fourier transform:

$$d(\omega) = \int_0^T \left(d(t) \times e^{-\gamma t} \right) e^{i\omega t} dt, \qquad (2.47)$$

where γ is the damping coefficient. These TDDFT simulations do not include energy dissipation. The damping can be thought of as modeling relaxation and energy dissi-

pation, where a molecule would physically emit energy as photons and relax back to the ground-state after being excited.

2.3 Mathematical models

In most cases it is not possible to represent continuous fields, operators, or solutions on a computer, even though they may be easily written down on paper in the language of differential calculus. In this case some form of discretization is necessary to expands these quantities in a finite basis that approximates the full solution within a desired accuracy. Mathematical models translate the physical equations into the mathematical form of linear algebra understood by a computer. This is an extremely important step from a computing perspective as there are generally a multitude of choices. Each can be associated with benefits and drawbacks, as well as presenting different computational complexities and bottlenecks. Here, the mathematical models employed by NESSIE for solving the equations associated with DFT and TDDFT are discussed.

2.3.1 Finite element discretization

Finite element is a method for solving partial differential equations [222, 97] and has been applied to a wide range of problems from structural analysis to electromagnetics. Here we limit our scope to a class of equations of the form

$$\Delta \Psi + c(r)\Psi(r) = f(r), \qquad (2.48)$$

which includes the time independent the Schrödinger equation [130, 154]. Finite element is a real space discretization that employs a mesh (equivalent to a graph) consisting of grid points (i.e. mesh nodes or graph vertices) and connections between nodes (i.e. graph edges) that define a collections of elements. A major advantage of finite element is the sparse matrix structure, which is a result of each vertex being connected only to its immediate neighbors. The finite element basis functions are defined locally for each grid point and are non-zero over a small subset of the entire domain. A 2-dimensional example of a single basis function can be seen in Figure 2.2. Furthermore, finite element is a very general approach that allows for mesh refinement at specific locations and can handle arbitrary boundary conditions. While Neumann boundary conditions are natural, both Dirichlet and periodic may be applied through modifications to the resulting system matrix and right-hand-side.

2.3.1.1 Mathematical formulation

The finite element method can be derived starting from the variational form of the problem

$$\int_{\Omega} \Delta \Psi \gamma(r) d\Omega + \int_{\Omega} c(r) \Psi(r) \gamma(r) d\Omega = \int_{\Omega} f(r) \gamma(r) d\Omega, \qquad (2.49)$$

where $\gamma(r)$ is an arbitrary test function and the integration is taken over the entire domain Ω . Any solution $\Psi(r)$ that satisfies (2.49) must also satisfy (2.48). The second derivative can be removed through Green's Identity. The first term in (2.49) becomes:

$$\int_{\Omega} \Delta \Psi \gamma(r) d\Omega = -\int_{\Omega} \nabla \Psi(r) \nabla \gamma(r) d\Omega + \int_{\Gamma} \left(\nabla \Psi(r) \cdot \hat{n} \right) \gamma(r) d\Gamma, \qquad (2.50)$$

where Γ is the boundary domain. This discretization naturally produces Neumann boundary conditions; setting the second term in (2.50) equal to zero implicitly defines $\nabla \Psi(r) \cdot \hat{n} = 0$. Both Dirichlet and periodic boundary conditions can be applied by modifying this third term in (2.50) as well as the system matrices. Next, the solution $\Psi(r)$ and test function $\gamma(r)$ are expanded in a local basis as,

$$\Psi(r) \approx \sum_{i}^{N} \psi_{i} \omega_{i}(r) \quad \gamma(r) \approx \sum_{i}^{N} v_{i} \omega_{i}(r), \qquad (2.51)$$

where ψ_i is the value of $\Psi(r)$ at the i^{th} grid point. The basis function $\omega_i(r)$ is also associated the i^{th} point and is non-zero only for neighboring elements.

The basis functions are defined so that $\omega_i(r)$ has a value equal to one at the i^{th} mesh node and a value equal to zero for all other grid points. They are also local in the sense that they are explicitly zero outside of the neighboring set of elements. The overlap integrals in (2.53) are then only non-zero for basis functions of neighboring grid points. Our finite element implementation uses up to a third degree polynomial basis. Higher order basis functions can decrease error in the solution, but also increase the problem size and number of overlapping basis functions (i.e. the number of non-zero elements per row in the system matrix). The integrals in equation (2.53) can be re-written as

$$\int_{\Omega} \omega_i(r)\omega_j(r)d\Omega = \sum_{e=1}^{\bar{N}_{elem}} \int_{\Omega_e} \omega_i(r)\omega_j(r)d\Omega_e, \qquad (2.52)$$

where the integration over the entire domain, for a given set of basis functions ω_i and ω_j , can be computed by integrating over a select subset of mesh elements Ω_e for which the overlap is non-zero. An example of a linear basis can be seen in Figure 2.2



Figure 2.2: Example of a finite element basis using piece wise linear functions for both 1-dimensional (taken from [154]) and 2-dimensional cases. Each basis function is equal to one at a single grid point and non-zero only over neighboring elements. Further refinement can be achieved using additional higher order polynomials terms in the basis (p-refinement) or a finer grid of points (h-refinement).

Now plugging the expressions in (2.51) into (2.49)+(2.50) we obtain:

$$-\sum_{i}^{N}\sum_{j}^{N}\psi_{i}v_{j}\times \left(\int_{\Omega}\nabla\omega_{i}\nabla\omega_{j}d\Omega + \int_{\Gamma}\nabla\omega_{i}\omega_{j}(r)d\Gamma + \int_{\Omega}c(r)\omega_{i}(r)\omega_{j}(r)d\Omega\right)$$
$$=\sum_{j}^{N}v_{j}\int_{\Omega}f(r)\omega_{j}(r)d\Omega. \quad (2.53)$$

Since v_j 's are arbitrary and exist on both sides of the equation they can be discarded, leaving only the known basis function behind. The only unknowns left are coefficients of the solution ψ_i .

The final result of (2.53) can be written as a matrix equation. The double summation over the basis functions on the left hand side of the equation, multiplied by the coefficients $\{\psi_1, ..., \psi_n\}$, represents a matrix A multiplied by a vector Ψ . Because the right-hand-side only has a single summation term it will result in a vector F. The solution Ψ can be found by solving a linear system:

$$A\Psi = F. \tag{2.54}$$

2.3.1.2 Application to Kohn Sham equations

The finite element method can also be applied an eigenvalue problems, such as the single particle Schrödinger equation of (2.11). Setting the right-hand-side of (2.48) equal to the right-hand-side of an eigenvalue problem,

$$f(r) = E_i \psi_i(r), \tag{2.55}$$

and using the variational form,

$$\int_{\Omega} E_i \psi_i(r) \gamma(r) d\Omega = E_i \sum_{i,j} \psi_j \int_{\Omega} \omega_i(r) \omega_j(r) d\Omega, \qquad (2.56)$$

will result in a generalize eigenvalue problem,

$$H\psi_i = E_i S\Psi,\tag{2.57}$$

where S is known as the overlap (or mass) matrix and the Hamiltonian,

$$H = L + U, \tag{2.58}$$

is the sum of the Laplace matrix L and an electrostatic matrix U that corresponds to the electrostatic potential in the finite element basis.

Both the Laplace matrix,

$$L_{ij} = \int_{\Omega} \nabla \omega_i(r) \nabla \omega_j(r) d\Omega, \qquad (2.59)$$

and Mass matrix,

$$S_{ij} = \int_{\Omega} \omega_i(r) \omega_j(r) d\Omega, \qquad (2.60)$$

can be formed once-and-for-all after defining the mesh. In general, they must be computed by evaluating an integral for each pair of overlapping basis functions. In one and two dimensional problems simple closed form expressions can be derived for these matrix elements. However, in three dimension these expressions become complicated and can instead be evaluated numerically, as is the case in NESSIE.

The electrostatic potential (i.e. from nuclei and electrons) is the quantity sought and converged upon in DFT, as there is a one-to-once correspondence with the charge density. This term is included in the calculation via the c(r) term of (2.53) and must be computed numerically at each SCF iteration. The electrostatic matrix U can be formed using the assemble technique and re-writing the integration over the entire domain can as a summation of integrations over each element:

$$U_{ij} = \int_{\Omega} c(r)\omega_i(r)\omega_j(r)d\Omega = \sum_{e=1}^{N_{elem}} \int_{\Omega_e} c(r)\omega_i(r)\omega_j(r)d\Omega_e.$$
 (2.61)

The integration is then performed for each element using a N-point Gaussian quadrature rule over the 3-dimensional tetrahedron. The same procedure is used in NESSIE for computing the Laplace and mass matrices for three-dimensional problems.

A question then become how many quadrature nodes should be used to compute the numerical integrations over each element. Using too few or too many quadrature nodes will have a negative effect on the solution accuracy. Counter intuitively, too many quadrature nodes will also have an adverse effect on the solution since evaluating a function at a Gauss point must be approximated through expansion of the polynomial basis, which includes additional error. Furthermore, the matrix elements of the Laplacian depend on gradients of the basis functions and are of a different order than the mass matrices, which do not. A more detailed discussion on generating the finite element mesh and matrix will be presented in Chapter 3.

2.3.1.3 Dirichlet boundary conditions

Flexibility of boundary conditions is a major benefit of finite element. Setting the boundary conditional in real-space methods, in general, is very intuitive since the basis coefficients are defined at a coordinate in space. This is opposed to the use a of delocalized basis where basis functional are functions defined over the entire space. Natural to finite element are Neumann boundary conditions, which specify a value for the gradient of the solution. Other boundary conditions, such as periodic and Dirichlet, are also possible and require modification to the Hamiltonian matrix and right-hand-side. Dirichlet boundary conditions force solutions to a specified value and are important in NESSIE for both the Kohn Sham eigenvalue problem and Poisson equation.

A boundary condition of zero is taken for the Kohn Sham eigenvalue problem. The computational domain must be made large enough that the wave functions, which decay very rapidly, vanish to zero at the boundary. Confinement effect can appear if the domain is made too small. The electrostatic potential calculated from the Poisson equation, however, decays much more slowly as 1/r. In this case the boundary values must be computed. In NESSIE this is accomplished by solving the integral equation at the boundary mesh nodes.

After computing the boundary value, the Dirichlet boundary condition must be set by modifying the matrix and right-hand-side. This is slightly different for the eigenvalue problem and Poisson equation, as the definition of their right-hand-sides differ. For the Poisson equation, and considering the linear system in (2.54), one can easily derive the method for setting the Dirichlet boundary conditions. It can be seen for a simple 4×4 case that re-writing this linear system as a set of linear equations and setting the $x_1 = \alpha$,

$$\begin{pmatrix}
x_1 & = \alpha \\
a_{22}x_2 + a_{23}x_3 + a_{2n}x_n = f_2 - a_{21}\alpha \\
a_{32}x_2 + a_{33}x_3 + a_{3n}x_n = f_3 - a_{31}\alpha \\
a_{42}x_2 + a_{43}x_3 + a_{44}x_4 = f_4 - a_{41}\alpha
\end{pmatrix},$$
(2.62)

requires operations to the matrix and right-hand-side vector. This will produce the matrix equation,

$$\begin{bmatrix} 1 & 0 & 0 & 0 \\ 0 & a_{22} & a_{23} & a_{24} \\ 0 & a_{32} & a_{33} & a_{34} \\ 0 & a_{42} & a_{43} & a_{44} \end{bmatrix} \begin{bmatrix} x_1 \\ x_2 \\ x_3 \\ x_4 \end{bmatrix} = \begin{bmatrix} \alpha \\ f_2 & - & a_{21}\alpha \\ f_3 & - & a_{31}\alpha \\ f_4 & - & a_{41}\alpha \end{bmatrix}, \quad (2.63)$$

that enforces the boundary condition. The following rules can be followed when setting the value of the i^{th} node for the general $n \times n$ case:

- (i) Set $a_{ii} = 1$, $a_{ij} = 0 \forall j \neq i$, $f_i = \alpha$ and
- (ii) Set $f_j = f_j a_{ji}\alpha$, $a_{j,i} = 0 \ \forall i \neq j$.

Specifying Dirichlet boundary condition for the Kohn Sham eigenvalue problem is slightly less intuitive. The boundary condition must be set by modifying the matrix pencil alone since the full right-hand-side depends on the unknown eigenvectors and eigenvalues. This is possible for the case of a zero boundary condition. Similar to the linear system, the boundary condition can be set by modifying the rows and columns of the Hamiltonian and mass matrices corresponding to boundary nodes. The procedure is the same for any eigenpair of the matrix pencil. Here, for simplicity, only a single eigenvalue and eigenvector pair $\{e, \Psi\}$ is considered. Modifying the eigenvalue problem $H\Psi = eS\Psi$ as,

$$\begin{bmatrix} \alpha & 0 & 0 & 0 \\ 0 & h_{22} & h_{23} & h_{24} \\ 0 & h_{32} & h_{33} & h_{34} \\ 0 & h_{42} & h_{43} & h_{44} \end{bmatrix} \begin{bmatrix} \psi_1 \\ \psi_2 \\ \psi_3 \\ \psi_4 \end{bmatrix} = e \begin{bmatrix} \beta & 0 & 0 & 0 \\ 0 & s_{22} & s_{23} & s_{24} \\ 0 & s_{32} & s_{33} & s_{34} \\ 0 & s_{42} & s_{43} & s_{44} \end{bmatrix} \begin{bmatrix} \psi_1 \\ \psi_2 \\ \psi_3 \\ \psi_4 \end{bmatrix}, \quad (2.64)$$

will reduce the equation at the boundary node to $\alpha \psi_1 = e\beta \psi_1$. This can only be true if $\alpha/\beta = e$. In NESSIE we choose $\alpha = \beta = h_{ii}$, which will guarantee $\psi_i = 0$ in the energy range associated with Kohn Sham eigenvalues of the ground state.

2.3.2 Muffin-tin domain decomposition

Domain decomposition partitions a global space into distinct regions. In terms of a real space finite element discretization, this corresponds to partitioning the mesh into multiple different sets of nodes and elements. The different regions must be connected at their interfaces. A non-overlapping domain decomposition would be equivalent to partitioning the matrix directly. In this case mesh nodes are specific to each partition and mesh elements are shared. NESSIE uses an overlapping domain decomposition, where a distinct mesh is created for each region and interface nodes are shared [131]. An example of the specific domain decomposition used in NESSIE can be seen in Figure 2.3. The global mesh is decomposed into N_{at} atom centered regions that directly surround each nucleus. The additional connecting region is known as the interstitial mesh and is the source of the term "muffin-tin", which it resembles in 2-dimensions. This sets the total number of meshes (and associated matrices) at $N_{at} + 1$. For the wave functions, interface points are formally defined to exist in the interstitial mesh. Currently all atoms use the same mesh, but the code could be updated to associate a different mesh with each type of atom. This could be necessary for heavy atoms that need high resolution to capture deep core electronic states.

For the atom-centered mesh, which has been chosen common to all atoms, a successive layers of polyhedra similar to the ones proposed in [129] is used. The atom-centered finite element mesh can be effectively built using a succession deltoidal icositetrahedron, and its rhombicuboctahedron dual. This discretization provides both tetrahedra of good quality, an arbitrary level of refinement - i.e. the distance between layers can be arbitrarily refined while approaching the nucleus, and the same number of surface nodes. The outer layer is consistently providing the same (relatively small) number of overlap nodes with the interstitial mesh at the muffin edges (i.e. $n_j =$ 26, 98, or 218 nodes respectively using linear, quadratic or cubic FEM). Consequently, the size of linear system in the interstitial region stays independent of the atomcentered regions system matrix size, and the approach can then ideally deal with full potential (all-electron).



Figure 2.3: Example of the muffin-tin domain decomposition used in NESSIE. A structured fine mesh surrounds each atomic nucleus. The interconnecting region left over is call the interstitial mesh. Note: generally interstitial mesh is more symmetric in \hat{x} , \hat{y} and \hat{z} , but is shown as narrow for esthetics.

One question is how to define the mesh to obtain an accurate solution. The accuracy of NESSIE results can depend on a variety of factors including the mesh. The mesh can be refined using p-refinement up to third degree polynomials, as discussed in the previous section. Accuracy can also be improved through h-refinement where

additional mesh nodes are added reducing the grid spacing and the size of each element. In NESSIE, h-refinement can be employed to either the atomic or interstitial meshes. For the atomic mesh, the distance between the first and third spherical shells that surround each atom define a layer of elements. This distance can be set using a coarse parameter. The distance between subsequent layers are generated as a geometric series, with the last distance greater than the fine parameter. For the interstitial mesh there exists a finer region directly surrounding the molecule and an outer courser region that allows wave function solutions to decay to zero.

The muffin-tin domain decomposition can lead to additional flexibility and benefits. A different basis can be used for the atom and interstitial domains. A hybrid approach using a combination of direct solvers for atom regions and iterative solvers for the interstitial domain can be used.

NESSIE has been upgraded to an MPI framework using this domain decomposition approach, where each atomic region (and associated matrix) is distributed over MPI processes. The interstitial mesh is known globally, but all mesh operations and numerical linear algebra operations on its associated matrix are performed in parallel with all MPI processes. For extreme scale problems, with many thousands of atoms, an iterative or hybrid approach is necessary. However, in this work where a maximum of one thousand atoms are considered, direct distributed memory solvers within the domain decomposition framework are used for solving both linear systems and eigenvalue problems.

2.3.2.1 Mesh, matrix and data distribution

Although the domain decomposition presents a natural distribution format for the atoms, the interstitial region contains the majority of mesh points and must also be partitioned. The final representation of the data must conform to the requirements of the eigenvalue solver PFEAST, which does not have a domain decomposition interface and instead distributes the eigenvectors of the full system by row. This subsection describes the distribution and ordering of the mesh and its associated matrices as well as any scalar data field defined on the mesh (e.g. eigenvectors, potential, electron density).

The global domain,

$$\Omega_g = \Omega_{it} \bigcup \Omega_{at}, \tag{2.65}$$

can be decomposed into overlapping interstitial Ω_{it} and atomic Ω_{at} regions. Here we defined Ω_{at} as the collection of atom subdomains $\Omega_{at}^{(i)}$:

$$\Omega_{at} = \left(\bigcup_{i=1}^{N_{at}} \Omega_{at}^{(i)}\right).$$
(2.66)

Each domain contains a certain set of mesh nodes,

$$\Omega = \{d_1, ..., d_n\},\tag{2.67}$$

defined in 3-dimensional space as $d_j = a_j \hat{x} + b_j \hat{y} + c_j \hat{z}$. A scalar field v(r) can then be defined on the domain as:

$$v(\Omega) = \{v(d_1), \dots, v(d_n)\}.$$
(2.68)

The field v will have a representation on the global domain Ω_g as v(r) or, equivalently, on both Ω_{it} and Ω_{at} as $v_{it}(r)$ and $v_{at}(r)$ with continuity and differentiability enforced at the interface.

In NESSIE, we actually store all data as v(r) on the global domain Ω_g . This is due to interfacing with the PFEAST eigenvalue solver (more in Section 2.3.2.2), which defines its own data format and distributes the eigenvectors 1-dimensional by row. Thus Ω_g can be defined however we see fit; first by stitching together Ω_{it} and Ω_{at} and then by reordering the mesh so that each MPI process in PFEAST contains the correct subset for solving the problem in parallel.

The full matrix structure for an unpermuted three atom system is show in Figure 2.4 with interface points for each atom defined to exist within Ω_{it} . Note that this matrix is never formed, but shown here to help defined the distribution format for the eigenvectors. The first large block corresponds to the interstitial region and the next three smaller blocks to the atoms.



Structure for 1 MPI Process

Figure 2.4: Matrix structure for a three atom (fictitious) system with one MPI process. The ordering of the global domain Ω_g is shown on the left, with the interstitial points (including interface) followed by the points associated with each atom region.

In general, the full domain can be distributed over the MPI processes. Each process p will contain a subset of the interstitial domain $\Omega_{it}^{(p)} \subset \Omega_{it}$ and a collection of atom domains $\{\Omega_{at}^{(p_1)}, \Omega_{at}^{(p_2)}, ..., \Omega_{at}^{(p_i)}\}$ assigned at the beginning of the code. The atoms are assigned to an MPI process circularly. The first atom is assigned to the first process (rank = 0), the second atom to the second process (rank = 1), and so on. If the number of atoms N_{at} is larger than the number of MPI process p then atom number p + 1 is given to the first process, atom number p + 2 is given to the second process, etc. The i^{th} atom is then assigned to process $i\% N_p$, where % is the modulo operator and N_p is the total number of MPI processes used to distribute the problem.

The interstitial domain is distributed evenly with one condition: $\Omega_{at}^{(i)}$ exists on the same MPI process as interface nodes between Ω_{it} and $\Omega_{at}^{(i)}$. This condition simplifies linear algebra operations so that no MPI communication is necessary between atom and interstitial domains, but has the drawback of reordering Ω_{it} depending on the number of MPI processes. The matrix structure for two and three MPI processes is shown in Figure 2.5 along with the ordering of Ω_g . An example of the data distribution



Figure 2.5: Matrix structure of three atoms system in Figure 2.4 reordered for three (left) and two (right) MPI processes. The global domain Ω_g , represented by the column vector to the right, is reordered and distributed by row over the processes. The column vector to the right represents the global domain Ω_g , which is reordered and distributed by row. Subsets of Ω_g assigned to each MPI processes are separated by a horizontal bar.

with three MPI process can be seen on the left of Figure 2.5. Here each process is assigned a subset ($\approx 1/3$) of Ω_{it} and a single atom. A load imbalance can take place if the total number of atoms is not evenly divisible by the number of MPI processes. This can be seen on the right of Figure 2.5 where both processes are given half of Ω_{it} , but the first process is assigned two atoms while the second process is assigned only one. This figure also highlights the difficulty to perform linear algebra operations on the interstitial domain. The atom blocks can be made completely independent and isolated to a specific MPI process, but the distributed interstitial matrix has many off-diagonal blocks representing connections between points assigned to a different MPI processes.

Data, such as the electron density, potential or eigenvectors, is defined on the nonoverlapping mesh defined as shown in Figures 2.4 and 2.5. The matrices, however, do not correspond exactly to what is shown in these figures. Separate matrices exist for each atom and for the interstitial region and the interfaces between them are known for both (i.e. as this is an overlapping method). Because the same mesh is reused for each atom, their Laplacian matrices are identical and only a single matrix must be stored per MPI process. The Hamiltonians, however, differ due to the potential U that is included and every MPI process creates a different distinct matrix for each of its atoms. The interstitial matrix is not comprised of independent domains and can not be distributed is the same manner as the atoms. Instead its distribution is defined to be 1-dimensional and by row to correspond to the PFEAST format, which will be descussed in Section 2.4.5. The interstitial matrix structure would be that same as in Figures 2.4 and 2.5 after removing all atom points.

2.3.2.2 Mesh, matrix and data ordering

The interstitial matrix is reordered depending on the number of MPI processes used to solve the matrix in parallel. Each process is assigned a set atoms as well as the interstitial points associated with the atom interfaces. Result at these points are needed to obtain the solution in the interstitial region. More about the linear system and eigenvalue solvers will be discussed in the next subsection. However, the purpose of this ordering is to allow the Schur complement matrix to be built without communication.

A permutation is recorded when the matrix is reordered to keep track of the Specific permutations are used for the atomic and interstitial meshes to generate matrices with a specific ordering. It is necessary to keep track of the interface points between the atomic and interstitial meshes for linear algebra and mesh operations. Additionally, the interstitial matrix distributed by row to each MPI process, so that linear algebra operations can be computed in parallel.

<u>The Atomic Mesh</u>

The atomic meshes have been permuted so that the interface points are grouped together at the beginning. The interface nodes in the mesh are given indices from 1 to the number of interface points N_s . These points in the atomic matrix are grouped together in a block within the first N_s rows; the solution vector for this region has the same data format.

If there are multiple atoms assigned to an MPI process each is independent and has its own dedicated memory. All of the interface points for the j^{th} atomic Hamiltonian are located within $H_{atom}(1 : N_s, :, j)$, a 3-dimensional matrix shown in FORTRAN notation. The Laplace matrices are identical for each atom (since there is no potential) and only a single matrix is created per MPI process. There is no atom specific solution array, only the globally distributed solution containing both atomic and interstitial points.

The Interstitial Mesh

The interstitial mesh is known globally on each MPI processes. The full matrix, however, is divided by row with each process is assigned a local piece. This data distribution must adhere to the format required within the distributed linear system solver, so no additional communication is necessary. The matrix size is forced to be divisible by P, the total number of processes. To guarantee this the interstitial matrix is appended with an identity block of up to P-1 additional non-zero elements. This has no effect on the solution and could be changed in the code, but does not seem to be necessary at this point.

Each MPI process is assigned a local piece of the interstitial matrix once-and-forall at the beginning of the code. The matrix is divided by row into equal pieces. The solution vector is also partitioned this way, by row. The starting and ending rows defined for each MPI process are stored in two variables within the interstitial mesh structure. These values can be accessed via a 'struct' within the code defined for the interstitial mesh as: mesh_it%Nstart and mesh_it%Nend, shown in FORTRAN notation.

The interstitial mesh contains overlap points at the interface of each atomic region. These overlap points must be treated more specifically. The boundary conditions for the interstitial mesh at these points are determined from the atom centered meshes. To avoid complex and possibly costly communication between processes, the interstitial mesh is permuted so that each MPI process owns the interface nodes that correspond to its assigned atoms. This means that the structure of the interstitial matrix will change for different number of MPI processes.

The exact location of these interface nodes is known only through a mapping vector. The location of the interface nodes in the interstitial mesh is given by map(i, j), where *i* is the known location of the interface node in the atomic mesh of the j^{th} atom. The mapping returns the location of the interface node within the global interstitial matrix. This must be accounted for when accessing this element in the code.

$$X_{atom-j}(i,:) = X_{itloc}(map(i,j) - mesh3d_{-i}t\%Nstart + 1,:)$$

$$(2.70)$$

2.3.3 Time propagation for TDDFT

Given an initial condition $\psi(x, t_0)$, the exact solution to the time dependent Kohn Sham equations is given by the expression,

$$\psi(r, t_f) = U(t_f, t_0)\psi(r, t_0), \qquad (2.71)$$

where

$$U(t_f, t_0) = \frac{1}{i\hbar} \tau \exp\left\{\int_{t_0}^{t_f} H(t)dt\right\},\tag{2.72}$$

is the time propagation operator and τ is a time-ordering operator.

With TDDFT the Hamiltonian gains its time dependence, in a non-linear way, from the electron density computed from the single particle time dependent wave functions. As the wave functions evolve, the electron density moves, which will effect both the Hartree potential and the exchange and correlation potentials.

For the case of a time independent Hamiltonian H, the time dependent wave functions can be determined directly by approximating the exponential of a matrix, which can be calculated in a number of way; commonly as a polynomial expansion or by diagonalizing the matrix. Importantly, the time ordering operator disappears from (2.72):

$$U(t_f, t_0) = \frac{1}{i\hbar} \exp\left\{\int_{t_0}^{t_f} Hdt\right\}.$$
(2.73)

A common approach with a time dependent Hamiltonian H(t) is to discretize the time domain fine enough so that it can be approximated as constant over some interval. In this case (2.73) can be modified,

$$\psi(r, t_f) = \frac{1}{i\hbar} \prod_{i=0}^{f-1} \exp\left\{\int_{t_i}^{t_{i+1}} H(t_i) dt\right\} \psi(r, t_0), \qquad (2.74)$$

to perform a series of time propagations over each interval. Now, the time propagation operator between time steps can be approximated or applied to perform the time evolution.

Although recent some work on NESSIE has used a direct diagonalization approach [52] in conjunction with the FEAST Eigenvalue Solver to compute the matrix exponential, it can be costly in both computation and memory. However, it is also

associated with some benefits: the ability to use larger time steps. It also has some potential for parallelization. The multiple levels of parallelism in FEAST can be taken advantage of in order to partially diagonalize the matrix.

The current massively parallel version of NESSIE uses instead the Crank Nicolson time propagation method [60]. This has been a common approach in real-time TDDFT [46, 45]. Crank Nicolson approximates the propagation operator as,

$$\tilde{U}(t_{i+1}, t_i) \approx \frac{1 - \frac{i}{2}H(t_i)\Delta t}{1 + \frac{i}{2}H(t_i)\Delta t},$$
(2.75)

where $\Delta t = t_{i+1} - t_i$. This is an implicit method that arises from a combination of forward and backward traveling Euler method. It results in a complex linear system to be solved at each time step:

$$\left(1 + \frac{i}{2}H(t_i)\Delta t\right)\psi(r, t_{i+1}) = \left(1 - \frac{i}{2}H(t_i)\Delta t\right)\psi(r, t_i).$$
(2.76)

The wave function for the next time step are computed from the Hamiltonian and wave functions of the current time step. In the context of NESSIE using a finite element discretization, there is a "mass matrix" S from the non orthogonal basis. In this case the previous equation can be updated straight-forwardly:

$$\left(S + \frac{i}{2}H(t_i)\Delta t\right)\psi(r, t_{i+1}) = \left(S - \frac{i}{2}H(t_i)\Delta t\right)\psi(r, t_i).$$
(2.77)

Crank Nicolson is often coupled with a predictor-corrector scheme, where the Hamiltonian is evaluated at the midpoint between time steps. However, it has been found that this makes little difference and is not used for any of the results presented in this dissertation.

2.4 Numerical algorithms

The mathematical modeling step describes the physics in terms of linear algebra. The resulting problems must then be solved with numerical algorithms. The most important of these, in the context of this dissertation, are solving linear systems and eigenvalue problems. Many different algorithms exist and may be appropriate depending on the storage format for the matrix (i.e. dense, sparse or banded) or the matrix properties. Furthermore, for large-scale problems, different use-cases and the physics behind the problem can determine the best choice for an algorithm.

2.4.1 Methods for sparse linear systems

Linear systems represent one of the most important linear algebra operations performed in NESSIE. Not only are they used to compute the Hartree potential by solving the Poisson equation, they are also the backbone of the FEAST eigenvalue solver used for the ground state Kohn Sham eigenproblem. The finite element method can be considered a linear scaling method since it produces sparse system matrices. This is a major benefit of the approach. Discretization methods that produce dense matrices can not be considered linear scaling approaches. This is due to the fact that the matrix scales as $O(n^2)$ with the number of basis functions. The dimension of sparse finite element matrices also scale as $O(n^2)$, but the number of non-zero elements per row is constant. The total number of non-zero elements in the matrix then scales as O(n).

In general, sparse linear system solvers can be classified as direct methods or iterative methods. Direct solvers can be used for small to medium sized problems and multi-frontal methods [64, 138] are the standard approach. Multiple software packages exist and include, most notably, MUMPS [15, 14], PARDISO [181, 180], UMFPACK [63, 62], and SuperLU [134]. The major drawback with direct methods is that they must explicitly factorize the matrix before the solution can be computed. This results in a large amount of fill-in, where the sparse matrix is augmented with dense blocks for which the LU factorization is performed.

On the other hand, because forming the matrix may not even be feasible, very large problems necessitate the use of iterative methods [176]. Iterative methods perform matrix-vector multiplications and do not need to factorize the matrix. However, sparse matrix-vector operations are far less efficient than for dense matrices since the non-zero structure is unknown and the cache cannot be meticulously managed.

Jacobi and Gauss-Seidel and SOR are the most basic iterative techniques. More important techniques are based on the use of a Krylov subspace to approximate the matrix inverse and include Arnodi's method [21], GMRES [175], Conjugate Gradient. To achieve good performance iterative methods must generally be combined with a preconditioning step. In many cases the preconditioning is achieved with the help of a direct solver that computes the LU factorization of a subset of the matrix; for example, a Jaccobi preconditioner factorizes the block diagonal components of the matrix, incomplete preconditioners partially factorize so that no additional non-zero elements are included from the sparse matrix, and other threshold methods further sparsify the matrix by dropping small non-zero entries. Many variations of iterative methods exist, including different types of Krylov techniques and preconditioners that are specific to a certain physical problem. These will not be discussed further since this work uses direct solvers exclusively.

2.4.2 The muffin-tin domain decomposition solver

When using domain decomposition, as with NESSIE, two general methods exist for solving linear systems. The first, which is not used in NESSIE, is the Schwartz method [188, 41]. This is an iterative techniques, different than an iterative solver, which defines overlapping domains. The solution in one region is used to define the boundary condition for other overlapping domains. The approach used in NESSIE [132, 131] is equivalent to the Schur complement technique. The Schur complement approach can be a direct method, where the solution in each independent region is used to form a set of complementary equations call the Schur complement. Both forming and solving the Schur complement are generally much more difficult than the solution in the independent domains. Many times the Schur complement is not explicitly formed and instead solved with an iterative technique. This has recently been applied to the context of linear systems that arise in FEAST [104]. However, in NESSIE the Schur complement is itself a sparse matrix and solved with a distributed memory sparse direct solver [113].

The Schur complement procedure arises by row reducing the matrix equation AX = Y to an upper block format, resulting in the equation

$$\begin{bmatrix} C & E \\ 0 & S \end{bmatrix} \begin{bmatrix} X_l \\ X_e \end{bmatrix} = \begin{bmatrix} Y_l \\ Y_e - E^T C^{-1} Y_l \end{bmatrix},$$
 (2.78)

where X_l and X_e refer to "local" and "external" pieces of the vector. The quantity

$$S := (D - E^T C^{-1} E) (2.79)$$

is known as the Schur complement. The solution to the linear-system can be obtained in a three step process.

(i) Solve:
$$CT = Y_l$$

(ii) Solve: $SX_e = Y_e - E^T T$
(iii) Solve: $CX_l = Y_l - E^T X_e$

The first and third steps can be trivially parallelized since C has a block diagonal structure:
$$\begin{bmatrix} C_1 & & \\ & C_2 & \\ & & \ddots & \\ & & & C_p \end{bmatrix} \begin{bmatrix} q_l^1 \\ q_l^2 \\ \vdots \\ q_l^p \end{bmatrix} = \begin{bmatrix} y_l^1 \\ y_l^2 \\ \vdots \\ y_l^p \end{bmatrix}.$$
(2.80)

The difficulty rises in steps (ii). In particular the formation of the Schur complement is non-trivial and requires an inversion of the C matrix, or equivalently the solution to CT = E. The solution T could be too large to form depending on the number of connections between subdomains. Also, the Schur matrix S is not block diagonal as was the case for C. Efficiently solving the Schur complement equation of step (ii) in parallel for X_e is much more demanding and requires communication between MPI processes.

In NESSIE, the independent blocks C_i correspond to the atom matrices. These are distributed amongst MPI processes as described in Section 2.3.2.2, along with the interstitial matrix. The Schur complement is formed from the interconnecting region, which is a 3-dimensional mesh. This results in a sparse Schur complement matrix, even when including the dense blocks of points that result from $E^T C^{-1} E$ of (2.79). Since the atom mesh has a small number of interface points, these dense blocks are small and the resulting Schur complement is still very sparse (see Figure 2.6). This is then solved with the cluster version of the PARDISO solver [105] included in Intel's Math Kernel Library.

The effect of this approach on the matrix can be seen in Figure 2.6 for a six atom system. This shows the full matrix on the left, which is never actually formed for the muffin solver. It contains six smaller dense blocks that correspond to each of the atoms. With the muffin solver each of these atom blocks would be an independent matrix. The interstitial matrix on the right, where the six atoms blocks have been replaced by 98×98 dense blue blocks labeled by Σ , is still very sparse since the interface with each atom contains only 98 points. Here we show the full interstitial



Figure 2.6: Structure of a six atom finite-element DFT matrix permuted for 2 L3 MPI processes (3 atoms per process). The left shows the matrix built from the full mesh including interstitial and atomic regions. The points that correspond to atomic regions are labeled at_1-at_6 . The interstitial matrix can be seen on the right, including boundary conditions $\Sigma_1 - \Sigma_6$ (in blue) that account for atomic points. The domain-decomposition approach compresses the large sparse set of atomic point on the left into the 98 × 98 dense blocks Σ on the right.

matrix, but in general it is also distributed over MPI processes by row. In this case it has been permuted for two MPI processes, with half of the interstitial matrix existing on each.

2.4.3 The FEAST algorithm

The FEAST algorithm [161] and associated software package [163, 162] is an accelerated subspace iterative technique for computing interior eigenpairs that makes use of a rational filter obtained from an approximation of the spectral projector. FEAST can be applied for solving both standard and generalized forms of Hermitian or non-Hermitian problems, and belongs to the family of contour integration eigensolvers [178, 179, 22, 98, 99, 24]. Once a given search interval is selected, FEAST's main computational task consists of a numerical quadrature computation that involves solving independent linear systems along a complex contour. The algorithm can exploit natural parallelism at three different levels: (i) search intervals can be treated separately (no overlap), (ii) linear systems can be solved independently across the quadrature nodes of the complex contour, and (iii) each complex linear system with multiple right-hand-sides can be solved in parallel. Within a parallel environment, the algorithm complexity becomes then directly dependent on solving a single linear system.

The FEAST algorithm utilizes spectral projection and subspace iteration to obtain selected interior eigenpairs. A Rayleigh-Ritz procedure is used to project matrices Aand B onto a reduced search subspace to form matrices

$$A_q = Q^H A Q \quad \text{and} \quad B_q = Q^H B Q. \tag{2.81}$$

Approximate eigenvalues $\tilde{\Lambda}$ and eigenvectors \tilde{X} of the original system (i.e. Ritzvalues and Ritz-vectors) can then be recovered from the solutions of the much smaller eigenvalue problem

$$A_q W_q = B_q W_q \Lambda_q \tag{2.82}$$

as

$$\tilde{X} = QW_q \quad \text{and} \quad \tilde{\Lambda} = \Lambda_q.$$
 (2.83)

Initializing \tilde{X}_m as a set of m random vectors and obtaining Q_m after QR factorization of \tilde{X}_m , results in a standard subspace iteration (i.e. power method) that converges linearly toward the dominant eigenpairs [174]. Many other sophisticated Krylov-based methods have also been developed to improve the convergence rate for the calculation of selected smallest, largest or interior eigenpairs [128, 116, 35, 191]. The subspace iteration technique, in turn, can be efficiently used for solving the interior eigenvalue problem when it is combined with filtering which aims to improve the convergence by increasing the gap between wanted eigenvalues and unwanted ones. It is well known that Ritz-pairs $(\tilde{X}_m, \tilde{\Lambda})$ converge toward the true eigenpairs (X_m, Λ) at a rate determined by the filter [174, 153, 197].

In theory, the ideal filter for the Hermitian problem would act as a projection operator $X_m X_m^H B$ onto the subspace spanned by the eigenvector basis, which can be expressed via the Cauchy integral formula,

$$\rho(\lambda) = \frac{1}{2\pi i} \oint_{\Gamma} dz (z - \lambda)^{-1}, \qquad (2.84)$$

as:

$$\rho(B^{-1}A) = \frac{1}{2\pi i} \oint_{\Gamma} dz (zB - A)^{-1}B = X\rho(\Lambda)(BX)^{-1}.$$
 (2.85)

Since the Cauchy integral goes to unity for points interior to the contour and zero for points exterior, $\rho(\Lambda)$ is a diagonal matrix with elements equal to one for eigenvalues that lie inside Γ and zero for all others. Thus,

$$\rho(B^{-1}A) \equiv X_m X_m^H B, \qquad (2.86)$$

where X_m are the eigenvectors associated with eigenvalues inside a given search interval delimited by the closed curve Γ . The projection operator for the Rayleigh-Ritz procedure can be taken as,

$$Q = \rho(B^{-1}A)Y,$$
 (2.87)

where Y is any set of linearly independent vectors that will be projected onto the eigenvector subspace.

In practice, the spectral projector must be approximated using a quadrature rule using n_e integration nodes and weights $\{(z_j, \omega_j)\}_{j=1,\dots,n_e}$:

$$\rho(B^{-1}A) \approx \tilde{\rho}(B^{-1}A) = \frac{1}{2\pi i} \sum_{j=1}^{n_e} \omega_j (z_j B - A)^{-1} B.$$
(2.88)

In this case, the Cauchy integral cannot be computed exactly and the diagonal elements of $\tilde{\rho}(\Lambda)$,

$$\tilde{\rho}(\lambda_i) = \frac{1}{2\pi i} \sum_{j=1}^{n_e} \omega_j (z_j - \lambda_i)^{-1}, \qquad (2.89)$$

are not exactly one or zero and depend on their proximity to the contour. Eigenvalues that are well inside or outside the contour will have values very close to one and zero, respectively. Eigenvalues near the contour edge, however, will have a significant component and will be represented in Q. In fact, this rational function,

$$\rho_a(z) = \frac{1}{2\pi i} \sum_{j=1}^{n_e} \frac{\omega_j}{(z_j - z)},$$
(2.90)

will define the convergence rate of FEAST.

In practice, the projector itself cannot be formed directly and instead its action onto a set of vectors must be computed:

$$Q_{m_0} = \tilde{\rho}(B^{-1}A)\tilde{X}_{m_0} = \sum_{j=1}^{n_e} \omega_j (z_j B - A)^{-1} B\tilde{X}_{m_0}, \qquad (2.91)$$

where we also consider a search subspace of size $m_0 \ge m$. The computation of Q_{m_0} amounts to solving a set of n_e complex shifted linear-systems

$$(z_j B - A)Q_{m_0}^{(j)} = B\tilde{X}_{m_0}$$
 with $Q_{m_0} = \sum_{j=1}^{n_e} \omega_j Q_{m_0}^{(j)}$. (2.92)

This matrix Q_{m_0} is then used as the Rayleigh-Ritz projection operator to form reduced matrices A_q and B_q of (2.81). If the exact spectral projector was known, solving the reduced eigenproblem in (2.82) will produce the exact eigenvalues $\tilde{\Lambda} = \Lambda_q = \Lambda$ and eigenvectors $\tilde{X} = QW_q = X$. However, since it is only approximated, the Ritz-values Λ_q and updated Ritz-vectors \tilde{X} are only an approximation to the true eigenpairs. Subspace iteration will then, in effect, tilt the subspace spanned by columns of \tilde{X}

Algorithm 1 The FEAST Algorithm

1: **input:** $A, B, \tilde{X}_{m_0}, \{(z_j, \omega_j)\}_{1,...,n_e}, \epsilon$ 2: while $(||A\tilde{X}_m - B\tilde{X}_m\Lambda_m|| > \epsilon)$ do 3: $Q_{m_0} = 0$ for $(j = 0; j < n_e; j = j + 1)$ do 4: $Q_{m_0}^{(j)} \leftarrow (z_j B - A)^{-1} B \tilde{X}_{m_0}$ 5: $Q_{m_0} \leftarrow Q_{m_0} + \omega_j Q_{m_0}^{(j)}$ 6: end for 7: $B_q = Q^H B Q$ $A_a = Q^H A Q$ 8: Solve $A_q W_q = B_q W_q \Lambda_q$ 9: $\tilde{X}_{m_0} = Q_{m_0} W_q \qquad \tilde{\Lambda} = \Lambda_q$ 10: 11: end while 12: output: $\tilde{X}_m, \tilde{\Lambda}_m$

toward the desire eigenspace. At convergence we will obtain $\tilde{X} = Q = X$ and $\tilde{\Lambda} = \Lambda$. The general outline can be seen in Algorithm 1 for computing *m* eigenpairs in a given search interval. The input \tilde{X} can be chosen as a set of m_0 random vectors or a previously calculated solution to a closely related problem.

2.4.4 From Hermitian FEAST to non-Hermitian FEAST

The work on DFT and TDDFT in this dissertation does not need to solve non-Hermitian eigenvalue problems. Although Hermitian matrices are much more common, electronic structure and quantum chemistry applications do often create non-Hermitian matrices,

$$A \neq A^H, \tag{2.93}$$

which are are not equal to their own conjugate transpose. These types of eigenvalue problems can arise in open systems [74, 130, 165], the use of absorbing boundary conditions [20, 152], complex scaling approaches [47], and computing complex band structures [125].

By allowing the search contour to be placed at arbitrary locations in the complex plane, the FEAST algorithm can be naturally extended to non-Hermitian problems which produce complex eigenvalues [114, 196]. The algorithm retains most of the properties of Hermitian FEAST including the multi-level parallelism. We note, however, a few theoretical and practical difficulties that distinguish the non-Hermitian eigenvalue problems from Hermitian ones, including: (i) defective systems, which must be treated using the Schur or Jordan forms; (ii) bi-orthogonality for dual right and left eigenvector subspaces; (iii) ill-conditioned eigenvalue problems that produce sensitive eigenvalues in finite precision arithmetic; (e.g. if a FEAST quadrature node lies near a complex eigenvalue).

The key point at which the non-Hermitian FEAST algorithm differs from the Hermitian one is the use of dual subspaces. Since the left and right eigenvectors do not necessarily lie in the same subspace, two separate projectors must then be calculated in order to recover both sets of vectors. A single sided algorithm where only the right subspace is used to project is also possible [195] but will not return a B-bi-orthogonal subspace of left and right eigenvectors, which can be of interest for many applications. The non-Hermitian algorithm is similar to its Hermitian counterpart. A comparison between the main numerical operations for the two algorithms is briefly outlined in Figure 2.7 (quantities associated with the left eigenvectors are written with a '^' symbol). A detailed description of the algorithm and software implementation can be found in the appendix.

2.4.5 **PFEAST:** three levels of MPI parallelism

Inherent to the FEAST algorithm are three separate levels of parallelism which will be denoted as L1, L2 and L3. At level L3, each linear system can be solved in parallel. Parallelism for the linear system solutions can easily be accomplished by using threaded linear system solvers such as the dense solvers in LAPACK, the PARDISO [180, 100] or MUMPS [13] packages for sparse systems, and SPIKE-SMP [145, 190] for banded matrices. This level has been upgraded. The FEAST computational kernel now allows for a distributed memory linear-system solver [113]. In this

Hermitian FEAST

Solving: $AX_m = BX_m\Lambda_m$ $A^H \hat{X}_m = B^H \hat{X}_m\Lambda_m^*$ **Solving:** $AX_m = BX_m\Lambda_m$ $[\Lambda_m]_{ii} \subseteq \mathcal{C}$ $[\Lambda_m]_{ii} \subseteq [\lambda_{min}, \lambda_{max}]$ Inputs: $A = A^H$, B hpd; $m_0 \ge m$; **Inputs:** A and B general; $m_0 \ge m$; $\{z_1,\ldots,z_{n_e}\}, \{\omega_1,\ldots,\omega_{n_e}\}$ $\{z_1, \ldots, z_{n_e}\}, \{\omega_1, \ldots, \omega_{n_e}\}$ $Y_{m_0} \leftarrow m_0$ initial vectors $Y_{m_0}, \hat{Y}_{m_0} \leftarrow m_0$ initial vectors; repeat repeat $Q_{m_0} = 0$
for $j = 1, n_e$ $Q_{m_0} = \widehat{Q}_{m_0} = 0$ for $j = 1, n_e$ $Q_{m_0}^{(j)} \leftarrow (z_j B - A)^{-1} B Y_{m_0};$ $\begin{array}{l} Q_{m_0}^{(j)} \leftarrow (z_j B - A)^{-1} B Y_{m_0}; \\ \widehat{Q}_{m_0}^{(j)} \leftarrow (z_j^* B^H - A^H)^{-1} B^H \widehat{Y}_{m_0} \end{array}$ $Q_{m_0} \leftarrow Q_{m_0} + \omega_j Q_{m_0}^{(j)}$ end $\begin{aligned} Q_{m_0} &\leftarrow Q_{m_0} + \omega_j Q_{m_0}^{(j)} \\ \widehat{Q}_{m_0} &\leftarrow \widehat{Q}_{m_0} + \omega_j^* \widehat{Q}_{m_0}^{(j)} \end{aligned}$ end $B_Q \leftarrow Q_{m_0}^H B Q_{m_0}$ Check B_Q hpd (resizing step) $A_Q \leftarrow Q_{m_0}^H A Q_{m_0}$ Solve $A_Q W = B_Q W \Lambda_Q$; $W^H B_Q W = I$ end $B_Q \leftarrow \widehat{Q}_{m_0}^H BQ_{m_0}$ Check B_Q non-singular (resizing step) $Y_{m_0} \leftarrow Q_{m_0} W$ $\begin{array}{l} A_Q \leftarrow \widehat{Q}_{m_0}^{\dot{H}} A Q_{m_0} \\ \text{Solve } A_Q W = B_Q W \Lambda_Q \text{ and} \end{array}$ **until** Convergence of Y_m , Λ_{Q_m} with $[\Lambda_{Q_m}]_{ii} \subseteq [\lambda_{min}, \lambda_{max}]$ $A_Q{}^H \widehat{W} = B_Q{}^H \widehat{W} \Lambda_Q^*; \ \widehat{W}^H B_Q W = I$ **Output:** $X_m \equiv Y_m \ (X_m^H B X_m = I_m);$ $Y_{m_0} \leftarrow Q_{m_0} W, \quad \widehat{Y}_{m_0} \leftarrow \widehat{Q}_{m_0} \widehat{W}$ $\Lambda_m \equiv \Lambda_{Q_m}$ **until** Convergence of $Y_m, \hat{Y}_m, \Lambda_{Q_m}$ with $[\Lambda_{Q_m}]_{ii} \subseteq \mathcal{C}$ Output: $X_m \equiv Y_m;$ $\hat{X}_m \equiv \hat{Y}_m \ (\hat{X}_m^H B X_m = I_m);$ $\Lambda_m \equiv \Lambda_{Q_m}$

Non-Hermitian FEAST

Figure 2.7: Brief outlook and comparison between the main numerical operations for the FEAST algorithms applied to the Hermitian and non-Hermitian problems. case, an additional level of threaded parallelism takes place within level L3 and is inherent to the specific MPI system solver implementation.



Figure 2.8: Overlapping MPI communicators for PFEAST. We refer to the collection of MPI processes within an MPI communicator as a "Communication-World". If the first level of parallelism was also used, this picture would then represent a single **L1**-Communication-World.

The second and third levels of parallelism overlap and must be managed. This results in two separate MPI-Communicators shown in Figure 2.8. We refer to the collection of MPI processes within an MPI communicator as a "Communication-World". At the second level, the **L2** communicator (defining each **L2**-Communication-World) is a carry-over from the previous FEAST distribution and specifies the mapping between MPI processes and quadrature nodes (i.e. linear systems). Each MPI process in **L2** maps directly to a set of quadrature nodes. Ideally, the number of members within each **L2** communicator will be equal to the number of quadrature nodes and each linear-system is solved in parallel. In this case, using a direct solver, each ma-

trix factorization $(z_i B - A)$ must be computed only once since subsequent PFEAST iterations solve a linear system at the same complex pivot, but with an updated set of right-hand-sides. The set of processes at the third level of parallelism (i.e. L3-Communication-World) is to be used by the distributed memory solver. In Figure 2.8 there are six total MPI processes P-0 through P-5. L2 communicator (MPI) ranks F-0, F-1 and F-2 map to contour integration nodes z_0 , z_1 and z_2 , where a linear system $(z_i B - A)Q = BY$ must be solved. Each integration node then owns exactly two L3 MPI processes S-0 and S-1 to be used by the distributed memory solver.

Dividing parallel resources among the second and third levels of parallelism results in a trade-off between memory and performance. The second level represents ideal linear scaling, since each linear-system can be solved independently. However, it also requires more memory. Each cluster of **L3** MPI processes (within an **L3**-Communication-World) will require a copy of the matrix. Placing more MPI processes at the third level of parallelism, in turn, will reduce the amount of memory required to store the matrix and eigenvector solutions (since they are distributed across the **L3** processes), which could become essential to many large-scale applications.

If the amount of memory required to store the matrix and eigenvector solutions is too large, it can be reduced using the first and third levels of parallelisms **L1** and **L3**. The **L1** level of parallelism subdivides the FEAST search interval resulting in fewer calculated eigenpairs per subinterval. As shown in Figure 2.9, this reduces the number columns in the eigenvector matrix per MPI process. Additional MPI processes at the **L3** level will allow the matrix and eigenvector solutions to be distributed by row. If both **L1** and **L3** are used each MPI process will contain a subset of columns and rows as seen in the bottom right of Figure 2.9.

All MPI solvers must adhere to a specific pre-defined data distribution, which can differ for each implementation. The solver defines a data distribution for both the matrix and rhs/solution vectors. PFEAST will require a predefined data distribution



Figure 2.9: Data distribution of input matrix A and eigenvector solutions X for different configurations of parallel resources. Additional MPI processes at the **L1** and **L3** levels reduce the memory required to store the matrix and solution. The **L1** level subdivides the PFEAST search interval and results in fewer number of eigenvectors calculated per node. Likewise, the **L3** level can be used to distribute X by row and A (as well as B for generalized problems) by the distributed memory solver format. MPI processes at the **L2** level would result in additional copies of both A and X.

for its kernel as well. The distribution format for PFEAST is defined for the eigenvectors X (as well as Q and Ritz-vectors \tilde{X}), and is independent of the distribution format of matrices A and B. The kernel then operates directly on the distributed vectors.

The eigenvectors data distribution has been defined to be 1-dimensional and by row. Each MPI process within an **L3** communicator stores and operates on a specific subset of rows and all corresponding columns. The choice of a 1-dimensional distribution can be justified by the fact that PFEAST calculates only a subset of m_0 eigenvectors; the number of rows n in the eigenvector matrix will, in general, be much larger than the number of columns (i.e. $n \gg m_0$). Additionally, the eigenvector columns can be distributed independently of the PFEAST kernel with the first level of parallelism **L1**. The data distribution will be the same for the right-hand-side vector supplied by PFEAST and linear-system solutions returned to PFEAST as they are directly related to the eigenvectors and have the same matrix dimensions. Since the MPI solver must operate on rhs/solution vectors in its own format, a reordering step with communication between the MPI processes within each **L3** communicator could be necessary.

The distribution of the eigenvectors is required to be the same for each of the L3 communicators; i.e. the same number of MPI processes is applied to each linearsystem and equivalent processes for different L2 communicators must correspond to the same eigenvector partition. Both the linear-system and eigenvectors solutions are, in general, dense and will be stored in a matrix using the following 1-dimensional distribution:

$$Q_k = \begin{bmatrix} q_1^T, & q_2^T, & \dots, & q_s^T \end{bmatrix}^T.$$
 (2.94)

Here the k^{th} L2 (MPI) rank has its solution vector distributed across s L3 MPI processes.

The data distribution will match for the members of each L2 communicator and the distributed Ritz-vectors Q can be found through an all-reduce operation; i.e. scaling each linear-system solution by the corresponding integration weight and summing the results. The PFEAST kernel then operates only on a local subset of rows q_i within Q_k . This row range is specified in two entries within the PFEAST parameter array and does not need to be equal for each L3 MPI process. Users can implement their own matrix multiplication routine or linear system solver as long as the result is placed back into the correct position within the 1-dimensionally distributed Q_k matrix.

It happens that, if the second level of parallelism L2 is used, a copy of Q will be stored for each L2 (MPI) rank. Subsequently, at convergence multiple copies of the eigenvectors $X = \tilde{X} = Q$ will also be stored across the **L2** communicator (i.e. for each **L3**-Communication-World). However, since PFEAST must compute two innerproducts of the form $A_q = Q^H A Q$ and $B_q = Q^H B Q$ further parallelization can be achieved. Moreover, since all columns of Q are known across the **L2** (MPI) ranks, only a small $m_0 \times m_0$ communication is necessary.

After the contour integration has been evaluated to find Q, all MPI processes from **L2** and **L3** become available. Both matrix multiplications $Y = \{AQ, BQ\}$ (outside kernel) and $\{A_q, B_q\} = Q^H Y$ (inside kernel) can take advantage of additional **L2** MPI processes. To make use of all MPI processes Q is subdivided and given a 2-dimensional decomposition. The matrix Q, which has already been distributed by row across the s MPI processes within an **L3** communicator, is now further distributed by column across the f **L2** processes. This results in the 2-dimensional decomposition

$$Q = \begin{bmatrix} q_{11} & q_{12} & \dots & q_{1f} \\ q_{21} & q_{22} & \dots & q_{2f} \\ \vdots & \vdots & \ddots & \vdots \\ q_{s1} & q_{s2} & \dots & q_{sf} \end{bmatrix}.$$
 (2.95)

Each block q_{ij} of matrix Q matches to exactly one MPI process. However, all columns of the matrix $q_i = [q_{i1}, ..., q_{if}]$ are known for each MPI rank in the **L2** communicator. That is, each MPI process within an **L2** communicator has in memory all the columns of the 1-dimensionally distributed q_i , but will only be performing the multiplication for one piece q_{ij} of the 2-dimensional distribution. The multiplication result for each q_{ij} must be placed back into the correct location within the matrix Y. After the multiplication is performed on all $s \times f$ MPI processes, the matrix $Y = \{AQ, BQ\}$ will have the same decomposition as (2.95) except that each block is known only on a single MPI process. The matrix multiplication is needed for three separate stages of the algorithm and the operations within the kernel, although transparent to the user, differ slightly. First, it is used to compute the Rayleigh-Ritz projection for $\{A_q, B_q\}$. In this case, the kernel leaves Y in its 2-dimensional distribution and the matrix multiplication has the form

$$A_{q} = Q^{H}AX = Q^{H}Y = \begin{bmatrix} q_{1}^{H} & q_{2}^{H} & \dots & q_{s}^{H} \end{bmatrix} \times \begin{bmatrix} y_{11} & y_{12} & \dots & y_{1f} \\ y_{21} & y_{22} & \dots & y_{2f} \\ \vdots & \vdots & \ddots & \vdots \\ y_{s1} & y_{s2} & \dots & y_{sf} \end{bmatrix}, \quad (2.96)$$

where each local chunk y_{ij} is known on a single MPI process only. The vector q_i , however, is known for all f MPI processes with the L2 communicator. When performing the dense multiplication $\{A_q, B_q\} = Q^H Y$ inside the kernel each $y_{i1}, ..., y_{if}$ must be multiplied with q_i^H . By storing the result of each $q_i^H y_{ij}$ to the correct location within $\{A_q, B_q\}$, only a small $m_0 \times m_0$ communication is necessary. Additionally, both AQ and BQ are needed in the computation of the eigenvector residuals. The result $Y = \{AQ, BQ\}$ is again left in its 2-dimensional distribution. This result is independent for each eigenvector and therefore for each MPI rank within an L2 communicator. And, since the L_1 norm is used, the final residual for each vector can be found by summing the scalar result within each of the **L3**-Communication-Worlds. After the residuals are determined, the final (scalar) values are communicated among all MPI processes. Also required is the right-hand-side for the linear-systems, which must be updated as Y = BQ at the beginning of each PFEAST iteration. Here y_i , the 1-dimensionally distributed piece of Y, is required for each MPI process within the L2 communicator and a reduction operations must be performed across the L2 communicator, summing the results for each L3-Communication-World.

2.4.5.1 **PFEAST** with a distributed-memory solver

Solving the linear-systems in (2.92) will be the dominant computational procedure performed in the eigenvalue calculation. Different "black-box" direct distributed memory solvers have been integrated with the PFEAST kernel. An interface has been created for two sparse direct solvers: the cluster version [105] of PARDISO [180] within the Intel Math Kernel Library [100] and MUMPS [13]. A banded interface has also been created for the SPIKE-MPI [167], although we do not present any results for banded matrices. Additionally, we highlight the versatility of the PFEAST kernel by describing how to integrate a Schur complement type solver.

Each interface requires a matrix-matrix multiplication routine specific to the distribution format accepted by the solver. To interface these solvers with the PFEAST kernel, linear-system solution and matrix multiplication results must be placed back into the correct row corresponding to the PFEAST distribution for the eigenvectors. Cluster-MKL-PARDISO offers multiple options for distributing the matrix and rhs/solution vector. The matrix and vectors can be distributed 1-dimensionally by row across the L3 MPI processes or stored locally on the head node. We have chosen to distribute both the matrix and the right-hand-side vector for the linear-system solver in order to match the eigenvector distribution. There is then no need for a reordering step since the PFEAST data distribution matches with the solver. MUMPS also offers multiple options for the matrix distribution and we have chosen to manually distribute the matrix by row to be consistent. However, MUMPS requires the rhs/solution to be fully constructed on the head node and an additional communication is then needed before and after the solve stage.

The PFEAST kernel can also be integrated within a domain decomposition framework. One approach is to divide the matrix A using a graph partitioner. The reordering will create a 2 × 2 block matrix

$$A = \begin{bmatrix} C & E \\ E^T & D \end{bmatrix}, \qquad (2.97)$$

where C contains the independent blocks and E and D contain the connections between them. Linear-systems possessing this block form can be solved with a Schur complement approach described in Section 2.4.2.

With PFEAST we are interested in a series of linear-systems in (2.92). The matrix $(z_j B - A)$ can be reordered to posses a block form in (2.97). The goal of domaindecomposition is then to obtain the solution to this linear systems in parallel using the Schur complement. The matrix reordering will implicitly partition vectors Q and $Y = B\tilde{X}$ (PFEAST inputs/outputs) into "local" and "external" pieces

$$Q \equiv \begin{bmatrix} Q_l \\ Q_e \end{bmatrix} \quad and \quad B\tilde{X} = Y \equiv \begin{bmatrix} Y_l \\ Y_e \end{bmatrix}.$$
(2.98)

It is then convenient to think of the "local" pieces Q_l and Y_l separately from the "external" Q_e and Y_e .

The matrix C has a block diagonal form and the solution to the linear-system $CQ_l = Y_l$ can be obtained in parallel. Each linear system $C_i q_l^i = y_l^i$ can be solved independently on a different **L3** MPI process. The vectors Q_l and Y_l should also be distributed across the MPI processes in the same manner so that C_i , q_l^i and y_l^i all exist on the same node.

The Schur complement solve in step (ii) should also be computed in parallel by distributing the vectors Q_e and Y_e among the **L3** MPI processes. The PFEAST kernel is independent of the ordering. Each piece of Q (resp. Y), local to the i^{th} **L3** MPI process would contain a subset of rows from both Q_l (resp. Y_l) and Q_e (resp. Y_e):

$$q_{i} = \begin{bmatrix} q_{l}^{i} \\ q_{e}^{i} \end{bmatrix} \quad with \quad Q = \begin{bmatrix} q_{1} \\ \vdots \\ q_{s} \end{bmatrix}.$$
(2.99)

The "local" components y_l would be extracted from the PFEAST work array and operated on for steps (i) and (iii). The solution q_l must then be placed back into the work array at the correct location (i.e to match (2.99)). Step (ii) would also have to extract a subset of the "external" points from the right-hand-side vector and place the solution back into the correct location.

2.4.5.2 **PFEAST** with the muffin-tin solver

Linking PFEAST and the muffin-tine solver is similar to what was described in the previous section. In this case the interstitial matrix becomes the Schur complement. The eigenvectors would then be distributed 1-dimensionally as shown in (2.99), with

$$q_{i} = \begin{bmatrix} q_{it}^{i} \\ q_{at_{1}}^{i} \\ q_{at_{2}}^{i} \\ \vdots \\ q_{at_{n}}^{i} \end{bmatrix}, \qquad (2.100)$$

where the i^{th} L3 MPI process contains one piece the corresponds to the distributed interstitial matrix q_{it}^i as well as additional points $q_{at_1}^i$ through $q_{at_n}^i$ corresponding to the set atoms that it was assigned. The interface between the interstitial region and the atoms is defined to exist in q_{it}^i . Thus, the size of each atom block $q_{at_1}^i$ is $n_{at} - N_s$, where n_{at} is the size of the atom matrix and N_s is the number of interface points, in the overlapping approach.

The PFEAST kernel accepts and returns the eigenvectors in this format. The muffin-tin solver described in Section 2.4.2 then must extract the interstitial and

atom points from the eigenvector and perform the three step procedure for solving a Schur complement-type problem.

Here, the general approach of the muffin-tin solver can be seen in Figure 2.10 when linked to PFEAST. FEAST must solve a linear system at each quadrature node with the muffin solver, each of which can be distributed over **L2** MPI processes. The muffin solver must then solve multiple linear systems itself as described in Section 2.3.2. The solution to a linear equation is computed for each of the atoms, where each is independent and can be solved in parallel using **L3** level of MPI. The solution to these problems is added into the interstitial matrix, which then becomes a Schur complement. All **L3** MPI processes are then used to solve a linear system corresponding to the interstitial matrix (i.e. with a distributed memory solver) and the final solution in this region is obtained. Another linear system must be solved for each atom, which can again be performed with perfect parallelism, to obtain the solution in atom regions.



Figure 2.10: Each linear system of FEAST is solved with the muffin solver. Linear systems for each atom matrices can be solved in parallelism, but must be solved twice: once to generated the Schur complement (interstitial matrix) and again to retrieve solution.

2.4.6 Mixing scheme for SCF convergence in DFT

The naive approach to updating the electron density at each SCF iteration is to use directly the result output from the Kohn Sham procedure. However, this approach may not converge as the initial guess for the density is usually far from the ground state solution, which will result in large oscillations between SCF iterations. Instead a mixing scheme can be employed. In this case two different electron densities must be defined for the k^{th} SCF iteration: the input density ρ_{in}^k used to construct the Kohn Sham Hamiltonian and the output density ρ_{out}^k computed from the wave functions as defined in Figure 2.1.

With simple mixing, the input electron density for the next iteration can be computed as,

$$\rho_{in}^{k+1} = (1-\beta)\rho_{in}^{k} + \beta\rho_{out}^{k}, \qquad (2.101)$$

where the parameter β is usually chosen less than 0.5. This, however, will converge very slowly.

In order to increase the convergence rate, more sophisticated methods have been developed for solving these fixed-point problems. Newton methods are impractical in electronic structure since the Jacobian matrix is unobtainable. Other quasi-Newton methods have been developed, notably by Anderson [16] and Broyden [37], which do not require the Jacobian or Hessian. These techniques were later refined in the context of SCF iteration by Pulay [169, 170] in the 1980s and have since been expanded upon [77, 68, 123, 213, 51]. They are also related to Krylov methods [177, 211] and GMRES [175]. For electronic structure calculations these iterative techniques are usually referred to as Direct Inversion of the Iterative Subspace (DIIS) methods.

The general idea is to build the input electron density as a linear combination of past densities and to compute the coefficient of expansion in a smart way. The approach used to obtain the results in this dissertation computes both an input mixing density,

$$\hat{\rho}_{in}^{k+1} = c_0 \rho_{in}^0 + c_1 \rho_{in}^1 + \dots + c_k \rho_{in}^k, \qquad (2.102)$$

and an output mixing density,

$$\hat{\rho}_{out}^{k+1} = c_0 \rho_{out}^0 + c_1 \rho_{out}^1 + \dots + c_k \rho_{out}^k, \qquad (2.103)$$

from the previous input and output densities. The input for the next iteration is chosen as a linear combination of the mixing densities:

$$\rho_{in}^{k+1} = (1-\beta)\hat{\rho}_{in}^{k+1} + \beta\hat{\rho}_{out}^{k+1}, \qquad (2.104)$$

where β is again referred to as the mixing parameter.

This approach can be truncated in order to keep the density subspaces small. We have seen that the inclusion of more densities in the mixing subspace does result in better convergence, but has diminishing returns. Keeping a history of twenty input and output densities seems to be more than sufficient. Better performance can also be obtained by choosing a larger value of β . Too large a value will result in instability, even with a DIIS method. However, as the density converges, the input and output densities become closer. A large improvement in convergence can be obtained by increasing the β parameter along the SCF iterations. In NESSIE we choose β to be proportional to the norm of the difference between input and output densities:

$$\beta = -0.1 \times \log_{10} \left| \left| \rho_{in}^{k} - \rho_{out}^{k+1} \right| \right|, \qquad (2.105)$$

where the value is also rounded up to the nearest tenth.

The coefficients $\{c_0, ..., c_k\}$ of (2.102) and (2.103) are the same for both the input and output mixing subspaces. They are computed by solving a $k \times k$ linear system with one right-hand-side,

$$M \times C = R, \tag{2.106}$$

where the i^{th} element of R is depends on the difference between the output and input densities of the current iteration k and previous iteration i,

$$r_i = \int_{\Omega} \left(\rho_{out}^k - \rho_{in}^k \right) \times \left[\left(\rho_{out}^k - \rho_{in}^k \right) - \left(\rho_{out}^i - \rho_{in}^i \right) \right] d\Omega, \qquad (2.107)$$

and each element m_{ij} of matrix M also take into account the densities of iteration j:

$$m_{ij} = \int_{\Omega} \left[\left(\rho_{out}^{k} - \rho_{in}^{k} \right) - \left(\rho_{out}^{i} - \rho_{in}^{i} \right) \right] \times \left[\left(\rho_{out}^{k} - \rho_{in}^{k} \right) - \left(\rho_{out}^{j} - \rho_{in}^{j} \right) \right] d\Omega. \quad (2.108)$$

CHAPTER 3 SIMULATION PROCESS

To perform linear response TDDFT calculations with NESSIE three steps must be completed. First, the unstructured finite element mesh is created. Next, the ground state DFT calculation is performed, to obtain the ground state electron density. Finally, the time dependent calculation can be run. Post processing of the time dependent electron density is also necessary to extract observables, such as the absorption and emission spectrum. This section will explain the step-by-step process of our modeling framework using a C_6H_6 benzene molecule as an example.

3.1 Mesh generation

The finite element muffin-tin discretization, outlined in Section 2.3.2, can be generated in a pre-processing stage. DFT and TDDFT simulations then read in the mesh data for each simulation. Higher order p-refinements, however, must be computed at run time. Only a few input parameters are necessary to set the size and granularity of both the interstitial and atomic domains. Interstitial and atomic meshes can be seen in figure 3.1. The surface and cross section of the atom mesh is shown on the top, with the right most plot a zoom. A cross section of the interstitial mesh is plotted on the bottom. With the muffin-tin domain decomposition, an empty region directly surrounding each atom is created in the interstitial mesh and resembles the benzene molecule. A zoom of the molecular region, which is more fine and has many more nodes, is also shown.



Figure 3.1: Atomic (top) and interstitial (bottom) meshes for the benzene molecule. The full cross section and zoom are shown for both, with atoms fitting into hole in the interstitial.

It is important to keep the interstitial mesh large enough so that the wave functions decay very close to zero at the boundary. Otherwise, the Dirichlet boundary conditions will result in a confinement effect. In Figure 3.2, the total energy and different eigenstates are plotted versus the size of the computational domain: varying the interstitial mesh offset from (2.0, 2.0, 2.0) to (16.0, 16.0, 16.0) angstroms. For too small a mesh, the ground state solutions will be inaccurate. As the size of the mesh is increased the ground state energies begin to flatted and remain constant. The reason we do not see monotonic decay here is due to the caclulation of the exact boundary condition for the Poisson equation. A mesh offset of (8.0, 8.0, 8.0) angstroms is used by default.



Figure 3.2: Ground state energies plotted vs. the size of the interstitial mesh offset. The total length in each direction is given by the interstitial mesh offset plus the size of the molecule (including muffin-tins). Confinement from the Dirichlet boundary conditions can have an effect on the ground state solutions if the mesh size is taken too small.

An additional question is how many quadrature nodes should be used for generating the finite element matrix. In higher dimensions the Gaussian quadrature points are no longer unique. We have used pre-generated quadrature nodes and weights. The basis functions are second degree polynomials for P2 and third degree polynomials for P3. Table 3.1 presents results for benzene using different numbers of quadrature nodes and reports the total energy E_T as well as the energy for the lowest core electron E_1 and the HOMO level E_H . The results suggest that 11 quadrature nodes (4th degree accuracy for the integration over each element) for P2 calculations and 24 nodes (8th degree) for P3 calculations are optimal since these lead to the lowest total energy.

# Gauss	4*	5^{*}	11	24	45	Ref. [129]
E_1	-265.63	-265.45	-266.39	-266.38	-266.32	-264.66
E_H	-7.02	-7.02	-7.04	-7.04	-7.04	-6.54
E_L	-1.82	-1.82	-1.86	-1.86	-1.86	
E_T	-6220.30	-6212.13	-6244.45	-6244.24	-6243.27	-6226.57
Benzene with P3 Mesh Refinement						
# Gauss	11*	24	45	70	126	Ref. [129]
E_1	-266.49	-266.49	-266.43	-266.36	-266.39	-266.38
E_H	-6.55	-6.55	-6.55	-6.55	-6.55	-6.53
E_L	-1.47	-1.47	-1.47	-1.47	-1.47	
E_T	-6262.91	-6263.41	-6262.44	-6261.38	-6261.76	-6262.57

Benzene with P2 Mesh Refinement

Table 3.1: Comparison between different accuracies for the numerical integration used to include the potential in the finite element Hamiltonian. The total energy (E_T) , lowest energy eigenstate (E_1) and HOMO level (E_H) , and LUMO level (E_L) are reported (in eV). Results were reported in [129] using a similar all-electron finite element basis and compared with the FHI-aims code with values: $E_T = -6263.83$, $E_1 = -266.44$ and $E_H = -6.53$. Note: P4 results in [129] matched very closely to FHI-aims. Note: SCF iteration did not converge for results marked with an asterisk.

3.2 DFT calculation

The ground state DFT problem is solved on the finite element mesh. Both threaded (OpenMP) and distributed (MPI) parallelism are used in the code. Because the main computation is solving the eigenvalue problems we define in NESSIE the same three levels of MPI parallelism (L1, L2, L3) that exist in FEAST. All of the mesh operations are parallelized over the three different levels. The user must specify how to distribute parallel resources and supply the definition of the FEAST eigenvalue search contours. At convergence the total energy and ground state eigen-spectrum shown in Figure 3.4 will be obtained obtain along with the ground state electron density plotted in Figure 3.3. Figure 3.3: Converged electron density for P2 finite element mesh. The top plot shows the density in the atomic mesh for a single hydrogen (left) and carbon (right) atom. The electron density in the full mesh (interstitial+atomic) is shown on the bottom. Hydrogen atoms have been clipped along the same plane as the interstitial mesh. As can be seen in the top plot, the electron density is much higher around carbon atoms than for hydrogen. The data in these regions has been replace by a volume plot for aesthetic purposes.





Figure 3.4: Ground state eigenvalues of benzene for P2 and P3 levels of mesh refinement. The total energy was calculated as -6244.44652890041 (eV) for P2 and -6263.41119295063 (eV) for P3.

3.2.1 SCF convergence

For the ground state calculation an Anderson mixing scheme outlined in Section 2.4.6 is employed to accelerate the convergence rate of the SCF procedure compared to simple mixing. Parameters for the Anderson mixing can have an effect on the convergence rate. The effect of beta mixing ratio β and the number of density mixing subspaces kept in memory can be seen in Figure 3.5. Here, because of the small number of electrons, very few mixing subspaces must be kept in memory. And, in general, incorporating additional subspaces will have diminishing returns, even for larger systems with thousands of electrons; although the effect is much more apparent for larger systems.

3.2.2 Parallelization of the search interval

For large scale molecules with more than a few hundred electrons, it becomes necessary to slice the FEAST search interval for the eigenvalue problem. This is a unique feature of the FEAST eigenvalue solver and a major benefit to electronic structure calculations. Performance benefits of parallelization at this level will be discussed in a later section. An example for C_6H_6 can be seen in Figure 3.6 that uses three 30% elliptical contours. One to compute the core states and two others



Figure 3.5: The effect of SCF parameters of on the convergence. The number of iterations are plotted until the total energy between consecutive SCF iterations reaches a relative convergence of 10^{-10} . For values of $\beta = 0.1$ and $\beta > 0.5$ the simple mixing (left) does not converge. With Anderson mixing (middle) the total energy converges much faster. For benzene only a few mixing subspaces must be kept in memory (right).

that each compute half of the valence states. This example was run on using the output of a previously converged ground state calculation and the search intervals were chosen for aesthetic purposes. Eigenvalue states can move substantially in a simulation started from a poor initial guess and contours should be chosen to cover then entire space (i.e. the end of one search interval should also be the beginning of the next). This approach is not necessary for a small molecule such as C_2H_6 and is shown here as an illustration. For a molecule with thousands of electrons this becomes mandatory and each interval would contain hundreds of eigenvalues.



Figure 3.6: Splitting the full search interval into three separate contours. The lowest (real) energy contour captures the core states while the other two target valence electrons. Each contour has eight quadrature nodes shown as circles. Symmetry allows the FEAST algorithm to only perform computations for the upper upper half of the contour.

3.3 TDDFT calculation

Real time TDDFT propagates the ground state wave functions directly in time. At the end of each DFT calculation, NESSIE saves the ground state potential as the sum of the Hartree and exchange and correlation terms. The ground state Kohn Sham Hamiltonian can be generated from this and the ionic potential. The wave functions are then recovered by solving an eigenvalue problem at the beginning of each TDDFT calculation. The electron density could also be stored and used to generate the potential and the matrix. However, the electron density must be computed directly at the quadrature nodes and, if stored at mesh nodes, would require an interpolation introducing error. The potential is used since it is defined at the mesh nodes and requires less storage space.

After solving the eigenvalue problem, the ground state wave functions are propagated directly in time using the Crank Nicolson operator of (2.75). This requires solving the linear system in (2.77) at each time step. As outlined in Section 2.2.2.2, the Hamiltonian gains its time dependence from the Hartree potential and exchange and correlation potentials that are functionals of the electron density. These must also be recomputed at each step. The time dependent Hartree potential is computed from the Poisson equation. Exchange and correlation terms must be computed with time instantaneous electron density and the same approximation used in the ground state calculation.

Observables can then be extracted from the time dependent density. In principle, all time dependent observables are functionals of the density, however, in practice the functional form is rarely known. Beside the electron density itself, the most straightforward observables to compute are the emission and absorption spectrum, which can be calculated directly from the induced dipole moment. The induced dipole moment is a measure of how far the electron density has moved away from the ground state value and an example can be seen for benzene in Figure 3.7 for an excitation polarized in the \hat{x} , \hat{y} and \hat{z} directions. The absorption spectrum from an impulse excitation can be calculated by taking a Fourier transform of the induced dipole moment, as described in Section 2.2.2.3. The absorption spectrum for each direction specific excitation can be seen in Figure 3.8 and the direction independent spectrum can then be computed as the average from \hat{x} , \hat{y} and \hat{z} .

Peaks in the absorption spectrum can be identified with specific phenomena. The electron dynamics for a specific peak can be investigated further by applying a sinusoidal stimulus specific frequency. This is a benefit of the real-time approach which gives direct access to the electron density at a resonant frequency.

3.3.1 Restarting the simulation

Larger molecules must perform a longer time propagation in order for the electron density to cycle through a sufficient number of periods. The movement of the electron density in TDDFT represents the movement of electrons. Quasiparticles and collective excitations, such as plasmons, will take longer to propagate back and forth within a



Figure 3.7: Induced dipole moment for C₆H₆. The three plots show the response when an impulse potential polarized in the \hat{x} (top), \hat{y} (middle) and \hat{z} (bottom) direction excites the system. Time steps of 5 attoseconds are used. The right graphics show the molecules orientation if the applied electric field was polarized from left to right along the page.

molecule of greater length. For large scale molecules, that can take days or weeks to simulate, it is beneficial to be able to restart the TDDFT simulation from an intermediate time step. Large scale calculation that are computationally expensive can be restarted as many times as needed to obtain an accurate result. This also can be a safe-guard against losing information due to hardware failure or user error.

To restart the TDDFT simulation the time dependent wave functions are written to a binary file in FORTRAN format. The time instantaneous density can be computed from the wave functions, which can then be used to generate the timedependent potential and Hamiltonian. Nothing else is needed to propagate the set of wave functions to the next time step. For generality, and to simplify some complexities in the code, the ground state wave functions are also stored and used to generate the ground state electron density, which is also needed to compute the induced dipole moment.



Figure 3.8: Oscillator streight of C_6H_6 for an impluse excitation polarized along the \hat{x} , \hat{y} and \hat{z} directions.

Here we show this capability for the benzene molecule. Figure 3.9 shows the induced dipole moment for a continuously executed simulation and for two separately executed simulations connected using a single restart. As can be seen, the results from the restart shown as circles and squares, fall exactly on the solid line of the continuously executed simulation. This only requires the wave functions to be saved at the end of a simulation. It allows longer simulation times to adhere to specific requirements of the computing system, which could have a maximum time limit.

3.3.2 Static core approximation

The oscillating frequencies of core and valence electronic states differ by orders of magnitude. Electromagnetic waves in the ultraviolet to visible range interact directly with valence states. Atomic core states interact with higher frequency X-Rays. With



Figure 3.9: Induced dipole moment for continuously executed TDDFT simulation (solid line) and a simulation with a single restart at time step $t = 2 \times 10^{-15}$ seconds. Results of the restarted calculation exactly match the one executed continuously with no restart.

a real-time approach in TDDFT, where the wave functions are propagated directly in time, a specific target frequency range must be chosen. The total simulated time must be chosen large enough to allow for multiple periods at the lowest target frequency. A time step must also be chosen to give sufficient accuracy. Because the time step is generally chosen as large as possible to minimize computation (i.e. the total number of time steps), it is not really possible to investigate the dynamics of both the core and valence within the simulation.

Large scale calculations with many electrons must propagate the Kohn Sham states in time. Since we are most interested in the UV-ViS range, and the time step is chosen as a few atto seconds, the dynamics of the core states cannot be captured accurately. We have employed a static core approximation, where the core states are held stationary at their ground state configuration, and only the valence states are time propagated. The total electron density $\rho(t) = \rho_{core} + \rho_{valence}(t)$ is used to compute the time-dependent potential and Hamiltonian, but only the valence component changes in time.

A comparison is shown in Figure 3.10 between a full TDDFT simulation, where core and valence states are time propagated, and a simulation employing the static Figure 3.10: Only valence states are propagated within the static core approximation. Core electronic states are held constant and do not move - their contribution to the potential is the same as in the ground state. Bottom figure is a zoom of the first femto second, where it can be seen that the two simulations look identical.



core approximation. The induced dipole moments are almost identical. Although it cannot be easily seen from the plot, the results from the static core do vary slightly from the full calculations by about 0.1 - 1%.

3.4 Comparison between P2 and P3

The discretization in NESSIE is usually improved with polynomial refinement (i.e. p-refinement). Edge refinement (i.e. h-refinement) can be accomplished for the interstitial mesh, but is generally reserved for at the atomic muffin-tine domains.

Ground state calculations require the use of a third degree polynomial (i.e. P3) basis to achieve accurate results for the total energy. Figure 3.11 shows the eigenvalue spectrum for both P2 and P3 levels of refinement. As can be seen the core and valence states are slightly different for P2 and P3, but the difference in total energy is more drastic.

The computed absorption spectrum for both P2 and P3 is shown in Figure 3.12. Time-dependent calculations are also slightly different for the two levels of refinement. Generally only P2 accuracy is used for TDDFT as it is much less expensive computationally and in good agreement with experiment.



Figure 3.11: Ground state eigenvalue spectrum for benzene with both P2 and P3 polynomial refinment for the finite element mesh. The total energy was calcualted as -6244.44652890041 (eV) for P2 and -6263.41119295063 (eV) for P3.



Figure 3.12: The absorption spectrum averaged over excitations in \hat{x} , \hat{y} and \hat{z} using both P2 and P3 polynomial refinement for the finite element mesh. The first major peak is located at 6.92 (eV) for P2 and 6.77 (eV) for P3.

3.5 Comparison between LDA and GGA

Both LDA and GGA functional can be employed in NESSIE. Here we compare Perdew Zunger [159] and PBE [157] functionals for both DFT and TDDFT. A P3 finite element basis is used to compare the ground state, as this is necessary to achieve good accuracy. Core electronic states are shifted down in energy for GGA while the HOMO level is shifted up. Results are in agreement with the all-electron package NWChem [5] for both core and valence states, as well as the pseudopotential code Octopus [6] where the same effect on the HOMO level was observed. The total energy is significantly larger (in magnitude) using the GGA functional.

The absorption spectrum for LDA and GGA are compared using a P2 finite element basis. Both LDA and GGA produce very similar results, with the fundamental peak located at 6.9 (eV) for both.


Figure 3.13: Ground state eigenvalue spectrum for benzene using both LDA and GGA functionals with P3 polynomial refinment for the finite element mesh. The total energy was calcualted as -6263.41119295063 (eV) for LDA and -6421.66928296016 (eV) for GGA.



Figure 3.14: The absorption spectrum averaged over excitations in \hat{x} , \hat{y} and \hat{z} using both LDA and GGA functionals with P2 polynomial refinement for the finite element mesh. The first major peak is located at 6.92 (eV) for LDA and 6.92 (eV) for GGA.

3.6 Visualizing the response density

The response density $\delta n(\omega, r)$ is the change in electron density due to an excitation at frequency ω . This can be visualized by applying a sinusoidal excitation at a given frequency, waiting for the density to reach a steady state where it oscillates at ω , and plotting the 4-dimensional data at a maximum and a minimum. Another approach is to compute $\delta n(\omega, r)$ directly. This is just the Fourier transform of the time varying electron density.

It is similar to the approach outlined in Section 2.2.2.3 for computing the photoabsorption spectrum from the induced dipole moment. In that case the imaginary part of the dipole Fourier transform was related to the photoabsortion cross section. Instead, the Fourier transform could be computed first for the electron density. Then taking the expected value would be exactly equivalent to the approach in the previous chapter.

This approach allows one to produce these plots for many different excitation frequencies at once. A set of frequencies can be selected and the Fourier transform for the electron density then computed on-the-fly while performing the time-dependent calculation. A exponential function,

$$\beta(t) = e^{-\gamma t},\tag{3.1}$$

multiplies the time varying electron density and acts as damping coefficient to simulat relaxation from the excitation back to the ground state. This must be used with the Fourier transform, which is taken as,

$$\delta n(r,\omega) = \int_0^T \left(n(r,t) - n_0(r) \right) \times \beta(t) \times e^{i\omega t} dt, \qquad (3.2)$$

to compute the response density for a given frequency. In practice the trapezoidal rule is used to approximate the integral.

Results from this approach are shown in Figures 3.15 and 3.16 for P2-LDA benzene excited with an impulse excitation polarized in the \hat{z} direction. Five values of ω (or equivalently E) have been chosen corresponding to peaks in the absorption spectrum. These including the fundamental frequency of 6.91 (eV), shown for the XY, YZ, and XZ planes, and four higher energy excitations at 10.04, 13.61, 16.20 and 17.74 (eV) shown only for the XZ plane.



Figure 3.15: P2-LDA response electron density for the first peak in absorption spectrum of benzene at 6.91 (eV) for an excitation in the \hat{z} direction. From left to right: XY, YZ and XZ planes.



Figure 3.16: P2-LDA response electron density for four different peaks in absorption spectrum of benzene shown in the XZ plane. Clockwise from the top left: excitations at 10.04, 13.61, 16.20 and 17.74 (eV) in the \hat{z} (upward) direction.

CHAPTER 4

SIMULATIONS RESULTS

4.1 Small molecules

The following presents a catalog of NESSIE results for select small molecules. Multiple comparisons are presented for each molecule. Ground-state results for the eigenvalue spectrum and total energy are given for both P2 and P3 polynomial refinement. Both LDA, using Perdew Zunger correlation, and GGA, under the PBE functional, are considered and compared. Time-dependent DFT results are also presented for P2 and P3, and for both LDA and GGA.

Results obtained with NESSIE have been compared (for both LDA and GGA) to ground state calculations with NWChem and Octopus. Both the eigenstates and total energies obtained using the "cc-PVQZ" basis set in NWChem were very close to values computed in NESSIE. Qualitative behaviour of the eigenspectrum when moving from LDA to GGA also matched. The Octopus code was also compared directly with NESSIE for the ground state, using the standard pseudopotentials included with the package and results matched qualitatively.

However, a direct comparison with other codes is not included here. It is difficult to perform a fair comparison of time-dependent calculation since many advanced options are available and the choice of pseudopotentials in Octopus and the atomic basis set in NWChem can have a significant effect on results. Therefore, instead, I compare directly to experimental data where possible. The results from NESSIE seem to match them quite well. The following presents, for each molecule, two tables and two graphs. Atom coordinates are listed in the top left table (in Angstroms). A selection of ground state Kohn Sham eigenvalue energies are given in the top right table(in electron volts). The eigenvalues include the lowest energy state E_1 and, if multiple types elements also exist (beside hydrogen), the first core state of additional element are listed as E_2 , E_3 etc. Also given are the HOMO and LUMO levels E_H and E_L , and the total energy E_T . The photoabsorption is plotted below for LDA (middle) and GGA (bottom), where each figure contains two results. One curve is for results obtained with a P2 finite element discretization and the other is for P3. These curves correspond to the photoabsorption averaged over excitations along each coordinate axis and required three separate simulations unless symmetries were present.

4.1.1 H₂

Atom Coordinates						
Type	X	Y	Ζ			
Н	0.0000	0.0000	-0.3705			
Η	0.0000	0.0000	0.3705			

	Ground-state Energies								
	LI	DA	GGA						
	P2 P3		P2	P3					
E_1	-10.4360	-10.2781	-10.5836	-10.4001					
$ E_L $	0.1014	0.1000	0.0820	0.0756					
E_T	-30.9021	-30.9530	-33.4798	-33.5977					

 $\mathbf{D}_{\mathbf{r}}$



Figure 4.1: Simulation results for H₂.

4.1.2 CH_4

Atom Coordinates								Ground-sta	te Energies
	Atom	Х	Y	Z]		LI	DA	GG
	С	0.00000	0.00000	0.00000]		P2	P3	P2
	Η	0.62758	0.62758	0.62758		E_1	-265.265	-265.577	-268.096
	Η	-0.62758	-0.62758	0.62758		E_H	-9.776	-9.486	-9.767
	Η	0.62758	-0.62758	-0.62758		E_L	-0.445	-0.355	-0.485
	Η	-0.62758	0.62758	-0.62758		E_T	-1087.78	-1091.62	-1117.89



GGA

P3

-268.429

-9.472

-0.400

-1121.72

Figure 4.2: Simulation results for CH₄.

4.1.3 H₂O

Atom Coordinates							
Atom	X	Y	Ζ				
Ο	0.0000	0.0000	0.0000				
Н	0.0000	0.0000	0.9600				
Н	0.0000	0.9292	-0.2410				

	Ground-state Energies								
	LI	DA	GGA						
	P2	P3	P2	P3					
E_1	-507.253	-506.516	-511.284	-510.481					
E_H	-7.948	-7.425	-7.857	-7.295					
E_L	-1.051	-0.931	-1.102	-0.958					
E_T	-2055.206	-2064.839	-2096.604	-2105.965					



Figure 4.3: Simulation results for H_2O .

4.1.4 CO

	Aton	n Coordii	nates
Atom	Х	Y	Ζ
С	0.0000	0.0000	-0.565
0	0.0000	0.0000	0.565

Ground-state Energies								
	LI	DA	GGA					
	P2	P2 P3		P3				
E_1	-510.512	-509.674	-514.310	-513.473				
E_2	-270.123	-269.679	-272.655	-272.243				
E_H	-9.874	-9.160	-9.821	-9.102				
E_L	-2.820	-2.291	-2.650	-2.068				
E_T	-3054.19	-3060.84	-3119.82	-3126.37				



Figure 4.4: Simulation results for CO.

4.1.5 C_2H_6

	Atom Coordinates						
Atom	Х	Υ	Ζ				
С	0.000	0.000	0.000				
С	0.000	0.000	1.520				
Н	1.020	0.000	-0.390				
Н	-0.510	0.000	-0.390				
Н	-0.510	0.000	-0.390				
Н	-1.020	0.000	1.920				
Н	0.510	0.000	1.920				
Н	0.510	0.000	1.920				

Ground-state Energies									
	LI	DA	GGA						
	P2	P3	P2	P3					
E_1	-265.439	-265.810	-268.235	-268.632					
E_H	-8.381	-8.122	-8.411	-8.140					
E_L	-0.549	-0.432	-0.557	-0.450					
E_T	-2143.72	-2151.52	-2201.18	-2208.92					



Figure 4.5: Simulation results for C_2H_6 .

4.1.6 SiH₄

Atom Coordinates					Ground-state Energies				
Type	X	Y	Ζ			LI	DA	GC	GA
Si	0.0000	0.0000	0.0000	j		P2	P3	P2	P3
Н	0.8544	0.8544	0.8544		E_1	-1771.43	-1771.97	-1778.94	-1779.58
H	-0.8544	-0.8544	0.8544		E_H	-8.9985	-8.5467	-9.0167	-8.5416
Н	-0.8544	0.8544	-0.8544		E_L	-0.7307	-0.4925	-0.7240	-0.4880
Η	0.8544	-0.8544	-0.8544		E_T	-7865.62	-7907.53	-7957.94	-7999.02



Figure 4.6: Simulation results for SiH₄.

4.1.7 Na₂

Atom Coordinates						
Type	Х	Y	Ζ			
Na	0.00	0.00	-1.50			
Na	0.00	0.00	1.50			

Ground-state Energies								
	LI	DA	GC	GA				
	P2 P3		P2	P3				
E_1	-1026.46	-1025.99	-1032.08	-1031.57				
E_H	-3.5875	-3.2543	-3.5804	-3.1729				
E_L	-2.1055	-1.8595	-2.1100	-1.7833				
E_T	-8757.60	-8785.56	-8883.42	-8911.29				



Figure 4.7: Simulation results for Na_2 .

 $4.1.8 \quad \mathrm{C_6H_6}$

	Ator	n Coordi	nates
Type	X	Υ	Ζ
С	1.3970	0.0000	0.0000
C	0.6985	0.0000	1.2098
C	-0.6985	0.0000	1.2098
C	-1.3970	0.0000	0.0000
C	-0.6985	0.0000	-1.2098
C	0.6985	0.0000	-1.2098
Н	2.4810	0.0000	0.0000
Н	1.2405	0.0000	2.1486
Н	-1.2405	0.0000	2.1486
Н	-2.4810	0.0000	0.0000
Н	-1.2405	0.0000	-2.1486
Н	1.2405	0.0000	-2.1486

	Ground-state Energies													
	LI	DA	GC	GA										
	P2	P3	P2	P3										
E_1	-266.388	-266.498	-269.092	-269.222										
E_H	-7.038	-6.551	-6.852	-6.350										
E_L	-1.858	-1.469	-1.654	-1.247										
E_T	-6244.45	-6263.41	-6402.97	-6421.67										



Figure 4.8: Simulation results for C_6H_6 .

4.1.9 Na₄

	Ator	n Coordin	atos			te Energies	gies		
T		V	7		LI	DA	GGA		
Type	Λ	ľ	L		P2	$\mathbf{P3}$	P2	$\mathbf{P3}$	
Na Na Na Na	$\begin{array}{c} 1.6030 \\ -1.6030 \\ 0.0000 \\ 0.0000 \end{array}$	$\begin{array}{c} 0.0000 \\ 0.0000 \\ 3.3135 \\ -3.3135 \end{array}$	$\begin{array}{c} 0.0000\\ 0.0000\\ 0.0000\\ 0.0000\\ \end{array}$	$ \begin{bmatrix} E_1 \\ E_H \\ E_L \\ E_T $	-1026.322 -2.9424 -2.4109 -17509.73	-1026.258 -2.7604 -2.2118 -17571.27	-1031.926 -2.9136 -2.4063 -17760.95	-1031.828 -2.6889 -2.1358 -17822.34	
				L_T	-11005.10	11011.21	-11100.55	-11022.04	



Figure 4.9: Simulation results for Na₄.

4.2 Discussion and comparison with experiment

Small molecule results will now be discussed and, where possible, compared directly to experiment. Most of the experimental values have been gathered from an online database [10] and references to their original publication are given in the text. Experimental results were gathered at room temperature. Although it is possible for TDDFT calculations to include temperature, it was not incorporated in the present simulations.

Comparing directly with experiment is the holy grail of numerical simulation. However, this brings additional difficulties as well. For some materials and molecules few experimental results are available. For others, there exists a vast library of results, which can differ significantly. It is also difficult to measure absorption and emission spectra experimentally and it is possible to miss electronic transitions entirely. A good example of this can be seen for the H_2 molecule in Section 4.2.1. Spectroscopy is an extremely complex field with electronic transitions coupled to vibrational modes (i.e. the Franck-Condon Principle). We are only interested in excited electronic states since the atoms in our simulation do not move. It can be difficult experimentally to decouple electronic and vibrational modes and thus to have a fair comparison with these results.

However, as will be shown in the next few pages, NESSIE TDDFT simulations still compare favorably with experiment. NESSIE results are plotted directly on top of the experimental values with no other fitting parameters and show some remarkable agreement.

4.2.1 H₂

It is well known that the first excited state of molecular hydrogen is within the range 11.19 to 11.37 eV. However, the experimental result for UV-ViS absorption [25] compared directly to NESSIE in Figure 4.11 misses this peak entirely. The measurement technique uses an incident beam of electrons and seems to excite other vibrational modes altering the spectrum. The higher energy absorption, however, seems to match extremely well.

Also shown in the figure are true electronic excited-states energies; from left to right: the first singlet, first triplet, and second singlet excited states [93], and the dissociation energy [172]. These states are known from other experiments and highlighted in Figure 4.10. Transitions from singlet to triplet states (i.e. from the ground

Figure 4.10: Experimental electronic excitation energies of H_2 copied from [93]. First singlet, first triplet and second singlet states are labeled and have values of 11.37eV, 11.87eV and 12.40eV, respectively.



state to the first triplet state) are forbidden as the total angular momentum must be conserved for an isolated system. However, our simulations do not include spin and it make sense that such a transition would be captured. The first peak for the LDA functional are close to 11.19eV, which is another value often referenced for the energy of the first excited state [182]. However, with the GGA functional the first peak shifts back toward the value of 11.37, which is shown in Figures 4.10 and 4.11.



Figure 4.11: Comparison between NESSIE and experimental values for H_2 .

4.2.2 CH₄

NESSIE simulation results for the photoabsorption spectrum of CH_4 are compared with experimental data: Experiment-A [50] and Experiment-B [106] in Figure 4.12. The lowest energy absorption peak at E = 9.76 eV in the experimental result is again shifted to the right compared to NESSIE. However, there also exist a second

peak at E = 10.4eV that seems to be shifted to the left compared to the simulated result. Again, as was the case with H₂ the higher energy spectra match extremely well. NESSIE does not include temperature, which could cause slower variation and smoothing of the experimental result which were performed at T = 298K. The maximum at $E \approx 13.6$ eV matches almost exactly with the P2 results from NESSIE, although with P3 this shift considerably.



Figure 4.12: Comparison between NESSIE and experimental values for CH_4 .

4.2.3 H₂O

In Figure 4.13, the photoabsorption spectrum of H_2O computed with NESSIE is compared directly to four different experimental results performed at room temperature: Experiment-A [56], Experiment-B [55], Experiment-C [112], and Experiment-D [49]. Again we seem very good agreement between experimental results and NESSIE simulations, which seems to capture the three major peaks. Again there is a small shift which worsens for the more accurate discretization.

For P2-LDA, the three peaks corresponding to Experiments A, B and C have values of E = 7.05 eV, E = 9.23 eV and E = 13.11 eV, respectively. Experimental values have been measured at E = 7.42 eV, E = 9.68 eV and E = 13.39 eV, again continuing the trend of NESSIE to underestimate. Overall, the general shape of the Experiment-D curve matches quite well and an additional small peak at $E \approx 11 \text{eV}$ is present in both the experimental and simulated results. The peak at $E \approx 17 \text{eV}$ can be seen in the results from both NESSIE and Experiment-C.

4.2.4 CO

NESSIE simulation results for the photoabsorption spectrum of CO are compared directly to experimental data from [48] in Figure 4.14. The first absorption peak looks to be captured well by the TDDFT simulation. This is also the case for the higher energy spectrum greater than 14eV.

However, it is not clear what is happening between 10 and 11 eV. With P2-LDA data it seems that NESSIE finds an additional resonance at E = 10.25 not detected by the experiment. Again, for P2-GGA, the resonance at E = 11.75 matches well with the experimental result, but the simulation also produces this intermediate peak in the absorption. Here, with P2-GGA, an additional resonance at 13eV emerges from the large peak at 14eV. This is made even more confusing by the P3 data (both LDA and GGA), where everything shifts down in energy and no longer lines up directly



Figure 4.13: Comparison between NESSIE and experimental values for H_2O .

with experimental data. So, it is not clear which resonance the second peak in the experimental result corresponds to or if anything has been missed in the experiment.

4.2.5 C_2H_6

Figure 4.15 compares the NESSIE simulation results for the photoabsorption spectrum of C_2H_6 directly to experimental data: Eperiment-A from [107] and Experiment-B from [139]. All of the simulated results match the experimental result qualitatively



Figure 4.14: Comparison between NESSIE and experimental values for CO.

reaching a broad maximum between 15 and 16eV. There is a small lower energy peak measured around E = 9.5eV. It is not clear what this corresponds to in the simulations, where we see three peaks at $E \approx 8.3$, $E \approx 9.5$ and $E \approx 10.3$ that exist for P2/P3 and LDA/GGA.

4.2.6 SiH₄

NESSIE simulation results for the photoabsorption spectrum of SiH_4 are compared directly to experimental data from [59] in Figure 4.16. Here, the experimental result



Figure 4.15: Comparison between NESSIE and experimental values for C_2H_6 .

does not have any distinguishable resonance aside from the maximum at E = 11The simulated results match the overall shape of the measure absorption spectrum well for all cases. The P2-LDA results has a maximum at E = 10.5eV, close to the experimental maximum. For P3-LDA this peak diminishes slightly in strength and the maximum simulated absorption shift to higher energy in more agreement with the experimental result. The GGA results are similar, but the absorption around

E = 11eV seems to be slighly diminished while two peaks at E = 14eV are less affected.



Figure 4.16: Comparison between NESSIE and experimental values for SiH₄.

4.2.7 C₆H₆

Figure 4.17 compares NESSIE simulation results for the photoabsorption spectrum of C_6H_6 directly to experimental data: Experiment-A from [65] and Experiment-B from [171]. In all cases, P2/P3 and LDA/GGA the first absorption peak is captured very well. The value of E = 6.9eV matches almost exactly with the experimental

values. Again, for P3 data, the peak shifts to lower energy, but is still a close match to the measure value. The higher energy spectrum match qualitatively with experiment. There is a small measured abosption peak around E = 9eV that matches the P2-LDA results from NESSIE. However, this peak shifts for P3 and GGA. There seems to be another measured peak at $E \approx 11\text{eV}$, which is also captured by NESSIE for all cases.



Figure 4.17: Comparison between NESSIE and experimental values for C_6H_6 .

4.3 Large-scale carbon nano-tubes

The parallel implementation of NESSIE is now used to investigate plasmon resonances in large-scale carbon nano-tubes (CNT). Results obtained previously with NESSIE have reported the existence of plasmons in metallic (3,3) single-wall CNTs up to seven unit cells [168]. The larger of the simulations presented in this section were not possible with the previous shared-memory version of the code.

A single unit cell of the (3,3) nano-tubes exhibits both 'A' and 'B' type carbon rings. Each tube is started with an 'H' hydrogen ring and terminated with both 'A' and 'H'. The coordinates of the first three rings 'H', 'A' and 'B' are presented in Table 4.1 from which all other atom coordinates can be determined.

Table 4.1: Coordinates $[X, Y, Z]^T$ in nanometers for three types of atom rings making up a (3,3) CNTs. Each unit cell contains both an 'A' and 'B' ring. In addition, the CNT is started with an 'H' and terminated with both 'A' and 'H' (i.e H-AB-AB...AB-A-H). Distance between rings can be determined from the values for Z. Rings are shown below, in the order 'H', 'A', 'B'. Technically, 'H' and 'B' have the exact same in-plane coordinates but are distinguished for clarity.

	•••	•			
	• •		•	•	
om 1	Atom 2	Atom 3	Atom 4	Atom 5	
4610 1190	$\begin{bmatrix} -0.37258\\ 2.05870 \end{bmatrix}$	$\begin{bmatrix} -2.09220 \\ 0.00000 \end{bmatrix}$	$\begin{bmatrix} 1.59660 \\ -1.35200 \end{bmatrix}$	$\begin{bmatrix} 1.04610\\ -1.81190 \end{bmatrix}$	

Type	Atom 1	Atom 2	Atom 3	Atom 4	Atom 5	Atom 6
Н	$\begin{bmatrix} 1.04610\\ 1.81190\\ 0.00000 \end{bmatrix}$	$\begin{bmatrix} -0.37258\\ 2.05870\\ 0.00000 \end{bmatrix}$	$\begin{bmatrix} -2.09220\\ 0.00000\\ 0.00000 \end{bmatrix}$	$\begin{bmatrix} 1.59660\\ -1.35200\\ 0.00000 \end{bmatrix}$	$\begin{bmatrix} 1.04610\\ -1.81190\\ 0.00000 \end{bmatrix}$	$\begin{bmatrix} 1.96920\\ -0.70671\\ 0.00000 \end{bmatrix}$
А	$\begin{bmatrix} 2.09220\\ 0.00000\\ 1.24860 \end{bmatrix}$	$\begin{bmatrix} 1.59660 \\ 1.35200 \\ 1.24860 \end{bmatrix}$	$\begin{bmatrix} -1.04610\\ 1.81190\\ 1.24860 \end{bmatrix}$	$\begin{bmatrix} -1.96920\\ 0.70671\\ 1.24860 \end{bmatrix}$	$\begin{bmatrix} -1.04610\\ -1.81190\\ 1.24860 \end{bmatrix}$	$\begin{bmatrix} 0.37258\\ -2.05870\\ 1.24860 \end{bmatrix}$
В	$\begin{bmatrix} 1.04610\\ 1.81190\\ 2.49720 \end{bmatrix}$	$\begin{bmatrix} -0.37258\\ 2.05870\\ 2.49720 \end{bmatrix}$	$\begin{bmatrix} -2.09222\\ 0.00000\\ 2.49720 \end{bmatrix}$	$\begin{bmatrix} -1.59660 \\ -1.35200 \\ 2.49720 \end{bmatrix}$	$\begin{bmatrix} 1.04610\\ -1.81190\\ 2.49720 \end{bmatrix}$	$\begin{bmatrix} 1.96920\\ -0.70671\\ 2.49720 \end{bmatrix}$

With the fully parallel code, simulations have now been run for CNTs up to forty unit cells. Table 4.2 briefly presents some physical attributes of the simulated molecules: 5-CNT, 10-CNT, 20-CNT and 40-CNT. The largest molecule contains 498 atoms and 2928 electrons and can be considered very large-scale in the context of DFT and TDDFT, especially for a full-core all-electron approach. The size of the discretization, which can be seen in Table 4.3 for each CNT molecule, grows linearly with the number of atoms.

N_u	N_{at}	N_e	L (nm)
5	78	408	1.2586
10	138	768	2.4927
20	258	1488	4.9944
40	498	2928	9.9888

Table 4.2: Some physical attributes of the family of (3,3) CNTs with the number of unit cells ranging from $N_u = 5$ to $N_u = 40$. The length L does not include an additional 'A' and 'H' rings that terminate the molecule. Also listed are the total number of atoms N_{at} and electrons N_e .

CNT	n _{at}	n_{it}	n_{total}
5	2,065	149,499	302,925
10	2,065	126,431	$397,\!877$
20	2,065	$233,\!696$	$741,\!182$
40	2,065	$446,\!559$	$1,\!426,\!125$

Table 4.3: Mesh size for the finite element discretization used for each CNT simultation. The total number of discretization points (and matrix size) is related to the size of the atomic mesh n_{at} and the size of the interstitial mesh n_{it} . The total number of points is given by: $n_{total} = N_{at} \times (n_{at} - n_s) + n_{it}$, where N_{at} is the number of atoms and $n_s = 98$ is the number of interface points between one atom and the interstitial region (for a P2 finite element). A reduced discretization in the interstitial region was used for the 10, 20 and 40 CNTs.

4.3.1 DFT Calculation

The major difficulty in the DFT calculation is solving the eigenvalue problem. The timing results for various stages an SCF iteration is presented in Table 4.4 for the family of (3,3) CNTs. Time results depend on the number of openMP thread and the number of MPI processes at each level of parallelism. Here we consider only the configuration of parallel resources used in the actual calculation. The time presented is averaged over all SCF iterations. The eigenvalue calculation take by far the most time for the 20 and 40 CNTs. On average, multiple FEAST iterations were necessary. Load balancing was an issue as the FEAST interval with core electrons generally converged in a single FEAST iteration, while intervals targetting valence states close to the continuum took more for many of the SCF steps. Also shown is the total number of SCF iterations to reach a convergence of 10^{-10} on the electron density.

CNT	OMP	L1	L2	L3	t_H	t_{KS}	t_{n_q}	t_{eig}	t_{it}	N_{it}
5	12	1	4	1	8	12	5	58	79	40
10	12	2	4	2	7	11	7	42	64	42
20	24	4	4	1	34	45	38	373	474	58
40	24	4	4	1	67	91	93	455	706	45

Table 4.4: Parallel configuration used for ground-state calculation of CNTs and timing results of various SCF operations. Listed are the number of openMP threads OMP and the number of MPI processes at each levels of MPI parallelism **L1**, **L2** and **L3**. Times are listed for computing the Hartree potential t_H , including the potential into the Hamiltonian t_{KS} and recomputing the electron density t_{n_q} . The reported time for solving the eigenvalue problem t_{eig} is the average time per SCF step, which varies depending on the number of FEAST iterations needed. Also reported are the total time per SCF iteration t_{it} and the total number of SCF iterations $N_i t$.

After convergence the ground state electron density is obtained along with the Kohn Sham potential. Table 4.5 lists the lowest energy eigenvalue, the HOMO and LUMO levels, and the total energy for all four CNT molecules.

4.3.2 TDDFT Calculation

The induced dipole moment for the CNTs can be seen in Figure 4.18. The staticcore approximation has been employed to reduce the number of Kohn Sham states propagated in time. This has very little effect on the result as the motion of inner core electrons is orders of magnitude faster than valence states and do not participate in

CNT	E_1	E_H	E_L	E_T
5	-267.73	-4.996	-4.640	-67713.28
10	-268.05	-4.823	-4.820	-128824.88
20	-268.07	-5.000	-4.799	-251350.58
40	-268.28	-4.955	-4.898	-496359.39

Table 4.5: Summary of the ground state calculation for the family of (3,3) CNTs. The lowest energy state E_1 , the HOMO and LUMO levels E_H and E_L and total energy E_T are shown.

	5-CNT	10-CNT	20-CNT	40-CNT
dt	0.01	0.01	0.01	0.01
Т	15	30	45	65
N_{it}	1500	3000	4500	6500
N_p	13	9	9	8

Table 4.6: TDDFT simulation parameters: time step dt and total simulation time T in femtoseconds, and the total number of time-steps N_{it} . Also listed is the approximate number of oscillations N_p of the main lowest frequency.

the simulation when using the time-step employed in these calculations. Here only the first fifteen femtoseconds of the simulation is shown. The time-dependent calculation is much more computationally expensive than the ground state calculation because of the large number of time steps needed to capture the lowest energy excitations.

An overview of the total simulation time and time step used to perform the Crank Nicolson propagation for each CNT can be seen in Table 4.6 along with the approximate number of oscillations of the main lowest frequency. The time interval must be increased with the length of the tube since the electrons take longer to travel the full length and perform a single oscillation for the lowest frequency excitation. This is shown in the figure, where the period grows much longer for the larger tubes. The entire simulation can be seen in Figure 4.18 for the 5-CNT, while the 40-CNT barely completes a single period over this time. However, frequency space TDDFT methods also have difficulties with large molecules since the number of electrons increases and a real-time approach is actually more scalable.



Figure 4.18: Induced dipole for 5, 10, 20 and 40 unit cell (3,3) carbon nanotubes. Only the first fifteen femtoseconds of each simulation are pictured.

Timing results for the TDDFT simulations can be seen in Table 4.7. Along with the parallel configuration used for the calculation, average times are presented for various operations computed at each time step. Also shown is the time per timestep. The table present performance tests for the 20-CNT with different parallel configurations on top. As can be seen, the third level of parallelism L3 is actually quite efficient for TDDFT. OpenMP threading is not as efficient for many of the matrix operations. Also, the full interstitial matrix must no longer be build on each MPI process and atom atom operations can be parallelized. The bottom shows the

	CP	PU	ON	AP I	$L1 \times I$	L2 L	3	t_H	t_{KS}	t_{n_q}	t_{pro}	$p_p \parallel t_\Delta$	t
	21	6	2	4	9	1	L	34	11	22	46	11	3
	21	6	1	2	9	د 2	2	12	6	10	25	53	3
	21	6	1	2	6		3	11	5	11	23	50)
	43	2	6	5	18	4	1	4	2	4	18	29)
	86	4	1	2	18	4	1	4	2	3	16	26	5
	<u> </u>												
CI	VT	C	PU	OM	$\mathbf{P} = L$	$1 \times L^2$	2	L3	t_H	t_{KS}	t_{n_q}	t_{prop}	$\ t_{\Delta t}$
Ę	5		24	3		2		4	6	3	2	17	29
1	0	4	18	6		4		4	4	2	3	11	23
2	0	4	32	6		18		4	4	2	4	18	29
4	0	4	32	12		18		2	21	14	19	48	103

Table 4.7: Top: Performance testing for different parallel configurations (and number of CPU cores) with the 20-CNT. Bottom: Parallel configuration used for TDDFT simulations and timing results of various SCF operations for all four CNTs. Listed are the number of openMP threads OMP and the number of MPI processes at each levels of MPI parallelism **L1**, **L2** and **L3**. Times are listed for computing the Hartree potential t_H , including the potential into the Hamiltonian t_{KS} , recomputing the electron density t_{n_q} , time propagation t_{prop} , and the time per time step $t_{\Delta t}$.

timing results for all four CNT molecules with the parallel configuration used to obtain the results. The total simulation time can be determined using the total number of time steps (see Table 4.6).

After computing the induced dipole, the photoabsorption spectrum can be computed via a Fourier transform as outlined in Section 2.2.2.3. Figure 4.19 shows the photoabsorption spectrum for the 10, 20 and 40 CNT on the same scale. Figures 4.20 and 4.21 show the oscillator strength vs. excitation energy for all four CNTs on a different scale so that the features can be more easily distinguished. Finally, a comparison can be seen in Figure 4.22 between a transverse excitation along the y-axis and the longitudinal excitations (shown in previous figures) along the z-axis for the 20-CNT. Here we see the major peaks in the absorption disappear for excitations less than 5eV. The inset shows a zoom from 2eV to 6eV where there exist activity at 3.25eV. We note that the band-to-band transition should not occur in perpendicular excitation, but the intensity of this peak is actually very low and may be due to some other effect. A new peak also forms around 6eV, which was previously reported for the 5-CNT [168], and is due to a characteristic shift of the π plasmon to higher energy.



Figure 4.19: Comparison between absorption spectrum for 10, 20 and 40 unit cell (3,3) carbon nano-tubes. Full energy range of the simulation is shown on the top. The bottom is a zoom of main region of interest.



Figure 4.20: Absorption spectrum for 5, 10, 20 and 40 unit cell (3,3) carbon nanotubes.

4.3.3 Discussion

After computing the photoabsorption spectrum, interesting excitations can be further investigated. The 4-dimensional surface plots of the response density $\delta n_q(\omega, r)$ can be calculated for specific resonances and used to determine the nature of the excitation (i.e. whether or not it is a plasmon). This required an additional simulation, where a set of resonances were identified beforehand and the response density was computed as the Fourier transform on-the-fly as outlined in Section 2.2.2.3. Figures



Figure 4.21: Main area of interest of the absorption spectrum for 5, 10, 20 and 40 unit cell (3,3) carbon nanotubes.

4.30, 4.31 and 4.32 at the end of this section present the 4-dimensional surface plots for interesting resonances in the 5, 10 and 20 CNTs below five electron volts.

Although we do not directly compute the recently proposed plasmon index [221] that can be used to distinguish a plasmon from a single-particle excitations, it is useful to transcribe the equation to gain more understanding of plasmon behaviour. The plasmon index is defined as,

$$\eta = \left| \frac{\int_{\Omega} \delta n(\omega, r) v_{ind}^*(\omega, r) dr}{\int_{\Omega} \delta n(\omega, r) v_{ext}^*(\omega, r) dr} \right|,\tag{4.1}$$



Figure 4.22: Comparison between a transverse (Y) and longitudinal (Z) excitations for the 20-CNT.

where v_{ind} is the induced electrostatic potential and v_{ext} is the external potential. Here, where the goal is to gain insight, the denominator can be safely ignored as it just is a normalization condition. The induced potential,

$$v_{ind}(\omega, r) = \int_{\Omega} \frac{\delta n(\omega, r')}{|r - r'|} dr', \qquad (4.2)$$

is the classical electrostatic potential due to the induced charge from a specific excitation. With TDDFT, this also includes the exchange and correlation energy [40]. As an aside, the numerator could be calculated directly as the Fourier transform of the time-dependent Hartree potential and exchange and correlation potentials (more specifically the difference from their ground state value). This would be similar to the procedure used here for computing the response density on-the-fly at each time step. In any case, the 4-dimensional surface plots can give some understanding of the induced potential and the plasmon index.

First let us consider excitation peak at 10 electron volts, which is common to all four CNTs. This is a $\pi + \sigma$ plasmons [168] and thus does not change with tube length.

The π plasmon can also be seen in later 4-dimensional plots for each CNT around E = 4.4eV. These plasmons can be detected experimentally [122, 121, 111, 151]. They can be excited by a transverse excitation since they are of bulk-type, but shift to high energies as seen in Figure 4.22. The surface plot is shown in Figure 4.23 and has a similar character for each of the molecules, which shows good agreement between the simulations.



Next let us consider the lowest energy excitation for each of the CNTs. Figure 4.24 shows the corresponding 4-dimensional surface plots. It was labeled as a single-particle excitation in [168] for the 5 CNT. However, this excitation also seems to resemble a surface plasmon in the 4-dimensional plots for the 10 and 20 CNT since a dipole moment exists across the tube, which results in a net potential (and a large value for the integral defining the plasmon index).

The velocity of this excitation for larger tubes also surpasses the Fermi velocity of 10^6 (m/s), as shown in Figure 4.25. The phase velocity is given by the equation,

$$v = \lambda f = 2Lf,\tag{4.3}$$




Figure 4.25: Velocity corresponding to the lowest energy peak in the absorption spectrum for the 5, 10, 20 and 40 unit cell CNT. As the length of the tube increases the speed increases past the Fermi velocity and the behavior seems to transform from a single-particle type excitation to a collective plasmonic excitation.

where f is the frequency of the excitation and the wavelength $\lambda = 2L$ is taken as twice the length of the CNT. As these plasmons are localized to the surface, a reasonable assumption is that they must travel back and forth the full length of the nano-tube to complete a single oscillation. This is further supported by the 4-dimensional plots.

These could be Luttinger-liquid plasmons. Luttinger-liquid theory [140, 200] describes quantum confined electrons in 1-dimension and, as such, describes the behaviour of electrons in carbon nano-tubes which are quasi 1-dimensional nanostructures. The theory predicts the existence of Luttinger-liquid plasmons with a velocity between $3 \times$ and $4 \times$ that of the Fermi velocity in single wall CNTs. In fact, the existence of Luttinger-liquid plasmons in CNTs has recently been verified experimentally [184] using scattering-type scanning near-field optical microscopy. The curve in Figure 4.25 plots plasmon velocity against CNT length for the lowest energy excitation of our simulations. It is beginning to approach a constant value and the slope suggests that the velocity for 80, 160 and 320 unit cell (3,3) CNTs will be 2.92, 3.21 and 3.3 $(\times 10^6 \text{ m/s})$, respectively. This is in close agreement with measured Luttinger-liquid plasmon velocities [184] in one micron tubes, which had values in the range of 3.3-4 $(\times 10^6 \text{ m/s})$. Note that the diameter should not significantly change the plasmon velocity according to the theory. The wavelengths for these observed plasmons were much longer than the length of our CNTs with a value of 100 nanometers or 75 nanometers depending on the exciting frequency (this was said to vary due to the frequency dependent dielectric constant of the substrate). Thus, the strong spatial confinement would not appear in our simulations. The wavelength, however, would be comparable to the length of a 320 unit cell tube where the velocity is predicted to have a constant value.

There is a second series of plasmon peaks identified for the 5-CNT in [168]. The first is located at 3.89eV, while the second at 4.76eV comes in the form of a double peak. These appear to be related to the peaks near 3.61eV in the 10-CNT (see Figure

4.21). Note that the 7-CNT also showed a similar double peak in [168]. However, for the longer tubes (20-CNT and 40-CNT) it becomes difficult to follow them as they would appear in the region dominated by the band-to-band transition which will be discussed next. This higher energy plasmon associated with the double peak quickly reaches fairly large plasmon velocities of more than 3 ($\times 10^6$ m/s) for the 7-CNT, as reported in [168]. The nature of these resonances is not clear as they could point to a small molecule effect specific to short CNTs and may not be related to the canonical Luttinger-liquid plasmon in the limit of very long tubes. This interpretation is consistent with [108] which showed that the Luttinger-liquid model can be applied to armchair CNTs if long-range electron-electron interactions are taken into account, while neglecting short-range interactions. This second series of plasmon resonances for shorter tubes could thus be influenced by short range interaction effects.

Another common peak at 3.25eV is also compared for all CNTs in Figure 4.19 which shows the 4-dimensional plots. This peak was labeled as a band-to-band transition in [168], which investigated nano-tubes up to seven unit cells. The band-to-band transition has been detected experimentally at 3.1eV [135] for (3,3) CNTs, which is in close agreement with other simulated resonances for the 20 (4D plot shown in Figure 4.32) and 40 unit cell tubes. The observed peak around 3eV for these tubes is also close to other *ab initio* simulations [189, 137] identifying the band-to-band transition. The character of this peak for the 10-CNT resembles the 5-CNT. However, the 20-CNT seems to undergo a transition and obtain a more plasmon-like characteristics. Excitonic effects have been attributed to resonances around 3eV in (3,3) CNTs [189] and have an extension of around 5 nanometers, which is very close to the size of the 20 CNT.

Four-dimensional plots for the π plasmon peak are now presented in Figure 4.27, which appear around $E \approx 4.4$ eV. Like the $\pi + \sigma$ plasmon, this can be excited by a transverse excitation but shifts to higher energy (around 6eV) as shown in Figure



4.22. For the larger tubes it becomes more difficult to distinguish characteristics from the 4-dimensional plot since the excitation is weaker compared to other resonances.



For the larger tubes, we note that two additional small sharp peaks appear in the gap between the main lowest frequency and the highly active area. A close up plot of this region is shown in Figure 4.28 for the 20 and 40 CNT. Also shown, in Figure 4.29, is the surface plot for the 20 CNT excitation at 1.53eV and 2.02eV, which seems to be quantized along the length of the tube. These peaks also resemble plasmons and are likely to be second and third harmonics of the lowest energy resonance, which have been witnessed in other TDDFT simulations of sodium chains [40]. Although



Figure 4.28: Small sharp resonances that appear in gap between the main lowest frequency and other excitations for the 20 and 40 unit cell CNTs.



Figure 4.29: Four dimensional surface plot taken at $\pm 2 \times 10^8$ of the 20 CNT. These peaks, shown in Figure 4.28, are in the gap between the main lowest frequency and other excitations. Top: response electron density for the peak at E = 1.53 eV. Bottom: response electron density for the peak at E = 2.02 eV.

we do not expect an exact integer multiple, the values of 1.53 and 2.02 are somewhat close to $2 \times$ and $3 \times$ the energy of the lowest excitation at 0.825eV for the 20-CNT. These resonances appear from a screening effect of the main plasmon resonance that hides the end of the nano-tube from other charge oscillations.



Figure 4.30: Surface plot of the response density $\delta n(w, r)$ taken at $\pm 10^9$ for the 5-CNT. Each plots is for a different ω corresponding to an energy E and frequency f in the range of Figure 4.21.



Figure 4.31: Surface plot of the response density $\delta n(w, r)$ taken at $\pm 10^9$ for the 10-CNT. Each plots is for a different ω corresponding to an energy E and frequency f in the range of Figure 4.21.



Figure 4.32: Surface plot of the response density $\delta n(w, r)$ taken at $\pm 10^9$ for the 20-CNT. Each plots is for a different ω corresponding to an energy E and frequency f in the range of Figure 4.21.

CHAPTER 5

DISCUSSIONS ON PERFORMANCE AND SCALABILITY

5.1 **PFEAST** eigenvalue solver

We now discuss how to place parallel resources across the three different levels of parallelism. The matrix systems used to gather results are a collection of Hamiltonians representing carbon nanotube (CNT) molecules of varying length (ranging from 54 to 246 atoms). These finite-element systems are generated from NESSIE and use second degree polynomial refinement and are generated from a 3-3 CNT unit cell. They are terminated with six hydrogen atoms at each end and follow a $(U_h|U_aU_b|...|U_aU_b|U_a|U_h)$ pattern (see a 3 unit cell CNT in Figure 5.1) with U_h representing the six hydrogen atoms and U_a and U_b different configurations of six atom carbon rings. Each carbon atom contains six electrons, which corresponds to three eigenmodes (since the electron spin is not explicitly taken into account). The number of sought eigenvalues m in a CNT system with n_u unit cells is given by

$$m \approx 6 + 18 + 36 \times n_u,\tag{5.1}$$

where the first six modes correspond to the twelve terminating hydrogen atoms, the next eighteen modes to the U_a carbon ring and the additional factor of thirty-six modes to the $|U_aU_b|$ unit cells.

The experiments were performed on the Mesabi Linux cluster at Minnesota Supercomputing Institute. The compute nodes feature two sockets, each with a twelve core 2.5 GHz Haswell E5-2680v3 processor. We define each MPI process as a single



Figure 5.1: 3-dimensional plot of a 3 unit cell CNT (3-CNT). The molecule is composed of 3 $|U_aU_b|$ units cells, includes an additional U_a ring, and is terminated with hydrogen atoms U_h at each end.

socket and allow the distributed memory solver to use 12 openMP threads per MPI process. The nodes are also equipped with 64 GB of system memory.

Although the PFEAST kernel could be interfaced with an iterative or hybrid solver, we consider only the application of three different sparse direct solvers. Both Cluster-MKL-PARDISO and MUMPS have been tested with the full matrices generated by stitching together the interstitial and atomic meshes within our finite-element electronic structure code (as seen in the left plot of Figure 2.6). Our application specific domain-decomposition solver (DD-Solver) reorders the matrix based on the number of L3 MPI processes, in order to reduce communication. For consistency the matrices have been permuted the exact same way for Cluster-MKL-PARDISO, MUMPS and the DD-Solver. All tests have been run with default parameters for the solvers operating in 'in-core' mode. Both Cluster-MKL-PARDISO and MUMPS decompose the full matrix by row over L3 MPI processes and contain interstitial and atomic points. The inputs to PFEAST will be consistent among the solvers with each L3 MPI process assigned the exact same subset of the eigenvector solutions, as long as the number of L3 MPI processes is divisible by the number of atoms.

The scalability of PFEAST will mainly depend on solving the complex linear systems and choosing how to distribute the parallel resources. This results in the trade-off between memory and performance mentioned in Section 2.4.5. If enough resources are available, it is advisable to place as many MPI processes at the second level of parallelism as possible. This level handles the independent linear systems and requires a copy of the matrix and solution on each L3 MPI process. It should result in close to ideal linear scaling. In contrast, if the matrix or solution can not fit into memory, it may be distributed using the third level of parallelism. The scalability then depends on the distributed memory solver.

5.1.1 Strong scalability of L2

The L2 scalability can be seen in Figure 5.2, where 3 MPI processes have been assigned to L3 and the number of clusters of L3 MPI processes (i.e. the number of L2-Communication-Worlds) is varied from 1 to 16 for a total of 3, 6, ..., 48 MPI processes (i.e. a total of 36 to 576 cores). These results have been gathered using the matrix system CNT-5 comprised of five unit-cells with a total of 78 atoms and a dimension of 302, 295. We note that the eigenvalue and eigenvectors obtained for this example correspond the true converged DFT solutions. The right graph in Figure 5.2 shows the ideal speedup in black that increases linearly with the number of L2-Communication-Worlds. The actual speedups (compared to a single L2-Communication-World) for the three solvers are plotted as points below the line. We see very close to linear scaling at this level. The total time to solve the eigenvalue problem is plotted on the left. Four PFEAST iterations were needed to reach the default convergence criteria of 10^{-12} on the eigenvector residuals. The factorization stage was only computed once for the case of 16 L2-Communication-Worlds, but did not result in super linear scaling due the 20 total seconds spent in the PFEAST kernel computing the solution to the reduced eigenvalue problem and performing communication.



Figure 5.2: Scalability of the second level of parallelism. The number of **L2** processes is increased from 1 to 16 while keeping the number of **L3** MPI processes at a constant value of 3 (for a total of 36 to 576 cores). This level is limited to 16 **L2** MPI processes since 16 quadrature points were used. These results are for the CNT-5 eigensystem with $m_0 = 600$ (226 eigenvalues). The left graph gives the total time to solve the eigenvalue problem. The speedup compared to 1 cluster of **L3** processes can be seen on the right.

5.1.2 Strong scalability of L3

The scalability of the third level of parallelism L3 is limited by the properties of the eigensystem and linear-system solver. The strong scalability of the DD-Solver, specifically, may be limited by the number atoms. All L3 MPI processes can work together in the solution to the interstitial matrix regardless of the number of atoms. But, if the number of L3 MPI processes exceeds the number of atoms, not all MPI processes will be used to solve distributed blocks (some will not have been assigned an atom). It can also be unbalanced if a different number of atoms are assigned to each MPI process. However, problems that require a distributed memory solver will usually target nanostructures with many atoms; usually far outnumbering the parallel resources. Figure 5.3 shows the strong scalability of the L3 level of parallelism within PFEAST. The matrix system from the CNT-5 molecule is used again with $m_0 = 600$ and the number of L3 MPI processes is varied from 1 to 32. We report the total time required to solve the eigenvalue problem (four PFEAST iterations to reach convergence) on the left. The speedup compared to 2 L3 MPI process can be seen on the right. Here we compare with 2 L3 MPI processes because of the communication overhead associated with the distributed memory solvers. The time should, ideally, drop by half each time the number of MPI processes is doubled. As can be seen in the graph, the efficiency of the L3 scaling is much worse than for L2.



Figure 5.3: Scalability of the third level of parallelism for 5-CNT system with $m_0 = 600$ (226 eigenvalues). The left graph the total time to solve the eigenvalue problem. Fully utilizing **L2** (with a total of 12,288 cores) these times could be reduced by a factor of 16. The scaling compared to 2 **L3** MPI processes can be seen on the right.

5.1.3 A fixed number of CPUs

Another situation common within the scientific computing community is where parallel resources are limited to a specific number processors. Here the three levels of parallelism within PFEAST can help to optimally utilize all parallel resources and achieve good performance. We only consider the second and third levels of parallelism L2 and L3. The question becomes how to divide the resources among the two levels. If 64 nodes are available on the system and 16 PFEAST contour points are used there are then five ways to distribute the resources across the L2 and L3 levels of parallelism. There is no restriction for the number of MPI processes applied to the L3 level. Each L2-Communication-World, however, should have the same number of linear systems for load balancing. With 16 contour nodes, the possible values are 1, 2, 4, 8, 16. In order to fully utilize the resources we choose the number of L3 MPI processes as 64, 32, 16, 8, 4. Table 5.1 presents the total time to solve the eigenvalue problem for a fixed number of MPI processes, but different distributions of parallel resources over the L2 and L3 levels. As we initially predicted, it is optimal to place as many resources as possible at the L2 level and the best performance happens for one linear system per L2-Communication-World.

Table 5.1: Time (in seconds) to solve the 5-CNT eigenvalue problem with $m_0 = 600$. The total number of MPI processes is held constant, but can be split between the second and third level of parallelism. More MPI processes at the **L3** level will reduce the memory required by each MPI process (i.e. the number of rows of the eigenvector and system matrices assigned to each process).

# L2	# L3	# Rows	DD-Solver	PARDISO	MUMPS
1	64	6260	639	982	923
2	32	10553	323	511	643
4	16	19139	172	315	332
8	8	38277	116	171	194
16	4	76554	104	115	125

5.1.4 L1 scalability

With the first level of parallelism L1, multiple contours can be solved in parallel to reduce m_0 (the number of right-hand-sides and projected eigenproblem dimension).

Table 5.2: Scaling of the **L1** level of parallelism. Together the three contours C1, C2 and C3 - defined by $(\lambda_{min} \ \lambda_{max})$ - are equivalent to the full contour C0. Each contour can be treated independently (in parallel) and calculates only a subset of eigenvalues. Sub-dividing the contour results in a speedup for the solve stage since the resulting linear-systems have fewer right-hand-sides (m_0) . The factorization and solve times for a single PFEAST iteration are recorded in the table. Results use the CNT-5 matrix system with a total of 13 MPI processes (156 cores), all placed at the L3 level.

	λ_{min}	λ_{max}	m_0	# Eig	Fact (s)	Solve (s)
C0	-500	-2	600	273	96	170
C1	-500	-100	200	66	94	59
C2	-100	-16	200	92	99	57
C3	-16	-2	200	115	96	56

If the reduced eigenvalue problem becomes too large it can not be solved with a traditional dense techniques and the contour must be sub-divided. Also, the L2 and L3 will eventually saturate and performance can not be improved placing resources at these levels of parallelism. Multiple contours at the first level L1 can then be used to increase performance by speeding up the solve stage. For this example we have again used the CNT-5 system with 78 atoms and present two separate cases. The first uses a single contour C0 to calculate all eigenmodes with $m_0 = 600$. The next case sub-divides the entire search interval into three separate pieces C1, C2 and C3 which can be calculated in parallel. This first contour C1 targets the core electronic modes within the system that are well separated from valance states. Two contours C2 and C3 are then used to calculate the valance modes, which are much more densely populated. The total time to solve the eigenvalue problem using the L1 level of parallelism would then depend on the slowest sub-contour. Because the separation between eigenvalues is small the second and third contours could exhibit slower convergence than C1 and require more PFEAST iterations. However, for the sake of comparison we only present the time of a single PFEAST iteration. Here we report the factorization time, which does not change much between the three contours, and the solve time for a single PFEAST iteration. Because m_0 is small, the time spent in the kernel is minimal for this example. Dividing the contour into three segments results in a speedup of around three for the linear system solve.

5.1.5 Weak scalability

To highlight the weak scalability of PFEAST, we present results for a variety of CNT systems. We consider only the L3 level of parallelism and place six atoms on each MPI process (each unit cell places an additional 12 atoms in the system). We then gather results, varying the length of the CNT up to 19 unit cells, using $9, 17, \ldots, 41$ L3 MPI processes (up to 492 cores) to optimally utilize the solver. The weak scalability can be seen in Figure 5.4. Large sparse eigenvalue problems usually attempt to calculate a percentage of the eigenpairs. However, with the L1 level of parallelism the number of eigenvalues (and right-hand-sides) can be held constant. To showcase the weak scalability we then use a constant value of $m_0 = 200$ for all matrix systems. The number atoms and system size are listed in the table for each molecule. The factorization and solve times for a single iteration of PFEAST are then plotted for all three solvers. We witness, for the factorization, a steeper increase in the time for the smaller systems. The factorization times then begin to taper off for larger systems. The solve time, however continue to increase for larger systems. This behavior is more pronounced for the DD-Solver and Cluster-MKL-PARDISO and is due to communication. MUMPS, however, ran out of system memory far before Cluster-MKL-PARDISO or the DD-Solver. These results show that the DD-Solver is not only faster (in total time), but has better weak scalability than the "black-box" solvers. Much larger molecular systems can then be handled through the domain-decomposition approach.

The total time to solve the eigenvalue problem will increase by an additional multiplication factor depending on the number of PFEAST iterations to reach con-



Figure 5.4: Weak scaling of factorization and solve stages (in seconds) for a single PFEAST iteration using 16 contour points (16 linear-systems solved in total using # L2=1) and 200 right-hand-sides. The matrix size is increased proportionally to the number of MPI processes. Reported timings could be reduced by a factor of 16 if L2 was fully utilized. MUMPS ran into memory issues for more than 198 atoms (757,934 size, 33 MPI).

vergence. However, if the **L2** level was fully utilized, resulting in a maximum of 7,872 cores, the absolute times could be reduced by a factor of 16 as discussed in Figure 6. In electronic structure applications, the number of wanted eigenvalues should also grow linearly with the size of the atomistic system. If the spectrum can be sliced uniformly at level **L1** as illustrated in Table III, and assuming an access to 'unlimited parallel resources' where the total number of **L3** MPI processes (and then cores) can grow linearly with the number of slices **L1**, the weak scalability and absolute timing results (that can also be divided by the number of **L2**) presented in Figure 8, shall then be preserved. Solving for all the eigenpairs of the largest system in Figure 6 that contains 246 atoms could use 5 slices each of block size 200, and result in $(\#L1) \times (\#L2) \times (\#L3) = 3,280$ total MPI processes (using #L1=5, #L2=16, #L3=41) or $3280 \times 12=39,360$ total cores (using 12 cores per MPI process).

5.2 Benchmarking full DFT and TDDFT calculations

Previous results have shown the scalability of the linear system solver and eigenvalue solvers used within the NESSIE code. These are the two major computations that take place in the DFT and TDDFT calculations. However, other operations can take a substantial amount of time. This includes computing the Hartree potential, adding the total potential into the Kohn Sham Hamiltonian and recomputing the electron density from the single particle wave functions. Each of operations has been parallelized over all three levels of MPI and with OpenMP directives.

Scalability for the major computational aspects of NESSIE are presented in Table 5.3 for five, ten and twenty unit cell CNTs containing 78, 138 and 258 atoms, respectively. Two separate results exist for each molecule. The first uses the standard NESSIE discretization. The second, labeled with a "-b", uses a coarser mesh in the molecular region and the interstitial matrix has approximately half the number of points.

The results have been gathered on a single compute node on Comet, which is comprised of two 12 core CPUs. Results presented in the table are for different configurations of N_{omp} and N_{mpi3} : the number of OpenMP threads and the number of MPI processes at L3, the third level of parallelism.

The reported total time corresponds to the observed time of a single SCF iteration without L1 or L2 and using a single FEAST contour point. Timing results for both DFT and TDDFT calculations can be extrapolated from these results. The number of FEAST contour points has been set explicitly to one for benchmarking purposes. This allows for time projections for both the eigenvalue computation and the Crank-Nicoloson time propagation, which must solve a single linear system per time step. All additional reported times correspond to operations existing in both DFT and TDDFT.

For the ground state Kohn Sham eigenvalue problem it is assumed that the L2 level of parallelism will be fully used and only the L3 level of parallelism must be considered. Additional MPI processes at L2 will have an effect on the matrix vector multiplication time t_{MV} and kernel time t_K , but not on times reported for the factorization t_F and solve t_S .

With Crank-Nicoloson the L2 parallelism level distributes the time dependent wave function (eigenvectors) and propagates them in parallel. This will not effect the factorization time, but the solve time will have ideal scaling at L1 and L2. Furthermore, only $M < M_0$ ground states need to be considered in this case, instead of the larger subspace needed with FEAST.

The time to compute the Hartree potential t_H includes calculating boundary conditions and solving the Poisson equation. The time to include the potential in the Hamiltonian t_{inc} must perform and integration over all mesh elements. And, the time to recomputing the electron density t_{n_q} sums over all occupied single particle wave functions. The times t_H , t_{inc} , t_{n_q} , t_K , and t_{MV} do not take L1 or L2 into consideration, which will have a performance benefit. The total reported time,

$$t_{total} = t_H + t_{inc} + t_n + t_F + t_S + t_{MV}, \tag{5.2}$$

is just the sum of the times for each operation. The projected time of and SCF iteration,

$$t_{SCF} = \frac{t_H + t_{inc} + t_{n_q}}{L1 * L2} + t_F + t_S + \frac{t_{MV} + t_K}{L2},$$
(5.3)

assumes eight FEAST contour points (L2 = 8) and perfect scaling at both L1 and L2 for these operations. The time for a single time step in the real-time TDDFT calculation $t_{\Delta t}$ will include t_H , t_{inc} , t_{n_q} and t_F , which will be the same as reported in the table for the SCF iteration. However, the solve time will be reduced by a factor of $M/(M_0 * L2)$, since fewer right-hand-sides must be considered (M for TDDFT compared to M_0 for SCF) and these can also be distributed over L2 MPI processes. A similar argument can be made for the matrix-vector multiplication here, but the matrices are complex (i.e. twice the size) and an additional factor of two is included in the projection.

$$t_{\Delta t} = \frac{t_H + t_{inc} + t_{n_q}}{L1 * L2} + t_F + \frac{t_S + 2 \times t_{MV}}{L2} \times \frac{M}{M_0}$$
(5.4)

Scalability for OpenMP can be seen from the (6, 1), (12, 1) and (24, 1) combinations of (N_{omp}, N_{mpi3}) . Not ideal, but still very good scaling is observed for all operations except recomputing the electron density. Scalability for L3 can be seen from the (6, 1), (6, 2) and (6, 4) and (12, 1) and (12, 2). Again, at this level fairly good scaling is seen, especially for smaller molecules. However, a major drawback of direct distributed memory linear system solvers can be witnessed. The solve time t_S has very poor scaling with L3, since communication must be performed. This is especially pronounced for larger molecules, particularly with the fine molecular grid. In many cases this performance hit negates all other gains. However, this level of parallelism is meant to address weak scalability and should be used when the matrix factorization can no longer fit into the memory of a single compute node.

Nomp	N_{mpi3}	t_H	t_{inc}	t_{n_q}	t_F	t_S	t_{MV}	t_K	t_{total}	t_{SCF}	$t_{\Delta t}$
1	1	110	24	24	138	105	92	21	514	266	158
6	1	28	12	27	36	20	32	9	154	65	43
6	2	19	8	12	23	46	23	5	136	75	29
6	4	14	5	6	18	34	11	2	90	55	22
12	1	19	9	27	24	12	23	8	122	43	29
12	2	14	5	12	18	27	12	4	92	48	22
24	1	16	7	25	20	9	25	8	110	36	25
10-CNT-3-3-b: $N_{at} = 138, N_e = 768, n = 383328, n_{it} = 111882, 2 \times (M_0 = 400)$							0)				
Nomp	N_{mpi3}	t_H	t_{inc}	t_{n_q}	t_F	t_S	t_{MV}	t_K	t_{total}	t_{SCF}	$t_{\Delta t}$
6	1	14	7	16	13	12	16	7	85	30	17
6	2	9	5	8	8	16	11	4	61	27	11
6	4	6	3	4	6	9	6	2	35	17	8
12	1	10	6	19	11	8	12	6	72	24	14
12	2	7	3	8	7	10	6	3	44	19	9
24	1	9	5	16	10	6	8	5	59	18	13
20-	CNT-3-3:	$N_{at} =$	258, Λ	$V_e = 148$	88, $n = 978312, n_{it} = 479826, 4 \times (M_0 = 500)$						
Nomp	N_{mpi3}	t_H	t_{inc}	t_{n_q}	t_F	t_S	t_{MV}	t_K	t_{total}	t_{SCF}	$t_{\Delta t}$
6	1	67	27	116	73	49	109	23	464	146	89
6	2	46	18	115	59	182	50	12	482	255	78
6	4	33	12	27	49	115	60	7	303	175	61
12	1	44	22	107	48	30	76	20	347	96	60
12	2	31	14	52	48	106	46	10	307	164	60
24	1	34	17	107	52	22	50	20	302	87	61
20-CNT-3-3-b: $N_{at} = 258, N_e = 1488, n = 713162, n_{it} = 205676, 4 \times (M_0 = 500)$											
Nomp	N_{mpi3}	t_H	t_{inc}	t_{n_q}	t_F	t_S	t_{MV}	t_K	t_{total}	t_{SCF}	$t_{\Delta t}$
6	1	35	17	108	26	38	37	17	278	76	35
6	2	21	11	105	21	50	26	9	243	80	30
6	4	15	7	18	12	29	15	5	101	45	16
12	1	23	14	79	21	17	28	14	196	47	27
12	2	15	8	35	14	21	15	7	115	40	18
24	1	19	12	74	19	13	18	13	168	38	24

10-CNT-3-3: $N_{at} = 138$, $N_e = 768$, n = 531244, $n_{it} = 259798$, $2 \times (M_0 = 400)$

Table 5.3: Performance results for major operations within a single NESSIE SCF iteration; from left to right: (t_H) computing the Hartree potential, (t_{inc}) including the potential into the Hamiltonian, (t_{n_q}) recomputing the electron density, and (t_F, t_S) and t_{MV} factorization, solve and mat-vec stages in FEAST. It is assumed c FEAST contours will be used (where $c \times M_0$ is specified under each test). Projections assume that L2 = 8 and ideal scaling of t_H , t_{inc} , t_{n_q} , and t_{MV} . The "-b" molecules use a coarser grid in the molecular region (set to 0.03) and have approximately half the number of interstitial points.

CHAPTER 6 CONCLUSION

This dissertation has presented a complete parallel framework for density functional theory and time-dependent density functional theory calculations. Specific contributions outlined in this dissertation are focused on parallel scalability for DFT and TDDFT. DFT is the general method applied to electronic structure calculation involving more than a handful of atoms. A real-space real-time approach for TDDFT is a scalable method for excited states calculations and can be applied to molecules containing up to a thousand atoms with the current NESSIE framework. The NESSIE code has been parallelized with a combination of MPI (for distributed memory) and OpenMP (for sharded memory).

Details on the modeling framework were presented in Chapter 2 including background on the physical model (DFT and TDDFT), the discretization (finite element combined with a muffin-tin domain decomposition) and numerical algorithms for solving linear algebra problems (mainly for eigenvalues and linear systems). An example for running the code was the subject of Chapter 3 and provides a step-by-step process for running the code and details for DFT and TDDFT calculations. Chapter 4 presented results for a collection of small molecules as well as a more detailed investigation of (3,3) carbon nano-tubes up to forty unit cells. Benchmarking of the codes performance and scalability was then shown in Chapter 5.

Scalability in NESSIE has been achieved through the creation of parallel numerical algorithms for linear systems and eigenvalue problems. A Schur complement linear solver has been developed for linear systems arsing from the domain decomposition. The PFEAST eigenvalue solver has also been created and integrated with the numerical simulation procedure for ground state DFT calculations. PFEAST is a stand-alone package that will be released in v4.0 of the FEAST Eigenvalue Solver project and contains three levels of MPI parallelism as well as an additional level of threading. In particular, it allows one to link a distributed memory linear solver with FEAST and is appropriate for domain decomposition approaches like the one employed in NESSIE. Since the eigenvalue problem in DFT is particularly difficult due to the number of eigenpairs that must be calculated, the FEAST algorithm is an ideal choice of algorithm as it allows for slices of eigenvalues and vectors to be computed independently. This feature is necessary for simulations of large scale molecules with more than a thousand electrons and used extensively for the large-scale results in this dissertation. As PFEAST is the backbone of NESSIE, all other aspects of the software package have been parallelized over the three levels of MPI inherent to the algorithm to achieve good strong scalability of all mesh and matrix operations.

Results for small molecules have also been presented and compared directly to experimental data, for which some remarkable agreement can be witnessed. The capability of the code for large scale molecules has been demonstrated for a set of carbon nano-tubes. These large-scale results were made possible by the weak scalability of NESSIE on distributed memory machines, where more computing resources were employed to solve larger problems. The investigation of plasmon resonances in carbon nano-tubes produced interesting results which seem to hint at the presence of a Luttinger-liquid.

Future directions in NESSIE could investigate X-ray absorption with little changes required to the approach or to the software. Photoabsorption in the infrared frequency range could also be handled within the real-space real-time framework. Simple vibrations could be studied with only small changes in the current NESSIE implementation, but challenges exist for a more general approach which would require mesh moving techniques to allow for the actual movement of atoms. Furthermore, the current NESSIE implementation can now investigate many interesting large-scale molecules with up to one thousand atoms. This opens up interesting research directions in the nanoengineering and biological application spaces.

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APPENDIX

NON-HERMITIAN FEAST

Details of the non-Hermitian FEAST algorithm will now be discussed. The FEASTv3.0 software package can not be applied to non-Hermitian matrices. Modifications to the algorithm are described here along with software implementation details. In the following, all quantities associated with the left eigenvectors will be written with a '^' symbol (e.g. \hat{X} , \hat{Y} and \hat{Q}).

Defining a search interval

A key feature of FEAST is the ability to calculate a subset of eigenvalues that exist within some interval. Figure 6.1 summarizes the different search contour options possible for both the Hermitian and non-Hermitian FEAST algorithms.

For the Hermitian case, the user must then specify a 1-dimensional real-valued search interval $[\lambda_{min}, \lambda_{max}]$. These two points are used to define a circular or ellipsoid contour C centered on the real axis and along which the complex integration nodes are generated. The choice of a particular quadrature rule will lead to a different set of positions for the nodes and associated quadrature weights i.e. (z_j, ω_j) . Since the eigenvalues are real, it is convenient to select a contour symmetric to the real axis (i.e. $C = C^*$) since the symmetry enables one to integrate over only half the contour (e.g. upper half).

With a non-Hermitian problem, it is necessary to specify a 2-dimensional search contour that surrounds the wanted complex eigenvalues. Circular or ellipsoid contours can also be used and they can be generated using standard options included into FEAST v3.0. These are defined by a complex midpoint λ_{mid} and a radius r for a circle (for an ellipse the ratio between the horizontal axis and vertical axis diameter can also be specified, as well as an angle of rotation). However, in some applications where the eigenvalues of interest belong to a particular subset in the complex plane, more flexibility for selecting a search contour with arbitrary shape could be needed. This option also lends itself to parallelism, where a large number of eigenvalues can be calculated by partitioning the complex plane into multiple contours. Consequently, a "Custom Contour" feature is also supported in FEAST v3.0 that allows for arbitrary quadrature nodes and weights.



Figure 6.1: Various search contour examples for the Hermitian and the non-Hermitian FEAST algorithms. Both algorithms feature standard elliptical contour options and the possibility to define custom arbitrary shapes. In the Hermitian case, the contour is symmetric with the real axis and only the nodes in the upper-half may be generated. n the non-Hermitian case, a full contour is needed to enclose the wanted complex eigenvalues.

Dual subspaces: left and right spectral projectors

In this section, in addition to the right spectral projector $\rho(B^{-1}A)$ defined in (2.85), we will be introducing a left spectral projector $\rho(AB^{-1})$. In order to simplify the discussion and offer some continuity with the notation used for the Hermitian FEAST, we are here assuming that B is non-singular. The non-Hermitian FEAST algorithm, however, is expected to work for cases where B is singular, assuming that the zB - A is non-singular in the expression of the spectral projector and the targeted eigenpairs are sufficiently well-conditioned. Since the non-Hermitian system may be defective, the filtering function is applied to the Jordan normal form:

$$\rho(B^{-1}A) = X\rho(J)X^{-1}.$$
(6.1)

When applied to each Jordan block J_k , the expression of the operator becomes [94]:

$$\rho(J_k) = \rho\left(\begin{bmatrix} \lambda_k & 1 & \dots & 0 \\ 0 & \lambda_k & \ddots & \vdots \\ \vdots & \ddots & \ddots & 1 \\ 0 & \dots & 0 & \lambda_k \end{bmatrix} \right) = \begin{bmatrix} \rho(\lambda_k) & \frac{\rho'(\lambda_k)}{0!} & \dots & \frac{\rho^{(m)}(\lambda_k)}{(m-1)!} \\ 0 & \rho(\lambda_k) & \ddots & \vdots \\ \vdots & \ddots & \ddots & \frac{\rho'(\lambda_k)}{0!} \\ 0 & \dots & 0 & \rho(\lambda_k) \end{bmatrix}$$
(6.2)

Using the Cauchy integral formula (2.84), the diagonal elements of J_k take the values one or zero, while all derivatives (i.e. off-diagonal elements) are zero. In practice, this may not be guaranteed with FEAST as the filter is approximated by the rational function (2.90). A generalization of the algorithm for addressing defective systems requires further studies, and our current FEAST non-Hermitian algorithm assumes that the Jordan form reduces to an eigenvalue decomposition. Consequently, we consider:

$$\rho(B^{-1}A) = X\rho(\Lambda)X^{-1} \equiv X\rho(\Lambda)\hat{X}^H B, \qquad (6.3)$$

where the left and right eigensubspaces satisfy the *B*-bi-orthonormal relationship i.e. $\hat{X}^H B X = I$. For the case of the Hermitian problem, $\hat{X} = X$ and the relation (2.86) can then be recovered. It is also important to mention the particular case of complex symmetric systems (i.e. $A = A^T$ and $B = B^T$) which leads to $\hat{X} = X^*$. In general, however, the left and right vectors are not straightforwardly related and they must be calculated explicitly.

From (2.84), (2.86), and (6.3), one can compute the right spectral projector $\rho(B^{-1}A)$ for the right eigenvector subspace X_m (i.e. $\rho(B^{-1}A)X_m = X_m$) as follow:

$$\rho(B^{-1}A) = \frac{1}{2\pi i} \oint_{\mathcal{C}} dz (zB - A)^{-1}B \equiv X_m \widehat{X}_m^H B.$$
(6.4)

For the treatment of the left eigenvector subspace solution of $\widehat{X}^{H}A = \Lambda \widehat{X}^{H}B$, it is first convenient to define the following eigenvalue decomposition:

$$\rho(AB^{-1}) = \widehat{X}^{-H} \rho(\Lambda) \widehat{X}^{H} \equiv B X \rho(\Lambda) \widehat{X}^{H}.$$
(6.5)

One can then construct the left spectral projector $\rho(AB^{-1})$ (i.e. $\widehat{X}_m^H = \widehat{X}_m^H \rho(AB^{-1})$) as:

$$\rho(AB^{-1}) = \frac{1}{2\pi i} \oint_{\mathcal{C}} dz B (zB - A)^{-1} \equiv B X_m \widehat{X}_m^H.$$
(6.6)

In FEAST, the projectors are formulated using the rational function ρ_a (2.90) along with the quadrature nodes and weights $(z_j, \omega_j)_{1 \le j \le n_e}$ that approximate the contour integrations in (6.4) and (6.6). The right and left subspaces Q_{m_0} and \hat{Q}_{m_0} are then obtained by applying the right and left projectors onto a set of m_0 vectors i.e.

$$Q_{m_0} = \rho_a (B^{-1}A) Y_{m_0} = \sum_{j=1}^{n_e} \omega_j (z_j B - A)^{-1} B Y_{m_0} \equiv X \rho_a(\Lambda) \widehat{X}^H B Y_{m_0}.$$
 (6.7)

and

$$\widehat{Q}_{m_0}^H = \widehat{Y}_{m_0}^H \rho_a(AB^{-1}) = \sum_{j=1}^{n_e} \omega_j \widehat{Y}_{m_0}^H B(z_j B - A)^{-1} \equiv \widehat{Y}_{m_0}^H B X \rho_a(\Lambda) \widehat{X}^H.$$
(6.8)

In practice, the calculation of both subspaces require solving a series of linear systems. For the right subspace,

$$Q_{m_0} = \sum_{j=1}^{n_e} \omega_j Q_{m_0}^{(j)}, \quad \text{with } Q_{m_0}^{(j)} \text{ solution of } (z_j B - A) Q_{m_0}^{(j)} = B Y_{m_0}, \tag{6.9}$$

which was already outlined in (2.92), and for the left subspace:

$$\widehat{Q}_{m_0} = \sum_{j=1}^{n_e} \omega_j^* \widehat{Q}_{m_0}^{(j)}, \quad \text{with } \widehat{Q}_{m_0}^{(j)} \text{ solution of } (z_j B - A)^H \widehat{Q}_{m_0}^{(j)} = B^H \widehat{Y}_{m_0}. \quad (6.10)$$

These numerical operations are also described in Figure 2.7. As a result of (6.7) and (6.8), Q_{m_0} (resp. \hat{Q}_{m_0}) is formed by a linear combinations of the columns of X_{m_0} (resp. \hat{X}_{m_0}). The Rayleigh-Ritz procedure involves the reduced matrices A_Q and B_Q formed by projecting on the right with a subspace containing the right eigenvectors Q_{m_0} , and projecting on the left with a subspace containing the left eigenvectors \hat{Q}_{m_0} . The resulting non-Hermitian reduced system can be solved using the QZ algorithm [147] in LAPACK [17] to yield the right and left eigenvectors W and \hat{W} defined in Figure 2.7. The long right (resp. left) Ritz vectors can then be recovered as $Y_{m_0} = Q_{m_0}W$ (resp. $\hat{Y}_{m_0} = \hat{Q}_{m_0}\hat{W}$) and used as initial guess subspaces for the next FEAST iterations until convergence.

Convergence for 2-dimensional search intervals

In our implementation of FEAST, the convergence criterion is satisfied if the norm of the relative residual associated with the eigenpairs (x_i, λ_i) and (\hat{x}_i, λ_i^*) , is found below an arbitrary threshold ϵ i.e.

$$\operatorname{res}_{i} = \max\left\{\frac{||Ax_{i} - \lambda_{i}Bx_{i}||_{1}}{||\alpha Bx_{i}||_{1}}, \frac{||A^{H}\widehat{x}_{i} - \lambda_{i}^{*}B^{H}\widehat{x}_{i}||_{1}}{||\alpha B^{H}\widehat{x}_{i}||_{1}}\right\} < \epsilon,$$
(6.11)

where the value of ϵ can be chosen typically equal to 10^{-13} if high accuracy is needed using double precision arithmetic. The parameter α is relative to the eigenvalue range in the search contour. The latter is defined differently for the Hermitian and non-Hermitian cases (as discussed in Section 6), and a non-zero value for α can be chosen as $\alpha = \max(|\lambda_{min}|, |\lambda_{max}|)$ for the Hermitian case and $\alpha = (|\lambda_{mid}| + r)$ for the non-Hermitian case.

As discussed in the introduction, the right/left eigenvectors associated with λ_i with $i = 1, \ldots, m$ (and hence all associated residuals res_i) are expected to converge

linearly along the FEAST subspace iterations at the rate: $|\rho_a(\lambda_{m_0+1})/\rho_a(\lambda_i)|$ for $i = 1, \ldots, m$. The convergence depends then on both the subspace size $m_0 \ (m_0 \ge m)$ and the accuracy of the rational filter ρ_a (2.90) that should ideally provide values very close to unity for eigenvalues on the interior of the search contour and zero elsewhere. Although quite effective, the Gauss-quadrature approach along a circular contour that was proposed in the original FEAST article [161], is clearly not the only possible choice for optimizing the convergence ratio. Three other options have already been considered for the Hermitian problem including [87]: (i) the trapezoidal rule; (ii) different contour shapes beside a circle such as a finely tuned flat ellipse; (iii) a new approximation of the spectral projector based on a Zolotarev approximant to the sign function which, after transformations, provides complex poles on the unit circle [223, 86]. Both Gauss and Zolotarev are well-suited choices for the Hermitian problem since they favor an accentuation of the decay of $|\rho_a|$ at the boundaries of the interval along the real axis. The trapezoidal rule is well-known for its exponential convergence property with the number of integration nodes n_e [203]. It is also expected to provide more consistency for capturing the complex eigenvalues of the non-Hermitian problem since, unlike Gauss and Zolotarev quadratures, the nodes are placed an equal distance apart along a circular contour. This leads to a similar decay for $|\rho_a|$ along arbitrary (equivalent) directions from the contour center.

Similarly to the Hermitian problem, the non-Hermitian FEAST algorithm also requires as input a search subspace of size m_0 chosen not smaller than the number of eigenvalues m within a given complex contour. If $m_0 \ (m_0 \ge m)$ is chosen too small, the ratio governing the convergence may be closer to one, leading to slow convergence. Alternatively, if m_0 is chosen too large, it may result in an unnecessary high number of right-hand-sides when solving the shifted linear systems in (6.9) and (6.10). We illustrate the dependence of the convergence rate on m_0 with two example with two examples. These tests use the QC324 matrix from the NEP collection [26]. A contour it has been created with a single eigenvalue λ_1 inside. The contour and few closest eigenvalues can be seen in Figure 6.2. The rational function $|\rho_a|$ which has been generated using a six-point trapezoidal rule is also shown in the figure (as a contour plot on the left and a 3-D surface plot on the right). Figure 6.3 shows the convergence of the relative residual norms for all the m_0 eigenvalues along the FEAST subspace iterations in the cases $m_0 = 2$ (left plot) and $m_0 = 4$ (right plot). For $m_0 = 2$, λ_3 is the closest eigenvalue outside of the search subspace and controls the convergence rate. Since this eigenvalue is relatively close to the contour, FEAST exhibits slow convergence. The case $m_0 = 4$, in turn, leads to drastic improvement in the convergence rate which benefits from the small values of $\rho_a(\lambda_5)$.

A typical recommended choice for the search subspace size is $m_o = 2m$. In practice, however, the exact number of eigenvalues m is unknown beforehand and the user must make an educated guess. Alternatively, m can also be estimated using, for example, the fast stochastic estimation procedure [66] that has been recently introduced in FEAST v3.0. It is important to note that in some situations slow convergence can result if the value of m_0 is only large enough to include the external eigenvalues bordering the contour. This problem can arise when the eigenvalues of interest are near a continuum or cluster of unwanted eigenvalues. With many unwanted eigenvalues



Figure 6.2: Value of the rational function, for the QC325 matrix, plotted as contour plot (left) and surface plot (right). The contour has been generated using a six-point trapezoidal rule and the FEAST 'custom contour' feature to position the quadrature nodes along a circular arc. The left plot includes the positions of the four closest eigenvalues. Only a single eigenvalue λ_1 is inside of the contour. More particularly, $|\rho_a(\lambda_1)| = 1.0000004$, $|\rho_a(\lambda_2)| = 1.7272309$, $|\rho_a(\lambda_3)| = 0.4206553$, $|\rho_a(\lambda_4)| = 3.6296209 \times 10^{-2}$, and $|\rho_a(\lambda_5)| = 6.9332547 \times 10^{-3}$. Note that λ_5 is not visible in the figure.

closely bordering the contour, it may not be possible to improve convergence by increasing the subspace size m_0 . In this case, using additional integration nodes to increase the accuracy of ρ_a may be necessary. A utility routine for calculating the rational function has also been included in FEAST v3.0 and can be used to investigate convergence for different contours and eigenvalue distributions.

Finally, and in contrast to the Hermitian problem where the contour nodes can be placed away from the eigenvalues (i.e. far enough from the real axis), a contour node could end up being located in the vicinity of a complex eigenvalue. In this case the rational function could take on values larger than one, and it then becomes possible for an eigenvalue outside of the contour to converge at a faster rate than the wanted eigenvalues inside. This is what is happening to λ_2 in Figures 6.2 and 6.3. If a quadrature node is located too close to an eigenvalue, however, it is likely to worsen the conditioning of the corresponding shifted linear system in (6.9) and (6.10), making the problem more challenging to solve using an iterative method.



Figure 6.3: Convergence of the residual norms (6.11) associated with eigenvalues λ_i in Figure 6.2. Two search subspace sizes are considered: $m_0 = 2$ (left plot) and $m_0 = 4$ (right plot). The dashed lines represent the theoretical linear convergence rate $|\rho_a(\lambda_{m_0+1})/\rho_a(\lambda_i)|$ which is perfectly matched by the values returned by FEAST. We note that the convergence of the wanted eigenvalue λ_1 is is considerably slower using the smaller size subspace $m_0 = 2$ since the eigenvalue λ_3 that governs the convergence rate for this case is too close to the search contour.

Resizing the subspace

The rank of the subspaces Q_{m_0} (6.7) and \widehat{Q}_{m_0} (6.8) is greater than or equal to the number of wanted eigenvalues m ($m_0 \ge m$) since their columns contain components from eigenpairs outside of the contour due to inaccuracies in the numerical integration. If m_0 is too great an overestimate of m, these matrices may be numerically rank deficient, and the reduced matrix $B_Q = \widehat{Q}_{m_0}^H B Q_{m_0}$ may be singular. Consequently, m_0 must be adjusted at runtime. Otherwise, the QZ algorithm used in the solution of the reduced system can produce infinite eigenvalue solutions [147]. Re-injecting these solutions into the subspace iteration would cause problems for the algorithm. The upper bound for the choice of m_0 should be the largest value before the subspaces become numerically rank deficient. One possible way to determine this threshold value consists of performing the spectral decomposition of B_Q and analyzing its eigenvalues. This approach is particularly well suited for our non-Hermitian algorithm since the spectral decomposition will be re-used at a later stage. It comes:

$$B_Q = V \Gamma \widehat{V}^H, \tag{6.12}$$

where Γ is the diagonal matrix for the eigenvalues $\{\gamma_i\}_{i=1,\dots,m_0}$, and V and \hat{V} are respectively the corresponding left and right bi-orthonormal eigenvector subspaces (i.e. $\hat{V}^H V = I_{m_0}$). For the Hermitian case where B and hence B_Q must be positive definite, this step can replaced by monitoring the failure of the Cholesky factorization of B_Q that could return a negative pivot. The position of the latter helped determining the threshold value for m_0 used to resize the subspace accordingly. For the non-Hermitian problem, the matrix B_Q is singular if there exists an eigenvalue equal to zero. In finite precision arithmetic, a zero eigenvalue must be characterized relatively i.e.

$$|\gamma_i| < \eta * \max(|\gamma_1|, ..., |\gamma_{m_0}|), \qquad (6.13)$$

where η is on the order of machine precision; e.g. 10^{-16} in double precision. If an eigenvalue γ_i is 16 orders-of-magnitude smaller than the maximum eigenvalue then it is out of range for the double precision arithmetic and is counted as a zero. The subspace size m_0 is reduced to \tilde{m}_0 such that B_Q has no eigenvalues satisfying (6.13). The spectral decomposition of B_Q is computed at each FEAST iteration, and as will be discussed in the next section, the resizing is performed in conjunction with a *B*-bi-orthonormalization for the subspaces $Q_{\tilde{m}_0}$ and $\hat{Q}_{\tilde{m}_0}$. The additional numerical cost of diagonalizing B_Q is on the order of (but less expensive than) the cost associated with the diagonalization of the reduced generalized system.

As a side remark, it interesting to note that using (6.7) and (6.8), B_Q can also be written as:

$$B_Q = (\hat{Y}_{m_0}^H BX) \rho_a^2(\Lambda) (\hat{X}^H BY_{m_0}).$$
(6.14)

Starting from the second FEAST iteration where the Ritz vectors Y_{m_0} and \hat{Y}_{m_0} not only are approximately spanned respectively by the true eigenvector subspaces X and \hat{X} but they also satisfy the property of B-bi-orthonormality (i.e. $\hat{Y}_{m_0}^H BY_{m_0} = I$ since $\widehat{W}^H B_Q W = I$ in Figure 2.7), it is possible to directly identify (6.14) with (6.12). It comes that $V = \hat{Y}_{m_0}^H BX$, $\hat{V}^H = \hat{X}^H BY_{m_0}$, and $\Gamma = \rho_a^2(\Lambda)$. The latter indicates that the eigenvalues of B_Q are related to the rational function ρ_a , and can then be used to estimate the convergence rate [197].

B-biorthonormalization of projectors

The intended result of FEAST is a set of *B*-bi-orthonormal vectors. However, the *B*-bi-orthogonality is not guaranteed after the contour integration due to numerical inaccuracies. This is especially pronounced in large problems in which many eigenvalues border the search contour. The contour integration could potentially include a large number of mixed states from the continuum in the subspaces Q_{m_0} (6.7) and \hat{Q}_{m_0} (6.8). In our numerical experiments, we have found that an explicit *B*-bi-orthonormalization of the FEAST subspaces Q_{m_0} and \hat{Q}_{m_0} helps improve the stability of the algorithm. Rather than performing a QR factorization or SVD of the subspaces, we are taking advantage of the eigen-decomposition of B_Q (6.12) that is already performed in FEAST as discussed in the previous section. From (6.12) and since $B_Q = \hat{Q}_{m_0}^H B Q_{m_0}$, it comes:

$$\Gamma = \hat{V}^H B_Q V = (\hat{V}^H \hat{Q}_{m_0}^H) B(Q_{m_0} V) \equiv (\hat{Q}_{m_0} \hat{V})^H B(Q_{m_0} V).$$
(6.15)

As a result, *B*-bi-orthonormal subspaces U_{m_0} and \hat{U}_{m_0} can be generated by updating the current subspaces Q_{m_0} and \hat{Q}_{m_0} as follows:

$$U_{m_0} = Q_{m_0} V \Gamma^{-1/2}, \qquad \widehat{U}_{m_0} = \widehat{Q}_{m_0} \widehat{V} \Gamma^{-H/2}.$$
(6.16)

As discussed in the previous section, the subspace size m_0 may have already been reduced to \tilde{m}_0 at this stage by allowing the eigenvectors in V and \hat{V} corresponding to the zero eigenvalues in Γ to be removed from the subspace. In practice, a subset of V and \hat{V} composed of \tilde{m}_0 column vectors can be easily extracted if the eigenpairs $\{\gamma_i, v_i \equiv V e_i, \hat{v}_i \equiv \hat{V} e_i\}_{i=1,...,m_0}$ are first sorted by decreasing values of $|\gamma_i|$. Denoting $V_{m_0 \times \tilde{m}_0}$ and $\hat{V}_{m_0 \times \tilde{m}_0}$ the subsets of the new V and \hat{V} subspaces restricted to their first \tilde{m}_0 columns, and $\Gamma_{\tilde{m}_0 \times \tilde{m}_0}$ the matrix of the first \tilde{m}_0 sorted eigenvalues, (6.16) becomes:

$$U_{\tilde{m}_{0}} = Q_{m_{0}} V_{m_{0} \times \tilde{m}_{0}} \Gamma_{\tilde{m}_{0} \times \tilde{m}_{0}}^{-1/2}, \qquad \widehat{U}_{\tilde{m}_{0}} = \widehat{Q}_{m_{0}} \widehat{V}_{m_{0} \times \tilde{m}_{0}} \Gamma_{\tilde{m}_{0} \times \tilde{m}_{0}}^{-H/2}.$$
(6.17)

Thereafter, the matrices of the reduced system can be obtained using a new Rayleigh-Ritz projection for A and B i.e. $B_U = \widehat{U}_{\widetilde{m}_0}^H B U_{\widetilde{m}_0}$ and $A_U = \widehat{U}_{\widetilde{m}_0}^H A U_{\widetilde{m}_0}$. In spite of our *B*-bi-orthonormalization procedure, the resulting B_U is not necessarily exactly identity, or even diagonal, due to numerical inaccuracies and finite precision arithmetic. However, this procedure is beneficial as a precursor to the QZ algorithm for solving the reduced generalized system, since it can help reducing the "overrepresentation" of the unwanted eigenpairs that lie close to the contour. The benefits of B-bi-orthonormalization can be seen in Figure 6.4. This test has uses the CSH4 matrix [47], an 801×801 complex scaled Hamiltonian from the BigDFT electronic structure code [79]. The eigenspectrum and the desired eigenvalues inside of a FEAST custom contour can be seen on the left side of Figure 6.4. One edge of the contour is parallel to the eigenvalue continuum. This results in a large number of mixed states after spectral projections in (6.7) and (6.8). Without bi-orthonormalization, the QZ algorithm fails to return a B-bi-orthogonal set of eigenvectors for large values of m_0 . The minimum obtained residual norm then degrades for larger subspace sizes. By employing bi-orthogonalization the QZ algorithm is more stable and is able to return a *B*-bi-orthogonal set. The minimum obtained norm remains constant for all m_0 values as shown in Figure 6.4 (right plot). Note that the B_Q matrix remains non-singular for all values of m_0 and no resizing operations have then been performed (i.e. $B_U \equiv B_Q$).



Figure 6.4: On the left: eigenvalue spectrum of CSH4. On the right: minimum obtained residual norm (6.11) after 20 FEAST iterations plotted in function of the subspace size m_0 . With bi-orthonormalization, the minimum norm stays relatively constant for all m_0 .

Removal of spurious solutions

In certain situations incorrect eigenvalues, so called spurious solutions, appear inside of the FEAST contour. These spurious eigenvalues do not converge. It is important to note that the corresponding spurious eigenvectors do not need to be explicitly removed from the search subspace to guarantee that the true solutions will converge as the FEAST iterations progress. Spurious solutions can be flagged a posteriori once FEAST has converged. This problem, however, leads to the practical issue of devising a suitable convergence test.

In FEAST v2.1 for the Hermitian case using Gauss quadrature along a circular contour, the true number of eigenvalues m could be obtained by counting the eigenvalues of B_Q (see (6.12) using $\hat{V} = V$) satisfying the condition $|\gamma_i| \leq 1/4$ [197, 76] (i.e. $|\rho_a(\lambda_i)| \leq 1/2$ from (6.14)) which guaranteed that λ_i is a true eigenvalue within $[\lambda_{min}, \lambda_{max}]$. Since FEAST v3.0 allows for a custom contour in the complex plane, it is not possible to perform a similar test by simply analyzing the values $|\gamma_i|$. A new strategy has been developed, which can be used to provide increasingly better estimates of the number of true eigenvalue solutions in the search subspace at each subsequent iteration.

By definition, if a Ritz eigenpair $(\lambda_i, y_i, \hat{y}_i)$ obtained after solving the reduced system is a genuine eigenpair of the matrix pencil (A,B), then $(\rho(\lambda_i), y_i, \hat{y}_i)$ yields eigenpairs of $\rho(B^{-1}A)$ (6.3) and $\rho(AB^{-1})$ (6.5). In practice, one can perform a comparison between a direct calculation of $\rho(\lambda_i)$ where λ_i is the Ritz value, and the value $\rho(\lambda_i)$ obtained by solving $\rho(B^{-1}A)y_i \simeq \rho(\lambda_i)y_i$ (which is only approximate if the Ritz vectors have not yet converged). A suitable choice for the function ρ should allow these two values for $\rho(\lambda_i)$ to differ significantly if λ_i is spurious, with the condition that $\hat{y}_i^H B \rho(B^{-1}A)y_i \simeq \rho(\lambda_i)$ can also be easily calculated. The choice of the approximate spectral projector ρ_a^2 (2.88) satisfies both conditions. Using (6.7) and (6.14):

$$\rho_a^2(\lambda_i) \simeq \hat{y}_i^H B \rho_a^2(B^{-1}A) y_i = \hat{y}_i^H B X \rho_a^2(\Lambda) \hat{X}^H B y_i \equiv [B_Q]_{i,i}, \qquad (6.18)$$

where $[B_Q]_{ii}$ denotes the i^{th} diagonal element of B_Q . Our identification procedure for the spurious solutions can then be summarized by the following three steps:

1. Compute the corresponding $\{\rho_a(\lambda_i)\}_{i=1,\dots,m_0}$ using (2.88) and the Ritz values solution of the reduced system $\{\lambda_i\}_{i=1,\dots,m_0}$; i.e.

$$\rho_a(\lambda_i) = \sum_{j=1}^{n_e} \frac{\omega_j}{z_j - \lambda_i},\tag{6.19}$$

- 2. Form the Ritz vectors and wait for the contour integration to be performed and B_Q constructed at the next FEAST iteration.
- 3. Compare the calculated values of $\rho_a(\lambda_i)$ with the corresponding diagonal values of B_Q (which are already sorted), and label λ_i as spurious if it satisfies the following inequality:

$$\left|\frac{\rho_a^2(\lambda_i) - [B]_{ii}}{\rho_a^2(\lambda_i)}\right| \ge \mu,\tag{6.20}$$

where μ is empirically chosen to be 10^{-1} . We have found that this criterion is both large enough to flag all the spurious solutions, and small enough to ensure that true solutions are not mislabeled as soon as they start converging.

Once a Ritz eigenpair is flagged as spurious, it is kept in the search subspace but it is not accounted for in the test for the residual convergence (6.11). On exit, however, FEAST uses a sorting procedure on the subspace to return the converged eigenpairs free from spurious solutions.

Complete non-Hermitian algorithm

The algorithm in Figure 6.5 provides a complete description of non-Hermitian FEAST. The algorithm is divided into six stages from initialization to convergence test, that further detail the different numerical operations in Figure 2.7. If the non-Hermitian eigenvalue problem is non-defective, FEAST is expected to converge and return the wanted eigenvalues associated with the *B*-bi-orthonormal right and left eigenvector subspaces. The convergence rate that was discussed in Section 6 depends on the quality of the filter to approximate the spectral projector, and the size of the search subspace (hence it depends on the number of the contour points n_e , and subspace size m_0).
Solving: $AX_m = BX_m\Lambda_m$ and $A^H\widehat{X}_m = B^H\widehat{X}_m\Lambda_m^*$ with $[\Lambda_m]_{ii} \subseteq \mathcal{C}$

Inputs: A and B general matrices in $\mathbb{C}^{n \times n}$; Search subspace size $m_0 \ge m$; Search contour nodes/weights $\{z_1, \ldots, z_{n_e}\}, \{\omega_1, \ldots, \omega_{n_e}\}$

0- Initialization

- **0.a** Choose m_0 independent vectors $Y_{m_0} = \{y_1, \ldots, y_{m_0}\}_{n \times m_0}$ (random or initial guess)
- **0.b** Choose m_0 independent vectors $\widehat{Y}_{m_0} = {\widehat{y}_1, \ldots, \widehat{y}_{m_0}}_{n \times m_0}$ (random or initial guess)

1- Contour Integration (optimization schemes detailed in Section 6)

For each pair (z_j, ω_j) **1.a** Solve: $(z_j B - A)Q_{m_0}^{(j)} = BY_{m_0} \longrightarrow Q_{m_0} = Q_{m_0} + \omega_j Q_{m_0}^{(j)}$ **1.b** Solve: $(z_j B - A)^H \widehat{Q}_{m_0}^{(j)} = B^H \widehat{Y}_{m_0} \longrightarrow \widehat{Q}_{m_0} = \widehat{Q}_{m_0} + \omega_j^* \widehat{Q}_{m_0}^{(j)}$

2- Spurious Detection

- **2.a** Form the projected matrix $B_Q = \widehat{Q}_{m_0}^H B Q_{m_0}$
- **2.b** Identify the number of spurious solutions m_s starting from the second FEAST iteration (Section 6)

3- Resize and B-bi-orthonormalization

- **3.a** Perform the spectral decomposition $B_Q = V \Gamma \widehat{V}^H$
- **3.b** Define new subspace dimension \widetilde{m}_0 if needed (Section 6)
- **3.c** Extract the \widetilde{m}_0 columns of V and \widehat{V} with largest magnitude entries of Γ_{m_0}
- **3.d** Form *B*-bi-orthonormal subspaces $U_{\widetilde{m}_0}$ and $\widehat{U}_{\widetilde{m}_0}$ (Section 6)

4- Rayleigh-Ritz Procedure

- **4.a** Form the matrices $B_U = \widehat{U}_{\widetilde{m}_0}^H B U_{\widetilde{m}_0}$ and $A_U = \widehat{U}_{\widetilde{m}_0}^H A U_{\widetilde{m}_0}$
- **4.b** Solve $A_U W = B_U W \Lambda_U$ and $A_U^H \widehat{W} = B_U^H \widehat{W} \Lambda_U^*$; with $\widehat{W}^H B_U W = I$
- **4.c** Compute $\rho_a(\lambda_i)$ (2.88) for the Ritz values $(\lambda_i = [\Lambda_U]_{ii})$; To be used by Step-**2.b**
- **4.d** Compute Ritz vectors $Y_{\widetilde{m}_0} = U_{\widetilde{m}_0}W$ and $\widehat{Y}_{\widetilde{m}_0} = \widehat{U}_{\widetilde{m}_0}\widehat{W}$

5- Convergence Test

- **5.a** Find the number of Ritz values m_r located inside the search contour
- **5.b** Compute the residuals (6.11) of the corresponding m_r eigenpairs
- **5.c** If convergence criteria is not reached for the $m = (m_r m_s)$ lowest calculated residuals, begin next iteration at Step-1 with $\tilde{m}_0 \to m_0$
- **5.d** Place the converged eigenpairs within the first m columns of Y_m , \hat{Y}_m and Λ_{Q_m} , and exit

Output: $X_m \equiv Y_m$; $\widehat{X}_m = \widehat{Y}_m$; $\widehat{X}_m^H B X_m = I_m$; $\Lambda_m \equiv \Lambda_{Q_m}$



Limitations of the algorithm

- Ill-conditioned linear systems In contrast to Hermitian FEAST which allows the selection of complex shifts (contour points) that are not located on the real axis, some of these shifts could potentially come close to a complex eigenvalue using non-Hermitian FEAST. Similar to a traditional (Hermitian or non-Hermitian) Arnoldi algorithm using shift-and-invert strategy, the resulting linear systems may become ill-conditioned. This can result in accuracy loss for all eigenvalues and, if the shift happens to be at the exact position of an eigenvalue, the linear system will also be singular. One practical solution of this problem consists of moving or dropping the problematic quadrature node (see, e.g. [149, 150]) which could be achieved by analyzing the eigenspectrum onthe-fly.
- **Defective system** Currently if the system is defective, the QZ algorithm used to solve the reduced system in Step-4b of Figure 6.5 will not produce a set of *B*-bi-orthogonal subspaces. In practice, the algorithm may still converge (without Step-2), but further studies are required to analyze the action of the approximate spectral projector on the Jordan form (6.1) and (6.2).
- **Ill-conditioned eigenvalue problem** Non-Hermitian systems can have poorly conditioned eigenvalues [81]. A well-known case is the real non-symmetric Grear matrix [202, 67] (e.g. with n = 100), which has extremely sensitive eigenvalues. There are some noticeable differences in the eigenvalues calculated using LAPACK-MATLAB, while comparing between the eigenvalue solutions of the matrix and its transpose. To get double precision accuracy for larger values of n, the computation must be carried out in higher-precision arithmetic. Interestingly, when FEAST operates on the Grear matrix (n = 100) or its transpose, the problem of sensitivity of the eigenvalues is not observed in any selected regions of the complex plane. For this matrix case, the projected reduced eigenvalue problem is then likely to be better conditioned than the original one. On the other hand, we have found that enforcing the condition of bi-orthogonality could affect the FEAST convergence for some other systems e.g. see the case of the QC2534 matrix discussed in [195]. Further studies are clearly needed to better understand the properties of FEAST when applied to ill-conditioned eigenvalue systems.

Reduced computational cost for matrix cases

Non-Hermitian eigenvalue problems come in three flavors: (i) complex general, (ii) real non-symmetric, and (iii) complex symmetric. The major computational task performed by FEAST is the numerical integration, where a set of linear systems must be solved along a complex contour. In the complex general case both Q_{m_0} and \hat{Q}_{m_0} are computed explicitly by solving the $2n_e$ (independent) linear systems defined in (6.9) and (6.10). It is important to note that most modern numerical libraries that include direct methods for solving linear systems supply a "conjugate transpose solve" feature as well (i.e. a linear system $A^H x = f$ can be solved using the factorization of A). Consequently, once the $(z_j B - A)$ matrices are factorized in (6.9), the system solves in (6.10) can be performed without re-factorizing the conjugate transpose of the matrices. Similarly using iterative methods, the conjugate transpose solve could be performed without factorizing twice the preconditioner. If such option is available, the contour integration in the most general case should involve only n_e (independent) factorizations and $2n_e$ (independent) solves with m_0 right hand sides. For the cases (ii) and (iii) above, it is possible to further reduce the workload by taking advantage of some additional matrix properties:

- **Complex symmetric** For the complex symmetric case $(A = A^T \text{ and } B = B^T)$, there exists a relationship between the left and right eigenvectors, which can be expressed as conjugate pairs i.e. $\hat{X} = X^*$. This allows the left subspace \hat{Q} to be expressed in terms of the right Q using the same simple relationship $\hat{Q} = Q^*$. Therefore \hat{Q} (6.10) does not need to be calculated, and only the n_e factorizations and n_e solves in (6.9) are then necessary.
- **Real non-symmetric** In general the treatment of the real non-symmetric case $(A = A^* \text{ and } B = B^*)$ is identical to the complex non-symmetric one. However, there exist savings for specific contours exhibiting symmetry across the real axis (i.e $C = C^*$). For this particular case, each integration node z_j with $j = 1, \ldots, n_e/2$ in the upper half of the complex plane has a conjugate pair z_j^* in the lower half. From the relationships:

$$(z_j B - A)^* = (z_j^* B - A)$$
 and $(z_j B - A)^H = (z_j^* B - A)^T$,

one can show that only the $n_e/2$ factorizations of $(z_j B - A)$ in the upper-half contour, along with $2n_e$ total solves, are needed to obtain both Q_{m_0} and \hat{Q}_{m_0} in (6.9) and (6.10).

Table 6.1 summarizes the number of factorizations and solves effectively needed to perform the full contour integration using a total of n_e nodes for problems with various symmetries. The cost of the Hermitian FEAST algorithm is also provided for reference.

Family of eigenvalue problems	(A,B) properties	#Factorizations	#Solves
Complex general	N/A	n_e	$2n_e$
Complex symmetric	$A = A^T, B = B^T$	n_e	n_e
Complex Hermitian with $\mathcal{C} = \mathcal{C}^*$	$A = A^H, B \text{ hpd}$	$n_e/2$	n_e
Real non-symmetric	N/A	n_e	$2n_e$
Real non-symmetric with $\mathcal{C} = \mathcal{C}^*$	N/A	$n_e/2$	$2n_e$
Real symmetric with $\mathcal{C} = \mathcal{C}^*$	$A = A^T, B \text{ spd}$	$n_e/2$	$n_e/2$

Table 6.1: Summary of the total number of factorizations and solves effectively needed by FEAST to perform the full contour integration using a total of n_e nodes. It is also assumed that the conjugate transpose solve feature is available for the system solver.

Validation and verification of the algorithm

Matrices obtained from the non-Hermitian eigenvalue problem (NEP) collection [26] and the MatrixMarket collection [34], have been used for testing and development. Our test parameters and results for a set of selected system matrices are provided in Table 6.2. The trapezoidal quadrature has been used by FEAST to obtain the results in Tables 2, 3, and 4. A subset of the eigenpairs has been targeted for each system matrix corresponding to the information provided in the NEP collection, if available. Only a few FEAST subspace iterations are needed for most systems to reach convergence.

Matrix	n	m_0	m	λ_{mid}	r	#Iteration
BFW782	782	44	22	(-5300, 300)	10000.0	2
BWM200	200	36	18	(-1200, 0.0)	60.0	2
CDDE5	961	140	70	(4.75, 0.0)	0.25	2
GRCAR	100	38	19	(0.3, 2.0)	0.5	4
QC324	324	72	37	(0.0, 0.0)	0.04	3
RBS480	480	112	56	(0.0, 0.5)	0.5	9
RW136	136	38	19	(1.0, 0.0)	0.5	5
TOLS340	340	16	8	(-60, 300)	30.0	3
TOLS4000	4000	144	72	(-60, 300)	233.0	8

Table 6.2: Non-Hermitian test cases. The contour is chosen as a full circle defined by the center and radius (λ_{mid}, r) using $n_e = 16$ trapezoidal quadrature nodes, and the convergence tolerance for the residual is set at 10^{-12} . The system size n, the subspace size m_0 , the final number of eigenvalues m found within the search contour, and number of FEAST iterations to reach convergence are also listed.

Performance results

Multiple search intervals

As mentioned previously, a major advantage of FEAST are the multiple levels of parallelism naturally contained within the algorithm. Here, we investigate the effect of using two levels of MPI parallelism by considering multiple search contours over the whole spectrum with multiple integration poles by contour. Although FEAST is generally applied to large sparse matrices, we consider only dense matrices in this section. The results aim to highlight the convergence properties and robustness of FEAST that remain independent of the linear-system solver. Load balancing and general scalability trends will remain valid for sparse and banded matrices as well as domain-decomposition methods. We note that a third level of MPI parallelism that would act on the individual linear systems using a distributed memory solver is also possible and considered elsewhere [113].

Multiple contours can be solved independently using the first level of parallelism of FEAST (overall orthogonality is also largely preserved [197, 75]). However, there is a threshold on the number of eigenvalues that can be calculated efficiently using a single FEAST contour. In practice m_0 should represent only a small percentage of the matrix size and it may not be suitable for m_0 to be larger than few thousands because of the $O(m_0^3)$ complexity of the reduced system solve. If enough parallel resources are available, however, the solution for an arbitrarily large number of eigenvalues can be obtained by partitioning the entire search domain into multiple contours. FEAST can then be applied to each in parallel with a reduced value for m_0 . An example of such partitioning is illustrated in Figure 6.6. The test uses the general complex FEAST dense interfaces on a 4000×4000 dense matrix constructed such that all eigenvalues exist within the unit disk. Two sets of contours are considered: First, squares with 4 trapezoidal intervals along each line segment for a total of 16 linear systems to be solved; Next, circles defined by 16 integration nodes. In all cases the size of the search subspace is set at $m_0 = 200$, and the criteria of convergence for the residual at 10^{-12} . At first we consider using only one MPI process per contour, so the 16 linear systems are solved one after another using the LAPACK dense solver. Table 6.3 reports the number of eigenvalues found in each contour, the number of FEAST iterations, and the total simulation times. Two simulation times are given, the fastest has been obtained using a new option offered in FEAST v3.0 that allows to save and reuse the factorization at each iteration (increasing the memory footprint by the number of integration nodes but removing the need to perform this costly step multiple times). Saving the factorization between FEAST iterations produced a $2-3\times$ speed improvement for all contours. As it can be observed from the number of FEAST iterations and the simulation times in Table 6.3, load balancing becomes an issue with some contours taking more than twice the time of the fastest converging contour. Since FEAST runs in parallel, its overall efficiency depends on the slowest converging contour (i.e Square 5 or Circle 3).

Even better performance can be achieved by taking advantage of another level of parallelism for solving the set of independent linear systems. In the general case, a

Contour N ^o	m	#Iterations	Time-1 (s)	Time-2 (s)
Square				
1	84	9	127	49
2	85	7	102	41
3	95	15	207	75
4	83	12	168	62
5	73	19	255	90
6	69	12	165	61
Circle				
1	120	4	62	28
2	129	8	111	45
3	137	11	150	58
4	118	8	112	45
5	109	6	86	37
6	104	4	61	28

Table 6.3: Timing results, number of eigenvalue m and number of iterations obtained using FEAST for each contour in Figure 6.6. We use $m_0 = 200$, $n_e = 16$ and one MPI process per contour. Two total times are reported by contour: Time-1 for FEAST normal use, and Time-2 that does not account for the cost of the multiple matrix factorizations along the FEAST iterations which are saved in memory. We note that the overall parallel FEAST efficiency is limited by the slowest individual performance on a single contour obtained here for either Square 5 or Circle 3.



Figure 6.6: A 4000×4000 dense matrix has been constructed such that all eigenvalues exist within the unit disk. Multiple FEAST contours, employing a trapezoidal quadrature, have been used to calculate a subset of the eigenvalues in parallel.

single factorization and two solves must be performed at each integration node. With a total of n_e factorizations and $2n_e$ solves, the simulation time could then potentially be reduced by a factor n_e or more (since the linear systems do not need to be refactorized at each iteration if n_e is equal to the #MPI processes). Table 6.4 presents scalability results for the 4000 × 4000 dense matrix considered in Figure 6.6 and Table 6.3.

For this dense example, one observes a maximum of $\sim 19 \times$ speed-up for a circular contour and $\sim 23 \times$ for a square contour compared to a single process when using 16 MPI. The super linear scaling is witnessed for both contours and is more pronounced for the square, which required a larger number of FEAST iterations to reach convergence.

Comparison with LAPACK

We now compare non-Hermitian FEAST directly to the LAPACK dense eigensolver. The entire spectrum of a 4000 size non-Hermitian matrix can be computed using the LAPACK in 220 seconds on 12 threads. Interestingly, the results in Tables 6.3 and 6.4 show that the FEAST dense interface targeting only a subset of the spectrum, is capable to perform always better than the LAPACK direct solver on the full spectrum (in particular when saving the factorization or by distributing the linear systems in parallel). Given enough parallelism, the entire spectrum can potentially

Contour N ^o	1	2	3	4	5	6	Speed-up
Square							
1 MPI	127	102	207	168	255	165	1.00
2 MPI	62	51	104	85	128	82	1.9
4 MPI	34	27	55	44	68	44	3.7
8 MPI	18	14	30	24	37	24	6.9
16 MPI	6	5	8	9	11	8	23.1
Circle							
1 MPI	62	111	150	112	86	61	1.00
2 MPI	31	56	76	57	43	31	1.9
4 MPI	16	29	40	30	124	17	3.7
8 MPI	9	16	22	16	13	9	6.8
16 MPI	4	5	8	6	4	3	19.5

Table 6.4: MPI scalability results for the system matrix and contours $(n_e = 16)$ considered in Figure 6.6 and Table 6.3. The first column indicated the cluster of MPI processes being used by each contour to distribute the linear systems. The last column indicates the speed-up performance associated with the slowest contour (Square 5 or Circle 3).

be divided into multiple independent non-overlapping 'square' contours, and the results in Table 6.4 suggests that all the 4000 eigenpairs could optimally be computed in 11 seconds using 16 * p MPI, with p the number of contours (i.e. 20 times faster than LAPACK). We note that a direct comparison with ScaLAPACK [33], the MPI version of LAPACK, would be more appropriate using a fixed number of MPI processors. In this study, however, we focus on the general scalability trend of the first two levels of FEAST parallelism that would remain unchanged using 'unlimited' MPI parallel resources (since FEAST could also be linked with a dense distributed-memory ScaLAPACK linear system solver [113]).

Table 6.3 shows that the time to reach convergence depends on the slowest converging FEAST contour. The previous results have used a FEAST contour consisting of the default value of $n_e = 16$ quadrature nodes. The number of quadrature nodes used within FEAST will affect the convergence rate and can be seen as a tuning parameter. Fewer quadrature nodes will generally increase the number of FEAST iterations to reach convergence. However, it could also reduce the total number of linear system factorizations/solves since fewer linear systems are required per FEAST iteration. Table 6.5 presents FEAST solution times for the slowest converging square contours using 4, 8 and 16 quadrature nodes. Even without saving the factorization, the performance of FEAST surpasses that of LAPACK for the case of 4 or 8 quadrature nodes. The best performance was achieved using $n_e = 8$, which converged in fewer iterations than $n_e = 16$. The seemingly slower convergence rate of $n_e = 16$ can be attributed to the need of additional FEAST iterations for automatically identifying the spurious solutions (i.e. Step 2 of Figure 6.5). Saving the factorization between

iterations results in a performance gain and all three cases become faster than LA-PACK. Even larger improvements can be made by distributing the linear systems as in Table 6.4. Timing results using an optimal number of MPI processes (i.e. $\#MPI = n_e$) for the three contours are also reported in Table 6.5. Fewer quadrature nodes will limit the amount of parallelism that can be applied at this level. Thus, the optimal value of n_e for the single node case may not result in the best performance using MPI.

n_e	Contour	#Iterations	#Solve	Time-1 (s)	Time-2 (s)	Time-MPI (s)
4	3	40	160	142	58	24
8	5	16	128	104	42	10
16	5	19	304	255	90	11

Table 6.5: Timing results using FEAST for the slowest converging square contour using $n_e = 4, 8, 16$ quadrature nodes. The definition of Time-1 and Time-2 is described in Table 6.3. Time-MPI uses the number of MPI processes equal to the number of quadrature nodes for optimal performance. For the single node cases, both $n_e = 4$ and $n_e = 8$ perform better than LAPACK, which took 220 seconds to compute the entire spectrum. All three values of n_e were faster than LAPACK when using MPI to distribute the linear systems (Time-MPI) or when saving the factorization (Time-2) using a single MPI processor.

A more comprehensive study is presented in Figure 6.7, where we consider the slowest converging square contour (i.e. from Figure 6.6) for matrices ranging in size from 1000 to 6000. The number of eigenvalues within each contour will increase with the size of the matrix. The value of m_0 varies proportionally in order to keep it approximately twice the size of the number of eigenvalues within each contour. Three cases are shown to highlight the performance of FEAST for different operational modes. Each graph reports the speedup of FEAST using $n_e = 4, 8$ and 16 quadrature nodes compared to LAPACK. The reported LAPACK times use 12 threads, which was optimal for all the matrices. When using a single MPI process (i.e. the linear systems are solved sequentially), the contours with 4 and 8 quadrature nodes are faster than LAPACK for all matrices, but the 16 node contour is slower for many of the sizes. When saving the factorization and reusing them at each subsequent iteration, however, FEAST is faster than LAPACK in all cases. Finally, using MPI to distribute the linear systems, FEAST is significantly, up to 35 times, faster than LAPACK. In all cases we see larger speedups when increasing the matrix size indicating better scaling for the dense linear system solves than the eigenvalue decomposition.

Comparison with ARPACK

We now compare the FEAST sparse interface directly to the ARPACK package that is also capable of obtaining a subset of the eigenspectrum by computing the m



Figure 6.7: The speedup ratio comparing the slowest converging FEAST contour to LAPACK for different matrix sizes. LAPACK using 12 threads took $\{13, 25, 127, 221, 578, 1025\}$ seconds to compute the entire spectrum for matrices of size $n_{mat} = \{1000, 2000, 3000, 4000, 5000, 6000\}$, respectively. The speedup ratio depends on the performance of LAPACK for the given matrix size, but also the number of FEAST iterations to reach convergence. Since the number of eigenvalues within each contour grows with the matrix size we have chosen values of $m_0 = \{50, 100, 150, 200, 250, 300\}$ to be proportional to n_{mat} . For cases $n_{mat} =$ $\{4000, 5000, 6000\}$ the 8 node FEAST contour needed fewer iterations to reach convergence and had better overall performance than when using 16 quadrature nodes.

eigenvalues closest to a given complex shift λ_{mid} . Analogous to the subspace size m_0 in FEAST, the number of basis vectors to use in the Krylov subspace must be specified. To perform the comparison we have integrated the same PARDISO solver used in FEAST into ARPACK through its reverse-communication-interface. Just like FEAST, ARPACK must solve many complex linear systems of the form (zB - A). However, unlike FEAST, ARPACK has only a single shift at $z = \lambda_{mid}$ and the matrix can then be factorized once and for all at the beginning of the process. The ARPACK RCI then supplies each right-hand-side to the solver one after another and uses the solutions to build a Krylov subspace. A fair comparison can be made by using a circular FEAST contour, picking the same λ_{mid} for each eigensolver and choosing the FEAST contour radius to include exactly m eigenvalues. The subspace size for both FEAST and ARPACK is kept at $m_0 = 2m$ for all examples.

The matrices used in this section to gather performance results have an eigenspectrum similar to what is shown in Figure 6.6. The eigenvalues are randomly distributed in the complex plane over the unit disk. The matrix structure, however, is no longer dense and instead has a block diagonal form:

$$A = \text{diag}\{A_1, A_2, \dots, A_k\}.$$
 (6.21)

The matrix A has a bandwidth determined by the size of each dense block A_i , which are formed in the same way as the matrices in the previous sub-section, but are com-

plex and symmetric instead of general complex. Consequently, the left eigenvectors are conjugate of the right eigenvectors. We note that for the general complex case, the FEAST interfaces would end up being twice slower (i.e. it would require twice the number of solves as reported in Table 6.1). Similarly, it would also be necessary for ARPACK to solve explicitly for the left eigenvectors, which would increase the ARPACK time by a factor of two and require the user to manually bi-orthogonalize the two eigenspaces. Finally, in all our numerical experiments, PARDISO is used as a general sparse solver and we do not take advantage of the special structure of the matrix (6.21).

In our experiments, the performance of FEAST is compared to ARPACK on a single shared memory node with 24 cores in Table 6.6. Six different matrices have been constructed and range in size from $n_{mat} = 25,000$ to $n_{mat} = 100,000$. Each matrix has a bandwidth $n_b = 10$ or $n_b = 40$. The amount of time to calculate 100, 200 and 400 eigenvalues is then recorded. We also vary the number of FEAST quadrature nodes from $n_e = 8$ to $n_e = 32$. For these matrices we see that increasing the number of contour points decreases the number of FEAST iterations to reach convergence and the best time is achieved at 32 quadrature nodes. In our particular example, the factorization time of the FEAST linear systems is negligible. We expect that those comparative results will hold using any other general sparse matrices only if it is possible to store the n_e FEAST factorizations on a single shared memory node. We note that ARPACK is in average two times faster than FEAST for all the recorded cases.

In contrast to FEAST, ARPACK must use the previous solution to generate the next basis vector and this process cannot be parallelized over the linear system solves. More linear systems must then be solved with ARPACK, but since each has only a single right-hand-side the total number of solution vectors is actually much less than FEAST. FEAST increases the total amount of work, but transforms the problem into one that can be broken down into independent blocks. While the current ARPACK implementation works at the level of BLAS-2, FEAST can take better advantage of both BLAS-3 and multi-threading over the cores. Interestingly, we have found out that FEAST can outperform ARPACK on a single node if MPI is used to distribute the linear systems. Since the parallel resources are limited for a single node (i.e. limited to 24 cores in our case), increasing the number of MPI processes will then decrease the number of openMP threads available to the linear system solver. Table 6.7 reports various FEAST timing results for different combinations of MPI processes and openMP threads using $n_e = 24$ contour points. The configuration that uses 4-MPI and 6 threads per MPI, in particular, appears to be as fast as, or faster than, ARPACK in 11 out of 18 examples.

Finally, each quadrature node of FEAST is independent and the associated linear systems can also be solved in parallel. Consequently, if more parallelism were available using multiple physical nodes each MPI process could be assigned a single linear system. Linear scaling should then be expected in agreement with the results of Table 6.4 (i.e. all the FEAST timing results in Table 6.6 would be divided by at least n_e using n_e nodes). This is the optimal configuration for FEAST and represents a level of parallelism not present within ARPACK. Indeed, while multiple shifts could be used

n_b	n _{mat}	8- n_e (s)	16- n_e (s)	24- n_e (s)	$32 - n_e$ (s)	ARPACK (s)
10	25,000	20	14	13	12	3
10	50,000	39	26	23	24	8
10	100,000	87	56	44	43	17
40	25,000	31	18	14	13	7
40	50,000	59	32	27	24	14
40	100,000	136	72	51	46	24
			m = 200 a	nd $m_0 = 4$	00	
n_b	n _{mat}	8- n_e (s)	16- n_e (s)	24- n_e (s)	$32 - n_e$ (s)	ARPACK (s)
10	25,000	59	32	24	22	8
10	50,000	92	55	42	39	29
10	100,000	182	114	87	80	45
40	25,000	82	44	30	26	17
40	50,000	161	81	57	51	39
40	100,000	352	195	134	117	56
			m = 400 at	nd $m_0 = 80$	00	
n_b	n_{mat}	$8-n_e$ (s)	16- n_e (s)	24- n_e (s)	$32 - n_e$ (s)	ARPACK (s)
10	25,000	174	88	62	55	47
10	50,000	328	149	127	97	84
10	100,000	670	341	290	217	159
40	25,000	251	121	98	72	57
40	50,000	477	259	158	168	101
40	100,000	1570	738	489	412	180

m = 100 and $m_0 = 200$

Table 6.6: Timing results for FEAST and ARPACK on a single shared memory node (24 cores). The different matrices are defined by their dimension n_{mat} and the size of the dense sub-blocks n_b . Results are presented for different number of eigenvalues m, which require a larger subspace size m_0 for FEAST and ARPACK. The total time (in seconds) to solve the eigenvalue problem is reported. A circular FEAST contour is used and the four reported times 8- n_e , 16- n_e , 24- n_e , and 32- n_e correspond to 8, 16, 24, and 32 quadrature nodes. The time for ARPACK to solve the problem is reported in the 'ARPACK' column.

in ARPACK along with a reduced subspace size, it would not be possible to define non-overlapping regions for breaking up the spectrum. Beside the impracticality of the implementation, it would lead to degradation of performances in load balancing, in particular.

n_b	n _{mat}	2- <i>mpi</i> (s)	4- <i>mpi</i> (s)	6- <i>mpi</i> (s)	12- <i>mpi</i> (s)	ARPACK (s)
10	25,000	6	6	7	8	3
10	50,000	11	11	14	16	8
10	100,000	22	22	28	31	17
40	25,000	8	7	8	9	7
40	50,000	17	14	16	17	14
40	100,000	33	28	32	34	24
			m = 200 a	nd $m_0 = 40$	00	
n_b	n _{mat}	2-mpi (s)	4-mpi (s)	6-mpi (s)	12- mpi (s)	ARPACK (s)
10	25,000	15	14	18	20	8
10	50,000	26	25	31	35	29
10	100,000	57	57	60	66	45
40	25,000	19	17	19	21	17
40	50,000	36	32	36	38	39
40	100,000	69	61	71	73	56
			m = 400 as	nd $m_0 = 80$	0	
n_b	n_{mat}	2-mpi (s)	4-mpi (s)	6-mpi (s)	12- <i>mpi</i> (s)	ARPACK (s)
10	25,000	41	42	54	65	47
10	50,000	79	79	101	117	184
10	100,000	147	137	173	205	159
40	25,000	62	48	70	68	57
40	50,000	106	79	95	106	101
40	100,000	224	168	166	196	180

m = 100 and $m_0 = 200$

Table 6.7: Timing results for FEAST ($n_e = 24$) using MPI to distribute the linear systems on a single shared memory node (24 cores). Increasing the number of MPI processes reduces the number of threads available to the linear system solver. The fastest times are highlighted in bold. Results for 24 MPI processes are not shown and performed significantly worse.

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