

**Ab-initio Study of Semi-conductor and Metallic
Systems: from Density Functional Theory to
Many Body Perturbation Theory**

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Chapter 1

Introduction

With the development of computing facilities, computational materials science has clearly emerged as an important field of condensed matter physics. In particular, ab-initio methods are nowadays indispensable for a thorough understanding of properties and phenomena of materials at the atomic scale. Among these, Kohn-Sham density functional theory (DFT)[1, 2] in the local density approximation (LDA)[3] or generalized gradient approximation (GGA)[4, 5] has been the main tool used by theoreticians for modelling the structural and electronic properties of materials. The basic idea of DFT is to replace the interacting many-electron problem with an effective single-particle problems. Therefore the computational costs were relatively low as compared to the traditional methods which were based on the complicated many electrons wave functions, such as Hartree-Fock(HF)[6, 7] theory and its descendants. DFT allows to calculate ground state properties of large systems: remarkable results have been achieved for ground state properties of a huge number of systems ranging from atoms and molecules to solids and surfaces.

A perfect crystal means that every atom of the same type locates in the correct position. However, most crystalline materials are not perfect: the regular pattern of atomic arrangement is interrupted by crystal defects. In fact, using the term "defect" is sort of a misnomer since these features are commonly intentionally used to manipulate the electronic properties of a material. One of the most important

defect type is the native point defect. A common application of these point defects is doping of semiconductor crystals, which controls many aspects of semiconductor behavior. For instance, group IV(e.g. Si, Ge, Sn) atoms taking places of III(e.g. Ga) atoms or VI(e.g. S, Se, and Te) atoms substituting V atoms(e.g. As) in III-V semiconductors will introduce extra valence electrons to the semiconductors, and the excess electrons increase the electron carrier concentration, which is important for the electrical properties of extrinsic semiconductors. On the experiment side, in order to obtain direct information about the local environment around the defects, an approach which has been exploited in the last decade is to cut the material along a cleavage plane, so that the defects originally inside the crystal are exposed on a surface, and these defects exposed on a surface or near the surface can be studied with the scanning tunneling microscopy (STM).

The first part of this work deals with Si doped GaAs. For the surface study of GaAs, the most interesting plane for GaAs is the (110) surface, which allows us to obtain usually large atomically flat surfaces without atomic reconstruction. Although Si doped GaAs systems have been intensively investigated theoretically and experimentally during the past several decades, a proper description of the properties of defect in GaAs systems such as the stabilities of charged defects and the detailed structure around a defect is still an area of active research. Within theory, these properties of defects can be analyzed from the structural optimization based on the DFT.

On the other hand, it turns out that the interaction between particles plays a very important role in physical properties. This indicates that the success of DFT is also accompanied by a number of serious problems. For instance, DFT underestimates the band gaps of semiconductors and insulators by some tens of percent due to the simplified treatment of electron-electron interaction. Apart from the too small gaps, the band dispersions are often reasonable, and hence, a simple rigid shift can be used to correct the band gaps of semiconductors and insulators. In any case, the DFT errors in s-p metals are probably less significant than the band gap errors in semiconductors and insulators. Therefore, the experimental band-structures are often compared with the results of calculations performed

within DFT. A much more serious problem of DFT arises when it is applied to calculate the electronic structures of some noble metals. An example is given in Ref.[8], where large discrepancies between measured Cu band structure and DFT results have been found. In this case, a simple rigid shift will lead to a ridiculous result since the band structure of Cu depends on considered band and k point. Moreover, DFT is mathematically represented by an hermitian hamiltonian, so that the corresponding single particle states have infinite lifetime.

In order to overcome the above difficulties encountered by DFT, a number of attempts have been made for improving the DFT. An exact theory for a system of interacting electrons is based on solving its many body Schrödinger equation. Unfortunately, the many body Schrödinger equation can not be solved exactly for most cases due to the nature of the electrons. The better description for the interacting electrons is to consider them as quasi-particles. The general notion was first introduced by L.D. Landau[9]. Landau's basic idea was that in a complicated system of strongly interacting particles, it may be still possible to describe the properties of the system in terms of the weakly interacting particles. A many body perturbation theory(MBPT)[10] treatment can deal with a weakly interacting system of particles, beginning with the non-interacting particles as the unperturbed state. The MBPT leads to the successful GW approximation(GWA), developed systematically by Hedin in 1965[11]. Within the MBPT, band energies can be obtained as the poles of the one-particle Green's function G , which involve the electron self-energy Σ (GWA). The self-energy can be formally expanded in terms of the dynamically screened Coulomb interaction W , the lowest term being iGW . Due to the high complexity and large computational requirements of many-body calculations, for a long time the applications of the GWA were restricted to the electron gas. With the rapid progress in computer power, applications to realistic materials eventually became possible about two decades ago. Numerous applications to semiconductors and insulators reveal that in most cases the GWA can yield good band structures. The success of the GWA in *sp* materials has prompted further applications to some complex metals with localized *d* electrons. For instance, full quasi-particle calculations within GW approximation for Ni[12]

has yielded a good description of band structure, except for the 6 eV satellite. Another successful calculations for Cu and Ag have been reported in Ref.[13] and Ref.[14], which are based on the plane wave basis sets. In order to describe properly the metals with localized d orbitals, one must also consider the contribution of d electrons to the valence band structure. This indicates that d electrons can not be frozen into the core part, but must be explicitly included into the valence. Apart from the d electrons, the semi-core electrons such as $3s$ and $3p$ for Cu and $4s$, $4p$ for Ag should be included into the valence[13, 14], since there is a large spatial overlap with the d electrons for these noble metals. All these will result in a large number of valence electrons in the calculation, which means large computational requirements for these metals, especially for calculations using plane wave basis sets. In this thesis, I will present a full quasi-particle bandstructure calculations of Cu and Ag using more efficient localized Gaussian basis sets instead of plane wave basis sets. In addition, it is possible to carry out a surface calculation for the noble metals within the GW approximation by using a localized Gaussian orbital basis.

Another interesting study for metals is the quasi-particle excitations, which play an important role in a rich variety of physical and chemical phenomena[15] such as energy transfer in photochemical reactions, desorption and oxidation of molecules at surfaces, spin transport within bulk metals, across interfaces, and at surfaces. One of the crucial properties of quasi-particle excitation is their lifetimes which determine the duration of these excitations. On the experiment side, linewidths of bulk excited electron states in metals have also been measured, with the use of photoelectron spectroscopy[16]. A more advanced tool for the study of both electron and hole lifetimes in the time domain became available by the advent of the time-resolved two-photon photoemission(TR-2PPE)[17, 18]. Within theoretical framework, the early investigations of quasi-particle lifetimes are based on the free-electron gas(FEG) model of Fermi-liquids[19]. Several other free-electron calculations of electron-electron scattering rates have also been carried out, for electron(hole) energies that deviate from the Fermi level, within the random-phase approximation (RPA)[20, 21] and with inclusion of exchange and correlation

effects[22, 23]. Nevertheless, detailed TR-TPPE experiments have reported large deviations of measured hot-electron lifetimes from those predicted within the FEG model. First-principles calculations of lifetimes in a variety of metals have been carried out only very recently [24, 25, 26, 27], by using the GW approximation of many-body theory. Here, I present a first-principles calculation of lifetimes for noble metals within the GW approximation based on the quasi-particle energies. The work based on the MBPT deals with bulk Cu, bulk Ag and Cu(100) using the GW approximation based on the localized Gaussian orbital basis sets.

In Chapter 2, the concept and progress of DFT are introduced. In chapter 3, an accurate, first-principles study of the electronic structure of Si doped GaAs within DFT is presented. The plane wave basis used to expand the Bloch wavefunction requires the use of pseudopotentials. We systematically studied the DX center in bulk GaAs and in GaAs(110), as well as the relative stability of Si_{Ga} defect with different charge states in different layers of GaAs(110). We report that the DX center is a metastable state in bulk GaAs and completely unstable in the top few layers of GaAs(110). Moreover, we found that the Si_{Ga}^- defect on the GaAs(110) surface is more stable compared to Si_{Ga}^- defects in the deep layers (below the surface), while the Si_{Ga}^+ defect on the GaAs(110) surface is more unstable, as compared to Si_{Ga}^+ defects in deep layers. The most interesting finding is that the extra charge of Si_{Ga}^- defect is mainly concentrated on the Si atom when the defect is exposed on the GaAs(110) surface. In addition, we studied the STM images of clean GaAs(110) and charged Si:GaAs(110) by employing the Tersoff-Hamann approximation[28]. The calculated STM images are in good agreement with the experimental results. We show that at the positive bias voltage the positively charged Si atom presents a bright feature while the negatively charged Si atom shows a dark feature.

In Chapter 4, we introduce the MBPT. In chapter 5, We present the calculations of quasiparticle bandstructures and lifetimes for noble metals Cu and Ag within the GW approximation. For Cu, both the calculated positions of the d bands and the width of the d bands is within 0.1 eV compared to the experimental results. For Ag, partial core correction should be included in the pseudopotential to get

reliable positions of the d bands. The calculated lifetime agree with the experiment in the energy region away from the Fermi level, but deviates from the experimental results near the Fermi level where short range interactions which GWA fails to describe play an important role. For a better description of the lifetime near the Fermi level, higher terms beyond the GW approximation in the many body perturbation theory need to be considered. In addition, the image potential state lifetimes in Cu(100) have been calculated using GW approximation based on the localized Gaussian orbital basis sets, and the calculated $n=1, 2$ image potential state lifetimes are in good agreement with experimental results.

Chapter 2

Density functional theory

2.1 Introduction

Quantum mechanics (QM) is the correct mathematical description of the behavior of electrons. In theory, QM can predict any property of an individual atom or molecule exactly. In practice, the QM equations have only been solved exactly for few electron systems. A number of methods ranging from semi-empirical to ab-initio(QM) approaches have been developed for approximating the solution for many electron systems. The former schemes usually need some parameters which are taken from or adjusted to experiments. Although semi-empirical calculations are much faster than their ab initio counterparts, if some parameters for semi-empirical simulations are not available, or some phenomena of a system are not yet known, one must rely only on ab initio calculations. The term ab initio is Latin for "from the beginning." This name is given to computations that are derived directly from theoretical principles with no inclusion of experimental data. The simplest type of ab initio electronic structure calculation is the Hartree-Fock (HF) scheme, which is based on a wave function in a form of one Slater determinant. Though the results of such calculations using HF scheme are reliable, the major disadvantage is that they are computationally intensive. An alternative scheme is DFT, which is based on the electron density rather than on the wave functions, and

commonly used to calculate the electronic structure of complex systems containing many atoms such as large molecules or solids. In the following sections I will mainly introduce DFT, which is based on the ab initio method. DFT has been developed more recently than other ab initio methods such as Hartree Fock theory and its descendants.

2.2 Schrödinger equation

An exact theory for a system of ions and interacting electrons is inherently quantum mechanical, which is based on solving a many-body Schrödinger equation of the form

$$\hat{H}_k \Psi_k(\{\mathbf{R}_I; \mathbf{r}_i\}) = E_k \Psi_k(\{\mathbf{R}_I; \mathbf{r}_i\}) \quad (2.1)$$

where \hat{H} is the hamiltonian of the system. and consists of the following terms

$$\begin{aligned} \hat{H} = & - \sum_{I=1} \frac{\hbar^2}{2M_I} \nabla_{\mathbf{R}_I}^2 - \sum_{i=1} \frac{\hbar^2}{2m_e} \nabla_{\mathbf{r}_i}^2 - \sum_{Ii} \frac{Z_I e^2}{|\mathbf{R}_I - \mathbf{r}_i|} + \frac{1}{2} \sum_{ij(j \neq i)} \frac{e^2}{|\mathbf{r}_i - \mathbf{r}_j|} \\ & + \frac{1}{2} \sum_{IJ(J \neq I)} \frac{Z_I Z_J e^2}{|\mathbf{R}_I - \mathbf{R}_J|} \end{aligned} \quad (2.2)$$

In the above equations: $\Psi_k(\{\mathbf{R}_I; \mathbf{r}_i\})$ is the many body wavefunction that describes the state of the system; E is the energy of the system; \hbar is Planck's constant divided by 2π ; M_I is the mass of ion I; m_e is the mass of the electron; Z_I is the valence charge of this ion.

In the Eqs. (2.2): the first and second terms are the kinetic energies of ions and electrons, respectively; Typically, the ions can be considered as moving slowly in space and the electrons responding instantaneously to any ionic motion due to the huge difference of mass between ions and electrons(three to five orders of magnitude), so the quantum-mechanical term for the kinetic energy of the ions can be omitted, and we take their kinetic energy into account as a classical contribution. This corresponds to the Born-Oppenheimer approximation. The

third term is the external potential experienced by electrons due to the presence of the ions; The fourth term is the Coulomb potential between electrons; The last term is the potential between ions(Madelung energy of the ions), which as far as the electron degrees of freedom are concerned is simply a constant. Finally, the hamiltonian of the system can be reduced to the following terms

$$\hat{H} = - \sum_{i=1} \frac{\hbar^2}{2m_e} \nabla_{\mathbf{r}_i}^2 - \sum_{Ii} \frac{Z_I e^2}{|\mathbf{R}_I - \mathbf{r}_i|} + \frac{1}{2} \sum_{ij(j \neq i)} \frac{e^2}{|\mathbf{r}_i - \mathbf{r}_j|} \quad (2.3)$$

Even with this simplification, however, solving for $\Psi_k(\{\mathbf{r}_i\})$ is an extremely difficult task, because of the nature of the electrons. If two electrons of the same spin interchange positions $\Psi_k(\{\mathbf{r}_i\})$ must change sign; this is known as the "exchange" property, and is a manifestation of the Pauli exclusion principle. Moreover, each electron is affected by the motion of every other electron in the system; this is known as the "correlation" property. Accurate results can be obtained for the energy levels and wave functions of two-electron atom by performing variational calculations. But this method becomes increasingly tedious when the number of the electrons increases in the atomic system. It is possible to produce a simpler, approximate picture, in which we describe the system as a collection of classical ions and essentially single quantum mechanical particles that reproduce the behavior of the electrons: this is the single-particle picture.

2.3 Thomas-Fermi model

In order to deal with many-electron systems, the Thomas-Fermi model was developed by Thomas and Fermi [29, 30] in 1927 after the introduction of the Schrödinger equation. This is the original DFT of quantum systems. In this model, a statistical model was employed to approximate the distribution of electrons in an atom. The basic idea in this approach is that electrons are distributed uniformly in phase space, and have no interacting with each other. Finally, the

kinetic energy was expressed as a functional of the electron density

$$T_{TF}[\rho] = C_F \int \rho^{\frac{5}{3}}(\mathbf{r}) d\mathbf{r} \quad (2.4)$$

where $C_F = \frac{3}{10}(3\pi^2)^{\frac{2}{3}}$, $C_F \cdot \rho^{\frac{5}{3}}$ is the kinetic energy density of the non-interacting homogeneous electron gas.

Combining this kinetic energy term with the classical expressions for the nucleus-electron and electron-electron interactions, which can both also be represented in terms of the electron density, the total energy of system can be expressed as

$$E_{TF}[\rho] = \frac{3}{10}(3\pi^2)^{\frac{2}{3}} \int \rho^{\frac{5}{3}}(\mathbf{r}) d\mathbf{r} - Z \int V(\mathbf{r})\rho(\mathbf{r}) d\mathbf{r} + \frac{1}{2} \int \int \frac{\rho(\mathbf{r}_1)\rho(\mathbf{r}_2)}{|\mathbf{r}_1 - \mathbf{r}_2|} d\mathbf{r}_1 d\mathbf{r}_2 \quad (2.5)$$

where Z is nuclear charge. Although the TF model represents a drastic approximation of the N -electron problem, it is very useful for the description of the general trends in the properties of atom. However, using the approximation for realistic systems more complex than an isolated atom yield poor quantitative predictions[31]. Nevertheless, introducing the one particle electron density to many-body systems was an important step for the calculations of many-electron systems, and the more sophisticated and accurate Hohenberg-Kohn theorems[32] were developed based on this idea.

2.4 Hohenberg-Kohn theorems

The name of DFT was called formally after the publication of two original papers of Hohenberg, Kohn and Sham[1, 32], referred to as the Hohenberg-Kohn-Sham theorem. Similar to Hartree-Fock theory[6, 7], DFT introduces three approximations: Born-Oppenheimer approximation, non-relativistic approximation and single-particle approximation. The largest error for the Hartree-Fock method is the single-particle approximation, and this can be reduced by the DFT. This theory has a tremendous impact on realistic calculations of the properties of molecules

and solids, and its applications to different problems continue to expand. A measure of its importance and success is that its main developer, W. Kohn (a theoretical physicist) shared the 1998 Nobel prize for Chemistry with J.A. Pople (a computational chemist). We will review here the essential ideas behind DFT.

Let us consider a system of N interacting electrons in a non-degenerate ground state associated with an external potential $V(\mathbf{r})$.

Lemma 1. *The ground state density, $\rho(\mathbf{r})$, uniquely determines the potential $V(\mathbf{r})$, within an additive constant.*

In order to prove this, assume that two different external potentials, $V(\mathbf{r})$ and $V'(\mathbf{r})$, give rise to the same charge density $\rho(\mathbf{r})$. I will show that this postulate is wrong.

The Hamiltonian of a system is given by

$$\hat{H} = \hat{T} + \hat{V} + \hat{W} \quad (2.6)$$

where the \hat{T} is the kinetic energy operator, \hat{W} represents the electron-electron interaction operator and \hat{V} denotes the external potential operator.

For a system with the kinetic energy T and the electron-electron interaction W , we assume that $V(\mathbf{r})$ and $V'(\mathbf{r})$ do not differ merely by a constant. Let E and Ψ be the total energy and wavefunction and E' and Ψ' be the total energy and wavefunction for the system with Hamiltonian H and H' , respectively.

$$\hat{H} = \hat{T} + \hat{V} + \hat{W} \quad (2.7)$$

and

$$\hat{H}' = \hat{T} + \hat{V}' + \hat{W} \quad (2.8)$$

Using the variational principle,

$$\begin{aligned}
 E < \langle \Psi' | \hat{H} | \Psi' \rangle &= \langle \Psi' | \hat{H}' + \hat{H} - \hat{H}' | \Psi' \rangle \\
 &= \langle \Psi' | \hat{H}' | \Psi' \rangle + \langle \Psi' | \hat{H} - \hat{H}' | \Psi' \rangle \\
 &= E' + \int \rho(\mathbf{r}) [V(\mathbf{r}) - V'(\mathbf{r})] d\mathbf{r}
 \end{aligned} \tag{2.9}$$

Similarly we can prove

$$E' < E - \int \rho(\mathbf{r}) [V(\mathbf{r}) - V'(\mathbf{r})] d\mathbf{r} \tag{2.10}$$

adding above two equations, we obtain

$$(E + E') < (E + E') + \int \rho(\mathbf{r}) [V(\mathbf{r}) - V'(\mathbf{r})] d\mathbf{r} - \int \rho(\mathbf{r}) [V(\mathbf{r}) - V'(\mathbf{r})] d\mathbf{r} \tag{2.11}$$

The sum of last two terms on the right hand-side is zero. Finally, we obtain the following relation

$$(E + E') < (E + E') \tag{2.12}$$

This is a contradiction, therefore we conclude that the assumption about the densities being the same is not correct. This proves that the ground state density uniquely determines the external potential $V(\mathbf{r})$, within an additive constant. So we conclude that the total energy of the system is a functional of the density, and is given by

$$E[\rho(\mathbf{r})] = \langle \Psi | \hat{H} | \Psi \rangle = F[\rho(\mathbf{r})] + \int V(\mathbf{r}) \rho(\mathbf{r}) d\mathbf{r} \tag{2.13}$$

where $F[\rho(\mathbf{r})]$ represents $\langle \Psi | \hat{T} + \hat{W} | \Psi \rangle$.

The energy functional can be expressed as

$$E[\rho] = F[\rho] + V[\rho] = T[\rho] + V[\rho] + W[\rho] \tag{2.14}$$

Therefore, the first theorem can be summarized by saying that the energy is a functional of the density. The second theorem establishes a variational principle:

Lemma 2. *The ground state energy can be obtained variationally: the density that minimises the total energy is the exact ground state density.*

The proof of the second theorem is also simple. According to theorem 1, a system with ground state density $\rho_0(\mathbf{r})$ determines uniquely its own external potential $V(\mathbf{r})$ and ground state wave function Ψ_0

$$E[\rho_0(\mathbf{r})] = \langle \Psi_0 | \hat{H} | \Psi_0 \rangle = F[\rho_0(\mathbf{r})] + \int V(\mathbf{r})\rho_0(\mathbf{r})d\mathbf{r} \quad (2.15)$$

Now consider any other density $\rho'(\mathbf{r})$, which necessarily corresponds to a different wave function Ψ' . This would lead to

$$E[\rho'(\mathbf{r})] = \langle \Psi' | \hat{H} | \Psi' \rangle = F[\rho'(\mathbf{r})] + \int V(\mathbf{r})\rho'(\mathbf{r})d\mathbf{r} > \langle \Psi_0 | \hat{H} | \Psi_0 \rangle = E[\rho_0(\mathbf{r})] \quad (2.16)$$

Thus the energy given by the equation 2.15 in terms of the Hohenberg-Kohn functional evaluated for the correct ground state density $\rho_0(\mathbf{r})$ is indeed lower than the value of this expression for any other density $\rho'(\mathbf{r})$.

It follows that if the functional $F[\rho(\mathbf{r})]$ was known, then by minimizing the total energy of the system, with respect to variations in the density function $\rho_0(\mathbf{r})$, one would find the exact ground state density and energy. Note that the functional only determines the ground state properties; it does not provide any guidance concerning excited states.

2.5 Kohn-Sham equations

The Hohenberg-Kohn theorem offers no practical guide to the explicit construction of the $F[\rho(\mathbf{r})]$ universal functional. For this purpose one still has to face the full intricacies of the many-body problem. Although there are some energy functionals for Coulomb systems derived with the theory of the homogeneous electron gas or in other more elaborated approaches, the situation cannot be considered satisfactory. Only with the approach introduced by Kohn and Sham[1] has been able

to calculate (not only) ground state properties of many-particle Coulomb systems with great accuracy.

In the following discussion we will define the density $\rho(\mathbf{r})$ and the one-particle and two-particle density matrices, denoted by $\gamma(\mathbf{r}, \mathbf{r}')$, $\Gamma(\mathbf{r}, \mathbf{r}'|\mathbf{r}, \mathbf{r}')$, respectively, as expressed through the many-body wavefunction:

$$\rho(\mathbf{r}) = N \int \Psi^*(\mathbf{r}, \mathbf{r}_2, \dots, \mathbf{r}_N) \Psi(\mathbf{r}, \mathbf{r}_2, \dots, \mathbf{r}_N) d\mathbf{r}_2 \dots d\mathbf{r}_N \quad (2.17)$$

$$\gamma(\mathbf{r}, \mathbf{r}') = N \int \Psi^*(\mathbf{r}, \mathbf{r}_2, \dots, \mathbf{r}_N) \Psi(\mathbf{r}', \mathbf{r}_2, \dots, \mathbf{r}_N) d\mathbf{r}_2 \dots d\mathbf{r}_N \quad (2.18)$$

$$\Gamma(\mathbf{r}, \mathbf{r}'|\mathbf{r}, \mathbf{r}') = \frac{N(N-1)}{2} \int \Psi^*(\mathbf{r}, \mathbf{r}', \mathbf{r}_3, \dots, \mathbf{r}_N) \Psi(\mathbf{r}, \mathbf{r}', \mathbf{r}_3, \dots, \mathbf{r}_N) d\mathbf{r}_3 \dots d\mathbf{r}_N \quad (2.19)$$

From the form of Eqs.(2.14) the energy functional contains three terms: the kinetic energy $T[\rho]$, the external potential $V[\rho]$ and the electron-electron interaction $W[\rho]$. The kinetic and electron-electron functionals are unknown. Using the expressions for the one-particle and two-particle density matrices, the explicit expression for $E[\rho]$ can be written as

$$\begin{aligned} E[\rho(\mathbf{r})] &= \langle \Psi | \hat{H} | \Psi \rangle = -\frac{\hbar^2}{2m} \int \nabla^2 \gamma(\mathbf{r}, \mathbf{r}')|_{\mathbf{r}'=\mathbf{r}} d\mathbf{r} \\ &+ \int \int \frac{e^2}{|\mathbf{r} - \mathbf{r}'|} \Gamma(\mathbf{r}, \mathbf{r}'|\mathbf{r}, \mathbf{r}') d\mathbf{r} d\mathbf{r}' + \int V(\mathbf{r}) \gamma(\mathbf{r}, \mathbf{r}) d\mathbf{r} \end{aligned} \quad (2.20)$$

Kohn and Sham[1] proposed a good approximation for reducing these expressions to a set of single-particle equations. They introduced a fictitious system of N noninteracting electrons to be described by the single-particle orbitals ϕ_i that appear in the Slater determinant. In this system the one-particle and two-particle density are known exactly from the orbitals:

$$\rho(\mathbf{r}) = \sum_i |\phi_i(\mathbf{r})|^2 \quad (2.21)$$

$$\gamma(\mathbf{r}, \mathbf{r}') = \sum_i \phi_i^*(\mathbf{r}) \phi_i(\mathbf{r}') \quad (2.22)$$

$$\Gamma(\mathbf{r}, \mathbf{r}' | \mathbf{r}, \mathbf{r}') = \frac{1}{2} [\rho(\mathbf{r})\rho(\mathbf{r}') - |\gamma(\mathbf{r}, \mathbf{r}')|^2] \quad (2.23)$$

With the help of above equations, the energy functionals of Eqs. (2.20) can take the following form

$$E[\rho(\mathbf{r})] = T^S[\rho(\mathbf{r})] + \frac{e^2}{2} \iint \frac{\rho(\mathbf{r})\rho(\mathbf{r}')}{|\mathbf{r} - \mathbf{r}'|} d\mathbf{r} d\mathbf{r}' + \int V(\mathbf{r})\rho(\mathbf{r}) + E^{XC}[\rho(\mathbf{r})] \quad (2.24)$$

In this expression, the first term represents the kinetic energy in the Slater determinant (i.e. this is not the true kinetic energy but is that of a system of non-interacting electrons, and hence the superscript S). Since the fictitious particles are non-interacting, the kinetic energy can be known exactly, and take the following form

$$T^S[\rho(\mathbf{r})] = \sum_i \langle \phi_i | -\frac{\hbar^2}{2m} \nabla_r^2 | \phi_i \rangle \quad (2.25)$$

The second term $V_H[\rho(\mathbf{r})]$ in Eqs. (2.24) is the bare Coulomb interaction (this term is separated out from the electron-electron interaction term $V_{ee}[\rho(\mathbf{r})]$ in Eqs. (2.20)); The last term in Eqs. (2.24) denotes the exchange-correlation term, which is simply the sum of the error made in using a non-interacting kinetic energy and the error made in treating the electron-electron interaction classically, and is expressed as

$$E^{XC}[\rho(\mathbf{r})] = (T[\rho(\mathbf{r})] - T^S[\rho(\mathbf{r})]) + (V_{ee}[\rho(\mathbf{r})] - V_H[\rho(\mathbf{r})]) \quad (2.26)$$

With the restriction condition $\int \rho(\mathbf{r}) d\mathbf{r} = N$, applying the variational principle

$$\frac{\delta(E - \epsilon_i \int \rho(\mathbf{r}) d\mathbf{r})}{\delta \phi_i} = \frac{\delta(E - \epsilon_i \int \rho(\mathbf{r}) d\mathbf{r})}{\delta \rho} \frac{\delta \rho}{\delta \phi_i} = 0 \quad (2.27)$$

to Eqs. (2.24), we arrive at the following single-particle equation through a variational argument

$$[-\frac{\hbar^2}{2m} \nabla_r^2 + V(\mathbf{r}) + e^2 \int \frac{\rho(\mathbf{r}')}{|\mathbf{r} - \mathbf{r}'|} d\mathbf{r}' + V_{XC}(\rho)] \phi_i = \epsilon_i \phi_i \quad (2.28)$$

where ϵ_i is the Lagrange multiplier, and a local multiplicative potential is introduced, which is the functional derivative of the exchange correlation energy with

respect to the density:

$$V_{XC}[\rho] = \frac{\delta E^{XC}[\rho(\mathbf{r})]}{\delta \rho(\mathbf{r})} \quad (2.29)$$

The single particle Eqs. (2.28) is the well known Kohn-Sham equation with the resulting density $\rho(\mathbf{r})$ and total energy E given by Eqs. (2.21) and Eqs. (2.24). The single particle orbitals ϕ_i that are their solutions are called Kohn-Sham orbitals. Kohn-Sham equation describes the behavior of non-interacting "electrons" in an effective local potential, which can be expressed as

$$V^{eff}[\mathbf{r}, \rho(\mathbf{r})] = V(\mathbf{r}) + e^2 \int \frac{\rho(\mathbf{r}')}{|\mathbf{r} - \mathbf{r}'|} d\mathbf{r}' + V_{XC}(\rho) \quad (2.30)$$

The effective potential is a function of the density, and hence depends on all the single-particle states. KS results from the Hohenberg-Kohn theorems that the ground state density uniquely determines the potential at the minimum (except for a trivial constant), so that there is a unique KS potential $V_{eff}(\mathbf{r})|_{min} = V_{KS}(\mathbf{r})$ associated with any given interacting electron system. We will need to solve these equations by iteration until we reach self-consistency, but this is not a significant problem. For the iteration, an initial charge density is needed. To obtain the charge density, an initial "guess" to the Kohn-Sham orbitals is needed. This initial guess can be obtained from a set of basis functions whereby the coefficients of expansion of the basis functions can be optimized. A more pressing issue is the exact form of $E^{XC}[\rho(\mathbf{r})]$. In the following section I will discuss the exchange-correlation functionals.

2.6 Local density approximation

In principle, the solution of the Kohn and Sham Eqs. (2.28) with the exact exchange-correlation potential would give a set of fictitious single particle eigenstates whose density of states equals that of the fully interacting system. Unfortunately, the exact exchange-correlation potential is not yet known. In practice, it is necessary to make approximations for this term. The generation of approximations for $E_{XC}[\rho(\mathbf{r})]$ has lead to a large and still rapidly expanding field of research. There

are now many different flavours of approximation for the exchange-correlation potential. One of the most widely used is the Local Density Approximation (LDA). Many approaches can yield local approximations to the XC energy. Overwhelming, however, successful local approximations are those that have been derived from the homogeneous electron gas (HEG) model. The principle of this approximation is to calculate the exchange and correlation energies per particle $\epsilon_{XC}[\rho(\mathbf{r})]$, of the homogeneous electron gas as a function of the density:

$$E_{XC}[\rho(\mathbf{r})] = \int \epsilon_{XC}[\rho(\mathbf{r})] \rho(\mathbf{r}) d\mathbf{r} \quad (2.31)$$

where $\rho(\mathbf{r})$ is the electronic density. This approximation is based on the assumption that the system locally appears as an homogeneous electron gas. The exchange-correlation energies per particle $\epsilon_{XC}[\rho(\mathbf{r})]$ can be separated into exchange and correlation contributions:

$$\epsilon_{XC}[\rho(\mathbf{r})] = \epsilon_X[\rho(\mathbf{r})] + \epsilon_C[\rho(\mathbf{r})] \quad (2.32)$$

While $\epsilon_X[\rho(\mathbf{r})]$ is an analytic function of $\rho(\mathbf{r})$ [33]

$$\epsilon_X[\rho(\mathbf{r})] = -\frac{3}{4}e^2 \left(\frac{3}{\pi}\rho(\mathbf{r})\right)^{\frac{1}{3}} \quad (2.33)$$

The functional form for the correlation energy density $\epsilon_C[\rho(\mathbf{r})]$ is unknown and has been simulated for the homogeneous electron gas in numerical quantum Monte Carlo calculations which yield essentially exact results (Ceperley and Alder, 1980) [3]. Various approaches, using different analytic forms for $\epsilon_C[\rho(\mathbf{r})]$, have generated several LDA's for the correlation functional, including: Vosko-Wilk-Nusair (VWN) [34], Perdew-Zunger (PZ81) [35], Cole-Perdew (CP) [36] and Perdew-Wang (PW92) [37]. All of these yield similar results in practice and are collectively referred to as LDA functionals.

The LDA has proven to be a remarkably fruitful approximation. Properties such as structure, vibrational frequencies, elastic moduli and phase stability (of similar structures) are described reliably for many systems. For this reason, the

LDA has been employed to describe the structure of GaAs(110) systems in the next chapter.

The common feature in this approach is that $E_{XC}[\rho(\mathbf{r})]$ depends on $\rho(\mathbf{r})$ in a local fashion, that is, $\rho(\mathbf{r})$ needs to be evaluated at one point in space at a time. For this reason they are referred to as the Local Density Approximation to Density Functional Theory. This is actually a severe restriction, because even at the exchange level, the functional should be non-local, that is, it should depend on \mathbf{r} and \mathbf{r}' simultaneously. It is a much more difficult task to develop non-local exchange-correlation functionals. More recently, a concentrated effort has been directed toward producing expressions for $E_{XC}[\rho(\mathbf{r})]$ that depend not only on the density $\rho(\mathbf{r})$, but also on its gradients[4, 5]. These expansions tend to work better for some cases, but still represent a local approximation to the exchange-correlation functional. More advanced method for the evaluation of the exchange-correlation part is based on the many body perturbation theory[10], which will be introduced in the third chapter.

2.7 Pseudopotential

Electrons in matter can be broadly categorised into two types: core electrons, which are strongly localized in the closed inner atomic shells, and valence electrons, which also extend outside the core. If core electrons are included completely in the calculation, a large number of basis functions(especially if a plane wave basis set is used) would be required due to the oscillations in the core regions which maintain orthogonality between valence and core electrons. As a result, all electron calculations demand a huge computational expense that is simply not practical. On the other hand, by realising that the electronic structure of the core-electrons remains largely unchanged in different chemical environments, and is also of minimal interest generally, the problems relating to the core-electrons can be overcome by use of the pseudopotential approximation. In this approach the ionic potential $V_{ion}(r)$ in the core region is replaced by a weaker pseudopotential $v_{ion}^{PS}(r)$. The

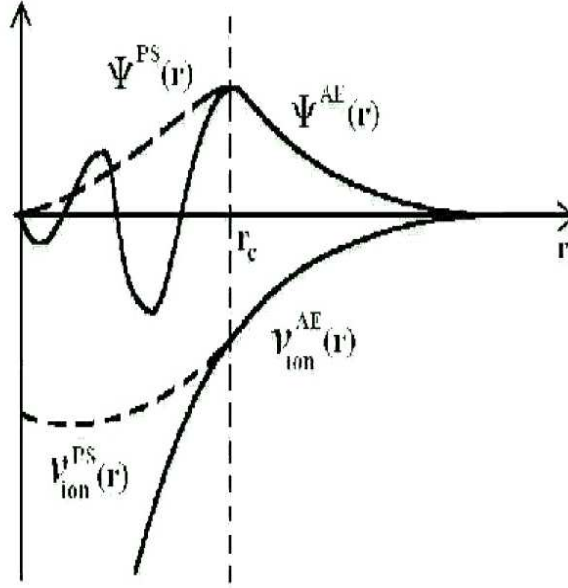


FIGURE 2.1: Schematic illustration of the pseudopotential concept. The solid lines show the all-electron wavefunction, $\Psi^{AE}(r)$ and ionic potential, $v_{ion}^{AE}(r)$, while the dashed lines show the corresponding pseudo-wavefunction, $\Psi^{PS}(r)$, given by the pseudopotential, $v_{ion}^{PS}(r)$. All quantities are shown as a function of distance, r , from the atomic nucleus. The cutoff radius r_c marks the point beyond which the all-electron and pseudo-quantities become identical.

corresponding set of pseudo-wavefunctions $\Psi^{PS}(r)$ and the all-electron wave functions $\Psi^{AE}(r)$ are identical outside a chosen cutoff radius r_c and so exhibit the same scattering properties, but $\Psi^{PS}(r)$ does not possess the nodal structure that cause the oscillations inside r_c . A schematic illustration of the pseudopotential concept is shown in Fig. 2.1. The choice of the r_c is very important. In general, the larger r_c will result in a softer pseudopotential (less cpu time in practical calculation), but also the less transferable. There is no well-defined answer for how to choose a r_c . In practice, r_c should be larger than the outermost node (if any) of the wavefunction for any given angular momentum. Usually there is one angular momentum that is harder than the others (in transition metals, the d state, in second-row elements N, O, F, the p state), and one should concentrate on this one and push outwards its r_c as much as possible. In addition, one should try to have not too different r_c for different angular momenta.

The majority of pseudopotentials used in DFT calculations are generated from all-electron atomic calculations by self-consistently solving the following radial

Schrödinger equation,

$$\left[-\frac{1}{2} \frac{d^2}{dr^2} + \frac{l(l+1)}{2r^2} - \frac{Z}{r} + V_H(r) + V_{XC}(r)\right]\psi_l^{AE}(r) = \varepsilon_l \psi_l^{AE}(r) \quad (2.34)$$

where $V_H(r)$ and $V_{XC}(r)$ are the Hartree and exchange-correlation potentials, and $\psi_l^{AE}(r)$ is the all-electron atomic wavefunction with angular momentum component l . Conventionally, the pseudopotential is then constructed by satisfying four general criteria: (i) the valence pseudo-wavefunction $\psi_l^{PS}(r)$ must be the same as $\psi_l^{AE}(r)$ outside a given cutoff radius r_c , (ii) the charge enclosed within r_c must be equal for the two wavefunctions,

$$\int_0^{r_c} r^2 |\psi_l^{PS}(r)|^2 dr = \int_0^{r_c} r^2 |\psi_l^{AE}(r)|^2 dr = 1 \quad (2.35)$$

This is commonly referred to as norm-conservation. (iii) $\psi_l^{PS}(r)$ must not contain any nodes and be continuous at r_c , as well as its first and second derivatives. Finally, (iv) the valence eigenvalues from all-electron and pseudopotential must be equal.

The pseudopotential is not uniquely constructed, indeed the above conditions permit a considerable amount of freedom when generating pseudo-wavefunctions, consequently many different ways have been developed for constructing pseudopotentials. Once a particular pseudo-wavefunction is created, the ionic pseudopotential is then obtained by inverting the radial Kohn-Sham Eqs. (2.34), giving,

$$V_{ion,l}^{PS}(r) = \varepsilon_l - V_H^{PS}(r) - V_{XC}^{PS}(r) - \frac{l(l+1)}{2r^2} + \frac{1}{2\psi_l^{PS}(r)} \frac{d^2 \psi_l^{PS}(r)}{dr^2} \quad (2.36)$$

where $V_H^{PS}(r)$ and $V_{XC}^{PS}(r)$ are calculated from pseudo-wavefunctions. A consequence of this procedure is that a separate pseudopotential must be generated for each angular momentum component l . The pseudopotential operator $V^{PS}(r)$ can be written in a semi-local form[38, 39] as

$$V_{ion}^{PS}(r) = V_{loc}^{PS}(r) + \sum_l \delta V_l^{PS}(r) \hat{P}_l \quad (2.37)$$

where $V_{loc}^{PS}(r)$ is a local potential and \hat{P}_l projects out the l th angular momentum component of the semi-local part $\delta V_l^{PS}(r)$

$$\delta V_l^{PS}(r) = V_{ion}^{PS}(r) - V_{loc}^{PS}(r) \quad (2.38)$$

Kleinman and Bylander[KB][40] observed that greater efficiency could be attained if the non-locality was not restricted to the angular momentum part, but if the radial component was also converted into a separable non-local form. Therefore in the Kleinman-Bylander approach, the semi-local form is converted into the fully non-local form $\delta V_{KB}^{PS}(r)$, given by,

$$\delta V_{KB}^{PS}(r) = \sum_l \frac{|\delta V_l^{PS}(r)\phi_l^0(r)\rangle\langle\phi_l^0(r)\delta V_l^{PS}(r)|}{\langle\phi_l^0(r)|\delta V_l^{PS}(r)|\phi_l^0(r)\rangle} \quad (2.39)$$

where the $\phi_l^0(r)$ are the atomic pseudo-wavefunctions calculated with $\delta V_l^{PS}(r)$. The Kleinman-Bylander form drastically reduces the computational resources in a pseudopotential calculation: for a plane-wave expansion of dimensionality N_{PW} , the semi-local form requires storage of $\sim (N_{PW}^2 + N_{PW})/2$ projections for each angular momentum state, whereas the corresponding KB pseudopotential evaluates just N_{PW} projections and simple multiplications.

2.8 Optimization method

In the beginning of the calculations atomic coordinates are either from experiments or classical simulations. Unfortunately, these atomic coordinates are usually not at equilibrium within DFT. Therefore, it is necessary to optimize the atomic coordinates. There are many optimization methods for first-principles calculations. Almost all optimization codes use what is called a quadratic approximation, namely they expand the dependence of energy in a form

$$E^* = E + \mathbf{g}^T \mathbf{s} + \frac{1}{2} \mathbf{s}^T \mathbf{B} \mathbf{s} \quad (2.40)$$

where E^* is the predicted energy for a step s from the current point; \mathbf{B} is the Hessian matrix. E and \mathbf{g} are the energy and gradient(negative of the force) calculated at the current point. According to the Hellmann-Feynman theorem [42, 43], the force can be expressed as

$$\mathbf{f} = -\frac{d}{d\mathbf{R}}\langle\Psi|\mathbf{H}|\Psi\rangle = -\langle\Psi|\frac{\partial\mathbf{H}}{\partial\mathbf{R}}|\Psi\rangle \quad (2.41)$$

where \mathbf{R} denotes the atomic positions, and \mathbf{H} is the hamiltonian of the system.

For the evaluation of the Hessian matrix, different algorithm use different approaches to this matrix. The most primitive is steepest descent, which takes \mathbf{B} as the unitary matrix so the algorithm will take a step along the direction of the force. Better algorithms such as conjugate gradient methods use some information about the previous step. By far and away the most common method is to exploit the Hessian, either by directly computing it (very CPU expensive for codes) or to create an estimate of it that improves as the calculation proceeds. The most successful approach is the Broyden-Fletcher-Goldberg-Shamo (BFGS)[41] update. In mathematics, the BFGS method is a method to solve an unconstrained nonlinear optimization problem. The BFGS method is derived from the Newton's method in optimization, a class of hill-climbing optimization techniques that seeks the stationary point of a function, where the gradient is zero. Newton's method assumes that the function can be locally approximated as a quadratic form in the region around the optimum, and use the first and second derivatives to find the stationary point. In Quasi-Newton methods the Hessian matrix of second derivatives of the function to be minimized does not need to be computed at any stage. The Hessian is updated by analyzing successive gradient vectors instead. Quasi-Newton methods are a generalization of the secant method to find the root of the first derivative for multidimensional problems. In multi-dimensions the secant equation is underdetermined, and quasi-Newton methods differ in how they constrain the solution. The BFGS method is one of the most successful members of this class, and I will give a very brief introduction to this optimization method.

If we take a step \mathbf{s}_k , the gradient will change from \mathbf{g}_k at the previous point to \mathbf{g}_{k+1} at the new point, and we can write

$$\mathbf{g}_k - \mathbf{g}_{k+1} = \mathbf{y}_k = \mathbf{B}_k \mathbf{s}_k \quad (2.42)$$

In principle, there are many ways to exploit this information. The method used in BFGS is to update the Hessian for the next step via:

$$\mathbf{B}_{k+1} = \mathbf{B}_k + \Delta \mathbf{B}_k \quad (2.43)$$

$$\Delta \mathbf{B}_k = -\frac{\mathbf{y}_k \mathbf{y}_k^T}{\mathbf{s}_k^T \mathbf{y}_k} + \frac{\mathbf{B}_k \mathbf{s}_k \mathbf{s}_k^T \mathbf{B}_k}{\mathbf{s}_k^T \mathbf{B}_k \mathbf{s}_k} \quad (2.44)$$

The procedure is then to solve for

$$\mathbf{s}_{k+1} = -\mathbf{B}_{k+1}^{-1} \mathbf{g}_k \quad (2.45)$$

move by \mathbf{s}_{k+1} , recalculate the gradient, update the Hessian \mathbf{B}_k and iterate. Often the first estimate for the Hessian is the unitary matrix, although it does not have to be and the better the initial guess is, the faster the algorithm will converge. Often the estimate of the Hessian will change rather a lot during the calculation, and at some locations can be rather bad. The power of the BFGS method is that experience over the last decade has indicated that in most cases it will correct itself rather quickly, and is therefore rather robust.

2.9 Expansions of Kohn-Sham orbitals

All ab initio electronic structure calculations employ expansions in basis sets of atomic orbitals. The two most common choices of basis function are plane waves and Gaussian type.

2.9.1 Plane wave basis set

As yet there has been no mention of how to handle the infinite number of interacting electrons moving in the static field of an infinite number of ions. Essentially, there are two difficulties to overcome: a wavefunction has to be calculated for each of the infinite number of electrons which will extend over the entire space of the solid and the basis set in which the wavefunction will be expressed will be infinite.

The ions in a perfect crystal are arranged in a regular periodic way (at 0K). Therefore the external potential felt by the electrons will also be periodic - the period being the same as the length of the unit cell L (That is, the external potential on an electron at \mathbf{r} can be expressed as $V(\mathbf{r}) = V(\mathbf{r} + L)$). This is the requirement needed for the use of Bloch's theorem. By the use of this theorem, it is possible to express the wavefunction of the infinite crystal in terms of wavefunctions at reciprocal space vectors of a Bravais lattice. The Bloch theorem states that wavefunction of an electron $\Psi_{i\mathbf{k}}(\mathbf{r})$, within a periodic potential, can be written as the product of a lattice periodic part $u_{i\mathbf{k}}(\mathbf{r})$ and a wavelike part $\exp(i\mathbf{k} \cdot \mathbf{r})$,

$$\Psi_{i\mathbf{k}}(\mathbf{r}) = \exp(i\mathbf{k} \cdot \mathbf{r})u_{i\mathbf{k}}(\mathbf{r}) \quad (2.46)$$

where the subscript i indicates the band index and \mathbf{k} is in the first Brillouin zone (BZ) and periodic; $u_{i\mathbf{k}}(\mathbf{r})$ can be expanded in a Fourier series :

$$u_{i\mathbf{k}}(\mathbf{r}) = \sum_{\mathbf{G}} c_{i,\mathbf{G}} \exp(i\mathbf{G} \cdot \mathbf{r}) \quad (2.47)$$

where $c_{i,\mathbf{G}}$ is the coefficient in the expansion. The above results show that the electron wavefunctions can be expanded in terms of a linear combination of plane waves,

$$\Psi_{i\mathbf{k}}(\mathbf{r}) = \sum_{\mathbf{G}} c_{i,\mathbf{k}+\mathbf{G}} \exp(i(\mathbf{k} + \mathbf{G}) \cdot \mathbf{r}) \equiv \sum_{\mathbf{q}} c_{i,\mathbf{q}} |\mathbf{q}\rangle (\mathbf{q} = \mathbf{k} + \mathbf{G}) \quad (2.48)$$

By the use of Bloch's theorem, the problem of the infinite number of electrons has now been mapped onto the problem of expressing the wavefunction in terms

of an infinite number of reciprocal space vectors within the first Brillouin zone of the periodic cell, \mathbf{k} . This problem is dealt with by sampling the Brillouin zone at special sets of \mathbf{k} -points discussed in Section 2.10. The electronic wavefunctions at each \mathbf{k} -point are now expressed in terms of a discrete plane wave basis set, which offers a complete basis set that is independent of the type of crystal and treats all areas of space equally. This is in contrast to some other basis sets which use localised functions such as Gaussians which are dependent on the positions of the ions.

Using a plane-wave basis set to expand the electronic wavefunctions in periodic systems leads to a particularly simple formulation of the Kohn-Sham equations in DFT. Accounting for the fact that the potential has the same lattice periodicity as $u_{i\mathbf{k}}(\mathbf{r})$

$$V_{eff}(\mathbf{r}) = \sum_{\mathbf{G}} V_{eff}(\mathbf{G}) \exp(i\mathbf{G} \cdot \mathbf{r}) \quad (2.49)$$

One arrives at the following expression for its matrix elements in the plane-wave basis:

$$\langle \mathbf{q}' | V_{eff}(\mathbf{r}) | \mathbf{q} \rangle = \sum_{\mathbf{G}} V_{eff}(\mathbf{G}) \langle \mathbf{q}' | \exp(i\mathbf{G} \cdot \mathbf{r}) | \mathbf{q} \rangle = \sum_{\mathbf{G}} V_{eff}(\mathbf{G}) \delta_{\mathbf{q}' - \mathbf{q}, \mathbf{G}} \quad (2.50)$$

Substituting these Bloch states into the Kohn-Sham equations, multiplying by $\langle \mathbf{q}' |$ from the left and integrating over \mathbf{r} gives a set of matrix equations for any given \mathbf{k}

$$\int d\mathbf{r} \langle \mathbf{q}' | \left\{ \left[-\frac{1}{2} \nabla^2 + V_{eff}(\mathbf{r}) \right] \sum_{\mathbf{q}} c_{i,\mathbf{q}} | \mathbf{q} \rangle = \varepsilon_i c_{i,\mathbf{q}} | \mathbf{q} \rangle \right\} \quad (2.51)$$

Finally, we obtain a reciprocal-space representation of the Kohn-Sham equations,

$$\sum_{\mathbf{G}'} \left[\frac{1}{2} |\mathbf{k} + \mathbf{G}|^2 \delta_{\mathbf{G}, \mathbf{G}'} + V_{eff}(\mathbf{G} - \mathbf{G}') \right] c_{i, \mathbf{k} + \mathbf{G}'} = \varepsilon_i c_{i, \mathbf{k} + \mathbf{G}} \quad (2.52)$$

For an exact calculation, the dimension of the plane-wave basis set should be infinite. Fortunately the plane-waves at the lower end of the kinetic energy range are most important, so a practical solution of KS equation can be obtained by truncating the basis set to a finite number of plane-waves. This is defined by the

kinetic cutoff energy E_{cut} ,

$$\frac{1}{2}|k + G|^2 \leq E_{\text{cut}} \quad (2.53)$$

Certain integrals and operations are much easier to carry out with plane wave basis functions, than with their localized counterparts. In practice, plane wave basis sets are often used in combination with pseudopotential, so that the plane waves are only used to describe the valence charge density. This is because core electrons tend to be concentrated very close to the atomic nuclei, resulting in large wavefunction and density gradients near the nuclei which are not easily described by a plane wave basis set unless a very high energy cutoff, (and therefore small plane wavelength), is used. Furthermore, all functions in the basis are mutually orthogonal, and plane wave basis sets do not exhibit basis set superposition error. Another important advantage of a plane wave basis is that it is guaranteed to converge to the target wave function if $E_{\text{cut}} \rightarrow \infty$ while there is no such guarantee for Gaussian type basis sets.

2.9.2 Gaussian orbital basis set

There are also some disadvantages with plane wave expansions. The principal of these is that an extremely large number of functions need to be used for the systems, expecially for some metals with localized d orbitals. This means that the time and memory requirements of such a code will be considerable. One way of avoiding this is to use a basis set of localised orbitals, such as Gaussian orbitals:

$$\psi_{i,k}(\mathbf{r}) = \sum_i C_i \phi_i(\mathbf{r}) \quad (2.54)$$

$$\phi_i(\mathbf{r}) = \frac{1}{\sqrt{\Omega}} \sum_{\mathbf{R}} \exp(i\mathbf{k} \cdot (\mathbf{R} + \tau_i)) f_i(\mathbf{r} - \mathbf{R} - \tau_i) \quad (2.55)$$

where

$$f_i(\mathbf{r}) = x^{n_1} y^{n_2} z^{n_3} Y_{lm}(\theta, \varphi) \exp(-\beta \mathbf{r}^2) \quad (2.56)$$

$Y_{lm}(\theta, \varphi)$ are spherical harmonics. The atomic position in the unit cell is τ_i and \mathbf{R} is a Bravais lattice vector. n_1 , n_2 and n_3 are integers, and specifies the orbital

character($n_1 + n_2 + n_3 = l$). If n_1 , n_2 and n_3 are all zero the function corresponds to a s orbital of spherical symmetry. Orbitals of p-symmetry correspond to one of these integers being unity and the others zero, whereas five d-like and one s-like orbital can be generated if $n_1 + n_2 + n_3 = 2$.

This expansion has the advantage that it is very efficient, efficiently applicable to all elements of the periodic table, and it is flexible (if we have one difficult atom, additional orbitals can be placed on just that atom so the overall speed of the calculation is not significantly affected). For example when modelling an element of a transition element as an impurity in silicon, higher angular momentum functions need to be placed on that atom. However, the rest of the system can be treated with the standard basis set.

Disadvantages include the fact that the functions can become over-complete (numerical noise can enter a calculation if two functions with similar exponents are placed on the same atom), that they are difficult to program (especially if high angular momentum functions are needed), that it is difficult to test or to demonstrate absolute convergence (many things can be changed: the number of functions, the exponents, the function centres).

2.10 k point samplings

The first Brillouin zone can be mapped out by a continuous set of points, throughout that region of reciprocal space (k-space). The occupied states at each k-point contribute to the electronic potential of the bulk solid. Since the set $\{\mathbf{k}\}$ is dense, there are an infinite number of k-points in the Brillouin zone at which the wavefunctions must be calculated. Therefore if a continuum of basis sets were required, the basis set for any calculation would still be infinite.

For this reason electronic states are only calculated at a set of k-points determined by the shape of the Brillouin zone compared to that of its irreducible part. The reason that this can be done is that the electronic wavefunctions at k-points

that are very close to each other will almost be identical. It is therefore possible to represent the electronic wavefunctions over a region of reciprocal space at a single k-point. This approximation allows the electronic potential to be calculated at a finite number of k-points and hence determine the total energy of the solid.

Methods have been devised for obtaining very accurate approximations to the electronic potential from a filled electronic band by calculating the electronic wavefunctions at special sets of k-points. The two most common methods are those of Chadi and Cohen[44] and Monkhorst and Pack[45]. I will mainly introduce the method of Chadi and Cohen, which has been employed for all calculations in this thesis.

We consider a smoothly varying periodic function, and expand it in a Fourier series:

$$g(\mathbf{k}) = f_0 + \sum_{i=m}^{\infty} g_m e^{i\mathbf{k} \cdot \mathbf{R}_m} \quad (2.57)$$

where the \mathbf{R}_m are lattice vectors. Let us assume another function $f(\mathbf{k})$ which has the complete symmetry T of the lattice:

$$f(\mathbf{k}) = \frac{1}{n_T} \sum_i g(T_i \mathbf{k}) \quad (2.58)$$

where T_i range over all the operations of the lattice point group T , and n_T is the number of elements in T . From $g(\mathbf{k})$ we can express $f(\mathbf{k})$ in the following form

$$\begin{aligned} f(\mathbf{k}) &= f_0 + \sum_{m=1}^{\infty} \sum_i \frac{1}{n_T} g_m e^{iT_i \mathbf{k} \cdot \mathbf{R}_m} \\ &= f_0 + \sum_{m=1}^{\infty} f_m A_m(\mathbf{k}) \end{aligned} \quad (2.59)$$

where

$$f_m = \frac{g_m}{n_T}, \quad (2.60)$$

and

$$A_m(\mathbf{k}) = \sum_{|\mathbf{R}|=C_m} e^{i\mathbf{k} \cdot \mathbf{R}_m} \quad (2.61)$$

In the above equation, the lattice vectors are ordered, and satisfy $0 < C_m \leq C_{m+1}$. $A_m(\mathbf{k})$ satisfies the following relations:

$$\frac{\Omega}{2\pi^3} \int_{BZ} A_m(\mathbf{k}) d\mathbf{k} = 0, \quad m = 1, 2, \dots \quad (2.62)$$

$$\frac{\Omega}{2\pi^3} \int_{BZ} A_m(\mathbf{k}) A_n(\mathbf{k}) d\mathbf{k} = N_n \delta_{mn} \quad (2.63)$$

$$A_m(\mathbf{k} + \mathbf{G}) = A_m(\mathbf{k}) \quad (2.64)$$

$$A_m(T_i \mathbf{k}) = A_m(\mathbf{k}) \quad (2.65)$$

$$A_m(\mathbf{k}) A_n(\mathbf{k}) = \sum_j a_j(mn) A_j(\mathbf{k}) \quad (2.66)$$

In the above equations Ω is the volume of the primitive cell, N_n is the number of lattice vectors, and \mathbf{G} is any lattice reciprocal lattice vector. The average value over the Brillouin zone of the f is given by

$$\bar{f} = \frac{\Omega}{2\pi^3} \int f(\mathbf{k}) d\mathbf{k} \quad (2.67)$$

Inserting Eqs. (2.59) into above equation, we can obtain $\bar{f} = f_0$. The perfect special \mathbf{k} -points, \mathbf{k}_0 , would have $A_m(\mathbf{k}_0) = 0$ for $m = 1, 2, \dots, \infty$, so that $\bar{f} = f_0 = f(\mathbf{k}_0)$. This point is called the "mean value point". In fact, such a point does not exist. The expansion coefficients f_m values markedly decrease for larger m so that the equation $A_m(\mathbf{k}_0) = 0$ should be satisfied for a finite value of m . As the $A_m(\mathbf{k})$ only depend on the lattice vectors, they can be determined for each crystal structure, independent of the actual function $f(\mathbf{k})$ that one wants to evaluate. Baldereschi [46] has obtained for the cubic crystals the point \mathbf{k}_0 , which satisfies $A_m(\mathbf{k}_0) = 0$ for a finite value of m , and yielded good electron charge density and energy in a number of diamond and zinc-blende crystals.

In order to satisfy the relation $A_m(\mathbf{k}_0) = 0$ for large values of m , one can choose a sequence of sets of points $\{\mathbf{k}\}$ and associated weights $\{W\}$ which satisfy the conditions

$$\sum_{i=1}^n W_i A_m(\mathbf{k}_i) = 0, \quad m = 1, 2, \dots, N \quad (2.68)$$

where N is a finite value.

$$\sum_{i=1}^n W_i = 1 \quad (2.69)$$

Using Eqs. (2.59), one can obtain

$$\sum_{i=1}^n W_i f(\mathbf{k}_i) = f_0 \sum_{i=1}^n W_i + \sum_{m=1}^N \sum_{i=1}^n W_i A_m(\mathbf{k}_i) f_m + \sum_{m=N+1}^{\infty} \sum_{i=1}^n W_i A_m(\mathbf{k}_i) f_m \quad (2.70)$$

Using Eqs. (2.69) and Eqs. (2.68), the above equation takes the following form

$$f_0 = \sum_{i=1}^n W_i f(\mathbf{k}_i) - \sum_{m=N+1}^{\infty} \sum_{i=1}^n W_i A_m(\mathbf{k}_i) f_m \quad (2.71)$$

Since the expansion coefficients f_m decrease rapidly for larger m , if N is large enough, we can neglect the second term of the above equation, and obtain the following relation

$$f_0 = \sum_{i=1}^n W_i f(\mathbf{k}_i) \quad (2.72)$$

Therefore, the average value over the Brillouin zone of the f can be represented by a special set of k-points.

The fact is used by Chadi and Cohen to generate a special set of k-points at which the reciprocal space should be sampled. The k-point set is generated as follows: picking two starting k-points \mathbf{k}_1 and \mathbf{k}_2 satisfying certain uniqueness conditions in the reciprocal space of the cell with point group operations $T = \{T_i\}$ then a new set of points can be generated by

$$\{\mathbf{k}_i\} = \mathbf{k}_1 + \{T_i\}\mathbf{k}_2 \quad (2.73)$$

The new set of points generated in this way can then be used in a similar process to generate larger sets. Through the symmetry operations, T_i , the new set of k-points can be "folded back" into the irreducible part. A normalised weighting factor can then be associated with each point with their ratios indicating the number of times that each point in the irreducible zone has been generated.

Chapter 3

DFT study of Si doped GaAs with different charge states

3.1 Introduction

Substitutional dopants in III-V semiconductors, such as Si atoms in GaAs, are of great interest for the applications in transistors, Schottky diodes, and doping superlattices which have been widely employed to control the electrical properties of semiconductors. Understanding the electronic properties of defects and the ability to control them is crucial for the performance of future microelectronic devices.

The DX center in bulk n-type (Al,Ga)As has been a topic of considerable interest for several decades [47, 48]. An enormous amount of work, either employing experimental methods [47, 48, 49, 50] or theoretical approaches [51, 52, 53, 54], have been done to understand the fundamental nature of the DX center in GaAs. The electrical and optical properties of the DX center are very well characterized experimentally, but very few experiments give the direct observation for the geometrical structure and charge state of DX center. Up to now, the model for the atomic structure of the DX center is primarily based on theoretical studies. The commonly accepted opinion is the model proposed by Chadi and Chang [51], who

pointed out that the DX center is related to Si impurities in GaAs and forms by electrons trapping on the shallow donor, displacing the substitutional Si away from one of its nearest-neighbor As atoms, along bond axis, breaking one bond and changing the bond configuration of Si from sp^3 to sp^2 . This configuration is metastable in bulk GaAs and stable in $\text{Al}_x\text{Ga}_{1-x}\text{As}$ alloys with $x \geq 0.22$ or under a hydrostatic pressure greater than ~ 20 kbar. The electron trapping involving a large lattice distortion for the DX formation lead to the negatively charged defect, which results in the lack of electron-paramagnetic-resonance(EPR)[55, 56, 57].

Although the stability of defects in bulk GaAs and in GaAs quantum dots[58, 59] have been extensively studied in the past, very few studies have been carried out to understand the stability of defects at or near the GaAs(110) surface and how the charge state of a defect affects the stability of the defect in the surface system. The DX center can exist in $\text{Al}_x\text{Ga}_{1-x}\text{As}$ alloys with $x \geq 0.22$, in bulk GaAs under a hydrostatic pressure greater than ~ 20 kbar or in GaAs quantum dots when the dot size is reduced to less than 14.5 nm in diameter[59], but nobody knows whether the DX center might also exist in the vicinity of the GaAs(110) surface so far.

On the experimental side, for the investigation of surface system, STM represents a unique tool and has been widely used to study the geometric and electronic structure of semiconductor surfaces[60]. It can directly show the local environment around the defects on the surface, such as deep level defects and defect complexes, whose structure is generally difficult to discern by other techniques. Bulk Si_{Ga} defects in GaAs(110) were investigated by many extended STM[62, 63, 64], and are not surface specific[61]. The defect in the deep layers of GaAs(110)(below the surface) remains some bulk physical features such as energy level, electronic structure, etc and the bulk defect can be loosely described as a point defect[62], but the reduced dimensionality and the interaction between the defect and the surface are expected to produce deviations from the bulk-like behavior. Capaz[64] studied the bulk and surface Arsenic antisite defects in GaAs(110) and found that defects display remarkably distinct properties depending on whether they are fourfold(below the surface) or threefold coordinated(at the surface). In this work, we

show that the stability of the charged defect depends on whether the Si_{Ga} defect is fourfold or threefold coordinated. Many bulk defects can be clearly analyzed by STM information. However, there are some properties, which are difficult to be directly observed by STM, such as the stability of a defect, the atomic structure of a defect in the deep layers(below the surface) and so on. Even today the interpretation of details around impurity atoms, defects, and adsorbates remains largely impossible without theoretical background. Ab initio total energy method is an essential tool to study these properties and interpret STM images. Within the theoretical method, the defect on the GaAs(110) surface has been studied by Wang et al.[65] and Duan et al. [66], but the bulk Si_{Ga} defects in the deep layers have not been studied yet.

In the present work we are especially interested in the DX center in GaAs(110) and the relative stabilities of the Si_{Ga} defects with different charge states in different layers of GaAs(110). In Sec. 3.2 we summarize the method and the computational details. In Sec. 3.3 we discuss the stability of the DX center in GaAs(110), and the relative stabilities of the Si_{Ga} with different charge states in GaAs(110), as well as the calculated STM images of neutral GaAs(110) and charged Si:GaAs(110). A summary of the main results of this chapter is given in Sec. 3.4.

3.2 Computational Method

The calculations are based on the ground state DFT within the local density approximation(LDA)[1], as implemented in the QUANTUM-ESPRESSO code[67]. We use norm conserving pseudopotentials to describe the electron-ion interactions. The valence wave functions are expanded in a plane wave basis set with the cutoff energy of 14 Ry. For the calculation of bulk GaAs, the Brillouin-zone integrations are performed using uniform k point grid of $6 \times 6 \times 6$ for self-consistent calculations and $8 \times 8 \times 8$ mesh for nonself-consistent calculations of densities of states. For

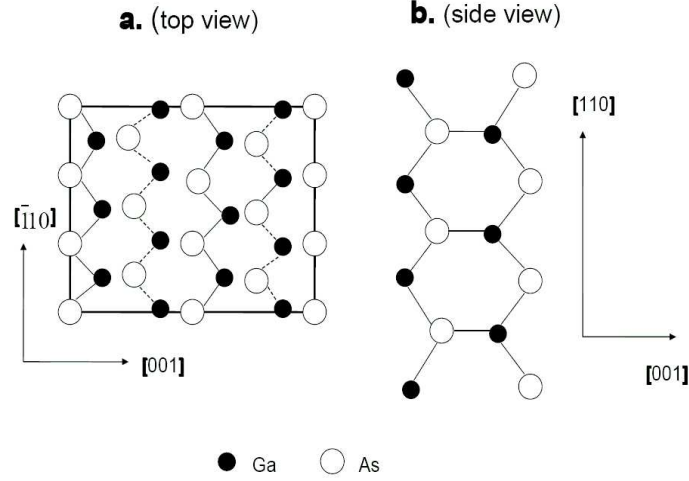


FIGURE 3.1: Top view of the supercells used to construct the clean and Si doped GaAs(110) surfaces. Open circles represent As atoms, solid circles denote Ga atoms.

the calculations of Si:GaAs(110), we use $5 \times 5 \times 1$ mesh for self-consistent calculations and $6 \times 6 \times 1$ mesh for nonself-consistent calculations of STM images. The calculation for bulk GaAs was performed using a $4 \times 4 \times 4$ unit supercell. Here all GaAs(110) geometries contain seven layers of atoms and five vacuum layers (about 12 Å for the vacuum region), and the supercells have 2×3 periodicity, as shown by the rectangle in Fig. 3.1. Since the upper and the lower surfaces of the slab are equivalent, our seven-layer slab allows to investigate the properties of substitutional Si in four positions (i.e. in the first, second, third, and fourth layer). We use a theoretical lattice constant for the calculations of bulk GaAs and Si:GaAs(110). The calculated lattice constant is 5.57 Å, in good agreement with experimental value of 5.65 Å. The structures are optimized using a technique of Broyden-Fletcher-Goldfarb-Shanno[41], which is a quasi-Newton method based on the construction of an approximated Hessian matrix at each system relaxation step. All the internal atoms are relaxed by minimizing the total energy and the quantum mechanical force until the changes in energy between two consecutive SCF steps are less than 0.1 meV and the force acting on each atom is less than 0.019 eV/Å. In order to determine the structure of DX center in bulk GaAs and in GaAs(110), we start to displace Si atom along $\langle 111 \rangle$ direction (i.e. along Si-As bond axis), break one Si-As bond, and obtain a first guessed DX center structure, then the

guessed structure is optimized by minimizing the total energy and the quantum mechanical force. Here all properties are referred to the optimized structures.

For the STM simulation, we employ Tersoff-Hamann approximation[28], which is based on quantum tunneling. When a conducting tip is brought very near to a metallic or semiconducting surface, a bias between the two can allow electrons to tunnel through the vacuum between them. According to Tersoff-Hamann approximation, the variation of the tunneling current with bias voltage V_{bias} is proportional to the local density of states(LDOS) of the sample at the tip position. Thus the energy integrated LDOS of occupied($V_{\text{bias}} < 0$) or unoccupied($V_{\text{bias}} > 0$) states in the energy range($E_f, E_f + V_{\text{bias}}$) contribute to the tunneling current, and can be directly compared with the contour map of constant current STM images. The tunneling current is given by the following equation

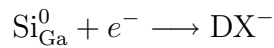
$$I(V) \propto \int_{E_f}^{E_f + V_{\text{bias}}} \rho(r_0, \varepsilon) d\varepsilon$$

where $\rho(r_0, \varepsilon)$ is the local charge density of all states encompassed by bias voltage, E_f is the Fermi level, and V_{bias} is the bias voltage.

3.3 Results and discussion

3.3.1 DX center in bulk GaAs and in GaAs(110)

According to Chadi[51, 52], DX center is formed from the following reaction



where Si_{Ga}^0 is a normal fourfold-coordinated substitutional Si atom on the Ga site shown in Fig. 3.2a, DX^- denotes the negatively charged broken bond configurations shown in Fig. 3.2(b-e), and e^- is a free electron in the conduction band.

Since it is difficult to accurately determine the energy levels of electrons and holes in a relatively small supercell, we only compare the total energies of defects with the same charge states. For the calculation of DX formation energy, here we employ the definition of Wei and Zhang[68]:

$$\Delta E(DX) = E(DX^-) - E(\text{Si}_{\text{Ga}}^-)$$

Where $E(DX^-)$ is the total energy of negatively charged DX^- center and $E(\text{Si}_{\text{Ga}}^-)$ is the total energy of fourfold-coordinated defect Si_{Ga} at the same charge state.

When DX center is formed, Si atom displaces away from one of its nearest-neighbor As atoms along the Si-As bond axis into a threefold-coordinated interstitial position, and hence, the bond configuration of Si atom changes from sp^3 to sp^2 . There are four possible configurations for the DX center in the bulk region of GaAs(110) and in bulk GaAs as can be seen from Fig. 3.2(a-e).

In bulk GaAs, the formation of the DX center involves a displacement of the Si atom by 1.09 Å along the Si-As bond axis, in good agreement with a previous calculation[52]. Configurations with the Si atom displacing along AB, AC, AD and AE directions, which are along the As→Si directions as can be seen from Fig. 3.2a, have the same formation energy due to the point-group symmetry of bulk GaAs, and the DX formation energy is calculated to be 0.22 eV. The positive DX formation energy indicates that the DX center is not stable in bulk GaAs.

For a surface system, on the other hand, the proximity of the surface has an important influence on the formation of the DX center. The DX formation energy and the broken Si-As bond length depend on the displacing direction of Si atom and Si position in GaAs(110). The formation energy ranges from 0.12 to 1.36 eV, and the Si-As distance from 3.27 to 4.08 Å, respectively, as can be seen from Table 3.1. Comparing to that in the bulk GaAs, the DX formation energy resulted from displacing Si atom in the first layer along AC direction is 0.12 eV, which is smaller than that of bulk GaAs. The smaller DX formation energy maybe originated from the two broken Si-As bonds(both bond lengths are 3.95 Å), which are parallel to the surface. But, it is still a metastable state due to the positive value of

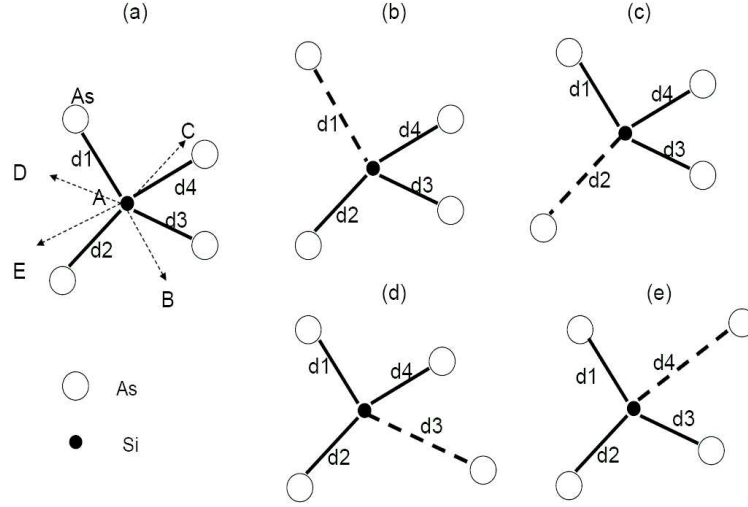


FIGURE 3.2: (a) Schematic view of the normal substitutional site(for GaAs(110) geometry, the motion of Si atom along AD or AE direction means that the displacing of Si atom is parallel to the surface). (b)-(e) Four possible DX center configurations with different broken Si-As bonds. Dash lines represent the broken Si-As bonds.

TABLE 3.1: Si-As bond length and formation energy of DX center in bulk GaAs:Si and GaAs(110):Si

Si positions	displacing	d1 (Å)	d2 (Å)	d3 (Å)	d4 (Å)	energy(eV)
Bulk	AB	3.46	2.39	2.39	2.39	0.22
The first layer of GaAs(110)	AB	only substitutional position				0.00
	AC	2.38	—	3.95	3.95	0.12
	AD	equivalent to AB				0.00
	AE	equivalent to AB				0.00
The Second layer of GaAs(110)	AB	3.95	2.46	2.48	2.48	0.81
	AC	2.39	4.08	2.38	2.38	1.36
	AD	only substitutional position				0.00
	AE	equivalent to AD				0.00
The third layer of GaAs(110)	AB	only substitutional position				0.00
	AC	2.39	3.41	2.39	2.39	0.93
	AD	2.39	2.39	3.49	2.39	1.15
	AE	2.39	2.41	2.39	3.49	1.15
The fourth layer of GaAs(110)	AB	3.53	2.39	2.39	2.39	1.12
	AC	2.39	3.53	2.39	2.39	1.12
	AD	2.39	2.39	3.27	2.39	0.95
	AE	2.39	2.39	2.39	3.27	0.95

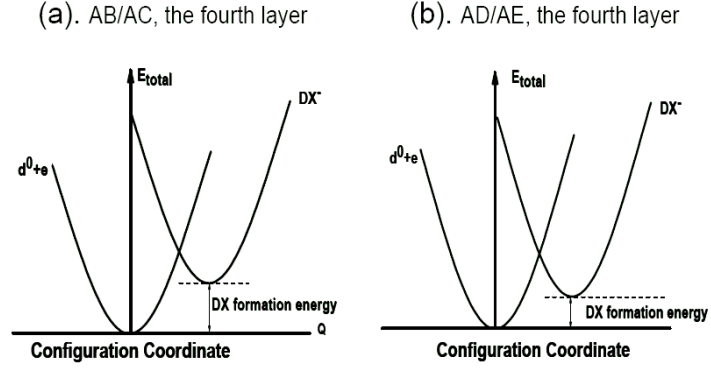


FIGURE 3.3: Configuration-coordinate diagrams for DX center in GaAs(110). (a) The DX center forms by displacing Si atom along AB/AC direction in the fourth layer of GaAs(110); (b) The DX center forms by displacing Si atom along AD/AE direction in the fourth layer of GaAs(110).

DX formation energy. However, the DX formation energies in the bulk region of GaAs(110) are significantly higher than those of bulk GaAs and resulted from displacing Si atom along AC direction in the first layer. The higher positive DX formation energies in the bulk region of GaAs(110) indicate that the DX centers are rather unstable near the GaAs(110) surface. This can be seen clearly from a simple configuration-coordinate diagram shown in Fig. 3.3. The lowest parabola in Fig. 3.3 represents the substitutional configuration for the Si atom in the fourth layer of GaAs(110), and the higher parabola in Fig. 3.3 denotes the interstitial configuration for the Si atom displacing along AB/AC(Fig. 3.3(a)) or AD/AE(Fig. 3.3(b)) in the fourth layer of GaAs(110). Displacing Si atom in the third or fourth layer along AD and AE directions, which are parallel to the surface, result in the same DX formation energies. Displacing Si atom in the fourth layer along AB and AC directions also lead to the same DX formation energies. However, if the Si atom locates in the first layer and is displaced along AB, AD or AE direction, it will move back to the substitutional position. Similarly, if the Si atom locates in the second layer and is displaced along the AD or AE direction or locates in the third layer and is displaced along AB direction, it will also move back to the substitutional position, which is different from the behavior of a Si atom in the fourth layer. The DX formation energy resulting from displacing Si atom in the second layer along AB direction(toward the surface) is 0.81 eV, which is smaller

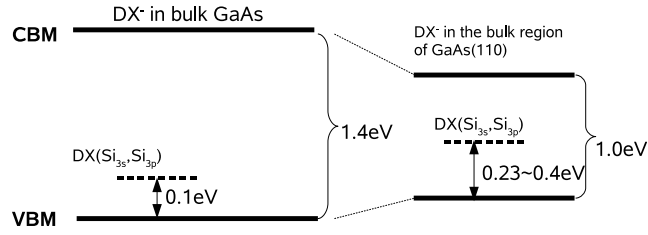


FIGURE 3.4: Schematic views of energy levels of DX in bulk GaAs and in GaAs(110).

than that of displacing Si atom in the same layer along AC direction (toward the bulk) by 0.55 eV, while the length of the unbroken Si-As bonds in the former case is larger than the latter by about 0.1 Å. Longer Si-As bond lengths means that a larger amount of negative charge is transferred to the Si atom which leads to larger Coulomb repulsion between Si atom and As atom.

When the DX center is formed in bulk GaAs, a localized electronic level, which is mainly formed by Si_{3s} and Si_{3p} orbitals from the broken Si-As bond, occurs in the band gap (the so-called DX state), and locates about 0.1 eV above the top of the occupied bands, as can be seen from Fig. 3.4. However, in the bulk region of GaAs(110), the DX state locates above the top of the occupied bands by about 0.23-0.4 eV depending on the Si position in GaAs(110) and the displacing direction of Si atom. Also the GaAs(110) surface has a smaller band gap than the GaAs bulk crystal. Both effects, i.e. the upward shift of the donor state and the decrease of the band gap in GaAs(110), increase the formation energy of the DX center and lower its stability as compared to its properties in bulk GaAs.

TABLE 3.2: Relative total energies($E_n - E_4$, $n=1,2,3,4$, E_n represents the total energy for Si atom in the different layers, and E_4 represents the total energy for Si atom in the fourth layer) of Si:GaAs(110) with different charge states

Si positions	negative(eV)	neutral(eV)	positive(eV)
$E_1 - E_4$	-0.50	0.00	0.30
$E_2 - E_4$	0.09	0.09	0.05
$E_3 - E_4$	0.05	0.01	-0.01
$E_4 - E_4$	0.00	0.00	0.00

3.3.2 Substitutional Si atom in GaAs(110) with different charge states

According to the last section, the DX center is unstable near GaAs(110) surface(i.e. no DX center near the GaAs(110) surface), so Si atom should be substitutional on the Ga site near the GaAs(110 surface)(i.e. Si_{Ga} defect is fourfold near the GaAs(110) surface, and all Si-As bond lengths are about 2.40 Å). In this section, we will discuss the stability of substitutional Si on Ga site with different charge states near the GaAs(110) surface. We will show that the charge state affects the stability of the system. For comparison, we set the total energy of a substitutional Si atom in the fourth layer as zero energy, all the total energies for substitutional Si atoms in the other layers are given relative to this layer. The calculated relative total energies of Si:GaAs(110) with different charge states are given in Table 3.2.

In this work Makov-Payne corrections[69] have been used for the computation of the charged systems. The total energies of charged system are calculated based on the neutral cell by the QUANTUM-ESPRESSO code. We add one electron to a system for the simulation of negatively charged system, and remove one electron from a system to simulate the positively charged system. It should be noted that in some cases the extra charge in the charged Si:GaAs(110) is not localized on the defect. For instance, once the defect locates in the deep layers, the extra charge is delocalized. For convenience, here Si_{Ga}^- or Si_{Ga}^+ means that the entire Si:GaAs(110) system is charged. For neutral system, when Si atoms replace Ga atoms in the second or third layer, the total energies are slightly higher than that of Si_{Ga}^0 defect

in the fourth layer with energy differences less than 0.1 eV. Although the bond configuration of the threefold-coordinated Si atom on the GaAs(110) surface is different from those of the fourfold-coordinated Si atoms in the deep layers, the total energy for Si_{Ga}^0 defect on the surface is same to that of the Si_{Ga}^0 defect in the fourth layer. It indicates that the bond configuration of the Si atom for neutral GaAs(110):Si has minor influence on the total energy. As a result the difference of the total energies between the substitutional Si atom in the first layer(at the surface) and in the deep layers(below the surface) may be expressed as

$$E(\text{Si}_{\text{Ga}}^0)_{\text{surface}} - E(\text{Si}_{\text{Ga}}^0)_{\text{deep layer}} \approx 0.0$$

For negatively charged Si:GaAs(110), the total energy for Si_{Ga}^- defects in the second layer and in the third layer are also slightly higher than that of Si_{Ga}^- defect in the fourth layer, and this behavior is similar to that of neutral system. However, once the Si_{Ga}^- defect is exposed on the surface, the total energy decreases by 0.4~0.5 eV with respect to those of Si_{Ga}^- defects in the deep layers. This indicates that Si atom prefers to replace Ga atom at the GaAs(110) surface when the system is negatively charged. The donor state, which lies in the band gap, is localized and primarily concentrated on the three-coordinated Si donor at the surface, as can be seen from Fig. 3.5. For positively charged system, the total energy for Si_{Ga}^+ defect in the second layer is larger than that of the Si_{Ga}^+ defect in the fourth layer only by 0.05 eV. However, once Si_{Ga}^+ defect is exposed on the surface, the total energy increases by 0.3 eV as compared to the Si_{Ga}^+ defect in the fourth layer, which is in contrast to the energy gain of the negatively charged Si:GaAs(110). It indicates that it is not stable for Si atom to substitute a Ga atom on the positively charged surface. As a result the difference of total energies between the substitutional Si atom in the first layer and in the deep layers are given by

$$E(\text{Si}_{\text{Ga}}^-)_{\text{surface}} - E(\text{Si}_{\text{Ga}}^-)_{\text{deep layer}} \approx -0.5\text{eV}$$

$$E(\text{Si}_{\text{Ga}}^+)_{\text{surface}} - E(\text{Si}_{\text{Ga}}^+)_{\text{deep layer}} \approx 0.3\text{eV}$$

As discussed above, the total energy of the charged Si:GaAs(110) will be signif-

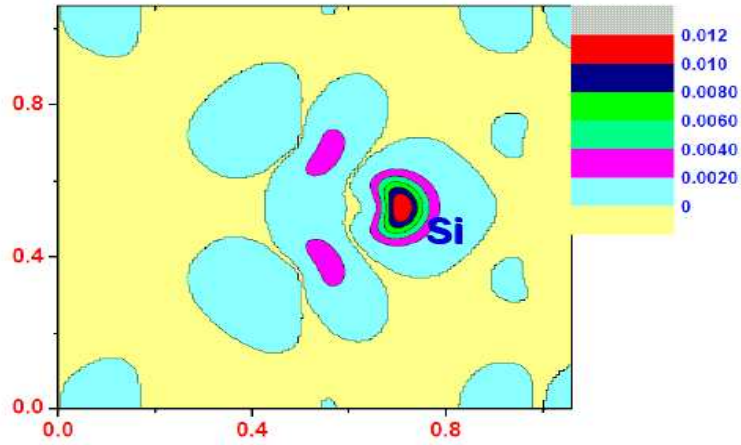


FIGURE 3.5: Contour plot of charge density $\rho(Si_{Ga}^-)$ for Si atom in the surface layer, associated with the donor state at the Γ point.

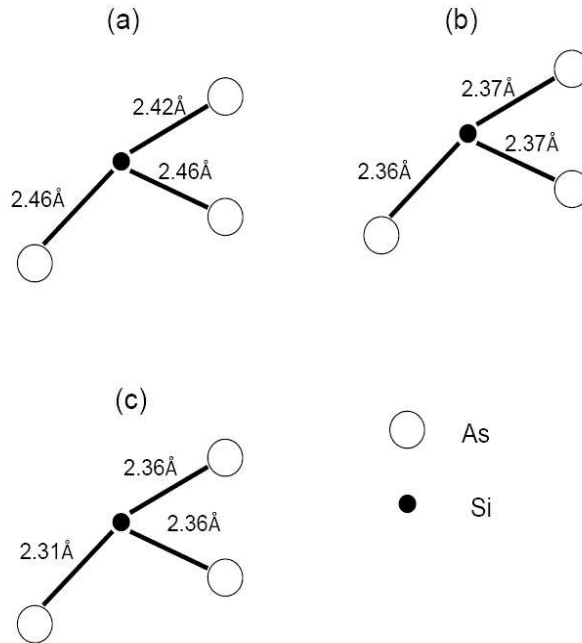


FIGURE 3.6: calculated AS-Si bond lengths for substitutional Si atom in the first layer(at the surface) with negative(a), neutral(b) and positive state(c)

icantly changed once Si atom is exposed on the surface. Here we will show that the extra charge of the charged Si:GaAs(110) is concentrated on the defect when Si atom replaces Ga atom on the surface which results in a quite different total energy with respect to the deep layers. In order to study the charge state of a defect exposed on the surface, the Si-As bond lengths are calculated, as can be seen from Fig. 3.6(a-c). For the neutral Si:GaAs(110), the calculated bond lengths are 2.36, 2.37 and 2.37 Å, respectively. For the negatively charged Si:GaAs(110),

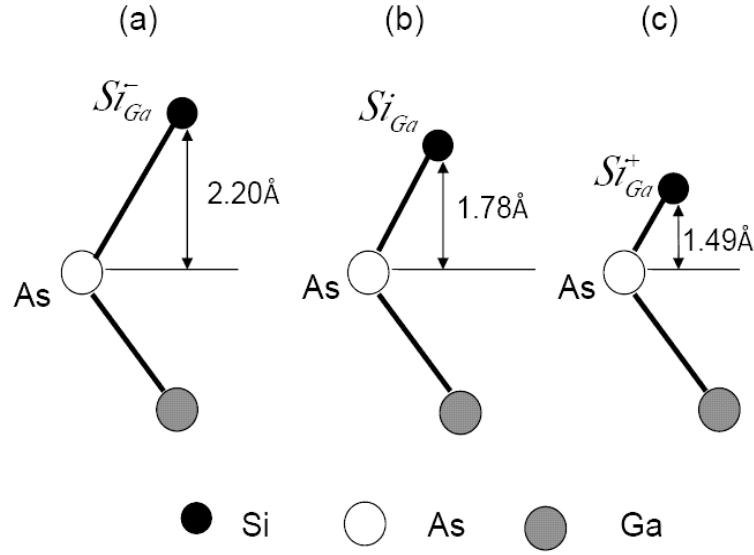


FIGURE 3.7: Relative height of substitutional Si atom in the first layer(at the surface)

the Si-As bond lengths are 2.46, 2.46 and 2.42 Å, respectively, larger than those of neutral Si:GaAs(110). Since the As atom in GaAs is negatively charged, the larger Si-As bond lengths for the negatively charged Si:GaAs(110) indicates that the extra negative charge is mainly concentrated on the Si atom, resulting in the negatively charged Si atom. The Coulomb repulsion between Si^- and As^- atom leads to the longer Si-As bond. The Si atom protrudes out of the surface in the negatively charged system as shown in Fig. 3.7. However, for the positively charged Si:GaAs(110), the Si-As bond lengths are 2.31, 2.36 and 2.36 Å, respectively, smaller than those of neutral Si:GaAs(110). This indicates that the extra positive charge is mainly concentrated on the Si atom, and the Coulomb attraction between Si^+ and As^- leads to the smaller Si-As bond length. The Si atom moves toward the bulk by 0.7 Å compared to that in the negative charged system as shown in Fig. 3.7.

3.3.3 STM simulation

In this section, we present the calculated STM images of the clean GaAs(110) and the charged Si:GaAs(110). It should be noted that some experiments give different

results since the doping concentration, significantly different tip and the measurement conditions have an important influence on the measured STM images[70]. Fig. 3.8 represents the calculated STM image of the clean neutral GaAs(110) at bias voltage of 2.0 V. Here the positive bias voltage indicates electrons tunnel from the probe into the conduction bands. The bright feature of the STM image corresponds to the large protrusion of the atoms and the dark feature corresponds to the small protrusion of atoms. The calculated STM image of clean GaAs(110) in Fig. 3.8 agrees with the experiments [70, 71] and previous theoretical results [65, 66]. The bright feature corresponds to the empty dangling-bond states of Ga atoms on the GaAs(110) surface.

The calculated STM images for positively charged donor at bias voltage of 1 V and 2 V are shown in Fig. 3.9 and Fig. 3.10, respectively. The STM image in Fig. 3.10 is similar to that of the clean GaAs(110), but the Si atom presents a larger protrusion compared to the Ga atom which is more apparent when the bias voltage is 1 V shown in Fig. 3.9. The large protrusion of the Si_{Ga}^+ defect at the positive bias voltage is similar to the experimental STM image obtained by Teichmann[72], who pointed out that the extra positive charge of the donor causes the bands to drop and results in an enhanced tunnel current in the Si atom site. This overcompensates the reduction of the current from the lower height of Si_{Ga}^+ (cf. Fig. 3.7) and corresponding larger distance to the tip. Fig. 3.11 gives the STM image for the negatively charged system at the bias voltage of 2 V. Different from the positively charged case, the Si_{Ga}^- defect on the surface presents a dark feature. As discussed in previous section, the extra negative charge is mainly concentrated on the Si atom for the negatively charged system, and the extra negative charge occupies the dangling bond of the Si atom. The completely occupied dangling bond leads to the small protrusion of the negative Si atom at the positive bias voltage.

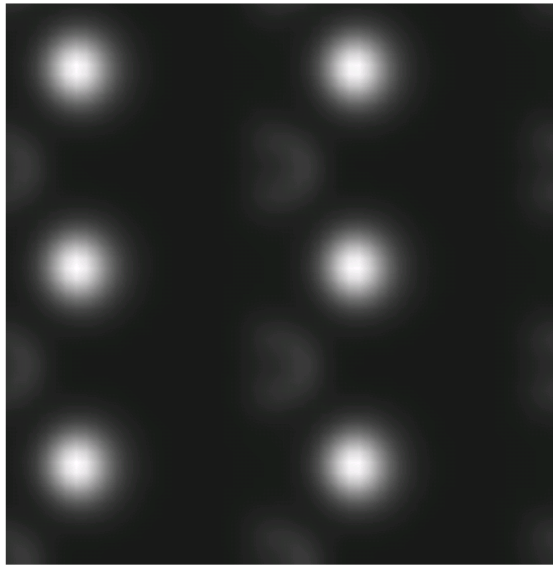


FIGURE 3.8: Calculated STM of clean GaAs(110) at a bias voltage of 2 V for neutral system

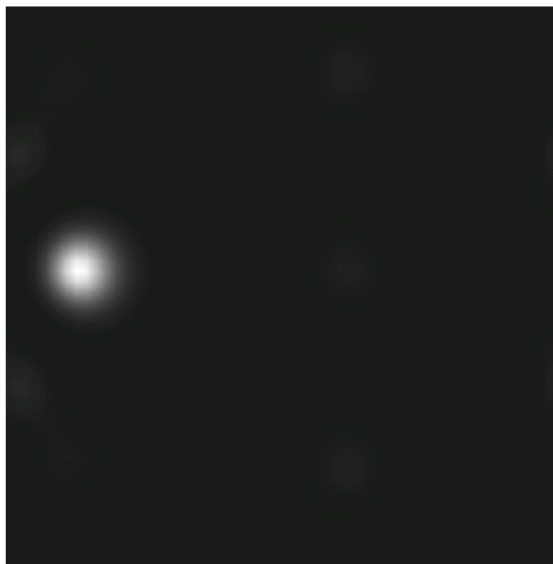


FIGURE 3.9: Calculated STM of Si doped GaAs(110) at a bias voltage of 1 V for positively charged Si(charge state: Si_{Ga}^+)

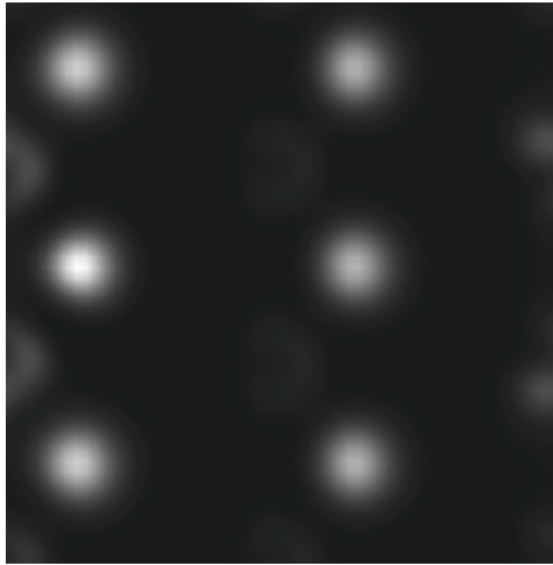


FIGURE 3.10: Calculated STM of Si doped GaAs(110) at a bias voltage of 2 V for positively charged Si(charge state: Si_{Ga}^+)

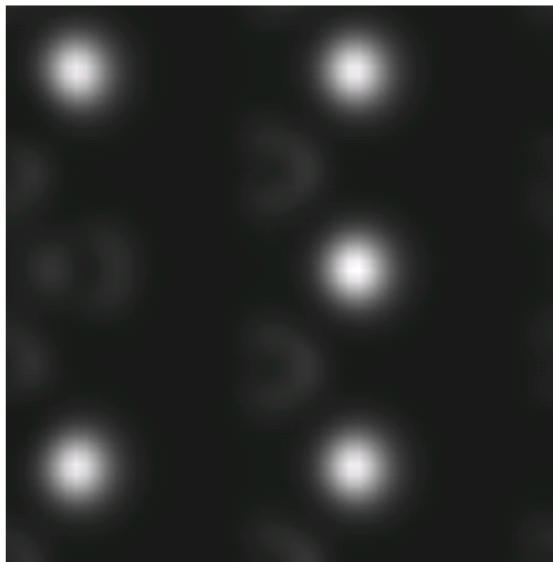


FIGURE 3.11: Calculated STM of Si doped GaAs(110) at a bias voltage of 2 V for negatively charged Si(charge state: Si_{Ga}^-)

3.4 Conclusions

In this work, firstly, we studied the DX center in bulk GaAs and in GaAs(110). In bulk GaAs, DX center is found to be a metastable state, in good agreement with previous calculation of Chadi[51]. However, for surface system, the DX center is found to be highly unstable, and the DX formation energy, which depends on Si position and displacing direction of Si atom in GaAs(110), varies from 0.8 eV to 1.3 eV. Secondly, we found that the extra charge is mainly concentrated on the defect for charged Si:GaAs(110) where Si atoms replace Ga atoms on the GaAs(110) surface. On the surface Si_{Ga}^- defect is stable, while Si_{Ga}^+ defect is unstable. In addition, we also reported the calculated STM images for clean GaAs(110) and the Si:GaAs(110) with different charge states which agree with experimental results. We found that the STM image switches from a bright feature to a dark feature when the charge state of Si_{Ga} defect on the surface changes from positive to negative at the positive bias voltage.

Chapter 4

Many body perturbation theory

4.1 Introduction

So far we have examined how one can justify the reduction of the many-body equation to a single particle problem. This can be done by introducing certain approximations. In this Chapter we show that excited-state properties can be accessed more directly with a purpose built method, the so-called many-body perturbation theory (MBPT) [10]. The one-particle band-structure paradigm itself has been deeply revised by the introduction of the concept of quasi-particles, which provides a proper interpretation of the results of photoemission experiments. Fig. 4.1 gives a schematic illustration. In direct photoelectron spectroscopy a photon with energy $\hbar\omega$ impinges on the sample and ejects an electron, whose kinetic energy E_{kin} is subsequently measured. The binding energy ϵ_i of this electron is given by the difference $\epsilon_i = E_{kin} - \hbar\omega$. In reality, they are correlated through the Coulomb interaction, and the ejection of an electron is always a many-body process. In this general sense ϵ_i equals the difference $\epsilon_i = E_0^N - E_i^{N-1}$ between the total energy E_0^N of the N particle ground state Ψ_0^N and the energy E_i^{N-1} of the $N - 1$ particle state Ψ_i^{N-1} that remains after the emission. Inverse photoelectron spectroscopy is the complementary process: electrons are injected into the sample, and the energy of the emitted photon is measured. The number of electrons in the system thus

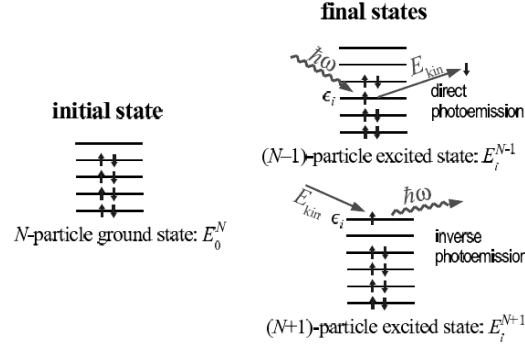


FIGURE 4.1: Schematic illustration of direct and inverse photoelectron spectroscopy. The measured energy difference $E_{kin} - \hbar\omega$ corresponds to $\epsilon_i = E_0^N - E_i^{N-1}$ in direct and $\epsilon_i = E_i^{N+1} - E_0^N$ in inverse photoelectron spectroscopy.

increases from N to $N + 1$, and we can identify $\epsilon_i = E_{kin} - \hbar\omega$ with the difference $\epsilon_i = E_i^{N+1} - E_0^N$ of the many electron systems.

The independent electron picture breaks down due to the strong Coulomb interaction originated from the fictitious single electron concepts. In fact, we can at least retain a nearly-independent particle picture if we consider quasi-particles instead of electrons (or holes). In the case of electron injection into a sample the repulsive Coulomb interaction creates a Coulomb hole around the additional electron (see Fig. 4.2). Analogously, if an electron leaves the system, its Coulomb hole also disappears. Relative to the ground-state N electron system, the addition (removal) of an electron in indirect (direct) photoelectron spectroscopy hence creates (annihilates) an ensemble consisting of the bare electron and its oppositely charged Coulomb hole. This ensemble behaves in many ways like a single-particle and is thus called "quasi-particle". Since the Coulomb hole reduces the total charge of the quasi-particle, the effective interaction between quasi-particles is screened and considerably weaker than the bare Coulomb interaction between electrons.

A better theoretical description of processes involving the ejection or injection of electrons is the many body perturbation theory (MBPT) that links the N particle with the $N \pm 1$ particle systems. In MBPT framework the key variable is the time ordered Green's function $G(\mathbf{r}_1, t_1, \mathbf{r}_2, t_2)$, also called propagator. The Green's

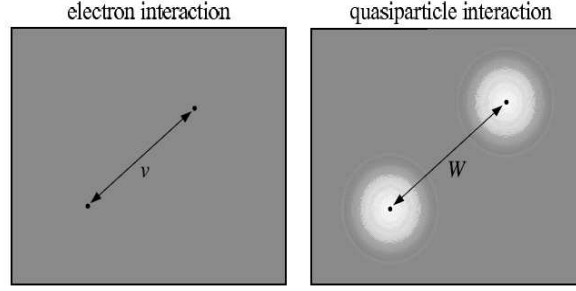


FIGURE 4.2: The coulomb hole around an electron together with the bare electron forms the quasi-particle, as can be seen from the figure in the right. Quasi-particles interact via a weak screened interaction W instead of the strong coulomb interaction v (see the figure in the left).

function contains the excitation energies and excitation lifetimes. Besides, we can directly obtain the ground-state electron density, the expectation values of one-particle operators and the ground-state total energy from it. The Green's function is hence capable of giving access to the same observables as the ground-state electron density. In contrast to the DFT expression $E(\rho)$, the functional $E(G)$ is even known exactly[73]. While the Green's function contains much more information than the electron density, it is also a more complicated function and thus rarely applied to ground-state properties.

4.2 The Green's function

In this section we introduce the time-ordered Green's function and examine its properties. The Green's function describes the probability amplitude for the propagation of an electron(hole) from position \mathbf{r}_1 at time t_1 to the position \mathbf{r}_2 at time t_2 . For the propagation of an additional electron from (\mathbf{r}_2, t_2) to (\mathbf{r}_1, t_1) in a many electron system, the time-ordered one-particle Green's function at zero temperature is defined as:

$$G^e(1, 2) = -\frac{i}{\hbar} \langle N, 0 | \hat{\psi}(1) \hat{\psi}^+(2) | N, 0 \rangle \theta(t_1 - t_2) \quad (4.1)$$

where $|N\rangle$ is the exact ground state of the interacting system. The $\hat{\psi}$ and $\hat{\psi}^+$ operators are annihilation and creation field operators within Heisenberg picture(they

contain the whole time dependence), respectively. Index 1 is shorthand for position, time and spin variables $(\mathbf{r}_1, t_1, \sigma_1)$. $\theta(t_1 - t_2)$ is the Heaviside step function defined by

$$\theta(t_1 - t_2) = \begin{cases} 1, & t_1 > t_2 \\ 0, & t_1 < t_2 \end{cases} \quad (4.2)$$

Similarly, we have the green's function

$$G^h(2, 1) = -\frac{i}{\hbar} \langle N, 0 | \hat{\psi}^+(2) \hat{\psi}(1) | N, 0 \rangle \theta(t_2 - t_1) \quad (4.3)$$

for the propagation of an additonal hole from (\mathbf{r}_1, t_1) to (\mathbf{r}_2, t_2) . For convenience, we combine the two expressions in one time-ordered Green's function

$$\begin{aligned} G(1, 2) &= G^e(1, 2) - G^h(2, 1) = -\frac{i}{\hbar} \langle N, 0 | \hat{T}[\hat{\psi}(1) \hat{\psi}^+(2)] | N, 0 \rangle \\ &= \begin{cases} -\frac{i}{\hbar} \langle N, 0 | \hat{\psi}(1) \hat{\psi}^+(2) | N, 0 \rangle, & t_1 > t_2 \\ \frac{i}{\hbar} \langle N, 0 | \hat{\psi}^+(2) \hat{\psi}(1) | N, 0 \rangle, & t_1 < t_2 \end{cases} \end{aligned} \quad (4.4)$$

where \hat{T} stands for the Wick time-ordering operator, which has the effect of ordering the operator with largest time on the left. The physically intuitive interpretation of the time-ordered Green's function can be emphasized from Eqs.(4.4). When t_1 is larger than t_2 , the Green's function describes the probability to find an electron in \mathbf{r}_1 with spin σ_1 at time t_1 , when an electron was added in \mathbf{r}_2 with spin σ_2 at time t_2 to the system in its ground-state. Instead, when t_1 is smaller than t_2 , the Green's function describes the probability to find a hole in \mathbf{r}_2 with spin σ_2 at time t_2 , when an electron was previously removed in \mathbf{r}_1 with spin σ_1 at time t_1 .

The one-particle Green's function G contains a great deal of useful information. According to G , we can calculate also the ground-state expectation value of any one-particle operator and the ground-state energy of the system. For example, the expectation value of electron denstiy $\rho(\mathbf{r}, t)$ can be expressed in terms of the Green's function as

$$\rho(\mathbf{r}, t) = \langle N, 0 | \hat{\phi}^+(\mathbf{r}, t) \hat{\phi}(\mathbf{r}, t) | N, 0 \rangle = -i\hbar G(\mathbf{r}, t, \mathbf{r}, t + \eta) \quad (4.5)$$

Here η is an infinitesimal positive number. It serves only to enforce the correct order of the field operators. Its unit should always be clear from the context; presently it is an infinitesimally small time.

Similarly, the Galitskii-Migdal formula[74] gives the ground-state total energy as a function of the one-particle Green's function only:

$$E = -\frac{i\hbar}{2} \sum_{\sigma_1, \sigma_2} \delta_{\sigma_1, \sigma_2} \int d\mathbf{r}_1 \lim_{r_2 \rightarrow r_1} \lim_{t_2 \rightarrow t_1^+} [i \frac{\partial}{\partial t_1} + h_0(\mathbf{r}_1)] G(1, 2) \quad (4.6)$$

where $h_0 = -\nabla^2/2 + v_{ext}$ is the one particle Hamiltonian. The one-particle excitation spectrum carried by the Green's function will be made obvious in the next section.

4.3 Lehmann representation

For convenience, I will employ atomic units(i.e. $\hbar = 1$) in the next part. The so-called Lehmann representation of the Green's function permits one to get insight into its physical meaning. It shows that the Green's function is closely related to the single-particle excitation energies of the system. If we insert the closure relation $\sum_i |N \pm 1, i\rangle \langle N \pm 1, i| = 1$ between the two field operators in Eqs. (4.4), where $\{|N \pm 1, i\rangle\}$ is the complete set of state vectors of the $N \pm 1$ particle system, then the Green's function depends only on the difference $\tau = t_1 - t_2$ in the case of time-independent Hamiltonian:

$$\begin{aligned} iG(\mathbf{r}_1, \mathbf{r}_2, \tau) &= \theta(\tau) \sum_i \langle N, 0 | \hat{\psi}(1) | N+1, i \rangle \langle N+1, i | \hat{\psi}^+(2) | N, 0 \rangle \\ &- \theta(-\tau) \sum_i \langle N, 0 | \hat{\psi}^+(2) | N-1, i \rangle \langle N-1, i | \hat{\psi}(1) | N, 0 \rangle \end{aligned} \quad (4.7)$$

In the above equation, if $\tau > 0$, only states with $N+1$ survive; if $\tau < 0$, only states with $N-1$ survive. The index i sums over all states of a given particle number.

For positive time ($\tau > 0$), the excitation energy of a system previously in its ground-state that, due to an additional electron, is now in the excited state i of $N + 1$, is given by

$$\epsilon_i^{N+1} = E_{N+1,i} - E_{N,0} \quad (4.8)$$

Similarly, for negative time $\tau < 0$, a hole energy in its excited state is given by

$$\epsilon_i^{N-1} = E_{N,0} - E_{N-1,i} \quad (4.9)$$

The time dependence of the matrix elements can be handled by introducing the Schrödinger picture:

$$\hat{\psi}_H(\mathbf{r}_1, t_1) = e^{i\lambda t_1} \hat{\psi}_S(\mathbf{r}_1) e^{-i\lambda t_1} \quad (4.10)$$

where λ corresponds to the excitation energy. $\hat{\psi}_H(\mathbf{r}_1, t_1)$ is field operator in the Heisenberg picture. $\hat{\psi}_S(\mathbf{r}_1)$ is the field operator in the Schrödinger picture. Therefore, the Green's function takes the following form

$$\begin{aligned} iG(\mathbf{r}_1, \mathbf{r}_2, \tau) &= \theta(\tau) \sum_i \langle N, 0 | \hat{\psi}_S | N+1, i \rangle \langle N+1, i | \hat{\psi}_S^\dagger(2) | N, 0 \rangle e^{-i\epsilon_i^{N+1}\tau} \\ &- \theta(-\tau) \sum_i \langle N, 0 | \hat{\psi}_S^\dagger(2) | N-1, i \rangle \langle N-1, i | \hat{\psi}_S(1) | N, 0 \rangle e^{-i\epsilon_i^{N-1}\tau} \end{aligned} \quad (4.11)$$

The minimum energy of the lowest conduction state is given by

$$\epsilon_c = E_{N+1,0} - E_{N,0} \quad (4.12)$$

Alternatively, the minimum energy of the highest valence state, which is required to remove an electron is given by

$$\epsilon_v = E_{N,0} - E_{N-1,0} \quad (4.13)$$

If the system is semiconductor or insulator, the band gap E_g is defined as the energy difference between the conduction and valence states

$$E_g = \epsilon_c - \epsilon_v = E_{N+1,0} + E_{N-1,0} - 2E_{N,0} \quad (4.14)$$

If the system is metallic system, the energy $\epsilon_c = \epsilon_v$ is the chemical potential μ . For semiconductor or insulator system, the chemical potential μ lies somewhere inside the gap. Therefore, if $\epsilon_i > \mu$, the energy is defined as $E_{N+1,i} - E_N$, if $\epsilon_i < \mu$, the energy is given by $E_N - E_{N-1,i}$. Consequently, we can define the so-called Lehmann amplitudes $f_i(\mathbf{r})$,

$$f_i(\mathbf{r}) = \begin{cases} \langle N, 0 | \hat{\psi}_S(\mathbf{r}) | N+1, i \rangle, & \epsilon_i > \mu \\ \langle N-1, i | \hat{\psi}_S(\mathbf{r}) | N, 0 \rangle, & \epsilon_i < \mu \end{cases} \quad (4.15)$$

The Green's function is then expressed as

$$iG(\mathbf{r}_1, \mathbf{r}_2, \tau) = \sum_i [\theta(\tau)\theta(\epsilon_i - \mu) - \theta(-\tau)\theta(\mu - \epsilon_i)] f_i(\mathbf{r}_1) f_i^*(\mathbf{r}_2) e^{-i\epsilon_i \tau} \quad (4.16)$$

Now the sum \sum_i is over electron and hole states. Fourier transformation of Eqs. (4.16) to the frequency axis using the Fourier transform of the Heaviside step function

$$\theta(\pm\tau) = \mp \frac{1}{2\pi i} \int_{-\infty}^{\infty} \frac{e^{-i\omega\tau}}{\omega \pm i\eta} d\omega \quad (4.17)$$

for $\eta \rightarrow 0^+$, one obtains the Lehmann representation of one particle Green's function

$$G(\mathbf{r}_1, \mathbf{r}_2, \omega) = \sum_i \frac{f_i(\mathbf{r}_1) f_i^*(\mathbf{r}_2)}{\omega - \epsilon_i + i\eta \text{sign}(\epsilon_i - \mu)} \quad (4.18)$$

The Green's function has poles at the true many particle excitation energies $\epsilon_i^{N\pm 1}$. The poles lie slightly above the real axis for frequencies below μ and slightly below the real axis for frequencies above μ , as shown in Fig. 4.3.

It must be emphasised that the concept of a quasi-particle is an interpretation of the experimentally observed many body spectrum in terms of one-particle-like excitations. The quasi-particle concept is used for various types of excitations, for instance electron-hole pairs (excitons) or vibrations (phonons). In the context of one-electron excitations, these quasiparticles can be viewed as a hole or an electron surrounded by its polarisation cloud. Quasi-particles have a finite lifetime due to dephasing, i.e., decay into other quasi-particles, because the true many-body

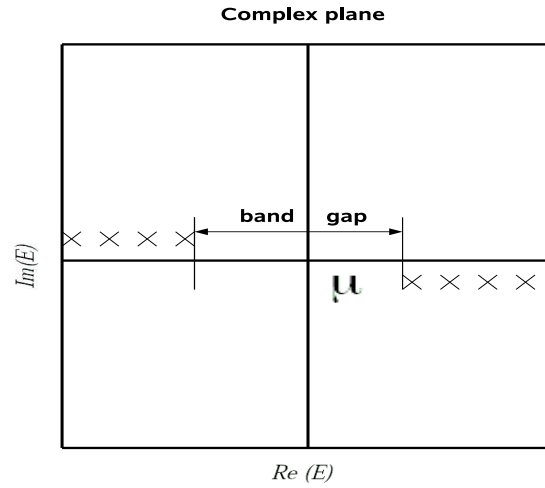


FIGURE 4.3: In the complex plane, the Green's function has poles(crosses) only along the real axis. The poles are located above the real axis for frequencies lower than the chemical potential μ , and below for frequencies greater than μ

eigenstates may contribute to more than one quasi-particle. In other words, a quasi-particle is not an eigenstate of the system but a superposition of eigenstates.

4.4 Spectral function

We apply the following relation

$$\lim_{\eta \rightarrow 0^+} \frac{1}{x + i\eta} = \wp \frac{1}{x} - i\pi\delta(x) \quad (4.19)$$

to the Eqs.(4.18), where $\wp \frac{1}{x}$ is the principal value of $\frac{1}{x}$, one can obtain the imaginary part of G :

$$\text{Im}G(\mathbf{r}_1, \mathbf{r}_2, \omega) = \pi \text{sign}(\mu - \epsilon_i) \sum_i f_i(\mathbf{r}_1) f_i^*(\mathbf{r}_2) \delta(\omega - \epsilon_i) \quad (4.20)$$

The spectral function is defined as

$$A(\mathbf{r}_1, \mathbf{r}_2, \omega) = \frac{1}{\pi} \text{sign}(\omega - \mu) \text{Im}G(\mathbf{r}_1, \mathbf{r}_2, \omega) = \sum_i f_i(\mathbf{r}_1) f_i^*(\mathbf{r}_2) \delta(\omega - \epsilon_i) \quad (4.21)$$

The spectral function A is a positive function of ω . From this expression one can immediately realize that the spectral function is hence the quantity directly related to the one-particle excitations, described by the Lehmann amplitudes f_i and the electron addition and removal energies ϵ_i . Moreover, the spectral function allows us to rewrite the Eqs. (4.18) as an integral over frequencies

$$G(\mathbf{r}, \mathbf{r}', \omega) = \int_{-\infty}^{\mu} \frac{A(\mathbf{r}, \mathbf{r}', \omega)}{\omega - \omega' - i\eta} d\omega' + \int_{\mu}^{\infty} \frac{A(\mathbf{r}, \mathbf{r}', \omega)}{\omega - \omega' + i\eta} d\omega' \quad (4.22)$$

Since

$$\begin{aligned} \sum_i f_i(\mathbf{r}_1) f_i^*(\mathbf{r}_2) &= \sum_i \langle N | \psi^+(2) | N-1, i \rangle \langle N-1, i | \psi(\mathbf{r}_1) | N \rangle \\ &+ \sum_i \langle N | \psi(\mathbf{r}_1) | N+1, i \rangle \langle N+1, i | \psi^+(\mathbf{r}_2) | N \rangle \\ &= \langle N | [\psi(\mathbf{r}_1), \psi^+(\mathbf{r}_2)]_+ | N \rangle \\ &= \delta(\mathbf{r}_1 - \mathbf{r}_2) \end{aligned} \quad (4.23)$$

The spectral function satisfies the normalization condition:

$$\int_{-\infty}^{\infty} A(\mathbf{r}_1, \mathbf{r}_2, \omega) d\omega = \delta(\mathbf{r}_1 - \mathbf{r}_2) \quad (4.24)$$

and from its diagonal one can extract important information such as the electronic density:

$$\rho(\mathbf{r}) = \int_{-\infty}^{\mu} A(\mathbf{r}, \mathbf{r}, \omega) d\omega \quad (4.25)$$

Since the connection between Green's function and spectral function has been demonstrated, now I will show how to calculate the spectra of photoemission experiments. In a photoemission experiment, the detector measures electrons that are extracted from a solid due to an incident photon, and the experimental observable in PES is the photocurrent. It is given by

$$I \sim \int d\mathbf{r}_1 \int d\mathbf{r}_2 \phi_{pe}^*(\mathbf{r}_1) \delta H(\mathbf{r}_1) A(\mathbf{r}_1, \mathbf{r}_2, E) \delta H(\mathbf{r}_2) \phi_{pe}(\mathbf{r}_2) \quad (4.26)$$

where \mathbf{r} comprises a spatial and a spin coordinate. ϕ_{pe} is a time-reversed damped LEED state describing the photoelectron that reaches the detector. $\delta H(\mathbf{r})$ is the perturbing field that excites the electron. Inserting Eqs.(4.25) into Eqs.(4.26) we obtain Fermis golden rule expression

$$I \sim \sum_i |\langle \phi_{pe} | \delta H | f_i \rangle|^2 \delta(E - \epsilon_i) \quad (4.27)$$

The magnitude of the transition matrix elements can vary considerably between different states and even become zero, in particular when the symmetry of the system defines selection rules. The photocurrent therefore reflects only a somewhat distorted picture of the electron density of states (DOS)

$$N^{el}(E) = \sum_i \delta(E - \epsilon_i) = \int d^3\mathbf{r} A(\mathbf{r}, \mathbf{r}, E) \quad (4.28)$$

4.5 Hedin's equations and the GW approximation

With the equation of motion for the Heisenberg creation and annihilation field operators $\hat{\psi}$ and $\hat{\psi}^\dagger$ (see Appendix(A.1)), the one-particle Green's function depends on the two particle Green's function, and is given by

$$[i \frac{\partial}{\partial t_1} - h_0(1)] G(1, 2) = \delta(1, 2) - i \int d3 v(1, 3) G_2(1, 3, 2, 3^+) \quad (4.29)$$

where $h_0 = -\frac{\nabla^2}{2} + V_{ext}$ is the one-particle term of the Hamiltonian. The two-particle Green's function G_2 describes the creation and annihilation of pairs of particles. Further deriving the equation of motion for G_2 , one would obtain higher orders of Green's function $G_2, G_3 \dots$. The physical meaning of this is that the electron hole pair created by the propagation of the first electron would induce a cascade of other electron-hole pairs. Therefore, the system of equations will become more and more complex with the increase of higher order Green's function. In fact, if one only needs to know the one particle Green's function, a good approximation,

which allows one to express the two particle Green's function in terms of one-particle ones, can be introduced. Similarly, if somebody want to calculate the absorption spectra(involving two-particle Green's function), one needs to find an approximation to replace the three-paritcle Green's function with the two-particle ones. This is the fundamental idea of many body perturbation theory, which provides a scheme to systematically find approximations for the higher-number of particle Green's functions in terms of the lower ones.

In this thesis, I am interested in the one particle Green's function. Within the idea of many body perturbation theory(MBPT), it allows one to get rid of the two particle Green's function in the Eqs.(4.29), while retaining only the terms dependent on one-particle Green's functions. This is achieved by introducing the Schwinger derivative technique[75]. The main physical idea of Schwinger is that the same polarization of the system could be obtained by a small time-dependent perturbing external potential $U_{per}(1, 2)$, which, in the spirit of the linear-response theory, will be made vanish at the end of the derivation. The following expansion presents the variation of G with respect to perturbation U_{per} (See Appendix(A.2) for detailed information):

$$\frac{\delta G(1, 2)}{\delta U_{per}(3)} = G(1, 2)G(3, 3^+) - G_2(1, 3, 2, 3^+) \quad (4.30)$$

Hence the two particle Green's function is expressed as terms of one particle Green's function by introducing the variation of G with respect to perturbation U_{per} . Inserting above equation in Eqs.(4.29), we can obtain:

$$[i\frac{\partial}{\partial t_1} - h_0(1) + i \int d3v(1, 3)G(3, 3^+)]G(1, 2) - i \int d3v(1, 3)\frac{\delta G(1, 2)}{\delta U_{per}(3)} = \delta(1, 2) \quad (4.31)$$

In the above equations $G(3, 3^+)$ is the electronic denstiy $\rho(3)$, so the Hartree potential $V_H(1)$ is described by the term $-i \int d3v(1, 3)G(3, 3^+)$. The self energy is defined by the last term of Eqs.(4.31)

$$i \int d3v(1, 3)\frac{\delta G(1, 2)}{\delta U_{per}(3)} = \int d3\Sigma(1, 3)G(3, 2) \quad (4.32)$$

The self energy describes all the effects of exchange and correlation in the system(i.e. all the interacting effects are includes in the self energy Σ). With the introduction of the Hartree potential V_H and the self energy Σ , Eqs.(4.31) can be then rewritten as:

$$[i\frac{\partial}{\partial t_1} - h_0(1) - V_H(1)]G(1, 2) = \delta(1, 2) + \int d3\Sigma(1, 3)G(3, 2) \quad (4.33)$$

The above equation is the so-called Dyson equation. The equation of motion of the non-interacting Green's function G_0 is given by

$$[i\frac{\partial}{\partial t_1} - h_0(1) - V_H(1)]G_0(1, 2) = \delta(1, 2) \quad (4.34)$$

The above equation shows that non-interacting Green's function is formally the functional inverse of the operator $(\omega - h_0 - V_H(1))$ in frequency space. So we can rewrite the Dyson Eqs.(4.33) as

$$G(1, 2) = G_0(1, 2) + \int d34G_0(1, 3)\Sigma(1, 3)G(4, 2) \quad (4.35)$$

In order to evaluate the self energy, by introducing $\int d5G^{-1}(4, 5)G(5, 2) = \delta(4, 2)$, the Eqs.(4.33) can be rewritten as

$$[i\frac{\partial}{\partial t_1} - h_0(1) - V_H(1)]G(1, 2) - i \int d345v(1^+, 3)\frac{\delta G(1, 4)}{\delta U_{per}(3)}G^{-1}(5, 2) = \delta(1, 2) \quad (4.36)$$

Inserting the relation

$$\frac{\delta F(1, 2)}{\delta G(3)} = - \int d45F(1, 4)\frac{\delta F^{-1}(4, 5)}{\delta G(3)} \quad (4.37)$$

to the last term in the left-hand-side of Eqs.(4.36), yields the self energy operator as

$$\Sigma(1, 2) = -i \int d34v(1^+, 3)G(1, 4)\frac{\delta G^{-1}(4, 2)}{\delta U_{per}(3)} \quad (4.38)$$

Now let us insert the chain rule

$$\frac{\delta F[G[H]](1)}{\delta H(2)} = \int d3\frac{\delta F[G](1)}{\delta G(3)}\frac{\delta G[H](3)}{\delta H(2)} \quad (4.39)$$

via V in the definition of self energy of Eqs.(4.38):

$$\Sigma(1, 2) = -i \int d345 v(1^+, 3) G(4, 2) \frac{\delta G^{-1}(1, 4) \delta V(5)}{\delta U_{per}(3) \delta V(5)} \quad (4.40)$$

where V is the local classical potential, and defined as the sum of the external perturbation and the Hartree potential:

$$V(1) = U(1) - i \int d2 v(1, 2) G(2, 2^+) \quad (4.41)$$

With the introduction of the irreducible vertex $\tilde{\Gamma}$

$$\tilde{\Gamma}(1, 2, 3) = -\frac{G^{-1}(1, 2)}{\delta V(3)} \quad (4.42)$$

and the definition of time ordered inverse dielectric function:

$$\epsilon^{-1}(1, 2) = \frac{\delta V(1)}{\delta U(2)} \quad (4.43)$$

In addition, one can introduce W , the dynamical screened Coulomb interaction, and defined by

$$W(1, 2) = \int d3 v(1, 3) \epsilon^{-1}(3, 2) \quad (4.44)$$

Finally, the self energy is expressed as

$$\Sigma(1, 2) = i \int d34 G(1, 4) W(3, 1^+) \tilde{\Gamma}(4, 2, 3) \quad (4.45)$$

Now let us further evaluate $\tilde{\Gamma}$, using Dyson equation $G^{-1} = G_0^{-1} - V - \Sigma$, $\tilde{\Gamma}$ is easily obtained:

$$\begin{aligned} \tilde{\Gamma}(1, 2, 3) &= \delta(1, 2) \delta(1, 3) + \frac{\delta \Sigma(1, 2)}{\delta V(3)} \\ &= \delta(1, 2) \delta(1, 3) + \int d45 \frac{\delta \Sigma(1, 2) G(4, 5)}{G(4, 5) \delta V(3)} \\ &= \delta(1, 2) \delta(1, 3) + \int d4567 \frac{\delta \Sigma(1, 2)}{\delta G(4, 5)} \\ &\times G(4, 6) G(7, 5) \tilde{\Gamma}(6, 7, 3) \end{aligned} \quad (4.46)$$

It should be noted that the chain rule (4.39) and derivative of the iverse (4.37) were used to obtain the above final expression of $\tilde{\Gamma}$. Inserting Eqs.(4.41) to Eqs. (4.43), the ϵ^{-1} can be then rewritten as

$$\begin{aligned}\epsilon^{-1}(1, 2) &= \frac{\delta(U - i \int d3v(1, 3)G(3, 3^+))}{\delta U(2)} \\ &= \delta(1, 2) + \int d3v(1, 3)P(3, 2)\end{aligned}\quad (4.47)$$

where

$$P(1, 2) = -i \frac{\delta G(1, 1^+)}{\delta U(2)} \quad (4.48)$$

is the reducible polarizability of the system. "reducible" means that the variation of $G(1, 1^+)$ is related to the bare external potential U . Alternatively, one can also define an irreducible polarizability

$$\tilde{P}(1, 2) = -i \frac{\delta G(1, 1^+)}{\delta V(2)} \quad (4.49)$$

The link between the two polarizabilities is made by using once more the chain rule (4.39) and the relation (4.47):

$$P(1, 2) = \tilde{P}(1, 2) + \int d34 \tilde{P}(1, 3)v(3, 4)P(4, 2) \quad (4.50)$$

By using the relation (4.37), one can obtain

$$\tilde{P}(1, 2) = -i \int d34 G(1, 3)G(4, 1)\tilde{\Gamma}(3, 4, 2) \quad (4.51)$$

Therefore, the problem of computing the interacting one-particle Green's function can be cast into five coupled equations in the framework of MBPT[10]. These equations are known as Hedin's equation, and expressed as:

$$\Sigma(1, 2) = i \int d34 G(1, 4)W(3, 1^+)\tilde{\Gamma}(4, 2, 3) \quad (4.52)$$

$$G(1, 2) = G_0(1, 2) + \int d34 G_0(1, 3)\Sigma(1, 3)G(4, 2) \quad (4.53)$$

$$\tilde{P}(1, 2) = -i \int d34 G(1, 3) G(4, 1) \tilde{\Gamma}(3, 4, 2) \quad (4.54)$$

$$W(1, 2) = v(1, 2) + \int d3d4 v(1, 3) \tilde{P}(3, 4) W(4, 2) \quad (4.55)$$

$$\begin{aligned} \tilde{\Gamma}(1, 2, 3) &= \delta(1, 2) \delta(1, 3) + \int d4567 \frac{\delta \Sigma(1, 2)}{\delta G(4, 5)} \\ &\times G(4, 6) G(7, 5) \tilde{\Gamma}(6, 7, 3) \end{aligned} \quad (4.56)$$

These equations involve the independent particle and full Green's function G_0 and G , the polarisation \tilde{P} , the bare and screened interaction v and W , the self-energy Σ , and the so-called vertex function Γ . For simplicity, we use the notation $1 = (\mathbf{r}_1, t_1, \sigma_1)$ for every pair of one spatial and one temporal variable. 1^+ indicates that the time argument has been increased by an infinitesimally small, positive amount.

Hedin's scheme provides directly a set of equations in terms of G and W and in particular an expansion of Σ in terms of W . Expansions in terms of v are known to fail: for instance, the second order term in v for the correlation energy of an homogenous electron gas is infinite[76]. The idea of using an interaction, namely W , weaker than v as coupling constant of the perturbative expansion was already introduced by Hubbard[77]. The screened Coulomb interaction is intuitively smaller than the bare one, as the interaction between two electrons being part of a system is reduced by the dielectric constant of the medium or, in other words, by the screening due to all the other electrons of the system.

This set of coupled equations cannot be solved directly due to the presence of the functional derivative in the definition of the vertex function. However, it is amenable to physically meaningful approximations. We will describe the equations in more detail before coming to the most important approximation, the GW approximation.

Eqs.(4.53) and Eqs.(4.55) have the same structure, known as Dyson equation. Σ represents the complicated correlation effects of a many particle system. G_0 and G correspond to the propagation of electrons or holes in the system, whereas v and W

can be interpreted as propagators for the quantum particles of the electric field. Thus, the polarisation function plays the role of the self-energy for the electric field particles, changing the bare Coulomb interaction into the screened Coulomb interaction.

Correspondingly, Eqs.(4.52) and Eqs.(4.54) have the same structure. They contain two propagators connected to one point of the interaction kernel and are connected via the vertex function to the other point of the kernel. This vertex function is a three-point kernel that describes all possible ways how a (dressed) electron is scattered when a (screened) electric field particle is created or annihilated.

When a charge is added to an electronic system, the first effect is the polarization of the system. By neglecting the electron-hole interaction(short range effect) in the electron-hole pairs that constitute the polarization and the exchange correlation effects in the induced interaction between the extra charge and the polarization charge of the system,(i.e. approximating the vertex function by its first term: $\tilde{\Gamma}(1, 2, 3) = \delta(1, 2)\delta(2, 3)$) one finds the GW approximation introduced by Hedin[11]

$$\Sigma(1, 2) = iG(1, 2)W(1^+, 2) \quad (4.57)$$

The GW approximation corresponds to neglecting vertex effects, i.e. to taking for the vertex the simplest expression:

$$\tilde{\Gamma}(1, 2, 3) = \delta(1, 2)\delta(2, 3) \quad (4.58)$$

Similarly, If one neglects the vertex effects in the polarization P , the random-phase approximation(RPA) for \tilde{P} is obtained:

$$\tilde{P}(1, 2) = -iG(1, 2)G(2, 1) \quad (4.59)$$

Even using the GW approximation to the self-energy, the Dyson equation is still a self-consistent equation in G that one could solve iteratively. However, this is rarely done. Instead of using the full Green's function G , the non-interacting

Green's function G_0 is used as a first approximation in Eqs.(4.57) and Eqs.(4.59). We use this approach, denoted as G_0W_0 in all the actual calculations.

4.6 Quasi-particle equations

In reality, the quasi-particle quantities are not calculated from the Dyson Eqs.(4.33). Let us reformulate the Dyson Eqs.(4.33). The Fourier transform of the Dyson Eqs.(4.33) gives:

$$[\omega - h_0(1) - V_H(1)]G(1, 2, \omega) - \int d3 \Sigma(1, 3, \omega)G(3, 2, \omega) = \delta(1, 2) \quad (4.60)$$

A practical solution of this equation is obtained by using a the biorthonormal representation[78],

$$G(\mathbf{r}_1, \mathbf{r}_2, \omega) = \sum_{\lambda} \frac{\phi_{\lambda}(\mathbf{r}_1, \omega) \tilde{\phi}_{\lambda}^*(\mathbf{r}_2, \omega)}{\omega - E_{\lambda}(\omega)} \quad (4.61)$$

with $\phi_{\lambda}(\mathbf{r}_1, \omega)$ and $\tilde{\phi}_{\lambda}^*(\mathbf{r}_2, \omega)$ being solutions of

$$[h_0(\mathbf{r}_1) + V_H(\mathbf{r}_1)]\phi_{\lambda}(\mathbf{r}_1, \omega) + \int d\mathbf{r}_2 \Sigma(\mathbf{r}_1, \mathbf{r}_2, \omega)\phi_{\lambda}(\mathbf{r}_2, \omega) = E_{\lambda}(\omega)\phi_{\lambda}(\mathbf{r}_1, \omega) \quad (4.62)$$

and

$$[h_0(\mathbf{r}_1) + V_H(\mathbf{r}_1)]\tilde{\phi}_{\lambda}^*(\mathbf{r}_1, \omega) + \int d\mathbf{r}_2 \Sigma'(\mathbf{r}_1, \mathbf{r}_2, \omega)\tilde{\phi}_{\lambda}^*(\mathbf{r}_2, \omega) = E_{\lambda}^*(\omega)\tilde{\phi}_{\lambda}^*(\mathbf{r}_1, \omega) \quad (4.63)$$

where $\Sigma'(\mathbf{r}_1, \mathbf{r}_2, \omega) = [\Sigma(\mathbf{r}_2, \mathbf{r}_1, \omega)]^*$. In the Lehmann representation in the thermodynamic limit one has a series of infinite close-lying poles on the real axis that merge to form a branch-cut. An alternative representation is given by a complex pole(i.e. the quasi-particle energy). A schematic comparison between the Lehmann representation and quasi-particle pole representation is given in Fig.4.4. The quasi-particle concept is based on the assumption that the full energy dependence of $G(\mathbf{r}_1, \mathbf{r}_2, \omega)$ can be described approximately with some well defined

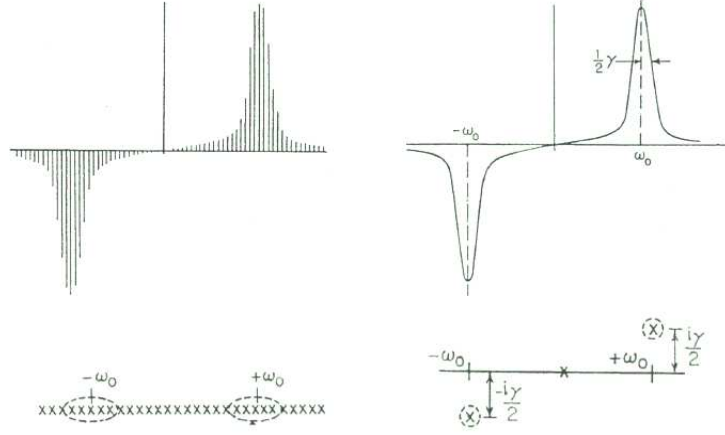


FIGURE 4.4: Schematic comparison for the Lehmann representation(left) and the quasi-particle pole representation(right). A series of close lying poles on the real axis in the Lehmann representation can be equivalently represented by a single pole in the complex plane. The real part of the pole corresponds to the the position of the peak, which represents the quasi-particle energies. The imaginary part of pole in the complex plane gives the width of peak, which is connected to the lifetime of excitation [79].

complex poles

$$G(\mathbf{r}_1, \mathbf{r}_2, \omega) \approx \sum_{\lambda} \frac{\phi_{\lambda}(\mathbf{r}_1, \epsilon_{\lambda}^{QP}) \tilde{\phi}_{\lambda}^*(\mathbf{r}_2, \epsilon_{\lambda}^{QP})}{\omega - \epsilon_{\lambda}^{QP}} \quad (4.64)$$

with the quasi-particle energies defined by

$$\epsilon_s^{QP} = E_{\lambda}(\epsilon_s^{QP}) \quad (4.65)$$

In this way, one finds the quasi-particle equation:

$$[h_0(\mathbf{r}_1) + V_H(\mathbf{r}_1)]\phi_s(\mathbf{r}_1) + \int d\mathbf{r}_2 \Sigma(\mathbf{r}_1, \mathbf{r}_2, E_s^{QP})\phi_s(\mathbf{r}_2) = E_s\phi_s(\mathbf{r}_1) \quad (4.66)$$

Strictly speaking, this equation has, in general, no real-valued solutions, because the nonlocal, frequency dependent potential Σ is, in general, not hermitian.

Chapter 5

Quasi-particle bandstructures and lifetimes in metallic systems

5.1 Introduction

In the last decade the electronic band structures and electron scattering in noble metals have been studied using experimental techniques[8, 80, 81] and a variety of theoretical methods [13, 14, 82, 83]. The GW approximation derived by Hedin in 1965 [11] has become the state-of-the-art approach to study the quasi-particle band-structures and lifetimes [13, 14, 82, 83] of a large number of systems going beyond the well-known Density Functional Theory(DFT).

It is well known that the usual local density approximation (LDA) [3] underestimates the band gaps for the semiconductors, which can be improved by using the so-called scissors operator(i.e. by rigidly shifting the conduction bands to higher energy). However, for the noble metals the quasiparticle corrections are both k-point and band dependent [13], so a scissors operator is useless. Another important drawback of the traditional DFT is that the exchange-correlation part is real, and hence infinit lifetime, whereas the lifetime of excited electrons and holes in metals is known to be finite.

A significant numerical problem when performing GW calculations for noble metals is given by the issue of basis set expansion. The presence of strongly localized d bands and the inclusion of semicore states for Cu and Ag make the Ab-initio calculations based on a plane wave expansion basis much heavier than those for semiconductors. For the case of Cu, plane waves with cutoff energy of at least 160 Ry (about 2700 plane waves) are needed for convergence [13]. GW calculations need larger computational requirements than DFT, in particular for the calculations of surfaces and bulk solids with occupied d orbitals [84]. It turned out that a Gaussian-orbital basis set instead of using plane-wave Fourier representations of the relevant two-point functions is efficient for the evaluation of GW self-energy corrections [84]. Another common simplification, i.e. the Plasmon-pole approximation(RPA) of the frequency dependence of $\epsilon(\omega)$ should be avoided for the GW calculation of Cu [13], which means that the dielectric matrix $\epsilon_{GG'}^{-1}(q; \omega)$ should be computed explicitly by performing the energy integral numerically. This imposes further heavy CPU load for the GW calculation with plane waves.

Most previous lifetime calculations are based on the DFT band energies [24, 25, 26, 27, 82]. Furthermore, the so-called 'energy shell' approximation with renormalization factor equal to unity was employed to obtain the imaginary part of the quasi-particle energies. In fact, the renormalization factor for noble metals varies between 0.57 and 0.77 [83]. The renormalization factor directly changes the quasiparticle energies, and also changes the quasiparticle lifetimes. Because of the large discrepancies between DFT band energies and experiments, the calculated lifetimes based on the DFT energies are not consistent with experiment. In order to compare the calculated lifetimes based on the DFT band energies with experiments, the DFT band energies have to be shifted rigidly(which, as mentioned above, is questionable in metals).

So far the quantitative understanding of excitation mechanisms and their relaxation channels is far from being complete. In many cases the GW approach correctly reproduces the changes of experimental electron relaxation times with the excitation energy [87]. However, large discrepancies between theoretical and experimental results occur for d holes in Cu and Ag. The lifetimes of holes at

the top of d bands as calculated in Ref. [82, 83] appear to be almost twice longer than the experimentally derived lifetimes. Ref. [86] argued that due to the shifted d bands in the DFT calculation the available phase space for Auger recombination is underestimated yielding calculated lifetimes which are too long. However, Ref. [83, 87] conclude that the drawbacks of the LDA band structures do not lead to qualitative changes of the quasiparticle lifetimes. The quasiparticle lifetime calculations based on the exact band energies have been performed for Cu using plane waves method [85]. However, because of large requirements of the computation, only several lifetimes at high symmetry points were calculated. Ref. [83] calculate the lifetimes based on the corrected local density approximation, but the d band threshold energies (1.7 eV for Cu, 3.2 eV for Ag) reported in this thesis are still close to the Fermi level, compared to the photoemission data about 2 eV for Cu and about 3.8 eV for Ag. So it is necessary to calculate the lifetimes based on the accurate quasiparticle band energies.

In this chapter we go beyond the 'energy shell' approximation to calculate the lifetimes of electrons and holes for Cu and Ag based on the accurate GW bandstructure energies, using the efficient localized Gaussian orbitals. Firstly, we discuss the quasiparticle bandstructures for Cu and Ag. Then we analyze the influence of the quasiparticle corrections to the lifetimes. Our calculation starts with a DFT calculation of the ground-state properties performed using norm-conserving pseudopotentials, then the calculated DFT band structure energies and wave functions are used within the GW approximation framework for the calculation of the self energy operator, whose real part yields the quasi-particle band energies and imaginary part yields the lifetimes.

A particular field of research is given by the lifetimes of metal surface states, which can be drastically different from the lifetimes of bulk states. Unfortunately, the calculation of the GW self-energy operator of a surface system (e.g. in a slab configuration) would be very time-consuming [88] and has been carried out so far only for the relatively simple Be(001) surface [89]. So far, The most previous works calculate image state lifetimes for the more complex noble-metal surfaces by employing a realistic one-dimensional model potential [90] instead of the full three

dimensional pseudopotential. Within this simple model, the surface corrugation is fully neglected (i.e. a single electron moving in an effective potential $V(z)$ is translationally invariant in the plane of the surface). Here, we report a full three-dimensional calculation for the image states lifetimes of Cu(100) by using the efficient localized Gaussian basis sets. With my knowledge, this is the first time for the evaluation of the image states of noble metal surfaces by performing a full three-dimensional calculation.

One particular aim of our work is to provide an approach to both band-structure energies and their lifetimes within one calculation within the GW approximation for the self energy. For computational purposes many calculations so far have focused on either the band structure (which can often be obtained within the plasmon-pole approximation of the GWA, leading to a real self energy and thus infinite lifetime) or on the lifetime (which can be done efficiently by considering a limited number of bands in the calculation). Here we show that both quantities are in fact obtained when the complex nature of the dielectric screening and the self energy are treated carefully.

This chapter is organized as follows. In Sec. 5.2, we discuss some details of the GWA and summarize the basic equations needed for the following discussions. In Sec. 5.3, we discuss the quasi-particle bandstructures for bulk Cu and Ag. In Sec. 5.4, we present the quasi-particle lifetimes for Cu and Ag based on the accurate quasiparticle energies. In Sec. 5.5, a full three-dimensional calculation for the image potential state lifetimes of Cu(100) is reported. A short summary concludes this chapter in Sec. 5.6.

5.2 Computational details and theoretical methods

The starting point of the calculation of band structures and lifetimes is given by a DFT calculation (referring to the electronic ground state), in which the Kohn-Sham equation

$$\left\{ -\frac{\hbar^2}{2m} \nabla^2 + V_{PS}(\mathbf{r}) + V_H(\mathbf{r}) + V_{XC}(\mathbf{r}) \right\} \Psi_{nk}(\mathbf{r}) = E_{nk} \Psi_{nk}(\mathbf{r}) \quad (5.1)$$

is solved.

In here, V_{PS} and V_H are the ions and Hartree potential, respectively. Since the d electrons have to be considered as valence electrons, the entire shell to which they belong must be included as valence states throughout the calculations [13, 91]. For Ag we employ a norm-conserving pseudopotential constructed along Hamann's scheme [38], based on the ionic configuration $4s^2 4p^6 4d^{10}$. For Cu, on the other hand, the transferability of such a pseudopotential seems not guaranteed. Therefore we employ a multireference pseudopotential (MRPP) constructed along Carlos' scheme [92], based on the configuration $3s^2 3p^6 3d^{10} 4s^1$. Gaussian orbitals are used to construct the LDA basis sets. We use 50 Gaussian orbitals of s, p, d and s* type for Cu atom and 60 Gaussian orbitals of s, p, d and s* type for Ag atom. The decay constants are 0.18, 0.50, 1.24, 3.2 and 8.5 for Cu, and 0.16, 0.50, 1.40, 3.00, 6.5 and 14.0 for Ag, respectively.

Based on the DFT results, the final treatment of band structures and lifetimes is done within many body perturbation theory (MBPT) [10]. Assuming that the electronic spectrum is still given by well-defined quasi particles, excited holes and electrons are described by an equation of motion

$$\begin{aligned} \left\{ -\frac{\hbar^2}{2m} \nabla^2 + V_{PS}(\mathbf{r}) + V_H(\mathbf{r}) \right\} \Psi_{nk}^{QP}(\mathbf{r}) \\ + \int \Sigma(\mathbf{r}, \mathbf{r}', E_{nk}^{QP}) \Phi_{nk}^{QP}(\mathbf{r}') d^3 \mathbf{r}' = E_{nk}^{QP} \Psi_{nk}^{QP}(\mathbf{r}) \end{aligned} \quad (5.2)$$

Compared to the KS equation, the local, real and hermitian exchange-correlation part is replaced by a non-local, complex and energy dependent self-energy operator. The self-energy operator results from electronic many-body effects beyond the DFT. In particular, it contains exchange and correlation effects (relevant for the band-structure energies) as well as Coulomb scattering effects (relevant for the resulting lifetimes). Consequently, the self-energy operator leads to complex quasi-particle energies, i.e. the real part denotes the band-structure energy while the imaginary part yields the inverse of the lifetime.

We calculate the self-energy with the GW approximation of many-body perturbation theory by retaining the first term in the series expansion of self-energy operator in terms of the screened Coulomb interaction W :

$$\Sigma(\mathbf{r}, \mathbf{r}', E) = \frac{i}{2\pi} \int e^{i\omega 0^+} G(\mathbf{r}, \mathbf{r}', E - \omega) W(\mathbf{r}, \mathbf{r}', \omega) d\omega \quad (5.3)$$

Equation (5.2) has to be solved self-consistently. It turns out that for many systems the DFT wave functions agree remarkably well with the final GW wave functions [93].

$$\Psi_{nk}^{DFT}(\mathbf{r}) = \Psi_{nk}^{QP}(\mathbf{r}) \quad (5.4)$$

Therefore, the self-energy operator can be constructed using the results of the LDA calculation (wave functions, energy spectrum, and electronic density) and an iterative treatment of equation (5.2) is not necessary. The quasi-particle energies are then simply given by

$$E_{nk}^{QP} = E_{nk}^{DFT} + \text{Re}[\Sigma_{nk}(E_{nk}^{QP})] - V_{XC}^{nk}, \quad (5.5)$$

where $\Sigma_{nk}(E_{nk}^{QP}) = \langle nk | \Sigma_{nk}(\mathbf{r}, \mathbf{r}', E_{nk}^{QP}) | nk \rangle$ and $V_{XC}^{nk} = \langle nk | V_{XC}(\mathbf{r}) | nk \rangle$.

Simultaneously the imaginary part of $\Sigma(E)$ can be considered, which yields the Coulomb-scattering lifetimes. The imaginary part of the self-energy corresponds to the quasi-particle lifetimes, and the probability of the scattering decay is given

by

$$\tau_{nk}^{-1} = 2|\text{Im}[\Sigma_{nk}(E_{nk}^{QP})]| \quad (5.6)$$

The numerical evaluation of the GW self-energy operator for the present case of metals, aiming at its real part and its imaginary part on equal footing, is more demanding than the common techniques for GW calculations which focus on the band structure alone, and is also more demanding than the evaluation of lifetimes alone.

One key quantity is given by the time-ordered polarizability of the system, $\tilde{P} = -iGG$, which we calculate from DFT-LDA quantities. For a periodic system it can be represented by basis functions $\chi_\beta(\mathbf{q}, \mathbf{r})$, leading to a matrix representation

$$\begin{aligned} \tilde{P}_{\beta\beta'}(\mathbf{q}, \omega) &= 2\frac{1}{V} \sum_{\mathbf{k}} \sum_m \sum_n M_\beta^{mn}(\mathbf{k}, \mathbf{q}) [M_{\beta'}^{mn}(\mathbf{k}, \mathbf{q})]^* \Theta(E_F - E_{m\mathbf{k}}) \Theta(E_{n,\mathbf{k}+\mathbf{q}} - E_F) \\ &\times \left[\frac{1}{E_{m\mathbf{k}} - E_{n,\mathbf{k}+\mathbf{q}} - \omega + i0^+} + \frac{1}{E_{m\mathbf{k}} - E_{n,\mathbf{k}+\mathbf{q}} + \omega + i0^+} \right] \end{aligned} \quad (5.7)$$

with spatial integrals $M_\beta^{mn}(\mathbf{k}, \mathbf{q}) = \int \psi_{m\mathbf{k}}^*(\mathbf{r}) \chi_\beta^*(\mathbf{q}, \mathbf{r}) \psi_{n,\mathbf{k}+\mathbf{q}}(\mathbf{r}) d^3r$. The summation is restricted to bands and \mathbf{k} -points such that $|m, \mathbf{k}\rangle$ is occupied and $|n, \mathbf{k} + \mathbf{q}\rangle$ is empty. The direct evaluation of Eqs. (5.7) can be tricky due to the careful balance between reciprocal-lattice points and frequencies, both of which have to be from a finite grid. In particular, the correct maintenance of charge neutrality (i.e. correct sum of the occupations of all states) is difficult to control. Furthermore, the replacement of the infinitesimally small imaginary part (from $i0^+$) by broadening is difficult to control. Last but not least, the direct evaluation of Eqs. (5.7) for all desired frequencies ω (usually several hundred) constitutes a computational bottleneck.

Instead, we first calculate the imaginary part of \tilde{P} [using $1/(a + i0^+) = 1/a - i\pi\delta(a)$], i.e.

$$\begin{aligned} \text{Im}\tilde{P}_{\beta\beta'}(\mathbf{q}, \omega) &= -2\pi\frac{1}{V} \sum_{\mathbf{k}} \sum_m \sum_n M_\beta^{mn}(\mathbf{k}, \mathbf{q}) [M_{\beta'}^{mn}(\mathbf{k}, \mathbf{q})]^* \\ &\times |f_{m\mathbf{k}} - f_{n,\mathbf{k}+\mathbf{q}}| \delta(\omega - (E_{m\mathbf{k}} - E_{n,\mathbf{k}+\mathbf{q}})) \quad . \end{aligned} \quad (5.8)$$

In here, the occupation numbers $f_{m\mathbf{k}}$ and $f_{n,\mathbf{k}+\mathbf{q}}$ result from the respective band-structure energy. The most important advantage of focusing on $\text{Im}(\tilde{P})$ first is that a given combination of $|m, \mathbf{k}\rangle$ and $|n, \mathbf{k} + \mathbf{q}\rangle$ contributes to a very small number of values for ω , only, thus reducing the total numerical demand by an order of magnitude or more.

In practice, an occupation number f is not taken from a sharp step at the Fermi level, but includes spectral broadening by a Gaussian broadening parameter, which corresponds to the finite \mathbf{k} -point grid employed in all calculations. Simply speaking, a state near the Fermi level can be expected to exhibit band-structure dispersion in \mathbf{k} (which is not explicitly considered due to the finite grid) and to cross the Fermi level somewhere in the vicinity of the grid point. It is thus not reasonable to attribute an occupation of just 0 (or 1) to a state slightly above (or below) the Fermi level, but the occupation function should be smeared out for states near the Fermi level. This procedure allows to mimic finer \mathbf{k} -point sampling than the one which is really employed, thus reducing the dependence of the final results on the \mathbf{k} -point grid. In addition, the δ -function with respect to ω must be replaced by a finite-width spectral function, as well. The evaluation using a finite \mathbf{k} -point grid yields sharp, discrete lines in the spectrum, which would be smeared out by continuous \mathbf{k} -point integration. Again, the effect of using \mathbf{k} -point integration instead of simple grid sampling can be mimicked by spectral broadening (smearing) of $\delta(\omega - (E_{m\mathbf{k}} - E_{n,\mathbf{k}+\mathbf{q}}))$. Typical values of the spectral broadening are 0.015 eV both for the occupation and the δ -function.

After calculating the imaginary part of \tilde{P} by carrying out the double summation of states $|m, \mathbf{k}\rangle$ and $|n, \mathbf{k} + \mathbf{q}\rangle$, its real part is simply obtained by a Kramers-Kronig transformation. All these operations are performed using a grid of 200 frequencies from zero to 100 eV for Cu, and a grid of 150 frequencies from zero to 100eV for Ag. Negative frequencies need not be considered explicitly since $\tilde{P}(-\omega) = \tilde{P}(\omega)$.

Regardless of the above-mentioned issues concerning band-structure summation, broadening etc., the polarizability must also be expanded in a real-space basis $\{\beta\}$ to represent its spatial properties. Here we employ a second ("auxiliary") basis

of Gaussian orbitals ("GW basis"). This GW basis set must contain functions of higher angular momenta than the LDA basis set [84]. We use 65 Gaussian orbitals of s, p, d, s*, f and g type for Cu atom and 80 Gaussian orbitals of s, p, d, s*, f and g type for Ag atom. A suitable choice of decay constants for these basis functions is given by 0.18, 0.65 and 2.65 for Cu, and 0.18, 0.65 and 3.00 for Ag, respectively.

After transforming the polarizability into the dynamically screened Coulomb interaction W , the self-energy operator is constructed. While its exchange part is quite simple to obtain, the correlation contribution to $\Sigma(E)$ is more difficult. Its expectation value for a given state is evaluated as

$$\begin{aligned}
\langle m\mathbf{k}|\Sigma_c(E)|m'\mathbf{k}\rangle &= \sum_{\mathbf{q},n,\beta\beta'} M_{\beta}^{mn}(\mathbf{k},\mathbf{q}) \left(M_{\beta'}^{m'n}(\mathbf{k},\mathbf{q}) \right)^* \frac{i}{2\pi} \int e^{-i\omega 0^+} \\
&\times \frac{W_{\beta\beta'}(-\mathbf{q},\omega) - v_{\beta\beta'}(-\mathbf{q})}{E - \omega - E_{n,\mathbf{k}+\mathbf{q}} + i0^+ \text{sign}(E_{n,\mathbf{k}+\mathbf{q}} - E_F)} d\omega \\
&= \sum_{\mathbf{q},n} \frac{i}{2\pi} \int e^{-i\omega 0^+} \\
&\times \frac{F_{mm'n}(\mathbf{k},\mathbf{q},\omega)}{E - \omega - E_{n,\mathbf{k}+\mathbf{q}} + i0^+ \text{sign}(E_{n,\mathbf{k}+\mathbf{q}} - E_F)} d\omega \quad (5.9)
\end{aligned}$$

with F containing the screening potential $W-v$ evaluated between states m , m' , and n . Owing to its origin from $\tilde{P}(\omega)$ (see above), F has a form (again within time-ordered Green-function theory) of [84]

$$F_{mm'n}(\mathbf{k},\mathbf{q},\omega) = \sum_j a_j \left(\frac{1}{\omega - (\omega_j - i0^+)} - \frac{1}{\omega + (\omega_j - i0^+)} \right) \quad (5.10)$$

(omitting the dependence of a_j and ω_j on m , m' , n , \mathbf{k} , and \mathbf{q}) having real amplitudes a_j and poles at frequencies $\pm(\omega_j - i0^+)$ with ω_j being real and positive. The self energy thus results as

$$\begin{aligned}
\langle m\mathbf{k}|\Sigma_c(E)|m'\mathbf{k}\rangle &= \sum_{\mathbf{q},n} \sum_j a_j \\
&\times \begin{cases} \frac{1}{E + (\omega_j - i0^+) - E_{n,\mathbf{k}+\mathbf{q}}} & \text{if } |n, \mathbf{k} + \mathbf{q}\rangle \text{ is occupied} \\ -\frac{1}{E - (\omega_j - i0^+) - E_{n,\mathbf{k}+\mathbf{q}}} & \text{if } |n, \mathbf{k} + \mathbf{q}\rangle \text{ is empty} \end{cases} \quad (5.11)
\end{aligned}$$

Consequently, the imaginary part of the self energy is given by

$$\begin{aligned} \text{Im} \langle m\mathbf{k} | \Sigma_c(E) | m'\mathbf{k} \rangle &= \sum_{\mathbf{q}, n} \sum_j a_j \\ &\times \begin{cases} \pi\delta(E + \omega_j - E_{n, \mathbf{k}+\mathbf{q}}) & \text{if } |n, \mathbf{k} + \mathbf{q}\rangle \text{ is occupied} \\ \pi\delta(E - \omega_j - E_{n, \mathbf{k}+\mathbf{q}}) & \text{if } |n, \mathbf{k} + \mathbf{q}\rangle \text{ is empty} \end{cases} \end{aligned} \quad (5.12)$$

which is immediately identified as

$$\text{Im} \langle m\mathbf{k} | \Sigma_c(E) | m'\mathbf{k} \rangle = \sum_{\mathbf{q}} \sum'_n \text{Im} F_{mm'n}(\mathbf{k}, \mathbf{q}, |E - E_{n, \mathbf{k}+\mathbf{q}}|) \quad (5.13)$$

with the sum over n restricted to such states $|n, \mathbf{k} + \mathbf{q}\rangle$ that are located between E and the Fermi level. On the one hand, this expression reflects the physical mechanism, i.e. the imaginary part of the self energy at energy E is related to scattering events (contained in $\text{Im}(W)$ and thus in $\text{Im}(F)$) involving final states between E and E_F (i.e. both quasi-electrons and quasi-holes are inelastically scattered towards the Fermi level). On the other hand, Eqs. (5.13) provides an efficient way of calculating the self energy: we first evaluate its imaginary part (for frequencies E on a real grid), from which the real part of Σ is obtained via Kramers-Kronig relation afterwards. Again, the evaluation of the sum in Eqs. (5.13), which includes or excludes states n according to their energy relative to E and E_F , is realised by employing spectral broadening of $E_{n, \mathbf{k}+\mathbf{q}}$, i.e. if $E_{n, \mathbf{k}+\mathbf{q}}$ is close to the Fermi level or close to E , it is included in the sum with fractional weight between 0 and 1, mimicing the dispersion of E_n in $\mathbf{k} + \mathbf{q}$.

5.3 Quasi-particle band-structures

In this section we discuss the band structures resulting from the quasi-particle theory outlined above.

TABLE 5.1: Calculated bandwidths and band structure energies (in eV) for Cu at high symmetry points. We show our DFT and GWA results in comparison. The values of the fifth column are GWA data from Ref. [13] using plane waves. The last column contains experimental data from Ref. [94].

	State	DFT	GWA	GWA	Exp.
		This Work		Ref. [13]	Ref. [94]
Positions	Γ_{12}	-2.31	-2.82	-2.81	-2.78
of d	X_5	-1.56	-2.09	-2.04	-2.01
bands	$L_3(2)$	-1.71	-2.24	-2.24	-2.25
Positions of	Γ_1	-9.22	-8.68	-9.24	-8.60
s/p bands	L_2	-1.07	-0.59	-0.57	-0.85
	$\Gamma_{12} - \Gamma_{25'}$	0.86	0.63	0.60	0.81
Width	$X_5 - X_3$	2.99	2.50	2.49	2.79
of d	$X_5 - X_1$	3.41	2.86	2.90	3.17
bands	$L_3^{(2)} - L_3^{(1)}$	1.47	1.27	1.26	1.37
	$L_3 - L_1$	3.43	2.74	2.83	2.91

5.3.1 Bulk Cu

Fig. 5.1 displays the band structure of Cu, as obtained within DFT (dashed curves) and within GWA (solid curves). In here, 50 Gaussian orbitals are used for representing the wave functions and 65 Gaussians for the polarizability and related quantities. Bandstructure energies for Cu at selected high symmetry points are given in Table 5.1, as well. For comparison, the table also contains data from a different quasiparticle calculation [13] (employing a plane waves basis) and from experimental photoemission data [94].

Exchange-correlation contributions to the self-energy arising from 3s and 3p core levels in copper are shown to be crucial [13], so we include the whole 3rd atomic shell into the valence in the pseudopotential generation.

As can be seen from Fig. 5.1, the calculated d-band threshold within DFT amounts to 1.5 eV to 1.7 eV, respectively, compared to quasiparticle data of 2.0 eV to 2.5 eV [95]. All *d* states are shifted to lower energy by 0.5-0.6 eV, leading to an excellent agreement with the experimental results for the positions of the d bands,

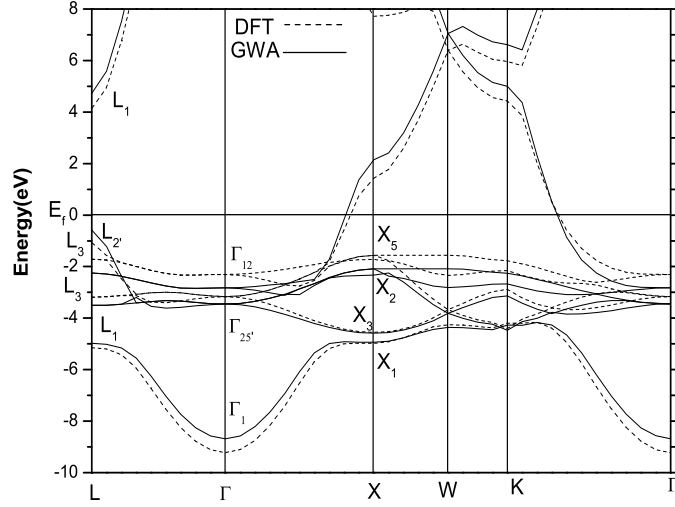


FIGURE 5.1: Calculated band structure for Cu along high-symmetry lines, obtained within DFT (dashed lines) and within GWA (solid lines).

as can be seen from Table 5.1. For the positions of d bands, our calculated results deviate from the experiments by less than 0.1 eV except for the L_2 point. It also can be seen from Table 5.1 that the calculated GWA band widths are significantly closer to the photoemission data than the DFT results. This indicates that Cu is very well described within the GWA. We also show our calculations and the calculated data using plane waves reported in Ref. [13] in comparison. For the widths of bands, our results employing Gaussian basis sets are close to the results of Ref. [13]. A big difference between our result and the result of Ref. [13] occurs at the Γ_1 point. The value of Ref. [13] is smaller than experiment by 0.64 eV, while our quasiparticle energy deviate from experiment only by 0.08 eV. All these results indicate that the localized Gaussian basis sets are efficient both for DFT and GWA calculations, especially for the GWA calculation of materials with occupied d orbitals or other semi-core states, which require a heavy CPU load with plane waves. For the case of Cu, we only need 65 Gaussian orbitals for convergence. We note in passing that for a given state the QP shift does not simply result from the difference between Σ and V_{xc} alone, but also includes the re-calibration of the Fermi level (which is shifted up by 0.27 eV due to $\Sigma - V_{xc}$).

We also find that the values of the quasiparticle corrections change with the

TABLE 5.2: Calculated bandwidths and band structure energies (in eV) for Ag at high symmetry points. We show our DFT and GWA results in comparison. The values of the fifth column are GWA data from Ref. [85] using plane waves. The last column contains experimental data from Ref. [96].

	Symmetry State	DFT DFT This	GWA GWA Work	Ref. GWA Ref. [14]	Exp. Exp. Ref. [96]
Positions of d bands	Γ_{12}	-3.73	-4.94	-4.81	-4.95
	X_5	-2.68	-3.80	-3.72	-3.97
	$L_3(2)$	-2.90	-4.05	-3.94	-4.15
Width of d bands	$\Gamma_{12} - \Gamma_{25}$	1.08	0.97	0.94	1.11
	$X_5 - X_3$	3.65	3.54	3.39	3.35
	$X_5 - X_1$	3.77	3.61	3.51	3.40
	$X_5 - X_2$	0.25	0.35	0.29	0.38
	$L_3^{(2)} - L_3^{(1)}$	1.92	1.89	1.85	1.99
	$L_3 - L_1$	3.52	3.26	3.17	2.94

different bands and k points, and at the L point the shifts even change sign for different valence bands, with QP corrections ranging from -0.53 eV to 0.52 eV. This indicates that the so-called scissors operator often used in LDA for semiconductor systems is not suitable for noble metals. These results are consistent with the results of Ref [13]. In short, the GW self-energy corrections lead to excellent agreement with the experimental data.

5.3.2 Bulk Ag

Our calculated DFT bandstructures and the full GWA bandstructures for Ag are shown in Fig. 5.2. Some characteristic energy differences relating mostly to the occupied bands are given in Table 5.2. This table also contains results from a different QP calculation [14] and photoemission data [96] for comparison.

Since the atomic structure of Ag is similar to the Cu, we also include the 4s and 4p states to the valence for Ag in the pseudopotential generation. Though the semicore states are included, we find out that partial core correction(PCC) [97] can not be neglected for Ag. The DFT band energies of Ag are changed by 0.2 eV

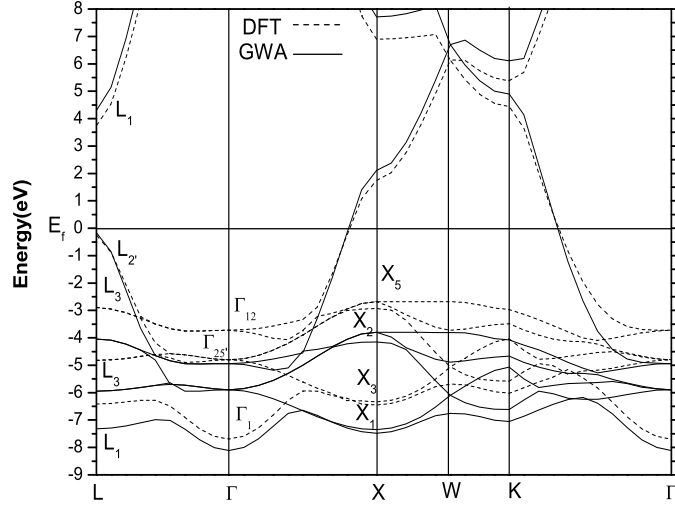


FIGURE 5.2: Calculated band structure for Ag along high-symmetry lines, obtained within DFT (dashed lines) and within GWA (solid lines).

with the inclusion of partial core correction. Our DFT band energy at X_5 point is -2.68 eV, and close to the result of -2.70 eV from a full potential calculation (within LMTO) [83]. Based on the DFT calculation, our calculated d band threshold at the X_5 point is about -3.80 eV within GWA, compared to photoemission data of 3.97 eV. Our quasiparticle results are in quite good agreement with the photoemission data.

Like in the case of Cu, the Fermi level observes a value of $\Sigma^{GW} - V_{xc}$ of about 1.0 eV, while the the top position of the d bands observes $\Sigma^{GW} - V_{xc} = -0.2$ eV. In total, this results in a downward of shift of about 1.2 eV of the top of the d bands, and leads to excellent agreement with available photoemission data, as can be seen from Table 5.2. It can be seen from Fig. 5.2 that the GW corrections do not act as a rigid shift of the whole occupied bandstructure with respect to the unoccupied part. This is similar to the situation in Cu. Note, however, that the bottom of the sp band (at Γ_1) is shifted downwards by 0.5 eV for Ag, which is different from the upwards shift in the case of Cu. For the bandwidths of d bands, GWA results are closer to experiment than the DFT results are.

We also show calculated data using plane waves reported in Ref. [14] for comparison. In spite of the different methods of calculation, our calculated results

are in good overall agreement with the results of Ref. [14] except for some minor differences. For example, our quasiparticle energies of the d bands deviate from photoemission data by about 0.1 eV, while the discrepancy is about 0.2 eV in Ref. [14]. For the case of Ag, we get converged results using only 80 Gaussian orbitals.

5.4 Quasi-particle lifetimes

Electron and hole dynamics in the bulk as well as at the various surfaces of a solid play a key role in a great number of physical and chemical phenomena [98, 99]. A new path for the study of both electron and hole dynamics in the time domain was opened by the development of the time-resolved two-photon photoemission (TR-TPPE) technique [17, 18]. Recently first principles calculations of the electron lifetimes in bulk noble metals have also been published in Refs. [24, 25, 26, 82]. However, many problems related to the approximations employed still remain unresolved. The common problem is that these calculations are based on the DFT band energies. Another problem is that the renormalization factors are neglected in many GW calculations such as Ref. [24, 25, 82]. In fact, the renormalization factor Z is quite far from the 1.0 used in the 'energy shell' approximation. We got an average value of $Z = 0.66$ for Cu, and $Z = 0.74$ for Ag. This modifies the energy of the excitation, and hence, the lifetime.

5.4.1 Bulk Cu

In Fig. 5.3 we compare our calculated lifetimes of electrons for Cu with the lifetimes of Ref. [27] and experimental data [100]. The dashed lines in Fig. 5.3 are the lifetimes of Ref. [27] obtained within the so-called Fermi liquid theory (FLT), the solid line represents the lifetimes of Ref. [27] calculated with plane waves, and the diamonds are the experimental data of Ref. [94].

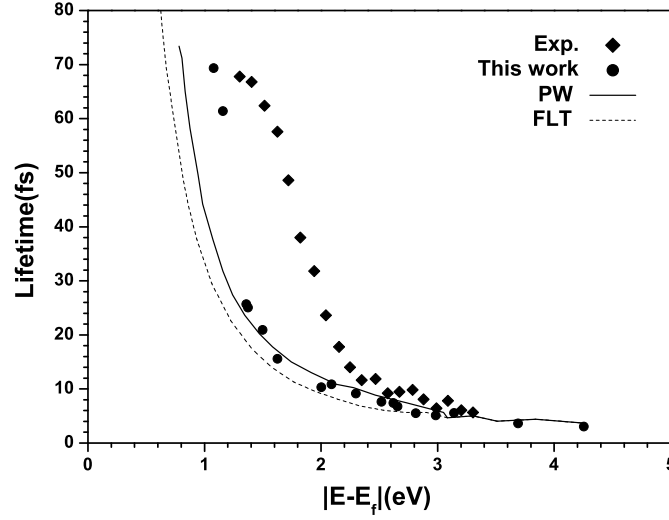


FIGURE 5.3: Calculated lifetimes of hot electrons in Cu along high-symmetry lines (circles) in comparison with experimental data (diamonds).[100] The solid and dashed lines denote theoretical results Ref. [27], obtained within GWA and within Fermi-liquid theory.

In spite of great differences in the methods of calculation, very good agreement is observed above 1.4 eV between our data and the lifetimes based on the plane waves. Compared to the lifetimes obtained with plane waves, our calculated lifetimes are closer to the experimental results at about 1.1 eV. Both our calculated electron lifetimes of hot electrons and the calculated lifetimes with plane waves are slightly larger than the results obtained within FLT. This is due to the influence of the d electrons which provide an additional screening mechanism that is absent in the jellium model. Below 2 eV the experimental data show a sudden increase. The origin of the disagreement between calculated results and experimental data was discussed in Ref. [27], and they explain that the increase is due to the interaction of the excited electron with its hole in the d bands. As can be seen from Fig. 5.1, the threshold of d bands is located about 2 eV below the Fermi level. The electrons at the top of the d bands can be excited by the photon energies widely used in 2PPE experiments. These excited electrons interact with the holes in d bands, and hence, the electron lifetimes increase at 2 eV for Cu. Gurtubay et al [101] compute the lifetimes for Cu with full inclusion of exchange and correlation(XC) effects based on the DFT band energies calculation. This approximation treats

on the same footing XC effects between pairs of electrons within the Fermi sea (screening electrons) and between the excited hot electron and the Fermi sea. This corresponds to including the higher order terms in the self-energy operator beyond the GWA. Electron-hole interaction effects, however, are still not included. Gurtubay et al conclude that the overall effect of short-range XC is small and GWT linewidths are close to their XC-free G_0W_0 counterparts. Our calculation, on the other hand, is based on accurate band energies and include the semicore states. So the only reason for the discrepancies between theoretical methods and experiments should be the interactions of the excited electrons with holes in the d bands, which are not included in the GW calculation.

First principle calculations based on the DFT predict an upper d-band edge that is located about 1.5 eV below the Fermi level, i.e., about 0.5 eV higher than observed by photoemission experiments [94]. This indicates that a rigid shift in the energy axis will occur for the hole lifetime calculations based on the DFT band energies. The direct influence on the calculated lifetimes based on the DFT band energies is that the calculated values of d-hole lifetimes cannot be directly compared with experiment. Our calculation is based on the accurate band energies, the hole energies in Cu are located about 2 eV below the Fermi level, and this is consistent with the threshold of d bands measured by photoemission experiment, as shown in Fig. 5.4. It can also be seen from Fig. 5.4 that d holes exhibit a longer lifetimes than sp electrons with the same excitation energy, in agreement with experiment. As can be seen from Fig. 5.1, the sp-like bands cross the d bands below the Fermi level, and a small overlap occurs between the sp states and the d states. This small overlap between sp bands and d bands is responsible for the increase of the hole lifetimes at the top of the d bands [82].

In order to find out the influence of quasiparticle corrections on the hole lifetimes, we compare our calculated d hole lifetimes with the lifetimes of Ref. [83] and experiments [86], as shown in Table 5.3. Ref. [83] are the results of GWA, based on the linear muffin-tin orbitals (LMTO) method. The last column are the experimental data.

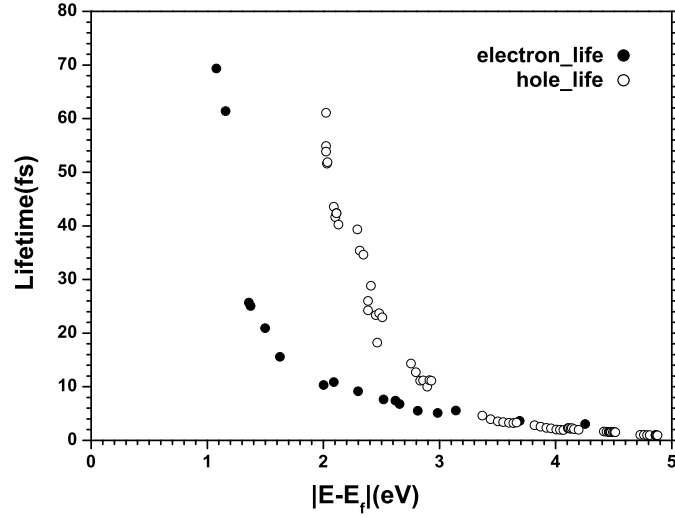


FIGURE 5.4: Calculated lifetimes of d holes in Cu, alongt high symmetry lines. For comparison, the electron lifetimes are also presented (solid circles); open circles represent the hole lifetimes;

TABLE 5.3: Lifetimes of d holes in Cu (in fs).

	Symmetry points	This work	Ref. [83]	Exp. [86]
Lifetime of d states	X_5	64.4	47.1	33 ± 7
	X_2	53.9	47.5	6.5
	$L_3(2)$	23.9	23.5	
	Γ_{12}	22.2	12.6	5.0
	Γ_{25}	6.2	3.2	3.7
	$L_3(1)$	6.0	3.02	
	X_3	1.7	0.6	2.8

Ref. [86] reported that the calculated lifetimes below 3 eV tend to approach those predicted by a FEG model of the solid, while the experimental show a distinct asymptotic behavior at the large binding energies. They attribute this discrepancy to the failure of the DFT to reproduce the actual band structure of Cu. As can be seen from Table 5.3, our calculated lifetime at Γ_{25} (about 3.58 eV below Fermi level) is larger than the experimental data. Both our calculated lifetime and the lifetime of Ref. [83] at X_3 (about 4.78 eV below the Fermi level) are smaller than the experimental data, especially, the calculated lifetime of Ref. [83] deviate from experiment by 2.2 fs. The LDA eigenvalues have been corrected

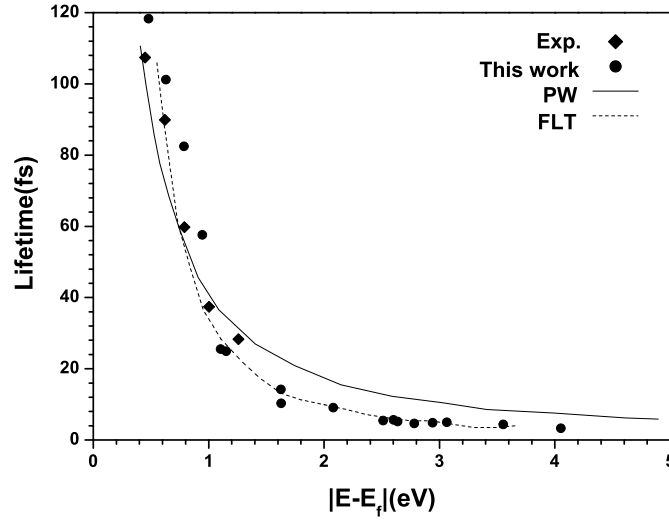


FIGURE 5.5: Calculated lifetimes of hot electrons in Ag, at high symmetry points. solid circles show our results; solid diamonds represent experiments [100]; solid line and dashed line denote the results of Ref. [27].

before entering the matrix elements of self-energy operator in Ref. [83]. As can be seen from Table 5.1 and Fig. 5.1, not only the positions of d bands are close to the experimental data, but also the d band width $X_5 - X_3$ is 2.63 eV, which is in good agreement with the experimental value of 2.79 eV. As can be seen from Table 5.2, all theoretical lifetimes above $\Gamma_{25'}$ are larger than experimental data. So the quasiparticle corrections to the DFT bandstructure energies are not essential reasons for the disagreements between theoretical lifetimes and experiments, and this is consistent with the conclusion of Ref. [87]. In order to accurately describe lifetimes of holes for noble metals, effects beyond the GW approximation should be included.

5.4.2 Bulk Ag

Our calculated electron lifetimes for Ag are given in Fig. 5.5 (solid circles). We compare our results with the results of Ref. [27] based on the plane wave bases sets, FLT results and experimental data [100].

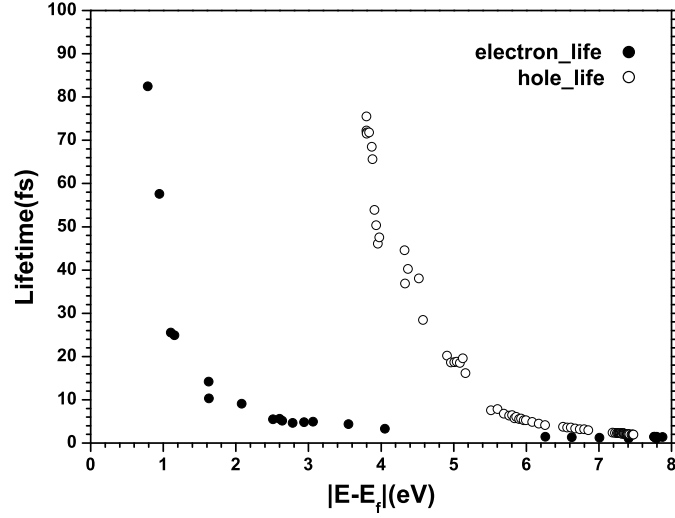


FIGURE 5.6: Calculated lifetimes of d holes in Ag, at high symmetry points. open circles represent the lifetime of holes; solid circles depict the lifetime of electrons.

Our calculated lifetimes of electrons in Ag are in good agreement with the results from literature. As can be seen from Fig. 5.5, our results are closer to the results obtained within FLT than the theoretical lifetimes from plane waves. As can be seen from Fig. 5.2, the d band threshold locates about 3.80 eV below the Fermi level. This indicates that for all excited electrons below 3.80 eV the relaxation of the hot electrons can not excite the electrons in the d bands, and lead to FLT behavior of Ag. Contrary to the case of Cu, the experimental data for Ag do not show a sudden increase at small energies, due to the fact that the d band threshold is about 3.80 eV below the Fermi level, as shown in Fig. 5.2, and no d electrons are excited for the photon energies widely used in 2PPE experiments. This indicates that the electron lifetimes of Ag are mainly attributed to the electron-electron interactions.

As can be seen from Fig. 5.6, the hole lifetimes of d states are located about 3.8 eV below the Fermi level, which is in agreement with the experimental threshold of d bands. It can also be seen from Fig. 5.6 that the hole lifetimes are larger than the electron lifetimes, which is in agreement with experiments [102], and this is similar to the case of Cu. The band structure in Fig. 5.2 presents a small overlap

TABLE 5.4: Lifetimes of d holes for Ag (in fs)

Symmetry points	This work	Exp. [102]
X_5	68.82	22
X_2	74.25	12
X_3	2.19	
X_1	1.95	2.0

between d and sp states at the top of d bands (3.8 eV below the Fermi level) and large overlaps between d bands at deeper energy below Fermi level. The small overlap yields a dramatic increase of the hole lifetimes, especially, at the top of d bands. As the d hole energy decreases, large overlap between d bands yields a rapid decrease of hole lifetimes.

As can be seen from Table 5.4, our calculated lifetime is about 68 fs at the upper edge of d bands (X_5 point, as shown in Fig. 5.2), while the experimental [102] value is 22 fs. Large discrepancy appears at X_2 point. For the states at X_1 , good agreement is observed between our calculated lifetime and lifetime measured by experiment. In spite of the large quasiparticle corrections of the d bands for Ag (1.2 eV), our calculated hole lifetimes for Ag still deviate from experiment, and this is similar to the case of Cu.

It is well known that the GW approximation accounts well for long-range screening. Short range interactions such as electron-hole and hole-hole interactions near the Fermi level may be crucial for the hole lifetimes in the case of noble metals and electron lifetimes in the case of Cu. With increasing distance to the Fermi level, the long range interactions dominate the scattering mechanism and thus lead to good agreement with the experiments.

5.5 Image potential state lifetimes in Cu(100)

At a metal surface one can find two types of electronic states which don't exist in a bulk metal. According to their charge density localization relative to the

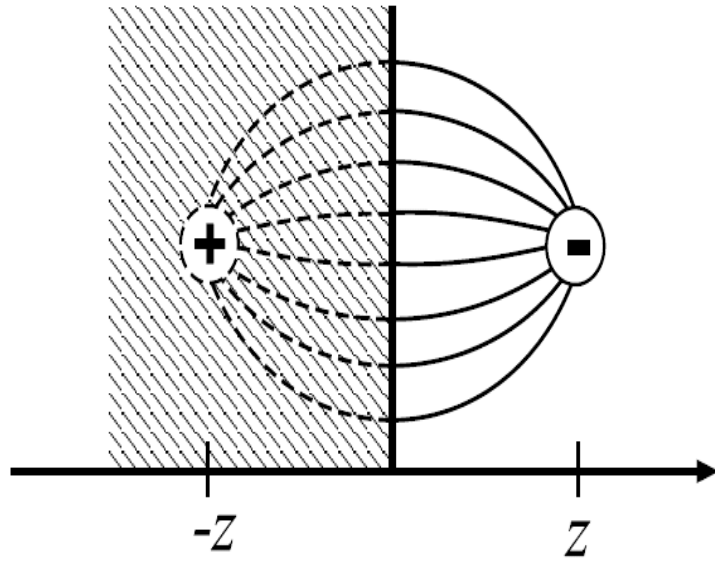


FIGURE 5.7: Sketch of the electric field lines for an electron at a distance z in front of a conducting metal surface (solid lines). A positive image charge is inside the solid at $-z$ (field lines are represented by dashed lines).

surface atomic layer, these states at a metal surface are intrinsic surface states and image-potential states, respectively. Tamm [103] and Shockley [104] predicted theoretically the so-called intrinsic surface states, which are localized mainly at the surface atomic layer. Echenique and Pendry [105] predicted the image potential states, which are localized mostly in the vacuum region if the bulk material has a band gap near the vacuum level (as given, e.g., for the Cu(100) surface).

The understanding of image potential states is rather simple. An electron at a distance z in front of a conducting metal surface moves in a Coulomb like attractive image potential, which can be expressed as

$$V(z) = -\frac{e^2}{8\pi\epsilon_0 z} \quad (5.14)$$

The attractive image potential is produced by the electron interaction with a positive (mirror image) charge at a distance z inside the metal, as can be seen from Fig. 5.7. If the metal has a band gap near the vacuum level, then an electron below the vacuum level usually transferred from the bulk by a laser may be trapped by the attractive potential. The electronic states of the trapped electron form a

hydrogen like Rydberg series with energies E_n

$$E_n = \frac{-0.85\text{eV}}{(n+a)^2}, n = 1, 2, \dots \quad (5.15)$$

where a is a quantum defect depending on both the energy gap position and width and also on the position of image state relative to the gap [105, 106].

5.5.1 Computational details and discussion of results

Here, the image potential state lifetimes for the Cu(100) surface have been evaluated within the many-body-perturbation theory. Our Cu(100) geometry contains 4 atomic layer and 26 vacuum layers, and the supercell has 1×1 periodicity. We evaluate the lifetimes from the imaginary part of the electron self energy, which we compute within the GW approximation. Similar to bulk Cu, localized Gaussian basis sets have been used for speeding up and reducing the computational cost of both DFT and GW calculations. These basis functions have the form of atomic orbitals, but are forced to be zero beyond a given confinement radius. For the image potential states of Cu(100), the $n = 1$ probability density has a maximum at 3.8 Å outside the crystal edge (the outermost atomic layer of the Cu(100)) [107]. In order to describe the image potential states in the vacuum region, we add ten 'virtual' atomic layers in the vacuum region above the surface, up to 23 Å in height. The layer distances between two 'virtual' atomic layers are the same as the layer distances of bulk region. Moreover, for the DFT and GWA calculations, we use the decaying p-type orbitals for every 'virtual' atom in the vacuum region. In addition, the LDA and GWA Gaussian basis sets designed for the bulk Cu calculation are used directly for the bulk region atom of Cu(100). In order to calculate the dynamically screened Coulomb interaction ($W = \epsilon^{-1}v$), one should calculate firstly the polarization \tilde{P} (see Sec.5.2), which is convoluted with the Coulomb interaction v yielding the dielectric function $\epsilon = 1 - v\tilde{P}$.

Finally, the image potential state energies and lifetimes of Cu(100) with quantum number $n = 1$ and $n = 2$ have been calculated. Our calculated quasi-particle

TABLE 5.5: Calculated image state lifetimes of $n = 1, 2$ and experimental results [108], as well as the theoretical results from literature using a model potential, at the Γ point.

quantum number	This work(fs)	Exp. [108](fs)	Ref. [107](fs)
$n = 1$	31	40 ± 6	30
$n = 2$	109	110 ± 10	132

corrections for image potential states are zero. This results in larger image potential state energies by 0.5 eV, as compared to experimental results, which are 4.04 eV and 4.45 eV for $n = 1$ and $n = 2$ states, respectively. The discrepancies between calculated image potential state energies and experimental results should be originated from the neglecting of the off-diagonal matrix elements of the self-energy Σ . Our calculated image potential state lifetimes of Cu(100) are presented in Table 5.5, together with experimental results and the theoretical results calculated from a model potential, and the calculated image state lifetimes using the pseudopotential methods are in agreement with the experimental results and the theoretical results by using a model one-dimensional potential. This also demonstrates that the quasi-particle corrections have minor influences on the lifetimes (See the discussion in the last section) For the one dimensional model potential method, ten parameters should be considered carefully to construct the one-dimensional potential. Most of the parameters are determined from the requirement of continuity of the potential and its first derivative everywhere in space, but four parameters of them are adjustable (i.e. the four independent parameters have to be fitted to the experimental parameters or the parameters from first-principles calculations). So the one-dimensional potential method is a semi-empirical method. For our method, taking advantage of the efficient Gaussian basis sets to expand the wave functions both for DFT and GWA, we can perform the image potential state lifetimes calculation using the pseudopotential instead of the semi-empirical one-dimensional potential.

5.6 Summary

In conclusion, we have performed the first ab initio calculations of quasiparticle bandstructures and lifetimes for noble metals Cu and Ag using efficient Gaussian orbital basis set. With the use of localized Gaussian orbitals, only less than 80 Gaussian orbitals both for the LDA basis sets and the GWA basis sets are needed, instead of thousands of plane waves. The traditional LDA calculations for Cu and Ag usually place the d bands too close to the Fermi level, especially for the Ag. The well-known discrepancies between experiments and DFT bandstructures are almost completely corrected by GW calculations. It turns out that Hedin's GW approximation is successful for the description of band-structures of noble metals.

Based on the accurate bandstructures, the lifetimes of electrons and holes in Cu and Ag have been calculated. Comparing with experimental data, the quasiparticle corrections do not lead to qualitative improvements of the quasiparticle lifetimes. It is well known that the GW approach well accounts for the long range screening, while short range interactions which are more important for noble metals are not well described. Therefore, it is probable that the improvement of the calculated lifetimes within GW approximation for d materials can be achieved by inclusion of higher terms of the many body perturbation, i.e., the short-range screening such as electron-hole and hole-hole interactions should be included when we evaluate the self-energy operator. In addition, our calculated image potential state lifetimes of Cu(100) with the quantum number $n = 1, 2$ using the pseudopotential method are in good agreement with experimental results.

Appendix A

Derivation of some equations in this thesis

A.1 The motion of Green's function

The many particle Hamiltonian can be written as

$$\hat{H} = \sum_i \left[-\frac{\nabla_i^2}{2m} + V_{ext}(\mathbf{r}_i) \right] + \frac{1}{2} \sum_{ij} v(\mathbf{r}_i, \mathbf{r}_j) \quad (\text{A.1})$$

where $V_{ext}(\mathbf{r})$ is the external potential generated by the atomic nuclei, $v(\mathbf{r}_1, \mathbf{r}_2) = e^2 / (4\pi\epsilon_0 |\mathbf{r}_1 - \mathbf{r}_2|)$ is the coulomb interaction, and ϵ_0 is the vacuum dielectric constant.

With the field operator introduced in Sec. 4.2. the above equation can be rewritten as

$$\hat{H} = \int \hat{\psi}^+(\mathbf{r}_1) h_0(\mathbf{r}_1) \hat{\psi}(\mathbf{r}_1) d\mathbf{r}_1 + \frac{1}{2} \int \hat{\psi}^+(\mathbf{r}_1) \hat{\psi}^+(\mathbf{r}_2) v(\mathbf{r}_1, \mathbf{r}_2) \hat{\psi}(\mathbf{r}_2) \hat{\psi}(\mathbf{r}_1) d\mathbf{r}_1 d\mathbf{r}_2 \quad (\text{A.2})$$

where h_0 is the one-particle term of Hamiltonian $h_0(\mathbf{r}) = -\nabla^2/2 + V_{ext}(\mathbf{r})$

In the Heisenberg picture, the equations of motion for the field operators are:

$$\begin{aligned} i\frac{\partial}{\partial t_1}\hat{\psi}(1) &= h_0(1)\hat{\psi}(1) + \int d2v(1,2)\hat{\psi}^+(2)\hat{\psi}(2)\hat{\psi}(1) \\ i\frac{\partial}{\partial t_1}\hat{\psi}^+(1) &= -h_0(1)\hat{\psi}^+(1) - \hat{\psi}^+(1) \int d2v(1,2)\hat{\psi}^+(2)\hat{\psi}(2) \end{aligned} \quad (\text{A.3})$$

The Green's function is given by

$$G(1,2) = -i[\theta(t_1 - t_2)\langle N|\hat{\psi}(1)\hat{\psi}^+(2)|N\rangle - \theta(t_2 - t_1)\langle N|\hat{\psi}^+(2)\hat{\psi}(1)|N\rangle] \quad (\text{A.4})$$

The derivative $\frac{\partial}{\partial t_1}$ of the first term in the right-hand-side of above equation is expressed as

$$\begin{aligned} &\frac{\partial}{\partial t_1} [\theta(t_1 - t_2)\langle N|\hat{\psi}(1)\hat{\psi}^+(2)|N\rangle] \\ &= [\frac{\partial}{\partial t_1}\theta(t_1 - t_2)]\langle N|\hat{\psi}(1)\hat{\psi}^+(2)|N\rangle + \theta(t_1 - t_2)\langle N|\frac{\partial}{\partial t_1}\hat{\psi}(1)\hat{\psi}^+(2) \end{aligned} \quad (\text{A.5})$$

Similarly, it is easy to obtain the derivative for the second term of Eqs.(A4). By using the following relation:

$$\frac{\partial}{\partial t_1}\theta(t_1 - t_2) = -\frac{\partial}{\partial t_1}\theta(t_2 - t_1) = \delta(t_1 - t_2) \quad (\text{A.6})$$

and summing up the derivatives of the two terms in Eqs.(A4), we can obtain:

$$\frac{\partial}{\partial t_1}G(1,2) = \delta(1,2) - i\langle N|T[i\frac{\partial}{\partial t_1}\hat{\psi}(1)\hat{\psi}^+(2)]|N\rangle \quad (\text{A.7})$$

where T stands for the Wick time-ordering operator(see sec. 4.2). Using the Eqs.(A2) for the field operators in the above equation, we can obtain:

$$\begin{aligned} \frac{\partial}{\partial t_1}G(1,2) &= \delta(1,2) - ih_0(1)\langle N|T[\hat{\psi}(1)\hat{\psi}^+(2)]|N\rangle \\ &\quad - i \int d3v(1,3)\langle N|T[\hat{\psi}^+(3)\hat{\psi}(3)\hat{\psi}(1)\hat{\psi}^+(2)]|N\rangle \end{aligned} \quad (\text{A.8})$$

We define the two particle Green's function as:

$$G_2(1, 2, 3, 4) = (-i)^2 \langle N | T[\hat{\psi}(1)\hat{\psi}(2)\hat{\psi}^+(4)\hat{\psi}^+(3)] | N \rangle \quad (\text{A.9})$$

Finally, the motion of Green's function is expressed as

$$[i\frac{\partial}{\partial t_1} - h_0(1)]G(1, 2) = \delta(1, 2) - i \int d3 v(1, 3)G_2(1, 3, 2, 3^+) \quad (\text{A.10})$$

A.2 Schwinger's functional derivative

If a system experiences a small time-dependent external perturbing potential $U_{per}(\mathbf{r}_1, \mathbf{r}_2, t)$ is given by

$$\hat{H}_{tot} = \hat{H} + \hat{H}_1 = \hat{H} + \int d\mathbf{r}_1 d\mathbf{r}_2 \hat{\psi}^+(\mathbf{r}_1, t^+) U_{per}(\mathbf{r}_1, \mathbf{r}_2, t) \hat{\psi}(\mathbf{r}_2, t) \quad (\text{A.11})$$

where $\hat{\psi}$ and $\hat{\psi}^+$ are field operators (see Sec. 4.2). The perturbation U_{per} will be made vanish at the end of the derivation.

Then the one particle Green's function becomes

$$G(1, 2) = -i \frac{\langle N | T[\hat{S}\hat{\psi}(1)\hat{\psi}^+(2)] | N \rangle}{\langle N | T[\hat{S}] | N \rangle} \quad (\text{A.12})$$

where \hat{S} is the time evolution operator, and is given by the following relation [76, 109]:

$$\begin{aligned} \hat{S}(t) &= \exp(-i \int_{-\infty}^{\infty} dt \hat{H}_1(t)) \\ &= \exp[-i \int_{-\infty}^{\infty} dt \int d\mathbf{r}_1 d\mathbf{r}_2 \hat{\psi}^+(\mathbf{r}_1, t^+) U_{per}(\mathbf{r}_1, \mathbf{r}_2, t) \hat{\psi}(\mathbf{r}_2, t)] \end{aligned} \quad (\text{A.13})$$

$U = 0$ corresponds to the definition of one-particle Green's function. The entire U dependence of this generalized Green's function lies in the time evolution operator

\hat{S} . Therefore:

$$\delta G(1, 2) = -i \frac{\langle N | T[\delta \hat{S} \hat{\psi}(1) \hat{\psi}^+(2)] | N \rangle}{\langle N | T[\hat{S}] | N \rangle} + G(1, 2) \frac{\langle N | T[\delta \hat{S}] | N \rangle}{\langle N | T[\hat{S}] | N \rangle} \quad (\text{A.14})$$

and the variation $\delta \hat{S}$ can be evaluated as

$$T[\delta \hat{S}] = -iT[\hat{S} \int_{-\infty}^{\infty} dt \int d\mathbf{r}_3 d\mathbf{r}_4 \hat{\psi}^+(\mathbf{r}_3, t^+) \delta U_{per}(\mathbf{r}_3, \mathbf{r}_4, t) \hat{\psi}(\mathbf{r}_4, t)] \quad (\text{A.15})$$

Inserting Eqs.(A15) to Eqs.(A14), one can get the following relation:

$$\delta G(1, 2) = - \int_{-\infty}^{\infty} dt \int d\mathbf{r}_3 d\mathbf{r}_4 [G_2(1, \mathbf{r}_4 t; 2, \mathbf{r}_3 t^+)] - G(1, 2) G(\mathbf{r}_4 t, \mathbf{r}_3 t^+) \quad (\text{A.16})$$

Finally, the final formula of the Schwinger's functional derivative is expressed as

$$\frac{\delta G(1, 2)}{\delta U_{per}(3, 4)} = G(1, 2) G(4, 3) - G_2(1, 4; 2, 3) \quad (\text{A.17})$$

Appendix B

Detailed information for the pseudopotential generation of Cu and Ag

B.1 Pseudopotential generation

One usually consider a configuration $3d^{10}4s^1$ for Cu in the DFT calculations. This configuration for Cu can yield good results for some quantities in the DFT calculations due to the large ($\simeq 66\text{eV}$) energy separation between the 3d and 3s states. However, the the 3d wave functions are compact and overlap significantly with the 3s and 3p semi-core states. This effect is larger if one performs a GW calculation. For instance, the inclusion of semi-core 3s and 3p into the valence leads to a very different values of $\langle \Sigma_x^{nk} - V_{xc}^{nk} \rangle$, and improves considerable the quasi-particle energies [13]. Therefore, an obvious solution would be include those 3s and 3p semi-core states as part of the valence in the pseudopotential generation scheme. That would require a somewhat larger energy cutoff in the plane wave expansion with respect to the 3d electrons and a larger number of electrons in the valence, corresponding to a manageable increase in the necessary computing resources. For the most pseudopotential program, one can only have a single pseudowave-function of

element	3s[Bohr]	3p[Bohr]	3d[Bohr]	4s[Bohr]	4p[Bohr]	4d[Bohr]
Cu(MRPP)	1.49	1.39	1.80	1.59	1.49	
Ag(Ham)				0.50	0.55	0.62

a given orbital with quantum number l , in order to warrant its nodeless structure. This indicates that if the 3s and 3p semi-core states are included in the valence for the pseudopotential generation of Cu, one has to consider an ionized configuration $3s^2 3p^6 3d^{10}$ instead of the complete configuration $3s^2 3p^6 3d^{10} 4s^1$. but pseudopotentials generated from ionized configurations tend to be less accurate. Here, I employ a multi-reference pseudopotential method[92], which allows us to consider a complete configuration $3s^2 3p^6 3d^{10} 4s^1 4p^0$ for the pseudopotential generation of Cu without further changes in bulk computational codes.

Similarly, the 4s and 4p semi-core states should also be included into the valence for the pseudopotential generation of Ag. Unfortunately, multi-reference pseudopotential scheme only yields good results for the 3d transition metals, so I have to use the traditional method to generate the pseudopotential for Ag(i.e. I consider an ionized configuration $4s^2 4p^6 4d^{10}$). In addition, I notice that the inclusion of partial core correction(PCC) [97] in the pseudopotential of Ag has an influence on the band energies in the DFT calculation of Ag.

The choice of matching radius r_c is very important. So many trials and tests with different cutoff radii have been done. In this thesis, the following cutoff radii are used for the Cu and Ag in the pseudopotential generation: Ham and MRPP represent the Hamann[38] and multi-reference schemes[92], respectively. All the pseudopotentials are norm-conserving, and have been generated by using a public available code APE(atomic pseudopotential generation)

B.2 Gaussian fitting

The pseudopotential generated by APE can be used directly for some plane wave codes. This pseudopotential for plane wave basis sets should be fitted for Gaussian

basis sets, since we employ a localized Gaussian basis sets in the quasi-particle bandstructures and lifetimes calculation.

$$V_l = V_{loc}(r) + \Delta\hat{V}_l \quad (\text{B.1})$$

The core potential is thought of as originating from Gaussian-type effective core charges:

$$V_{loc}(r) = -\frac{Z_v}{r} \left\{ \sum_{i=1}^2 c_i \text{erf}[(\sqrt{\alpha_i^{loc}})r] \right\} \quad (\text{B.2})$$

where Z_v is the Valence charge, $c_1 + c_2 = 1$, erf is the error function, given by

$$\text{erf}(x) = \frac{2}{\pi} \int_0^x \exp(-t^2) dt \quad (\text{B.3})$$

The remaining potential is expanded in Gaussian-type functions as

$$\Delta\hat{V}_l = \sum_{i=1}^n (A_i + A_{i+n}) e^{-\alpha_i^l \cdot r^2} \quad (\text{B.4})$$

where A_i and A_{i+n} are the linear coefficients, which correspond to the decay constants α_i^l for each angular momentum l value. Here I set $n = 5$, and $n = 4$ for Cu and Ag, respectively.

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Eidesstattliche Erklärung

Hiermit erkläre ich, dass ich die vorliegende Arbeit selbstständig und ohne fremde Hilfe verfasst, andere als die von mir angegebenen Quellen und Hilfsmittel nicht benutzt und die den benutzten Werken wörtlich oder inhaltlich entnommenen Stellen als solche kenntlich gemacht und keine vorherigen Promotionsversuche unternommen habe.

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