Phys 435

TOPICS IN CONDENSED MATTER

University of Waterloo

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Disclaimer

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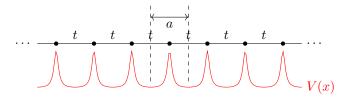
1 Introduction

The condensed in condensed matter physics refers to the fact that a condensed material is anything that is not a gas (i.e. liquid or solids). In this class we will be mostly interested in solids. The main qualifying feature of solids are their rigidity, (i.e. a solid has not only a fixed volume, like a liquid, but also a fixed shape). Rigidity is a consequence of the fact that atoms in solids form an ordered crystal lattice. Localized to each atom is a bunch of electrons; some in inner orbitals and some in the outer orbitals. Electrons in outer orbitals may be weakly bound to their respective atoms and be able to move around the lattice structure. These electrons are called valence electrons.

2 Effects of the Electronic Structure Topology

2.1 Toy Model of a Solid

A large part of solid state physics is related to the motion of valence electrons in a periodic potential of ionized atoms of the crystal lattice. Consider the simplest toy model of a solid: a one dimensional material with uniform spacing among the electrons with periodic potential. Let the uniform spacing between neighboring electrons be denoted a and we refer the region within distance a/2 of each atom as the **primitive unit cell** or simply unit cell of the lattice. In this model (an in most of the lattice models considered in the class) we will have the working assumption that each electron is localized to an atom. This can be understood if the periodic potential V(x) is very large between the atoms and nearly zero at the location of each atom.



Now consider an electron moving in this potential. Quantum mechanically we anticipate any electron motion to be due to the effects of quantum tunneling. Specifically, the is expected that if an electron were to tunnel at all, it would be most likely to tunnel to its nearest neighboring atoms with some amplitude t. If we let $|n\rangle$ denote the state of an electron localized to the n-th atom $(n \in \mathbb{Z})$ at position na in this lattice we can consider two possible ways any electron can tunnel away from $|n\rangle$.

$$|n\rangle \xrightarrow{t} |n+1\rangle$$
 transition to the right

$$|n\rangle \xrightarrow{t} |n-1\rangle$$
 transition to the left

Therefore, one can write the Hamiltonian for this system as follows,

$$H = -t \sum_{n} \left(\underbrace{|n+1\rangle\langle n|}_{\longrightarrow} + \underbrace{|n-1\rangle|n\rangle}_{\longleftarrow} \right)$$
 (2.1)

Where the preceding minus is simply convention. Notice that we are omitting the localized terms in this Hamiltonian because we can take the potential energy to be zero.

$$\langle n|H|n\rangle = \langle n|V|n\rangle = 0$$
 Since $V(na) \equiv 0$

In order to find the energy eigenstates of the Hamiltonian eq. (2.1) we need to find states $|\psi\rangle$ such that $H|\psi\rangle = E|\psi\rangle$. Evidently, eq. (2.1) is non-diagonal in the $|n\rangle$ basis. Instead consider a different basis indexed by k such that,

$$|n\rangle = \sum_{k} |k\rangle\langle k|n\rangle \tag{2.2}$$

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If this new basis represents momentum states, then $\langle k|n\rangle$ has a specific form. Motivated by free electrons, $\langle k|x\rangle \propto e^{-ikx}$ we map x to the position in the crystal:

$$x \mapsto na$$

Therefore the coefficients in eq. (2.2) have similar form,

$$\langle k|n\rangle \propto e^{-ikna}$$
 (2.3)

The normalization factor comes from completeness,

$$\sum_{n} |\langle n|k\rangle|^2 = 1 \tag{2.4}$$

Heretofore, no restrictions have been made on the range of n. Evidently, a real substance will possess finite, albeit many, unit cells (atoms). Let $n \in \mathbb{Z}$ be an integer but equip this model with periodic boundary conditions such that there are N distinct position states representing the number of unit cells. This periodic boundary condition can be written as,

$$|n\rangle = |n+N\rangle \tag{2.5}$$

Letting $\gamma \in \mathbb{R}$ be the normalization coefficient in eq. (2.3), eq. (2.4) becomes,

$$1 = \sum_{n=1}^{N} \left| \gamma e^{ikna} \right|^2 = N\gamma^2 \implies \gamma = \frac{1}{\sqrt{N}}$$

Consequently, eq. (2.2) can be written as follows,

$$|n\rangle = \frac{1}{\sqrt{N}} \sum_{k} e^{-ikna} |k\rangle \tag{2.6}$$

Additionally, the periodic boundary condition of eq. (2.5) gives,

$$|n\rangle = |n+N\rangle = \frac{1}{\sqrt{N}} \sum_k e^{-ik(n+N)a} |k\rangle = \frac{1}{\sqrt{N}} \sum_k e^{-ikna} |k\rangle$$

Therefore $e^{-ikNa} = 1$ which gives,

$$k = \frac{2\pi m}{Na} \qquad m \in \mathbb{Z} \tag{2.7}$$

Therefore k must be discrete. Note that this assumption was already made above when we insisted that eq. (2.2) as a new basis for the same Hilbert space. In fact, eq. (2.6) is simply a Fourier transform that is commonly called an **inverse lattice Fourier transform**. Additionally, combining this with dimensionality of $|n\rangle$ gives the following constraint on the values of k,

$$-\frac{\pi}{a} \le k < \frac{\pi}{a}$$

Effectively corresponds to demanding that there are N possible distinct values of k.

$$m \in \left\{ \frac{-N}{2}, \frac{-N+1}{2}, \dots, \frac{N-1}{2}, \frac{N}{2} \right\}$$

$$k \in \left\{ \frac{2\pi}{Na} \frac{-N}{2}, \frac{2\pi}{Na} \frac{-N+1}{2}, \dots, \frac{2\pi}{Na} \frac{N-1}{2}, \frac{2\pi}{Na} \frac{N}{2} \right\}$$
(2.8)

In order to substitute eq. (2.6) into eq. (2.1), first take the dual of eq. (2.6),

$$\langle n| = \frac{1}{\sqrt{N}} \sum_{k} e^{ikna} \langle k| \tag{2.9}$$

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Therefore eq. (2.1) becomes,

$$H = -t\frac{1}{N} \sum_{n} \sum_{k,k'} \left\{ \left(e^{-ik(n+1)a} |k\rangle \right) \left(e^{ik'na} \langle k'| \right) + \left(e^{-ik(n-1)a} |k\rangle \right) \left(e^{ik'a} \langle k'| \right) \right\}$$

$$H = -t\frac{1}{N} \sum_{n} \sum_{k,k'} \left\{ e^{-ik(n+1)a} e^{ik'na} |k\rangle \langle k'| + e^{-ik(n-1)a} e^{ik'a} |k\rangle \langle k'| \right\}$$

$$(2.10)$$

By eq. (2.7), (k - k') is an integer multiple of $2\pi/Na$. Letting that integer be μ we have the following identity:

$$\frac{1}{N} \sum_{n=1}^{N} e^{i(k-k')na} = \frac{1}{N} \sum_{n=1}^{N} e^{i2\pi\mu n/N} = \delta_{\mu} = \delta_{k,k'}$$
 (2.11)

Therefore eq. (2.10) simplifies greatly,

$$H = -t \sum_{k,k'} \delta_{k,k'} \left\{ e^{-ika} |k\rangle \langle k'| + e^{ika} |k\rangle \langle k'| \right\}$$

$$H = -t \sum_{k} \left\{ e^{-ika} + e^{ika} \right\} |k\rangle \langle k|$$

$$H = -2t \cos(ka) \sum_{k} |k\rangle \langle k|$$
(2.12)

Therefore eq. (2.12) is the diagonalized form of eq. (2.1). By diagonalizing the Hamiltonian we were able to determine the energy levels $\varepsilon(k)$ of the various states,

$$\varepsilon(k) = -2t\cos(ka) \tag{2.13}$$

The momentum space index k is the wave-vector with $p = \hbar k$ where p is an ordinary linear momentum. Additionally, t acts as a tunneling coefficient that dictates a tunneling rate (up to a constant \hbar) for the electrons in the solid. Unlike free particles, the momentum k in this toy model is confined to a discrete region.

$$-\frac{\pi}{a} \le k < \frac{\pi}{a}$$

This interval is called the **first Brillouin zone**. To highlight this difference, we sometimes refer to p in this model as the **crystal momentum** of the electron in state $|k\rangle$.

Moreover, the periodic boundary conditions restrict k to take on discrete and finite values pursuant to eq. (2.8). Note that L = Na is the size of the crystal, N is the number of atoms and a is the interval between two atoms in the solid.

The states that diagonalized the Hamiltonian are called **Bloch states** and are denoted $|k\rangle$ where is given by the inverse crystal Fourier transform (inverse of eq. (2.6)),

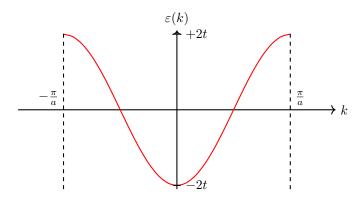
$$|k\rangle = \frac{1}{\sqrt{N}} \sum_{n} |n\rangle e^{ikna}$$

Which is a **lattice Fourier transform**. Since eq. (2.13) takes on the form of a cosine with amplitude 2t the accessible energy levels are confined to the finite interval,

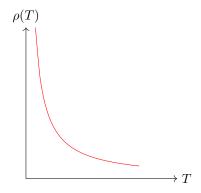
$$\varepsilon_{\min} = -2t \qquad \varepsilon_{\max} = +2t$$

The interval has a width of $\varepsilon_{\text{max}} - \varepsilon_{\text{min}} = 4t$ and is referred to as the allowed energy band.

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Given eq. (2.8), there are N distinct Bloch vector states $|k\rangle$. Suppose now that we have 1 electron per atom (instead of 1 electron in total). We recall the **Pauli principle** which states that only 1 electron can occupy a given state in a band. If there is one electron per atom and N total atoms, there are N electrons. If we recall the fact that electrons are spin-1/2 fermions, each momentum state corresponds to two distinct spin states. Therefore in total there are 2N states accessible to the electrons and therefore the band is half-filled.



Of course, this holds when the electrons in the material are at zero temperature where the electrons occupy all of the lowest possible energy states. For a half-filled energy band, all of the *negative* energy states are filled and the positive energy states are vacant. This separation defines the **Fermi energy** for this system which occurs at the interface between filled energy states and unfilled energy states ($\varepsilon_F = 0$ in this problem). In order to find the state that corresponds to this upper limit one needs to solve,

$$\varepsilon(k) = \varepsilon_{\rm F} = 0 \implies -2t\cos(ka) = 0$$

Therefore the value of k that solves this equation is,

$$k = \pm \frac{\pi}{2a} = \pm k_{\rm F}$$

Where $k_{\rm F}$ is given a special name: the **Fermi wave-vector** (Fermi momentum).

 $|k| < k_{\rm F}$: filled states $|k| > k_{\rm F}$: empty states

The **Fermi surface** defines the surface in momentum space separating the filled states from the unfilled states. Most of the observed properties of metals follow from the existence of the Fermi surface.

This concept is so important that it is worth measuring the volume in momentum space corresponding to the filled states. This is the volume enclosed by the Fermi surface. In this example, the volume in momentum space is characterized by the $2k_{\rm F}$ interval. Since $k=2\pi m/Na$, the volume per single k-value is $2\pi/Na$.

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Letting n = 1/a be the linear density of electrons per unit length, we can compute the total number of electrons:

$$N = nNa = 2\frac{2k_{\rm F}}{2\pi/Na}$$

Which allows one to calculate $k_{\rm F}$ in terms of the electron density n,

$$k_{\rm F} = \frac{\pi}{2}n$$

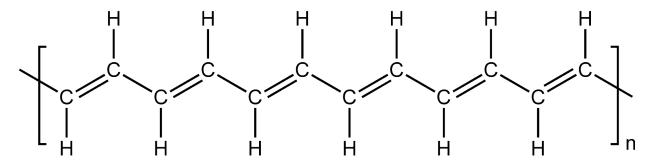
Which makes sense for this model because there is one electron per atom making n = 1/a. This result is called **Luttinger's theorem** which states that the volume enclosed by the Fermi surface (sometimes called the Fermi sea) is directly proportional to the electron density.

2.2 Polyacetylene

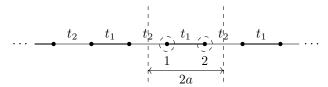
In section 2.1 we considered a one dimensional toy model of electrons in a solid. This model is highly idealized and is fair from a description of a real material. It is possible however to modify the model analyzed in section 2.1 to consider the case of the one dimensional polymer called polyacetylene. Polyacetylene consists of weakly-interacting chains of CH units.

$$C: 1s^2 2s^2 2p^2$$
 4 valence electrons

Electrons found in the inner orbitals of the atoms are tightly bound to their nuclei and therefore only the valence electrons in carbon are free to travel throughout the lattice.



Three of the valence electrons are engaged in bonding with neighboring carbon and hydrogen atoms while the 4-th one is free to move around. As it turns out, the bonds between carbon atoms possess alternating tunneling amplitudes.



In Polyacetylene, the tunneling amplitude between the carbon atoms joined by a double bound is slightly higher $t_1 > t_2$. This is referred to as **Peierls instability**.

Since this deviation is only slight, let $2\delta t = t_1 - t_2$ be a small perturbation such that,

$$t_1 = t + \delta t \qquad t_2 = t - \delta t \tag{2.14}$$

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Another important difference is that the primitive unit cell has size 2a (instead of a). This two atom basis can be arranged on a lattice with lattice constant 2a. This is referred to as a **Bravais lattice**. Notationally, we can describe any crystal as,

$$crystal = (basis, Bravais lattice)$$

To describe an electron in this lattice, we need two indices: one to describe the unit cell the electron is in (n) and one to describe which atom (1 or 2) the electron is located at within the unit cell. Therefore the Hamiltonian needs to characterize a number of possible transitions,

 $\begin{array}{ll} |n,1\rangle \xrightarrow{t_1} |n,2\rangle & \text{transition to the right within a unit cell} \\ |n,2\rangle \xrightarrow{t_1} |n,1\rangle & \text{transition to the left within a unit cell} \\ |n,1\rangle \xrightarrow{t_2} |n-1,2\rangle & \text{transition to the left across a unit cell} \\ |n,2\rangle \xrightarrow{t_2} |n+1,1\rangle & \text{transition to the right across a unit cell} \end{array}$

Just as before in section 2.1, we only consider the tunneling of electrons for nearest neighbors. This will be a recurring motif of this class which is formally called the **tight-binding model**.

$$H = \sum_{n} \{-t_1|n, 2\rangle\langle n, 1| - t_2|n - 1, 2\rangle\langle n, 1| - t_1|n, 1\rangle\langle n, 2| - t_2|n + 1, 1\rangle\langle n, 2|\}$$
(2.15)

Just as before, we need to write $|n,\alpha\rangle$ in the momentum basis (a new basis for each $\alpha\in\{1,2\}$),

$$|n,\alpha\rangle = \sum_{k} |k,\alpha\rangle\langle k,\alpha|n,\alpha\rangle$$
 (2.16)

where $\langle k, \alpha | n, \alpha \rangle$ has a specific form. Motivated by free electrons, $\langle k | x \rangle \propto e^{-ikx}$ we map x to the position in the crystal:

$$x \mapsto n \cdot 2a$$

Noting the factor of 2 because the unit call is twice as long as before. The coefficients in eq. (2.16) have similar form,

$$\langle k, \alpha | n, \alpha \rangle \propto e^{-i2kna}$$
 (2.17)

The normalization factor comes from completeness,

$$\sum_{n} |\langle n, \alpha | k, \alpha \rangle|^2 = 1 \tag{2.18}$$

Analogously to section 2.1, let $n \in \mathbb{Z}$ be an integer but equip this model with periodic boundary conditions such that there are N distinct positions for the electron for each $\alpha \in \{1,2\}$ representing the number of unit cells. It is important to note that here N does not represent the total number of atoms; N is the total number of unit cells (equivalently the total number of distinct $|n,\alpha\rangle$ states for $fixed\ \alpha$). Moreover the extra factor of 2 in the exponential is due to the increases unit cell size. Therefore the periodic boundary conditions for each α are,

$$|n,\alpha\rangle = |n+N,\alpha\rangle \tag{2.19}$$

Letting $\gamma \in \mathbb{R}$ be the normalization coefficient in eq. (2.17), eq. (2.18) becomes,

$$1 = \sum_{n=1}^{N} \left| \gamma e^{i2kna} \right|^2 = N\gamma^2 \implies \gamma = \frac{1}{\sqrt{N}}$$

Consequently, eq. (2.16) can be written as follows,

$$|n,\alpha\rangle = \frac{1}{\sqrt{N}} \sum_{k} e^{-i2kna} |k,\alpha\rangle$$

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Additionally, the periodic boundary condition of eq. (2.19) gives,

$$|n,\alpha\rangle = |n+N,\alpha\rangle = \frac{1}{\sqrt{N}} \sum_k e^{-i2k(n+N)a} |k,\alpha\rangle = \frac{1}{\sqrt{N}} \sum_k e^{-i2kna} |k,\alpha\rangle$$

Therefore $e^{-i2kNa} = 1$ which gives,

$$k = \frac{\pi m}{Na} \qquad m \in \mathbb{Z} \tag{2.20}$$

Therefore k must be discrete (an assumption made above to write down eq. (2.16)). Additionally, combining this with dimensionality of $|n, \alpha\rangle$ gives the following constraint on k,

$$-\frac{\pi}{2a} \le k < \frac{\pi}{2a}$$

Note the Brillouin zone for Polyacetylene is half the width of the Brillouin zone for the toy model discussed in section 2.1 (again this is due to the unit cell being twice the size). Thus ensures there are N states of the form $|k, \alpha\rangle$.

Substituting these results into eq. (2.15) gives,

$$H = -t_{1} \frac{1}{N} \sum_{n} \sum_{k_{1},k_{2}} |k_{1},2\rangle \langle k_{2},1| e^{-2ik_{1}na} e^{2ik_{2}na}$$

$$-t_{2} \frac{1}{N} \sum_{n} \sum_{k_{1},k_{2}} |k_{1},2\rangle \langle k_{2},1| e^{-2ik_{1}(n-1)a} e^{2ik_{2}na}$$

$$-t_{1} \frac{1}{N} \sum_{n} \sum_{k_{1},k_{2}} |k_{1},1\rangle \langle k_{2},2| e^{-2ik_{1}na} e^{2ik_{2}na}$$

$$-t_{2} \frac{1}{N} \sum_{n} \sum_{k_{1},k_{2}} |k_{1},1\rangle \langle k_{2},2| e^{-2ik_{1}na} e^{2ik_{2}(n-1)a}$$

$$(2.21)$$

$$H = \frac{1}{N} \sum_{n} \sum_{k_1, k_2} |k_1, 1\rangle \langle k_2, 2| \left\{ -t_1 e^{i2(k_2 - k_1)na} - t_2 e^{-i2k_1 a} e^{i2(k_2 - k_1)na} \right\} + \cdots$$

$$\cdots + \frac{1}{N} \sum_{n} \sum_{k_1, k_2} |k_1, 2\rangle \langle k_2, 1| \left\{ -t_1 e^{i2(k_2 - k_1)na} - t_2 e^{+i2k_1 a} e^{i2(k_2 - k_1)na} \right\}$$
(2.22)

By eq. (2.20), $(k_2 - k_1)$ is an integer multiple of π/Na . Letting that integer be μ we have the following identity:

$$\frac{1}{N} \sum_{n=1}^{N} e^{i2(k_2 - k_1)na} = \frac{1}{N} \sum_{n=1}^{N} e^{i2\pi \mu n/N} = \delta_{\mu} = \delta_{k_1, k_2}$$

Therefore eq. (2.22) simplifies greatly,

$$H = -t_1 \sum_{k} \{ |k, 2\rangle \langle k, 1| + |k, 1\rangle \langle k, 2| \}$$

$$-t_2 \sum_{k} \{ |k, 2\rangle \langle k, 1| e^{2ika} + |k, 1\rangle \langle k, 2| e^{-2ika} \}$$
(2.23)

At this point eq. (2.23) has been diagonalized with respect to the momentum index k (it is a simple sum over k). It is not however diagonalized completely. To diagonalize this Hamiltonian further, we can write $H = \sum_k H(k)$ where H(k) is the summand of eq. (2.23). Henceforth, the dependence of k can be omitted from the states $|k,\alpha\rangle$ for convenience.

$$H(k) = -\left(t_1 + t_2 e^{-i2ka}\right)|1\rangle\langle 2| - \left(t_1 + t_2 e^{+i2ka}\right)|2\rangle\langle 1|$$

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Notice that H(k) can be written as a 2×2 matrix as follows,

$$H(k) = {1 \atop 2} \left(\begin{array}{cc} 1 & 2 \\ 0 & -(t_1 + t_2 e^{-i2ka}) \\ -(t_1 + t_2 e^{+i2ka}) & 0 \end{array} \right)$$
 (2.24)

Since H(k) represents a two level system, we can therefore write H(k) as a sum over the three Pauli matrices with the following spin-correspondence (which we refer to as a **pseudo-spin**),

$$|1\rangle = |\uparrow\rangle \qquad |2\rangle = |\downarrow\rangle$$

The Pauli matrices are,

$$\sigma_x = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix} = |1\rangle\langle 2| + |2\rangle\langle 1|$$

$$\sigma_y = \begin{pmatrix} 0 & -i \\ i & 0 \end{pmatrix} = -i(|1\rangle\langle 2| - |2\rangle\langle 1|)$$

$$\sigma_z = \begin{pmatrix} 1 & 0 \\ 0 & -1 \end{pmatrix} = |1\rangle\langle 1| - |2\rangle\langle 2|$$

$$(2.25)$$

Upon close inspection of eq. (2.24), it becomes easy to identity the Pauli's within H(k). Explicitly the t_1 term is easy,

$$H(k) = \epsilon \sigma_z - t_1 \sigma_x - t_2 \begin{pmatrix} 0 & e^{-i2ka} \\ e^{+i2ka} & 0 \end{pmatrix}$$

The remaining decomposition follows from $e^{\pm ix} = \cos(x) \pm i \sin(x)$

$$H(k) = \epsilon \sigma_z - t_1 \sigma_x - t_2 \cos(2ka)\sigma_x - t_2 \sin(2ka)\sigma_y$$

Alternatively it is possible to finding this correspondence by making use of (pseudo-)ladder operators of eq. (2.25),

$$|1\rangle\langle 2| = \frac{1}{2}(\sigma_x + i\sigma_y)$$
 $|2\rangle\langle 1| = \frac{1}{2}(\sigma_x - i\sigma_y)$

We therefore define d(k) as an effective magnetic field such that,

$$\mathbf{d}(k) = \begin{pmatrix} d_x(k) \\ d_y(k) \\ d_z(k) \end{pmatrix} = \begin{pmatrix} -t_1 - t_2 \cos(2ka) \\ -t_2 \sin(2ka) \\ 0 \end{pmatrix}$$
(2.26)

Therefore $H(k) = d(k) \cdot \sigma$. This system is *equivalent* to a spin-1/2 particle in a magnetic field of the form d(k). The eigenstates of H(k) correspond to $\langle \sigma \rangle$ along d or in the opposite direction to d.

$$\varepsilon_{-}(k) = -|\boldsymbol{d}(k)|$$
 $\varepsilon_{+}(k) = +|\boldsymbol{d}(k)|$

Using eq. (2.14),

$$d_x(k) = -(t + \delta t) - (t - \delta t)\cos(2ka)$$

$$d_y(k) = -(t - \delta t)\sin(2ka)$$

$$d_z(k) = 0$$
(2.27)

This is contrasting to the previous model; before we had a *single* energy band pursuant to eq. (2.13), but here we have *two energy bands*. This result can be generalized to the following observation:

Every atom in the unit cell will give rise to a separate band.

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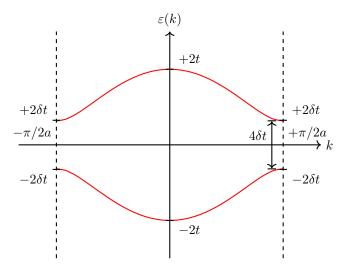
Explicitly, $|d(k)|^2$ can be calculated from eq. (2.27),

$$|\mathbf{d}(k)|^2 = d_x^2(k) + d_y^2(k) + d_z^2(k) = 4t^2 \cos^2(ka) + 4\delta t^2 \sin^2(ka)$$

Which makes the two bands have the form,

$$\varepsilon_{\pm}(k) = \pm 2\sqrt{t^2 \cos^2(ka) + \delta t^2 \sin^2(ka)}$$

Which when plotted is,

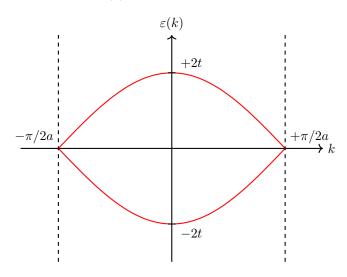


Notice that the first Brillouin zone is $\frac{\pi}{2a} \le k \le \frac{\pi}{2a}$ is *smaller* than it was previously. In this model, there is one electron per atom and thus two electrons per unit cell. Therefore there are 2N electrons. Accounting for spin degeneracies, there are 2N accessible states per energy band. At zero temperature, the entire lower band is filled and the upper band is empty. This makes polyacetylene an **insulator** because it requires an energy of at least $4\delta t$ in order to transition an electron across the Fermi surface.

In the limit of the $\delta t \to 0$ one should expect to recover eq. (2.13) from section 2.1,

$$\lim_{\delta t \to 0} \varepsilon_{\pm}(k) = \pm 2 \lim_{\delta t \to 0} \sqrt{t^2 \cos^2(ka) + \delta t^2 \sin^2(ka)} = \pm 2t \cos(ka)$$

As expected. Notice however the plot of $\varepsilon(k)$ when $\delta t = 0$ differs from the one depicted in section 2.1,



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The resolution between these two plots is to notice that in the polyacetylene model, the unit cell had with 2a instead of a. Therefore in momentum space, the plotted values for k are bounded by $-\pi/2a \le k \le \pi/2a$ instead of $-\pi/a \le k \le \pi/a$. If one were to unravel the upper band into regions where $|k| > \pi/2a$ one would recover the initial model.

Heretofore, we have assumed that $\delta t > 0$. If however we consider $\delta t < 0$, then we transition from $t_1 > t_2$ to $t_2 > t_1$ pursuant to eq. (2.14). And our diagram is adjusted

Therefore, the ground state of polyacetylene is doubly degenerate, corresponding to the two signs of δt . As it turns out, this difference in sign is related to topological properties in momentum space. Upon examination of eq. (2.26) once can notice how d depends on δt . How does d(k) change as k goes from $-\pi/2a$ to $\pi 2a$?

$$d_x(k) = -(t + \delta t) - (t - \delta t)\cos(2ka)$$
$$d_y(k) = -(t - \delta t)\sin(2ka)$$

Checking specific points,

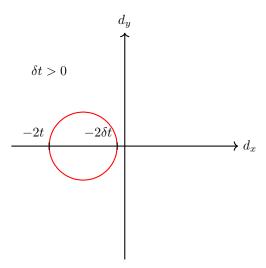
$$d_x\left(\pm \frac{\pi}{2a}\right) = -(t+\delta t) + (t-\delta t) = -2\delta t$$

$$d_y\left(\pm \frac{\pi}{2a}\right) = 0$$

$$d_x(0) = -(t+\delta t) - (t-\delta t) = 2t$$

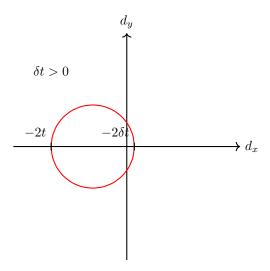
$$d_y(0) = 0$$

Plotted the case of $\delta t > 0$ in d_y, d_x space,



It can be seen that when $\delta t > 0$, the curve in d_x, d_y does not enclose the origin $\mathbf{d} = 0$. Recalling that $\varepsilon_{\pm}(k) = \pm |\mathbf{d}(k)|$, as long as $\mathbf{d}(k) \neq 0$, there exists a energy gap between the two bands ε_{+} and ε_{-} . Therefore for $\delta t > 0$ polyacetylene is an ordinary insulator. However, in the case where $\delta t < 0$,

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It can be seen that the origin |d| = 0 is enclosed. This *topological* difference is what leads us to call the $\delta t < 0$ case as a **topological insulator** as apposed to an **ordinary insulator**.

Effectively, there is no possible way to continuously deform the Hamiltonian (i.e. changing δt) without merging the two energy bands; the transition from $\delta t > 0$ to $\delta t < 0$ permits the band gap to close temporarily.



An observable feature of topological insulators becomes evident for the edge states at zero energy (i.e. in the middle of a band gap). In this case of polyacetylene, these states that are closest to the band gap are those that live on the edge of the Brillouin zone where $|k - \pi/2a|$ is small. In order to explore the physics of these regions, define δk such that,

$$k = \frac{\pi}{2a} + \delta k$$

Where δk is small. In this case, the Hamiltonian $H(k) = d(k) \cdot \sigma$ can be expanded out explicitly,

$$d_x(k) = d_x \left(\frac{\pi}{2a} + \delta k\right)$$

$$= -(t + \delta t) - (t - \delta t)\cos(\pi + 2\delta ka)$$

$$= -(t + \delta t) + (t - \delta t)\cos(2\delta ka)$$

$$= -(t + \delta t) + (t - \delta t) + \mathcal{O}\left((\delta k)^2\right)$$

$$\simeq -2\delta t$$

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Similarly for d_y ,

$$d_y(k) = d_y \left(\frac{\pi}{2a} + \delta k\right)$$

$$= -(t - \delta t) \sin(\pi + 2\delta ka)$$

$$= (t - \delta t) \sin(2\delta ka)$$

$$= (t - \delta t) 2\delta ka + \mathcal{O}\left((\delta k)^3\right)$$

$$\approx 2t\delta ka$$

Therefore,

$$H(\delta k) = -2\delta t \sigma_x + 2ta\delta k \sigma_y$$

In order to draw familiarity with systems that have been studied previously, let use rename a number of terms,

$$2\delta t = m$$
 $2ta = \hbar v_F$

Where m has units of energy and v_F has units of velocity. The Hamiltonian can now be written as,

$$H = -m\sigma_x + \hbar v_F \delta k \sigma_y$$

Since $\hbar \delta k$ has units of momentum, we simply label it p. Then the Hamiltonian can be written as,

$$H = v_F p \sigma_y - m \sigma_x$$

Which is identical to the Dirac equation (1D) for a *relativistic* particle of $mass\ m$ and velocity v_F in place of the speed of light c. These velocity v_F is called the **Fermi velocity**.

$$\varepsilon_{\pm}(p) = \pm \sqrt{v_F^2 p^2 + m^2}$$

Consider the interface between 2 polyacetylene samples $\delta t > 0$ and $\delta t < 0$. In the position basis $p \to -i\hbar \partial_x$ we can write our Dirac Hamiltonian as,

$$H = -i\hbar v_F \sigma_y \partial_x - m(x)\sigma_x$$

Where m has an x dependence.

$$H\psi = E\psi$$

With E = 0 localized near x = 0.

$$[-i\hbar v_F \sigma_u \partial_x - m(x)\sigma_x]\psi(x) = 0$$

Therefore we should look for a solution of the form,

$$\psi(x) = e^{f(x)} \sigma_y |z\rangle$$

Where $|z\rangle$ is written in σ_z basis,

$$|z\rangle = z_{\uparrow}|\uparrow\rangle + z_{\perp}|\downarrow\rangle$$

Using $\sigma_y^2 = \mathbf{1}$ and $\sigma_x \sigma_y = i \sigma_z$ multiply by both sides by σ_y ,

$$\left[-i\hbar v_F \frac{\mathrm{d}f}{\mathrm{d}x} + im(x)\sigma_z\right] e^{f(x)}|z\rangle = 0$$

Dropping the exponential $e^{f(x)}$,

$$\left[-i\hbar v_F \frac{\mathrm{d}f}{\mathrm{d}x} + im(x)\sigma_z \right] |z\rangle = 0$$

Adjusting the constant coefficients,

$$\left[\frac{\mathrm{d}f}{\mathrm{d}x} - \frac{m(x)}{\hbar v_F} \sigma_z\right] |z\rangle = 0 \tag{2.28}$$

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Recalling that,

$$\sigma_z|\uparrow\rangle = +|\uparrow\rangle$$
 $\sigma_z|\downarrow\rangle = -|\downarrow\rangle$

Assume $|z\rangle = |\uparrow\rangle$,

$$\left[\frac{\mathrm{d}f}{\mathrm{d}x} + \frac{m(x)}{\hbar v_F}\right]|\uparrow\rangle = 0$$

Therefore,

$$\frac{\mathrm{d}f}{\mathrm{d}x} = -\frac{m(x)}{\hbar v_F}$$

Can be solved using,

$$f(x) = -\frac{1}{\hbar v_F} \int_0^x \mathrm{d}x' m(x')$$

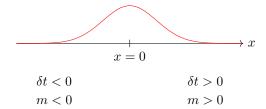
Where m(0) = 0. Altogether, the solution ψx can be written,

$$\psi(x) = e^{-\frac{1}{\hbar v_F} \int_0^x \mathrm{d}x' m(x')} \sigma_y |\uparrow\rangle$$

Since $\sigma_y |\uparrow\rangle = i |\downarrow\rangle$,

$$\psi(x) = ie^{-\frac{1}{\hbar v_F} \int_0^x dx' m(x')} |\downarrow\rangle$$

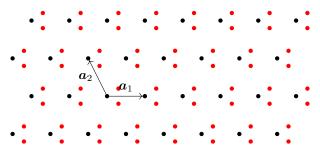
Of course an analogous solution exists for $|z\rangle = |\downarrow\rangle$.



2.3 Higher Dimensions

A crystal is an infinite, periodic array of identical groups of atoms,

crystal = (basis, Bravais lattice)



A Bravais lattice is all points in \mathbb{R}^n ($\mathbb{R}^1, \mathbb{R}^2, \mathbb{R}^3$, etc.) such that,

$$R = n_1 a_1 + n_2 a_2 + n_3 a_3 \tag{2.29}$$

Where a_1, a_2, a_3 are the primitive translation vectors of the Bravais lattice. Moreover the coefficients n_1, n_2, n_3 are integers,

$$n_1, n_2, n_3 = 0, \pm 1, \pm 2, \pm 3, \dots$$

Typically, the choice for a_i is not unique. However, the best choice for the primitive transition vectors is typically the one that exhibits the most symmetry.

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The **primitive unit cell** is the parallelepiped defined by the primitive translation vectors. A primitive unit cell is also not unique. A special type of unit cell, usually the most convenient, is the **Wigner-Seitz unit cell**. A Wigner-Seitz unit cell is the region of space about a Bravais lattice point that are closer to this point than to any other Bravais lattice point. As an example, the first Brillouin zone is the Wigner-Seitz cell in reciprocal (momentum) space. Analogously to eq. (2.29) which indexes lattice points in position space, we define the **reciprocal lattice** and all points in momentum space G such that,

$$G = m_1 b_1 + m_2 b_2 + m_3 b_3$$

Provided that we have the following correspondence between R and G,

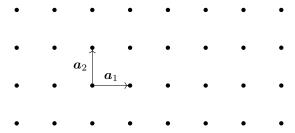
$$e^{i\mathbf{G}\cdot\mathbf{G}} = 1$$

This is equivalent to the following,

$$\boldsymbol{b}_i \cdot \boldsymbol{a}_j = 2\pi \delta_{ij} \quad \forall i, j$$

2.4 Electronic Structure for Square Lattice

The square lattice in 2 dimensions has two primitive basis vectors a_1, a_2 ,



Using Cartesian coordinates,

$$\mathbf{a}_1 = a\hat{\mathbf{x}}$$
 $\mathbf{a}_2 = a\hat{\mathbf{y}}$

The reciprocal lattice has the following form,

$$oldsymbol{b}_1 = rac{2\pi}{a} oldsymbol{\hat{x}} \qquad oldsymbol{b}_2 = rac{2\pi}{a} oldsymbol{\hat{y}}$$

The Hamiltonian for this system in the nearest neighbor approximation can be written as the sum over all possible transitions with tunneling coefficient t.

$$H = -t \sum_{\mathbf{R}, \mathbf{a}} (|\mathbf{R} + \mathbf{a}\rangle \langle \mathbf{R}| + |\mathbf{R}\rangle \langle \mathbf{R} + \mathbf{a}|)$$
(2.30)

Where $a \in \{a_1, a_2\}$. Notice that we only need to sum over terms of the form involving R + a and not R - a. If we were to include those terms, we would double-count all each of the possible transitions. In order to diagonalize eq. (2.30) we use the familiar Fourier transform but here we apply it as a vector change of coordinates,

$$|m{R}
angle = \sum_{m{R}} |m{k}
angle \langle m{k}|m{R}
angle$$

Where $\langle \mathbf{k} | \mathbf{R} \rangle$ is normalized in the usual fashion,

$$\langle \boldsymbol{k} | \boldsymbol{R} \rangle = \frac{1}{\sqrt{N}} e^{i \boldsymbol{k} \cdot \boldsymbol{R}}$$

Therefore,

$$|{\bf R}\rangle = \frac{1}{\sqrt{N}} \sum_{\bf R} |{\bf k}\rangle e^{i{\bf k}\cdot{\bf R}}$$

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In the reciprocal lattice, k can be written as,

$$\mathbf{k} = k_1 \mathbf{b}_1 + k_2 \mathbf{b}_2$$

Where k_1 and k_2 are bounded,

$$-\frac{1}{2} \le k_1, k_2 < \frac{1}{2}$$

Therefore,

$$\mathbf{k} \cdot \mathbf{R} = (k_1 \mathbf{b}_1 + k_2 \mathbf{b}_2) \cdot (n_1 \mathbf{a}_1 + n_2 \mathbf{a}_2)$$
$$\mathbf{k} \cdot \mathbf{R} = 2\pi (k_1 n_1 + k_2 n_2)$$

Using the $|\mathbf{k}\rangle$ basis, eq. (2.30) becomes,

$$H = -t \sum_{\mathbf{R}, \mathbf{a}} \sum_{\mathbf{k}, \mathbf{k}'} \left(|\mathbf{k}\rangle \langle \mathbf{k}'| e^{-i\mathbf{k}\cdot(\mathbf{R}+\mathbf{a})} e^{i\mathbf{k}'\cdot\mathbf{R}} + |\mathbf{k}\rangle \langle \mathbf{k}'| e^{-i\mathbf{k}\cdot\mathbf{R}} e^{i\mathbf{k}'\cdot(\mathbf{R}+\mathbf{a})} \right)$$
(2.31)

Using a multi-dimensional version of eq. (2.11),

$$\frac{1}{N} \sum_{\boldsymbol{R}} e^{-i(\boldsymbol{k} - \boldsymbol{k}') \cdot \boldsymbol{R}} = \delta_{\boldsymbol{k}, \boldsymbol{k}'}$$

Therefore eq. (2.30) becomes,

$$H = -t \sum_{\boldsymbol{k},\boldsymbol{a}} |\boldsymbol{k}\rangle \langle \boldsymbol{k}| \left(e^{-i\boldsymbol{k}\cdot\boldsymbol{a}} + e^{i\boldsymbol{k}\cdot\boldsymbol{a}}\right)$$

Recognize a cosine term,

$$H = -2t \sum_{\mathbf{k}, \mathbf{a}} |\mathbf{k}\rangle \langle \mathbf{k}| \cos(\mathbf{k} \cdot \mathbf{a})$$

The energy levels (as indexed by \boldsymbol{k}) can now be read off easily,

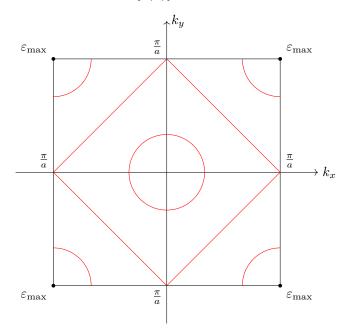
$$\varepsilon(\mathbf{k}) = -2t \sum_{\mathbf{a}} \cos(\mathbf{k} \cdot \mathbf{a}) = -2t [\cos(ak_x) + \cos(ak_y)]$$

Clearly, $\varepsilon(\mathbf{k})$ has minimum when $k_x = k_y = 0$. Therefore,

$$\min\{\varepsilon(\mathbf{k})\} = \varepsilon_{\min} = -4t$$

Likewise at the Brillouin zone corners,

$$\max\{\varepsilon(\mathbf{k})\} = \varepsilon_{\max} = 4t$$



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Expanding k near the minimum,

$$\varepsilon(\mathbf{k}) \simeq -4t + ta^2(k_x^2 + k_y^2) = -4t + ta^2k^2$$
 (2.32)

Notice that this looks like the dispersion of a free electron. Recall that for a free electron,

$$\varepsilon(\mathbf{k}) = \frac{\hbar^2 k^2}{2m}$$

Therefore we can define an effective mass m^* as follows,

$$\varepsilon(\mathbf{k}) - \varepsilon_{\min} = ta^2 k^2 = \frac{\hbar^2 k^2}{2m^*}$$

Where the effective mass is,

$$m^* = \frac{\hbar^2}{2ta^2}$$

In conclusion, electrons near the minimum energy level ε_{\min} behave like free particles with effective mass m^* .

The Fermi surface in this model can be calculated by setting $\varepsilon(\mathbf{k}) = \varepsilon_{\rm F}$,

$$\varepsilon(\mathbf{k}) = -4t + \frac{\hbar^2 k^2}{2m^*} = \varepsilon_{\mathrm{F}}$$

Therefore,

$$\frac{\hbar^2 k^2}{2m^*} = \varepsilon_{\rm F} - \varepsilon_{\rm min}$$

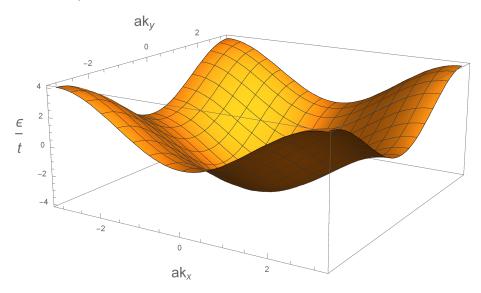
Defines the equation of a circle in k_x, k_y space with radius,

$$k_F = \sqrt{\frac{2m^*}{\hbar^2}(\varepsilon_{\rm F} - \varepsilon_{\rm min})}$$

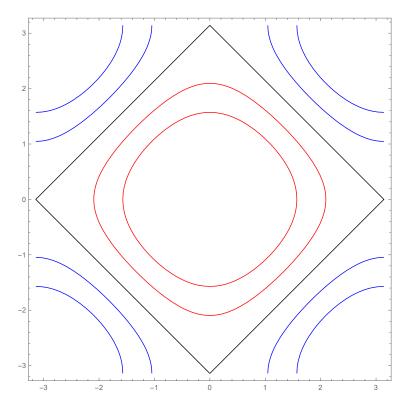
Recall from eq. (2.32) that this equation only defines the Fermi surface near the minimum energy level. In general,

$$\varepsilon(\mathbf{k}) = -2t(\cos(k_x a) + \cos(k_y a))$$

For a half-filled band, $\varepsilon_{\rm F}=0$, the Fermi surface is depicted above. The dispersion relation is,



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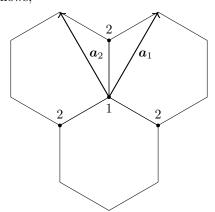


2.5 Graphene

The primitive unit vectors for Graphene are,

$$a_1 = \frac{a}{2} \left(\hat{\boldsymbol{x}} + \sqrt{3} \hat{\boldsymbol{y}} \right) \qquad a_2 = \frac{a}{2} \left(-\hat{\boldsymbol{x}} + \sqrt{3} \hat{\boldsymbol{y}} \right)$$
 (2.33)

Where the lattice structure is as follows,



$$\boldsymbol{b}_{1} = \frac{2\pi}{a\sqrt{3}} \left(\sqrt{3}\hat{\boldsymbol{x}} + \hat{\boldsymbol{y}} \right) \qquad \boldsymbol{b}_{2} = \frac{2\pi}{a\sqrt{3}} \left(-\sqrt{3}\hat{\boldsymbol{x}} + \hat{\boldsymbol{y}} \right)$$
(2.34)

The nearest neighbor Hamiltonian becomes,

$$H = -t \sum_{\mathbf{R}} (|\mathbf{R}, 2\rangle \langle \mathbf{R}, 1| + |\mathbf{R} - \mathbf{a}_1, 2\rangle \langle \mathbf{R}, 1| + |\mathbf{R} - \mathbf{a}_2, 2\rangle \langle \mathbf{R}, 1| + \text{h.c.})$$

$$(2.35)$$

In the reciprocal lattice,

$$|\boldsymbol{R}, \alpha\rangle = \frac{1}{\sqrt{N}} \sum_{\boldsymbol{R}} |\boldsymbol{k}, \alpha\rangle e^{-i\boldsymbol{k}\cdot\boldsymbol{R}}$$

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Therefore the Hamiltonian becomes,

$$H = -\frac{t}{N} \sum_{\mathbf{R}} \sum_{\mathbf{k}, \mathbf{k}'} \left(|\mathbf{k}, 2\rangle \langle \mathbf{k}, 1| e^{-i\mathbf{k} \cdot \mathbf{R}} e^{i\mathbf{k}' \cdot \mathbf{R}} + |\mathbf{k}, 2\rangle \langle \mathbf{k}, 1| e^{-i\mathbf{k} \cdot (\mathbf{R} - \mathbf{a}_1)} e^{i\mathbf{k}' \cdot \mathbf{R}} + |\mathbf{k}, 2\rangle \langle \mathbf{k}, 1| e^{-i\mathbf{k} \cdot (\mathbf{R} - \mathbf{a}_2)} e^{i\mathbf{k}' \cdot \mathbf{R}} + \text{h.c.} \right)$$

$$H = -t \sum_{\mathbf{k}} \left[|\mathbf{k}, 2\rangle \langle \mathbf{k}, 1| \left(1 + e^{i\mathbf{k}\cdot\mathbf{a}_1} + e^{i\mathbf{k}\cdot\mathbf{a}_2}\right) + |\mathbf{k}, 1\rangle \langle \mathbf{k}, 2| \left(1 + e^{-i\mathbf{k}\cdot\mathbf{a}_1} + e^{-i\mathbf{k}\cdot\mathbf{a}_2}\right) \right]$$

We have now diagonalized H in terms of the momentum wave-vector index k. Letting $H = \sum_{k} H(k)$ where,

$$H(\mathbf{k}) = -t \left[|2\rangle\langle 1| \left(1 + e^{i\mathbf{k}\cdot\mathbf{a}_1} + e^{i\mathbf{k}\cdot\mathbf{a}_2} \right) + |1\rangle\langle 2| \left(1 + e^{-i\mathbf{k}\cdot\mathbf{a}_1} + e^{-i\mathbf{k}\cdot\mathbf{a}_2} \right) \right]$$

Invoking pseudo-spin states $|1\rangle = |\uparrow\rangle$ and $|2\rangle = |\downarrow\rangle$ we can write,

$$H(\mathbf{k}) = \mathbf{d}(\mathbf{k}) \cdot \boldsymbol{\sigma}$$

Where d(k) is,

$$d^{x}(\mathbf{k}) = -t[1 + \cos(\mathbf{k} \cdot \mathbf{a}_{1}) + \cos(\mathbf{k} \cdot \mathbf{a}_{2})]$$

$$d^{y}(\mathbf{k}) = -t[\sin(\mathbf{k} \cdot \mathbf{a}_{1}) + \sin(\mathbf{k} \cdot \mathbf{a}_{2})]$$

$$d^{z}(\mathbf{k}) = 0$$

The eigenvalues of $H(\mathbf{k})$ are then the usual,

$$\varepsilon_{\pm}(\mathbf{k}) = \pm |\mathbf{d}(\mathbf{k})|$$

Since we have 1 electron per carbon atom, there are 2 electrons per unit cell which means that the ε_{-} band is completely filled while the ε_{+} is empty. For this configuration, the conduction band is empty and thus graphene is an insulator.

In order to analyze the electronic structure of graphene, we need to rewrite k which is expressed via the reciprocal lattice vectors b_1, b_2 in terms of the Cartesian vectors a_1, a_2 .

$$\boldsymbol{k} = k_1 \boldsymbol{b}_1 + k_2 \boldsymbol{b}_2$$

$$\mathbf{k} \cdot \mathbf{a}_1 = (k_1 \mathbf{b}_1 + k_2 \mathbf{b}_2) \cdot \mathbf{a}_1 = 2\pi k_1$$
 $\mathbf{k} \cdot \mathbf{a}_2 = 2\pi k_2$

Recalling eqs. (2.33) and (2.34),

$$oldsymbol{a}_1 = rac{a}{2} \Big(\hat{oldsymbol{x}} + \sqrt{3} \hat{oldsymbol{y}} \Big) \qquad oldsymbol{a}_2 = rac{a}{2} \Big(-\hat{oldsymbol{x}} + \sqrt{3} \hat{oldsymbol{y}} \Big)$$

$$oldsymbol{b}_1 = rac{2\pi}{a\sqrt{3}}\Big(\sqrt{3}oldsymbol{\hat{x}} + oldsymbol{\hat{y}}\Big) \qquad oldsymbol{b}_2 = rac{2\pi}{a\sqrt{3}}\Big(-\sqrt{3}oldsymbol{\hat{x}} + oldsymbol{\hat{y}}\Big)$$

We can now write k_x, k_y in terms of k_1, k_2 ,

$$k_x = \frac{2\pi}{\sqrt{3}a}(k_1 - k_2)$$
 $k_y = \frac{2\pi}{\sqrt{3}a}(k_1 + k_2)$

Inverting this system gives,

$$k_1 = \frac{a}{4\pi} \left(k_x + \sqrt{3}k_y \right)$$
 $k_2 = \frac{a}{4\pi} \left(-k_x + \sqrt{3}k_y \right)$

Therefore,

$$d^{x}(\mathbf{k}) = -t[1 + \cos(2\pi k_1) + \cos(2\pi k_2)]$$

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$$= -t \left[1 + 2\cos\left(\frac{k_x a}{2}\right) \cos\left(\frac{\sqrt{3}k_y a}{2}\right) \right]$$

Similarly for $d^y(\mathbf{k})$,

$$d^{y}(\mathbf{k}) = -2t \cos\left(\frac{k_{x}a}{2}\right) \sin\left(\frac{\sqrt{3}k_{y}a}{2}\right)$$

By examining these equations it can be seen that there are two points (and only two points) in the first Brillouin zone where both d^x and d^y vanish simultaneously. We label these points as \mathbf{k}_{\pm} ,

$$\mathbf{k}_{+} = (k_{+,x}, k_{+,y}) = \left(\frac{4\pi}{3a}, 0\right)$$

$$\mathbf{k}_{-} = (k_{-,x}, k_{-,y}) = \left(-\frac{4\pi}{3a}, 0\right)$$

At either of these points,

$$|d(\mathbf{k}_{\pm})| = \sqrt{d^{x^2}(\mathbf{k}) + d^{y^2}(\mathbf{k})} = 0$$

Therefore, the gap between the ε_+ band and the ε_- closes at only two points. Since graphene is an insulator and does not have a Fermi surface (meaning its *not* a metal) we refer to graphene, and materials that share these features, as **semimetals**.

The Brillouin zone of graphene can be derived using eq. (2.52) and eq. (2.54). First, rewrite k which is typically defined in terms of the reciprocal lattice vectors b_1, b_2 in terms of the Cartesian vectors a_1, a_2 .

$$\mathbf{k} = k_1 \mathbf{b}_1 + k_2 \mathbf{b}_2 \qquad \mathbf{b}_i \cdot \mathbf{a}_j = 2\pi \delta_{ij}$$

$$\mathbf{k} \cdot \mathbf{a}_1 = (k_1 \mathbf{b}_1 + k_2 \mathbf{b}_2) \cdot \mathbf{a}_1 = 2\pi k_1$$
 $\mathbf{k} \cdot \mathbf{a}_2 = 2\pi k_2$

Therefore using eq. (2.54) yields,

$$k_1 = \frac{a}{4\pi} \left(k_x + \sqrt{3} k_y \right)$$

$$k_2 = \frac{a}{4\pi} \left(-k_x + \sqrt{3} k_y \right)$$
(2.36)

Recalling eq. (2.53),

$$egin{align} oldsymbol{a}_1 &= rac{a}{2} \Big(\hat{oldsymbol{x}} + \sqrt{3} \hat{oldsymbol{y}} \Big) & oldsymbol{a}_2 &= rac{a}{2} \Big(-\hat{oldsymbol{x}} + \sqrt{3} \hat{oldsymbol{y}} \Big) \ oldsymbol{b}_1 &= rac{2\pi}{a\sqrt{3}} \Big(\sqrt{3} \hat{oldsymbol{x}} + \hat{oldsymbol{y}} \Big) & oldsymbol{b}_2 &= rac{2\pi}{a\sqrt{3}} \Big(-\sqrt{3} \hat{oldsymbol{x}} + \hat{oldsymbol{y}} \Big) \end{aligned}$$

We can now write k_x, k_y in terms of k_1, k_2 ,

$$k_x = \frac{2\pi}{a}(k_1 - k_2)$$
 $k_y = \frac{2\pi}{\sqrt{3}a}(k_1 + k_2)$

Using eq. (2.50),

$$1 = e^{iN_i \mathbf{k} \cdot \mathbf{a}_i} \implies N_i \mathbf{k} \cdot \mathbf{a}_i = N_i 2\pi k_i = 2\pi m_i \implies N_i k_i = m_i$$

Where $m_i \in \mathbb{Z}$ is an integer and $i \in \{1, 2\}$.

Moreover, since there are N_i unit cells in the a_i direction we have,

$$-\frac{N_i}{2} \le m_i < \frac{N_i}{2} \qquad m_i \in \mathbb{Z}, \forall i \in \{1, 2\}$$

Therefore,

$$-\frac{N_i}{2} \leq N_i k_i < \frac{N_i}{2} \qquad \forall i \in \{1,2\}$$

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$$-\frac{1}{2} \le k_i < \frac{1}{2} \qquad \forall i \in \{1, 2\}$$

This result is intuitive. The first Brillouin zone only extends halfway in the direction of b_1 , b_2 or combinations thereof. These inequalities yield the following,

$$|k_1|, |k_2| < \frac{1}{2}$$

$$\mathbf{k} = k_x \hat{\mathbf{x}} + k_y \hat{\mathbf{y}}$$

Where the norms of b_i are independent of i.

$$b^{2} = |\mathbf{b}_{1}|^{2} = |\mathbf{b}_{2}|^{2} = 4\frac{(2\pi)^{2}}{3a^{2}} = \frac{16\pi^{2}}{3a^{2}}$$
$$\frac{1}{2} \le \frac{\mathbf{k} \cdot \mathbf{b}_{i}}{b^{2}} = k_{i} \le \frac{1}{2}$$

Therefore,

$$-\frac{1}{2} \le \frac{1}{b^2} \left(\frac{2\pi}{a\sqrt{3}} \left(\sqrt{3}\hat{\boldsymbol{x}} + \hat{\boldsymbol{y}} \right) \right) \cdot \left(k_x \hat{\boldsymbol{x}} + k_y \hat{\boldsymbol{y}} \right) \le \frac{1}{2}$$
$$-\frac{1}{2} \le \frac{3a^2}{4(2\pi)^2} \left(\frac{2\pi}{a\sqrt{3}} \left(\sqrt{3}k_x + k_y \right) \right) \le \frac{1}{2}$$
$$-\frac{4}{\sqrt{3}} \pi \le a \left(\sqrt{3}k_x + k_y \right) \le \frac{4}{\sqrt{3}} \pi$$

Similarly,

$$-\frac{4}{\sqrt{3}}\pi \le a\Big(-\sqrt{3}k_x + k_y\Big) \le \frac{4}{\sqrt{3}}\pi$$

And the positive sum of these two inequalities gives,

$$-\frac{4}{\sqrt{3}}\pi \le ak_y \le \frac{4}{\sqrt{3}}\pi$$

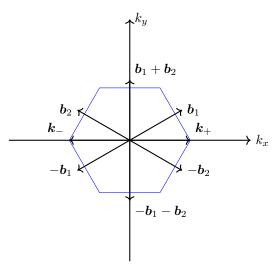
Which does not constrain the first Brillouin zone any further. Instead we invoke,

$$\frac{1}{2} \leq \frac{\boldsymbol{k} \cdot (\boldsymbol{b}_1 + \boldsymbol{b}_2)}{b^2} \leq \frac{1}{2}$$

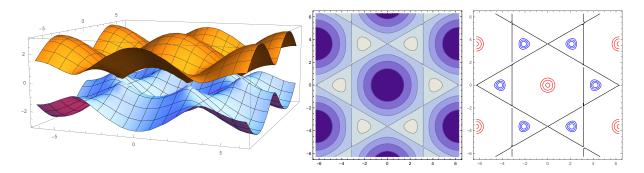
Which becomes,

$$-\frac{4}{\sqrt{3}}\pi \le 2ak_y \le \frac{4}{\sqrt{3}}\pi$$

Therefore the Brillouin zone is fully characterized.



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The first Brillouin zone for graphene forms a hexagonal lattice with the corners obtained from \mathbf{k}_{\pm} by adding $\pm b_1, \pm b_2$,

$$G = m_1 \boldsymbol{b}_2 + m_2 \boldsymbol{b}_2$$

In graphene, the upper and lower band are *not* separated by a band gap (sometimes $\varepsilon_{+} > \varepsilon_{-}$ and sometimes $\varepsilon_{+} < \varepsilon_{-}$). The crossing points are precisely the two points where $\mathbf{k} = \mathbf{k}_{\pm}$. If we expand $H(\mathbf{k})$ near \mathbf{k}_{\pm} to first order,

$$k_x = k_{x,\pm} + \delta k_x$$
 $k_y = \delta k_y$

To first order,

$$d_{\pm}^{x}(\delta \mathbf{k}) \simeq \pm \frac{\sqrt{3}}{2} ta \delta k_{x}$$

$$d_{\pm}^{y}(\delta \boldsymbol{k}) \simeq \frac{\sqrt{3}}{2} ta \delta k_{y}$$

For both the x and y directions, the leading coefficient $\frac{\sqrt{3}}{2}ta$ has units of speed times \hbar . We again call this the Fermi velocity,

$$v_{\rm F} = \frac{\sqrt{3}ta}{2\hbar}$$

Then we can write the Hamiltonian as,

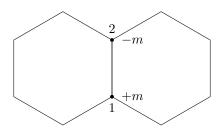
$$H_{+}(\mathbf{k}) = \hbar v_{\mathrm{F}}(\pm k_{x}\sigma^{x} + k_{y}\sigma^{y}) \tag{2.37}$$

Where here k_x and k_y refer to δk_x and δk_y . This is done by shifting coordinates such that $\mathbf{k}_{\pm} = 0$ for each of the +,- cases. Upon examination of eq. (2.37), eq. (2.37) takes the form of a 2D Dirac Hamiltonian for a massless relativistic particle in 2D. The energies near the touching points \mathbf{k}_{\pm} is,

$$\varepsilon(\mathbf{k}) = \pm \hbar v_{\rm F} k$$

Where $k = \sqrt{k_x^2 + k_y^2}$. This means that $\varepsilon(\mathbf{k})$ has linear dispersion near \mathbf{k}_{\pm} .

One question we can ask is what can we do to graphene in order to open a gap between k_- and k_+ . In order words, how can we give the electron in eq. (2.37) an effective mass. This can be accomplished by introducing a discrepancy between the atoms labeled 1 and the atoms labeled 2 within a unit cell. As an example, we can add energies $\pm m$ to the atoms as follows,



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Which effectively corresponds to adding the following term to the original Hamiltonian,

$$H' = m \sum_{\boldsymbol{R}} (|\boldsymbol{R}, 1\rangle \langle \boldsymbol{R}, 1| - |\boldsymbol{R}, 2\rangle \langle \boldsymbol{R}, 2|)$$

Applying the Fourier transform yields the following,

$$H' = m \sum_{\mathbf{k}} (|\mathbf{k}, 1\rangle \langle \mathbf{k}, 1| - |\mathbf{k}, 2\rangle \langle \mathbf{k}, 2|)$$

Which further adjusts the dispersion relationship,

$$H'(\mathbf{k}) = m(|1\rangle\langle 1| - |2\rangle\langle 2|) = m\sigma^z$$

Therefore,

$$H_{\pm}(\mathbf{k}) = \hbar v_{\mathrm{F}}(\pm k_x \sigma^x + k_y \sigma^y) + m\sigma^z$$
$$\varepsilon_{\pm}(\mathbf{k}) = \pm \sqrt{\hbar^2 v_{\mathrm{F}}^2 k^2 + m^2}$$

This opens up a band gap of size 2m. Therefore clean graphene (m = 0) is a semimetal due to inversion symmetry but becomes an insulator for m > 0.

Returning to the case of m = 0,

$$H_{\pm}(\mathbf{k}) = \hbar v_{\rm F}(\pm k_x \sigma^x + k_y \sigma^y)$$

The ' \pm ' can be thought of as a two component degree of freedom. Because there are two components, we can think of ' \pm ' as another pseudo-spin value τ .

$$\begin{split} \tau^z &= |+\rangle\langle +|-|-\rangle\langle -|\\ \tau^y &= -i(|+\rangle\langle -|-|-\rangle\langle +|)\\ \tau^x &= |+\rangle\langle -|+|-\rangle\langle +| \end{split}$$

Which means that the Hamiltonian can be written without the '±' as a single matrix,

$$H(\mathbf{k}) = \hbar v_{\rm F} (k_x \tau^z \sigma^x + k_y \sigma^y) \tag{2.38}$$

2.5.1 Spin-Orbit Interactions

What are the symmetries of eq. (2.38)? We must have the following:

- 1. Time Reversal Symmetry
- 2. Inversion Symmetry

In order to facilitate discussions, we introduce two operators; the inversion (parity) operator π and time-reversal operator Θ . The symmetries of eq. (2.38) can we written,

$$\pi^{\dagger} H(\mathbf{k}) \pi = H(\mathbf{k}) \tag{2.39}$$

$$\Theta H(\mathbf{k})\Theta^{-1} = H(\mathbf{k}) \tag{2.40}$$

Since τ^z refers to the two band touching points (coordinates in momentum space), τ^z changes sign under both symmetries,

$$\pi^{\dagger} \tau^z \pi = -\tau^z \tag{2.41}$$

$$\Theta \tau^z \Theta^{-1} = -\tau^z \tag{2.42}$$

Furthermore, $k_{x,y,z}$ represent momentum and is thus time and parity odd,

$$\pi^{\dagger} k_{x,y,z} \pi = -k_{x,y,z}$$

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$$\Theta k_{x,y,z} \Theta^{-1} = -k_{x,y,z}$$

Therefore we must conclude that σ^x is even under these symmetries and σ^y is odd pursuant to eqs. (2.39) and (2.40),

$$\pi^{\dagger} \sigma^{y} \pi = -\sigma^{y}$$

$$\Theta \sigma^{y} \Theta^{-1} = -\sigma^{y}$$

$$\pi^{\dagger} \sigma^{x} \pi = +\sigma^{x}$$

$$\Theta \sigma^{x} \Theta^{-1} = +\sigma^{x}$$

This is intuitive because σ refers to the spatial location of atoms 1 and 2 oriented in the y direction. If we were to include the massive term $m\sigma^z$ we break symmetry in eqs. (2.39) and (2.40),

$$\pi^{\dagger} \sigma^z \pi = -\sigma^z \qquad \Theta \sigma^z \Theta^{-1} = \sigma^z \tag{2.43}$$

Specifically, $m\sigma^z$ breaks inversion symmetry. This is intuitive because m adds spatial discrepancies between 1 and 2. Can we open a gap in graphene without breaking time-reversal or inversion symmetries?

Interestingly, this lack of symmetry is an artifact of the simplistic model introduced by eq. (2.35). If we include next-nearest neighbor terms, it is possible to recover symmetries. Upon examination of eq. (2.43) and eqs. (2.41) and (2.42) we find that $\tau^z \sigma^z$ is invariant under inversion,

$$\pi^{\dagger} \tau^z \sigma^z \pi = \tau^z \sigma^z$$

But $\tau^z \sigma^z$ is not invariant under time-reversal,

$$\Theta \tau^z \sigma^z \Theta^{-1} = -\tau^z \sigma^z$$

Other component that is missing from our model is the real spin of the electron (not pseudo-spin) S. Spin S is parity even and time-reversal odd,

$$\pi^{\dagger} S \pi = S \qquad \Theta S \Theta^{-1} = -S$$

$$S^{z} = |\uparrow\rangle\langle\uparrow| - |\downarrow\rangle\langle\downarrow|$$

$$S^{y} = -i(|\uparrow\rangle\langle\downarrow| - |\downarrow\rangle\langle\uparrow|)$$

$$S^{x} = |\uparrow\rangle\langle\downarrow| + |\downarrow\rangle\langle\uparrow|$$

Upon adding this term, $\tau^z \sigma^z S^z$ maintains the desired symmetries,

$$\pi^{\dagger}\tau^{z}\sigma^{z}S^{z}\pi = \tau^{z}\sigma^{z}S^{z} \qquad \Theta\tau^{z}\sigma^{z}S^{z}\Theta^{-1} = \tau^{z}\sigma^{z}S^{z}$$

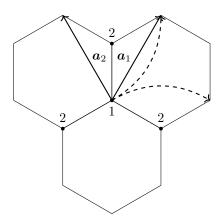
This was discovered by Kane and Mele (2005) and their paper gave rise to the field of topological insulators. Therefore the true graphene Hamiltonian can be written as,

$$H(\mathbf{k}) = \hbar v_{\rm F} (k_x \tau^z \sigma^x + k_y \sigma^y) + \Delta_{\rm SO} \tau^z \sigma^z S^z \tag{2.44}$$

Where $\Delta_{\rm SO}$ is referred to as the **spin-orbit interaction**. The origin of the $\Delta_{\rm SO}$ term is due to the following process. When the valence electrons have the ability to tunnel between atoms, some of the atoms in the crystal lattice become ionized. The electric field of ionized atoms is felt as a magnetic field in the rest frame of the electrons. Effectively this magnetic field has strength $\mathbf{B} = \mathbf{v} \times \mathbf{E}/c$ which tends to be small as $B \propto c^{-1}$. Additionally, \mathbf{E} grows for heavier and heavier atoms. This magnetic field acts on the spin in the usual manner $H \propto \mathbf{B} \cdot \mathbf{S}$. This is encapsulated in eq. (2.44) as $\tau^z \sigma^z$ is coupled with the spin S^z .

In order to formally introduce the $\Delta_{SO}\tau^z\sigma^zS^z$ term into the Hamiltonian, we need to include next nearest-neighbor tunneling $(1 \to 1 \text{ and } 2 \to 2 \text{ tunneling})$,

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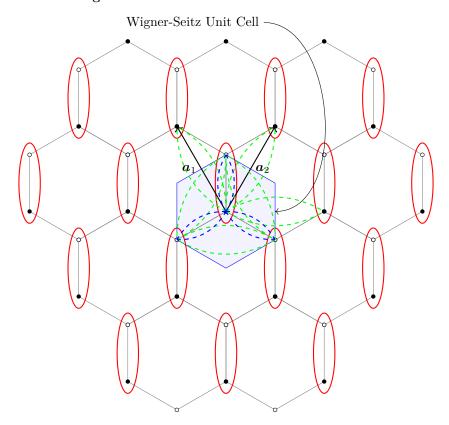


The additional terms to eq. (2.35) can be expressed as,

$$H = \cdots - it'(|\mathbf{R} + \mathbf{a}_1 - \mathbf{a}_2, 1\rangle\langle \mathbf{R}, 1| - |\mathbf{R}, 1\rangle\langle \mathbf{R} + \mathbf{a}_1 - \mathbf{a}_2, 1|)\cdots$$

Which has tunneling amplitude it'. The intermediate negative sign is due to the tunneling amplitude being imaginary which switches signs under conjugation.

2.5.2 Second Nearest-Neighbors



In the graphene lattice depicted above, each unit cell located at $\mathbf{R} = n_1 \mathbf{a}_1 + n_2 \mathbf{a}_2$ where $n_1, n_2 \in \mathbb{Z}$ has two atoms $\alpha \in \{\circ, \bullet\}$.

$$\boldsymbol{a}_1 = \frac{a}{2} \Big(\hat{\boldsymbol{x}} + \sqrt{3} \hat{\boldsymbol{y}} \Big) \qquad \boldsymbol{a}_2 = \frac{a}{2} \Big(-\hat{\boldsymbol{x}} + \sqrt{3} \hat{\boldsymbol{y}} \Big)$$

An electron localized in the atom $\alpha \in \{\circ, \bullet\}$ at R is denoted as $|R, \alpha\rangle$.

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There are 6 possible nearest-neighbor transitions illustrated in blue above. These transitions, with intensity t_1 can be written as the nearest-neighbor Hamiltonian H_1 ,

$$H_1 = -t_1 \sum_{\boldsymbol{R}} \{ |\boldsymbol{R}, \circ\rangle \langle \boldsymbol{R}, \bullet| + |\boldsymbol{R} - \boldsymbol{a}_1, \circ\rangle \langle \boldsymbol{R}, \bullet| + |\boldsymbol{R} - \boldsymbol{a}_2, \circ\rangle \langle \boldsymbol{R}, \bullet| + \text{h.c.} \}$$

There are 12 possible next-nearest-neighbor transitions illustrated in green above. These transitions, with intensity t_2 can be written as the nearest-neighbor Hamiltonian H_2 (note that these transitions were all defined in the same direction; this is important when we invoke $t_2 = i\tilde{t}_2, \tilde{t}_2 \in \mathbb{R}$ later),

$$\begin{split} H_{2,\circ} &= -t_2 \sum_{\boldsymbol{R}} \{ |\boldsymbol{R} + \boldsymbol{a}_1, \circ\rangle \langle \boldsymbol{R}, \circ| + |\boldsymbol{R} - \boldsymbol{a}_2, \circ\rangle \langle \boldsymbol{R}, \circ| + |\boldsymbol{R} - \boldsymbol{a}_1, \circ\rangle \langle \boldsymbol{R} - \boldsymbol{a}_2, \circ| + \text{h.c.} \} \\ H_{2,\bullet} &= -t_2 \sum_{\boldsymbol{R}} \{ |\boldsymbol{R} + \boldsymbol{a}_1, \bullet\rangle \langle \boldsymbol{R}, \bullet| + |\boldsymbol{R}, \bullet\rangle \langle \boldsymbol{R} + \boldsymbol{a}_2, \bullet| + |\boldsymbol{R} - \boldsymbol{a}_1, \bullet\rangle \langle \boldsymbol{R} - \boldsymbol{a}_2, \bullet| + \text{h.c.} \} \end{split}$$

Altogether,

$$H = H_1 + H_2 = H_1 + H_{2,\bullet} + H_{2,\circ} \tag{2.45}$$

In order to diagonalize eq. (2.45) with respect to R, we may switch to momentum space,

$$|\mathbf{R}, \alpha\rangle = \sum_{\mathbf{R}} |\mathbf{k}, \alpha\rangle \langle \mathbf{k}, \alpha | \mathbf{R}, \alpha\rangle \qquad \forall \alpha \in \{\bullet, \circ\}$$
 (2.46)

Where $\langle \mathbf{k}, \alpha | \mathbf{R}, \alpha \rangle$ is defined and normalized independent of α . Motivated by free electrons, $\langle k | x \rangle \propto e^{-ikx}$ we map x to the position in the crystal:

$$x \mapsto n_1 \mathbf{a}_1 + n_2 \mathbf{a}_2 = \mathbf{R}$$

$$k \mapsto k_1 \frac{\mathbf{a}_1}{a} + k_2 \frac{\mathbf{a}_2}{a} = \mathbf{k}$$
(2.47)

Therefore,

$$\langle \mathbf{k}, \alpha | \mathbf{R}, \alpha \rangle \propto e^{i\mathbf{k}\cdot\mathbf{R}}$$
 (2.48)

The normalization factor comes from completeness,

$$\sum_{\mathbf{R}} |\langle \mathbf{k}, \alpha | \mathbf{R}, \alpha \rangle|^2 = 1 \tag{2.49}$$

Heretofore, no restrictions have been made on the range of \mathbf{R} . Evidently, a real substance will possess finite, albeit many, unit cells. To emulate this feature, equip this model with periodic boundary conditions such that there are N_i distinct unit cells in the \mathbf{a}_i direction. Therefore for all α , \mathbf{R} ,

$$|\mathbf{R}, \alpha\rangle = |\mathbf{R} + N_1 \mathbf{a}_1, \alpha\rangle = |\mathbf{R} + N_2 \mathbf{a}_2, \alpha\rangle$$
 (2.50)

Letting $\gamma \in \mathbb{R}$ be the normalization coefficient in eq. (2.48), eq. (2.49) becomes,

$$1 = \sum_{\boldsymbol{R}} \left| \langle \boldsymbol{k}, \alpha | \boldsymbol{R}, \alpha \rangle \right|^2 = \sum_{n_1 = 1}^{N_1} \sum_{n_2 = 1}^{N_2} \left| \gamma e^{i \boldsymbol{k} \cdot \boldsymbol{R}} \right|^2 = N_1 N_2 \gamma^2 \implies \gamma = \frac{1}{\sqrt{N_1 N_2}}$$

Consequently, eq. (2.46) can be written as follows:

$$|\mathbf{R}, \alpha\rangle = \frac{1}{\sqrt{N_1 N_2}} \sum_{\mathbf{k}} e^{i\mathbf{k} \cdot \mathbf{R}} |\mathbf{k}, \alpha\rangle$$
 (2.51)

Additionally, the periodic boundary condition of eq. (2.50) gives,

$$|\mathbf{R}, \alpha\rangle = |\mathbf{R} + N_i \mathbf{a}_i, \alpha\rangle = \frac{1}{\sqrt{N_1 N_2}} \sum_{\mathbf{k}} e^{i\mathbf{k}\cdot(\mathbf{R} + N_i \mathbf{a}_i)} |\mathbf{k}, \alpha\rangle = \frac{1}{\sqrt{N_1 N_2}} \sum_{\mathbf{k}} e^{i\mathbf{k}\cdot\mathbf{R}} |\mathbf{k}, \alpha\rangle$$

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Therefore $e^{iN_i \mathbf{k} \cdot \mathbf{a}_i} = e^{i2N_i k_i a} = 1$ which gives,

$$2N_i k_i a = 2\pi m_i \qquad m_i \in \mathbb{Z}, \forall i \in \{1, 2\}$$

$$k_i = \frac{\pi m_i}{N_i a} \qquad m_i \in \mathbb{Z}, \forall i \in \{1, 2\}$$

$$(2.52)$$

Therefore each of the k_i 's are discrete. Summing eq. (2.45) over the momentum space k, k' and using the following identity,

$$\frac{1}{N_1 N_2} \sum_{\boldsymbol{R}} e^{-i \left(\boldsymbol{k} - \boldsymbol{k}'\right) \cdot \boldsymbol{R}} = \delta_{\boldsymbol{k}, \boldsymbol{k}'}$$

Yields,

$$\begin{split} H_1 &= -t_1 \sum_{\pmb{k}} \left\{ |\pmb{k}, \diamond\rangle \langle \pmb{k}, \bullet| \left\{ 1 + e^{-i\pmb{k}\cdot\pmb{a}_1} + e^{-i\pmb{k}\cdot\pmb{a}_2} \right\} + \text{h.c.} \right\} \\ H_{2, \diamond} &= -t_2 \sum_{\pmb{k}} \left\{ |\pmb{k}, \diamond\rangle \langle \pmb{k}, \diamond| \left\{ e^{i\pmb{k}\cdot\pmb{a}_1} + e^{-i\pmb{k}\cdot\pmb{a}_2} + e^{i\pmb{k}\cdot(\pmb{a}_2 - \pmb{a}_1)} \right\} + \text{h.c.} \right\} \\ H_{2, \bullet} &= -t_2 \sum_{\pmb{k}} \left\{ |\pmb{k}, \bullet\rangle \langle \pmb{k}, \bullet| \left\{ e^{i\pmb{k}\cdot\pmb{a}_1} + e^{-i\pmb{k}\cdot\pmb{a}_2} + e^{i\pmb{k}\cdot(\pmb{a}_2 - \pmb{a}_1)} \right\} + \text{h.c.} \right\} \end{split}$$

Therefore $H = \sum_{\mathbf{k}} H(\mathbf{k})$ gives,

$$H(\mathbf{k}) = \sum_{\mathbf{k}} \left(-t_1 \left[|\circ\rangle\langle \bullet| \left\{ 1 + e^{-i\mathbf{k}\cdot\mathbf{a}_1} + e^{-i\mathbf{k}\cdot\mathbf{a}_2} \right\} \right] - t_2 \left[(|\bullet\rangle\langle \bullet| + |\circ\rangle\langle \circ|) \left\{ e^{i\mathbf{k}\cdot\mathbf{a}_1} + e^{-i\mathbf{k}\cdot\mathbf{a}_2} + e^{i\mathbf{k}\cdot(\mathbf{a}_2 - \mathbf{a}_1)} \right\} \right] \right) + \text{h.c.}$$

Since $H(\mathbf{k})$ is a two-level system $\{|\bullet, \circ\rangle\}$ we invoke a pseudo-spin denoted by the Pauli matrices σ .

$$\begin{aligned} & \boldsymbol{\sigma} = (\sigma^x, \sigma^y, \sigma^z) \\ & \sigma^z = |\bullet\rangle \langle \bullet| - |\circ\rangle \langle \circ| \\ & \sigma^y = -i(|\bullet\rangle \langle \circ| - |\circ\rangle \langle \bullet|) \\ & \sigma^x = |\bullet\rangle \langle \circ| + |\circ\rangle \langle \bullet| \end{aligned}$$

Therefore $|\bullet\rangle\langle \circ|$ can be written as follows,

$$|\bullet\rangle\langle \circ| = \frac{1}{2}\{\sigma^x + i\sigma^y\} \qquad |\circ\rangle\langle \bullet| = \frac{1}{2}\{\sigma^x - i\sigma^y\}$$

Since $t_1 \in \mathbb{R}$, this implies,

$$\begin{split} |\diamond\rangle\langle\bullet| \left\{1 + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_1} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_2}\right\} + \text{h.c.} &= |\diamond\rangle\langle\bullet| \left\{1 + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_1} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_2}\right\} + |\bullet\rangle\langle\circ| \left\{1 + e^{i\boldsymbol{k}\cdot\boldsymbol{a}_1} + e^{i\boldsymbol{k}\cdot\boldsymbol{a}_2}\right\} \\ &= \frac{1}{2}(\sigma^x - i\sigma^y) \left\{1 + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_1} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_2}\right\} + \frac{1}{2}(\sigma^x + i\sigma^y) \left\{1 + e^{i\boldsymbol{k}\cdot\boldsymbol{a}_1} + e^{i\boldsymbol{k}\cdot\boldsymbol{a}_2}\right\} \\ &= \sigma^x(1 + \cos(\boldsymbol{k}\cdot\boldsymbol{a}_1) + \cos(\boldsymbol{k}\cdot\boldsymbol{a}_2)) + \sigma^y(\sin(\boldsymbol{k}\cdot\boldsymbol{a}_1) + \sin(\boldsymbol{k}\cdot\boldsymbol{a}_2)) \end{split}$$

In contrast, if t_2 is imaginary $t_2^* = -t_2$ which means $t_2 = i\tilde{t}_2$,

$$t_{2}\underbrace{(|\bullet\rangle\langle\bullet|+|\circ\rangle\langle\circ|)}_{\mathbf{1}} \Big\{ e^{i\boldsymbol{k}\cdot\boldsymbol{a}_{1}} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_{2}} + e^{i\boldsymbol{k}\cdot(\boldsymbol{a}_{2}-\boldsymbol{a}_{1})} \Big\} + \text{h.c.}$$

$$= i\tilde{t}_{2} \Big\{ e^{i\boldsymbol{k}\cdot\boldsymbol{a}_{1}} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_{2}} + e^{i\boldsymbol{k}\cdot(\boldsymbol{a}_{2}-\boldsymbol{a}_{1})} \Big\} + \text{h.c.}$$

$$= i\tilde{t}_{2} \Big\{ e^{i\boldsymbol{k}\cdot\boldsymbol{a}_{1}} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_{2}} + e^{i\boldsymbol{k}\cdot(\boldsymbol{a}_{2}-\boldsymbol{a}_{1})} \Big\} + \text{h.c.}$$

$$= i\tilde{t}_{2} \Big\{ e^{i\boldsymbol{k}\cdot\boldsymbol{a}_{1}} + e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_{2}} + e^{i\boldsymbol{k}\cdot(\boldsymbol{a}_{2}-\boldsymbol{a}_{1})} - e^{-i\boldsymbol{k}\cdot\boldsymbol{a}_{1}} - e^{i\boldsymbol{k}\cdot\boldsymbol{a}_{2}} - e^{-i\boldsymbol{k}\cdot(\boldsymbol{a}_{2}-\boldsymbol{a}_{1})} \Big\}$$

$$= 2\tilde{t}_{2} \Big\{ \sin(-\boldsymbol{k}\cdot\boldsymbol{a}_{1}) + \sin(\boldsymbol{k}\cdot\boldsymbol{a}_{2}) + \sin(\boldsymbol{k}\cdot(\boldsymbol{a}_{1}-\boldsymbol{a}_{2})) \Big\}$$

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In order to write $H_{2,\bullet}$ in terms of $d(k) \cdot \sigma$, recognize that if the electrons have spin,

$$S = (S^x, S^y, S^z)$$

$$S^z = |\uparrow\rangle\langle\uparrow| - |\downarrow\rangle\langle\downarrow|$$

$$S^y = -i(|\uparrow\rangle\langle\downarrow| - |\downarrow\rangle\langle\uparrow|)$$

$$S^x = |\uparrow\rangle\langle\downarrow| + |\downarrow\rangle\langle\uparrow|$$

Then the eigenstates can become the following,

$$|\mathbf{R}, \alpha\rangle \mapsto |\mathbf{R}, \alpha_{\sigma}, \alpha_{S}\rangle \qquad \alpha_{\sigma} \in \{\diamond, \bullet\}, \alpha_{S} \in \{\uparrow, \downarrow\}$$

Then is is possible to write the following,

$$\mathbf{1} = S^z \sigma^z$$

Therefore,

$$d^{x}(\mathbf{k}) = -t_{1}\{1 + \cos(\mathbf{k} \cdot \mathbf{a}_{1}) + \cos(\mathbf{k} \cdot \mathbf{a}_{2})\}$$

$$d^{y}(\mathbf{k}) = -t_{1}\{\sin(\mathbf{k} \cdot \mathbf{a}_{1}) + \sin(\mathbf{k} \cdot \mathbf{a}_{2})\}$$

$$d^{z}(\mathbf{k}) = 2\tilde{t}_{2}S^{z}\{\sin(-\mathbf{k} \cdot \mathbf{a}_{1}) + \sin(\mathbf{k} \cdot \mathbf{a}_{2}) + \sin(\mathbf{k} \cdot (\mathbf{a}_{1} - \mathbf{a}_{2}))\}$$

In terms of the x, y components of k,

$$\mathbf{k} = k_x \hat{\mathbf{x}} + k_y \hat{\mathbf{y}}$$

$$\mathbf{a}_1 = \frac{a}{2} \left(\hat{\mathbf{x}} + \sqrt{3} \hat{\mathbf{y}} \right) \qquad \mathbf{a}_2 = \frac{a}{2} \left(-\hat{\mathbf{x}} + \sqrt{3} \hat{\mathbf{y}} \right)$$
(2.53)

$$\mathbf{k} \cdot \mathbf{a}_{1} = (k_{x}\hat{\mathbf{x}} + k_{y}\hat{\mathbf{y}}) \left(\frac{a}{2} \left(\hat{\mathbf{x}} + \sqrt{3}\hat{\mathbf{y}}\right)\right) = \frac{a}{2} \left(k_{x} + \sqrt{3}k_{y}\right)$$

$$\mathbf{k} \cdot \mathbf{a}_{2} = (k_{x}\hat{\mathbf{x}} + k_{y}\hat{\mathbf{y}}) \left(\frac{a}{2} \left(-\hat{\mathbf{x}} + \sqrt{3}\hat{\mathbf{y}}\right)\right) = \frac{a}{2} \left(-k_{x} + \sqrt{3}k_{y}\right)$$
(2.54)

Trig identities give,

$$d^{x}(\mathbf{k}) = -t_{1} \left\{ 1 + 2 \cos\left(\frac{ak_{x}}{2}\right) \cos\left(\frac{\sqrt{3}ak_{y}}{2}\right) \right\}$$
$$d^{y}(\mathbf{k}) = -2t_{1} \left\{ \cos\left(\frac{ak_{x}}{2}\right) \sin\left(\frac{\sqrt{3}ak_{y}}{2}\right) \right\}$$
$$d^{z}(\mathbf{k}) = -2\tilde{t}_{2}S^{z} \left\{ -2\sin\left(\frac{ak_{x}}{2}\right) \cos\left(\frac{\sqrt{3}ak_{y}}{2}\right) + \sin(k_{x}a) \right\}$$

The energy dispersion relations are,

$$\varepsilon(\mathbf{k}) = \pm |\mathbf{d}(\mathbf{k})|$$

$$\varepsilon_{\pm}(\mathbf{k}) = \pm \left(t_1^2 \left\{1 + 2\cos\left(\frac{ak_x}{2}\right)\cos\left(\frac{\sqrt{3}ak_y}{2}\right)\right\}^2 + \cdots + 4t_1^2 \left\{\cos\left(\frac{ak_x}{2}\right)\sin\left(\frac{\sqrt{3}ak_y}{2}\right)\right\}^2 + \cdots + 4\tilde{t}_2^2 s^{z^2} \left\{-2\sin\left(\frac{ak_x}{2}\right)\cos\left(\frac{\sqrt{3}ak_y}{2}\right) + \sin(k_x a)\right\}^2\right)^{1/2}$$

$$(2.55)$$

By examining these equations it can be seen that there are two points (and only two points) in the first Brillouin zone where both d^x and d^y vanish simultaneously (at least in the case where $\tilde{t}_2 = 0$). We label these points as \mathbf{k}_{\pm} ,

 $\mathbf{k}_{+} = (k_{+,x}, k_{+,y}) = \left(\frac{4\pi}{3a}, 0\right)$

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$$\mathbf{k}_{-} = (k_{-,x}, k_{-,y}) = \left(-\frac{4\pi}{3a}, 0\right)$$

Therefore, in the limit where $\mathbf{k} \simeq \mathbf{k}_{\pm}$,

$$\mathbf{k} = \mathbf{k}_{\pm} + \delta \mathbf{k}_{\pm}$$

Therefore in the limit that $\delta \mathbf{k}_{\pm} \simeq \mathbf{0}$,

$$\begin{aligned} \cos(ak_x) &= \cos(ak_{\pm,x} + a\delta k_{\pm,x}) \\ &= \cos(ak_{\pm,x}) \cos(a\delta k_{\pm,x}) - \sin(ak_{\pm,x}) \sin(a\delta k_{\pm,x}) \\ &= \cos\left(\frac{4\pi}{3}\right) \cos(a\delta k_{\pm,x}) - \sin\left(\pm\frac{4\pi}{3}\right) \sin(a\delta k_{\pm,x}) \\ &= -\frac{1}{2} \cos(a\delta k_{\pm,x}) \pm \frac{\sqrt{3}}{2} \sin(a\delta k_{\pm,x}) \\ &\simeq -\frac{1}{2} \pm \frac{\sqrt{3}}{2} a\delta k_{\pm,x} \\ \sin(ak_x) &= \sin(ak_{\pm,x} + a\delta k_{\pm,x}) \\ &= \cos(ak_{\pm,x}) \sin(a\delta k_{\pm,x}) + \sin(ak_{\pm,x}) \cos(a\delta k_{\pm,x}) \\ &= \cos\left(\frac{4\pi}{3}\right) \sin(a\delta k_{\pm,x}) + \sin\left(\pm\frac{4\pi}{3}\right) \cos(a\delta k_{\pm,x}) \\ &= -\frac{1}{2} \sin(a\delta k_{\pm,x}) \mp \frac{\sqrt{3}}{2} \cos(a\delta k_{\pm,x}) \\ &= -\frac{1}{2} a\delta k_{\pm,x} \mp \frac{\sqrt{3}}{2} \\ \cos\left(\frac{ak_x}{2}\right) &= \cos\left(\frac{ak_{\pm,x}}{2} + \frac{a\delta k_{\pm,x}}{2}\right) \\ &= \cos\left(\frac{ak_{\pm,x}}{2}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) - \sin\left(\frac{ak_{\pm,x}}{2}\right) \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= \cos\left(\frac{ak_{\pm,x}}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) - \sin\left(\frac{a\delta k_{\pm,x}}{3}\right) \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} + \frac{\sqrt{3}}{2} \frac{a\delta k_{\pm,x}}{2} \\ &= \cos\left(\frac{ak_{\pm,x}}{2}\right) \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{ak_{\pm,x}}{2}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= \cos\left(\frac{2\pi}{3}\right) \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{ak_{\pm,x}}{2}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= \cos\left(\frac{2\pi}{3}\right) \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{2\pi}{3}\right) \cos\left(\frac{a\delta k_{\pm,x}}{2}\right) \\ &= -\frac{1}{2} \sin\left(\frac{a\delta k_{\pm,x}}{2}\right) + \sin\left(\frac{a\delta$$

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Therefore to first order,

$$d_{\pm}^{x}(\mathbf{k}) \simeq -t_{1} \left\{ 1 + 2 \left(-\frac{1}{2} \mp \frac{\sqrt{3}}{2} \frac{a \delta k_{\pm,x}}{2} \right) \right\}$$

$$d_{\pm}^{y}(\mathbf{k}) \simeq -2t_{1} \left\{ \left(-\frac{1}{2} \mp \frac{\sqrt{3}}{2} \frac{a \delta k_{\pm,x}}{2} \right) \frac{\sqrt{3} a \delta k_{\pm,y}}{2} \right\}$$

$$d_{\pm}^{z}(\mathbf{k}) \simeq 2\tilde{t}_{2} S^{z} \left\{ -2 \left(-\frac{1}{2} \frac{a \delta k_{\pm,x}}{2} \mp \frac{\sqrt{3}}{2} \right) + \left(-\frac{1}{2} a \delta k_{\pm,x} \pm \frac{\sqrt{3}}{2} \right) \right\}$$

Simplifying

$$d_{\pm}^{x}(\mathbf{k}) \simeq \pm \frac{at_{1}\sqrt{3}\delta k_{\pm,x}}{2}$$
$$d_{\pm}^{y}(\mathbf{k}) \simeq \frac{at_{1}\sqrt{3}\delta k_{\pm,y}}{2}$$
$$d_{\pm}^{z}(\mathbf{k}) \simeq \pm 3\sqrt{3}t_{2}S^{z}$$

This new degree of freedom, namely whether or not the electron is near k_+ or k_- can be expressed as another pseudo-spin τ ,

$$\begin{split} \boldsymbol{\tau} &= (\tau^x, \tau^y, \tau^z) \\ \tau^z &= |+\rangle \langle +|-|-\rangle \langle -| \\ \tau^y &= -i(|+\rangle \langle -|-|-\rangle \langle +|) \\ \tau^x &= |+\rangle \langle -|+|-\rangle \langle +| \end{split}$$

Moreover a unique constant emerges as the Fermi velocity $v_{\rm F}$,

$$\hbar v_{\rm F} = \frac{at_1\sqrt{3}}{2}$$

Thus $H(\mathbf{k})$ becomes,

$$H(\mathbf{k}) = \sigma^x d^x(\mathbf{k}) + \sigma^y d^y(\mathbf{k}) + \sigma^z d^z(\mathbf{k})$$

= $\hbar v_F (\tau^z \sigma^x \delta k_x + \sigma^y \delta k_y) + 3\sqrt{3} \tilde{t}_2 \sigma^z \tau^z S^z$

Finally, suppress the notation ' δ ' and define the spin orbital coupling Δ_{SO} .

$$\Delta_{SO} = 3\sqrt{3}\tilde{t}_2$$

Therefore near $k \simeq k_{\pm}$,

$$H(\mathbf{k}) = \hbar v_{\rm F}(\tau^z \sigma^x k_x + \sigma^y k_y) + \Delta_{\rm SO} \sigma^z \tau^z S^z$$

This Hamiltonian has the desired invariant spin terms $\sigma^z \tau^z s^z$ and should have been expected due to symmetry due to symmetries.

2.5.3 Spin-Orbit Interactions Analyzed

In order to study the effect Δ_{SO} in eq. (2.44), we will include a massive term $m\sigma^z$ and study the following,

$$H(\mathbf{k}) = \hbar v_{\rm F} (k_x \tau^z \sigma^x + k_y \sigma^y) + \Delta_{\rm SO} \tau^z \sigma^z S^z + m \sigma^z$$

Specifically for our analysis, we will look at the affect on $S^z=1$ spin up (\uparrow) spins. The associated Hamiltonian for $S^z=1$ has two bands $\tau^z=\pm 1$,

$$\{H(\mathbf{k})\}_{\tau^z=+1} \equiv H_+(\mathbf{k}) = \hbar v_{\rm F}(k_x \sigma^x + k_y \sigma^y) + (m + \Delta_{\rm SO})\sigma^z$$

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$$\{H(\mathbf{k})\}_{\tau^z=-1} \equiv H_+(\mathbf{k}) = \hbar v_{\rm F}(-k_x\sigma^x + k_y\sigma^y) + (m - \Delta_{\rm SO})\sigma^z$$

Therefore the dispersion relations are distinct for each value of τ^z ,

$$\{\varepsilon(\mathbf{k})\}_{\tau^z=+1} \equiv \varepsilon_+(\mathbf{k}) = \pm \sqrt{\hbar^2 v_{\rm F}^2 \left(k_x^2 + k_y^2\right) + \left(m + \Delta_{\rm SO}\right)^2}$$

$$\{\varepsilon(\mathbf{k})\}_{\tau^z=-1} \equiv \varepsilon_-(\mathbf{k}) = \pm \sqrt{\hbar^2 v_{\rm F}^2 \left(k_x^2 + k_y^2\right) + \left(m - \Delta_{\rm SO}\right)^2}$$

$$(2.56)$$

In the very large m limit $m \gg \Delta_{\rm SO}$, the energy cost (+m) for electrons being localized at sites of type 1 is much higher than the energy cost (-m) associated with sites of type 2. Therefore all of the electrons will localized to sites of type 2. Therefore in this case, the electrons can not propagate between sites because the atoms of type 1 are in the way. We refer to these as **non-topological insulators** or sometimes *atomic* insulators.

In contrast to this, in the region where $m \approx \Delta_{\rm SO}$ we have interesting behaviors for eq. (2.56). Since the band gap associated with the contribution from $\varepsilon_{-}(\mathbf{k})$ is $2|m - \Delta_{\rm SO}|$, the band gap closes when m transitions between $m > \Delta_{\rm SO}$ and $m < \Delta_{\rm SO}$.

- $m > \Delta_{SO}$: Trivial insulator
- $m = \Delta_{SO}$: Transition point/semimetal
- $m < \Delta_{SO}$: Topological insulator

We have already seen this behavior for graphene in which m = 0 defined a semimetal, and m > 0 defined a topological insulator form of graphene. Recall the Hamiltonian of graphene had the following form,

$$H(\mathbf{k}) = \mathbf{d}(\mathbf{k}) \cdot \boldsymbol{\sigma}$$

The unit vector of d is,

$$oldsymbol{\hat{d}}(oldsymbol{k}) = rac{oldsymbol{d}(oldsymbol{k})}{|oldsymbol{d}(oldsymbol{k})|}$$

Which can be thought of as a map from the first Brillouin zone (k) to the surface of a unit sphere \hat{d} . Interpreting the mappings in the way always one to understand the topological differences between ordinary and topological insulators.

Near the band touching points k_{\pm} we have,

$$\begin{split} \hat{\boldsymbol{d}}(\boldsymbol{k}_{+}) &= \frac{\left(\hbar v_{\mathrm{F}} k_{x}, \hbar v_{\mathrm{F}} k_{y}, m + \Delta_{\mathrm{SO}}\right)}{\sqrt{\hbar^{2} v_{\mathrm{F}}^{2} \left(k_{x}^{2} + k_{y}^{2}\right) + \left(m + \Delta_{\mathrm{SO}}\right)^{2}}} \\ \hat{\boldsymbol{d}}(\boldsymbol{k}_{-}) &= \frac{\left(-\hbar v_{\mathrm{F}} k_{x}, \hbar v_{\mathrm{F}} k_{y}, m - \Delta_{\mathrm{SO}}\right)}{\sqrt{\hbar^{2} v_{\mathrm{F}}^{2} \left(k_{x}^{2} + k_{y}^{2}\right) + \left(m - \Delta_{\mathrm{SO}}\right)^{2}}} \end{split}$$

Clearly for the case of $m > \Delta_{SO}$,

$$d(\mathbf{k}_{-}) \cdot \hat{\mathbf{z}} = m - \Delta_{SO} > 0$$

Therefore $d^z(\mathbf{k}_-)$ is always positive. This is also true for $d^z(\mathbf{k}_+)$. Therefore $\mathbf{d}(\mathbf{k})$ only covers the upper hemisphere of the unit sphere as \mathbf{k} covers the whole Brillouin zone.

In difference to this is the case for $m < \Delta_{SO}$. $d^z(\mathbf{k}_-)$ is always negative which means that $\hat{\mathbf{d}}(\mathbf{k})$ is able to cover the entire sphere.

Given this observation, we can define the following **topological invariant** that characterizes the difference between topological insulators and ordinary insulators. This quantity is the integer $n \in \{0,1\}$ defined as,

$$n = \frac{1}{4\pi} \int_{BZ} d^2k \hat{\boldsymbol{d}} \cdot \left(\partial_{k_x} \hat{\boldsymbol{d}} \times \partial_{k_y} \hat{\boldsymbol{d}} \right) = \begin{cases} 1 & \text{Topological Insulators} \\ 0 & \text{Normal Insulators} \end{cases}$$

Where $\int_{\rm BZ}$ refers to an integration over the entire Brillouin zone (BZ).

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2.5.4 Topological/Non-topological Interfaces

Topological insulators behave exactly like insulators except that electrons near the band touching points have the potential to transition to the upper band and permit conduction. In order to study this in detail, it is sufficient to consider the Hamiltonian component associated with the topological terms,

$$H_{-}(\mathbf{k}) = \hbar v_{\rm F}(-k_x \sigma^x + k_y \sigma^y) + (m - \Delta_{\rm SO})\sigma^z$$

Where the coefficient in front of σ^z differs in sign depending on whether or not the insulator is topological or not. Let us assume that we have a material that transitions from a topological insulator to an ordinary insulator by letting m(x) denote the coefficient in front of σ^z as it varies along the x-direction.

$$m(x) = m - \Delta_{SO}$$

Since k_x no-longer corresponds to plane wave solutions to the Schrodinger equation, we replace k_x with its representation in position space,

$$k_x \mapsto -i\frac{\partial}{\partial x}$$

And solve $H(k_y)$,

$$H(k_y) = i\hbar v_{\rm F} \frac{\partial}{\partial x} \sigma^x + \hbar v_{\rm F} k_y \sigma^y + m(x) \sigma^z$$

As a preliminary exercise, we can solve the Schrodinger equation for the case of $k_y = 0$. Note this will be nearly identical to eq. (2.28),

$$H(0) = i\hbar v_{\rm F} \frac{\partial}{\partial x} \sigma^x + m(x)\sigma^z \tag{2.57}$$

For the zero energy eigenstate we require $H\psi=0$. This suggests the following ansatz solution (for some unspecified $|z\rangle$),

$$\psi(x) = e^{f(x)}\sigma^x|z\rangle \tag{2.58}$$

Substituting eq. (2.58) into eq. (2.57) and making use of the fact that $\sigma^z \sigma^x = i \sigma^y$,

$$\left(i\hbar v_{\rm F}\frac{\mathrm{d}f}{\mathrm{d}x} + im(x)\sigma^y\right)e^{f(x)}|z\rangle = 0$$

Therefore,

$$\left(\frac{\mathrm{d}f}{\mathrm{d}x} + \frac{m(x)}{\hbar v_{\mathrm{F}}}\sigma^{y}\right)|z\rangle = 0$$

If we choose $|z\rangle$ such that $\sigma^y|z\rangle = |z\rangle$.

$$\frac{\mathrm{d}f}{\mathrm{d}x} = -\frac{m(x)}{\hbar v_{\mathrm{F}}} \implies f(x) = -\frac{1}{\hbar v_{\mathrm{F}}} \int_{0}^{x} \mathrm{d}x' m(x')$$

Therefore,

$$\psi(x) = e^{-\frac{1}{\hbar v_{\rm F}} \int_0^x \mathrm{d}x' m(x')} \sigma^x |z\rangle$$

What is $\sigma^x |z\rangle$?

$$\sigma^y \sigma^x |z\rangle = -\sigma^x \sigma^y |z\rangle = -\sigma^x |z\rangle$$

Therefore $\sigma^x|z\rangle$ is an eigenstate of σ^y with eigenvalue -1. Therefore,

$$\psi(x) = e^{-\frac{1}{\hbar v_{\rm F}} \int_0^x \mathrm{d}x' m(x')} |\sigma^y = -1\rangle \tag{2.59}$$

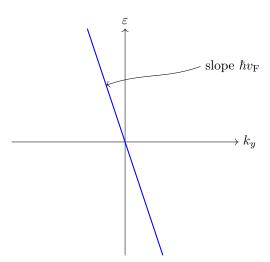
This is the zero-energy eigenstate of the Hamiltonian localized at x = 0 for the special case of $k_y = 0$. The eigenstate of eq. (2.59) is know in high energy physics as the Jackiv-Relli solution. If we consider cases where $k_y \neq 0$ we have,

$$H(k_u)\psi(x) = \hbar v_F k_u \sigma^y \psi(x) = -\hbar v_F k_u \psi(x)$$

Which means $\psi(x)$ is an eigenstate of $H(k_y)$ with energy $-\hbar v_F k_y$. The dispersion relation is,

$$\varepsilon(k_y) = -\hbar v_{\rm F} k_y \tag{2.60}$$

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We refer to this type of dispersion as **chiral dispersion**. To understand why we refer to this as chiral dispersion, recall some properties of relativistic quantum mechanics. The velocity of free electrons is,

$$\varepsilon(\mathbf{k}) = \frac{\hbar^2 k^2}{2m} \tag{2.61}$$

This can be derived using classical intuition,

$$v = \frac{p}{m} = \frac{\hbar k}{m} \tag{2.62}$$

Of course, eq. (2.61) and eq. (2.62) only hold for free electrons. More generally, we can define the **group velocity**,

$$\boldsymbol{v} = \frac{1}{\hbar} \boldsymbol{\nabla}_{\boldsymbol{k}} \varepsilon(\boldsymbol{k}) \tag{2.63}$$

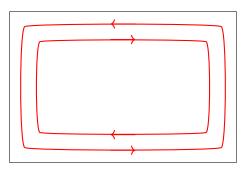
For eq. (2.60) we have,

$$\boldsymbol{v} = -v_{\mathrm{F}}\hat{\boldsymbol{y}}$$

Which means that all the electrons move in one direction. This is in contrast with an ordinary conductor where electrons have the ability to propagate in point directions subject to an external potential.

2.5.5 Quantum Spin Hall Effect

Using alternating external potentials V_y we can construct and consider a 2D metal where the electrons move in alternating directions.



This metal produces the quantum spin hall effect. We can define the current density j,

$$\boldsymbol{j} = -en\boldsymbol{v}$$

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Where n is the density of the electrons which can be written as a integral over momentum space. Doing so gives the current where the electrons move to the right as,

$$I_R = -e \int v_{\rm F} n_{\rm F}(\varepsilon_k) \frac{\mathrm{d}k_x}{2\pi}$$

Where $k = 2\pi n/L$ and $2\pi/L$ is the volume in momentum space corresponding to a single quantum state.

$$\frac{1}{L}\frac{\mathrm{d}k_x}{\frac{2\pi}{L}} = \frac{\mathrm{d}k_x}{2\pi}$$

If we convert this integral from an integral over momentum space to and integral over energies,

$$\varepsilon = \hbar v_{\rm F} k_x \implies \mathrm{d} k_x = \frac{\mathrm{d} \varepsilon}{\hbar v_{\rm F}}$$

Therefore the current is,

$$I_R = -ev_F \frac{1}{2\pi\hbar v_F} \int_{-\infty}^{\varepsilon_{FR}} d\varepsilon = -\frac{e}{h} \int_{-\infty}^{\varepsilon_{FR}} d\varepsilon$$

Where ε_{FR} is the Fermi energy for the right-traveling electrons. The current for the electrons traveling to the left is opposite in sign,

$$I_L = \frac{e}{h} \int_{-\infty}^{\varepsilon_{FL}} \mathrm{d}\varepsilon$$

Therefore the net current is,

$$I = I_R + I_L = -\frac{e}{h} \int_{\varepsilon_{FL}}^{\varepsilon_{FR}} d\varepsilon = -\frac{e}{h} (\varepsilon_{FR} - \varepsilon_{FL}) = -\frac{e^2}{h} V_y$$

The conductivity is,

$$\frac{I}{V_u} = -\frac{e^2}{h}$$

The interesting feature of this conductivity is that is only depends on the fundamental constants e, h. As it turns out, measurements of this conductivity represent our most precise measurements of e and h. An excellent external resource is a book by R.B. Laughlin called *Different Universe*. An important thing to remember that this analysis only applies for a particular spin value. In this case we analyzed spin up states $S^z = +1$. In general we have,

$$\sigma_{xy}^{\uparrow} = -\frac{e^2}{h}$$
 $\sigma_{xy}^{\downarrow} = +\frac{e^2}{h}$

Which means that we can construct a spin current,

$$\sigma_{xy}^S = \frac{1}{2} \left(\sigma_{xy}^{\uparrow} - \sigma_{xy}^{\downarrow} \right) = -\frac{e^2}{h}$$

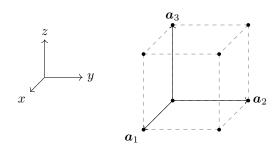
Effectively, if we have an equal number of electrons moving to the right and the left but each direction is biased to a particular spin, we can have a buildup of spin up electrons on the right and spin down electrons on the left. This is the analogue of *spin voltage*.

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2.6 Bravais Lattices in 3D

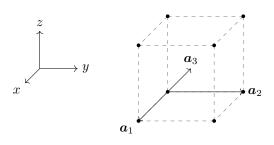
There are a number of typical Bravais lattices in three dimensions. The simplest one is the simple cubic.

1. Simple Cubic:



$$\boldsymbol{a}_1 = a\hat{\boldsymbol{x}} \quad \boldsymbol{a}_2 = a\hat{\boldsymbol{y}} \quad \boldsymbol{a}_3 = a\hat{\boldsymbol{z}}$$

2. Body-centered Cubic (bcc):

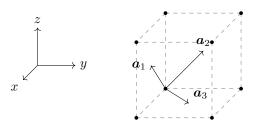


$$a_1 = a\hat{x}$$
 $a_2 = a\hat{y}$ $a_3 = \frac{a}{2}(\hat{x} + \hat{y} + \hat{z})$

An alternative choice of primitive translation vectors that does not assign a bias to a particular vector is,

$$m{a}_1 = rac{a}{2}(-\hat{m{x}} + \hat{m{y}} + \hat{m{z}}) \quad m{a}_2 = rac{a}{2}(\hat{m{x}} - \hat{m{y}} + \hat{m{z}}) \quad m{a}_3 = rac{a}{2}(\hat{m{x}} + \hat{m{y}} - \hat{m{z}})$$

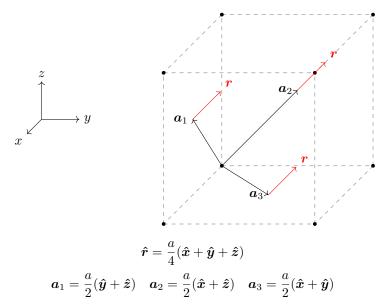
3. Face-centered Cubic (fcc):



$$a_1 = \frac{a}{2}(\hat{y} + \hat{z})$$
 $a_2 = \frac{a}{2}(\hat{x} + \hat{z})$ $a_3 = \frac{a}{2}(\hat{x} + \hat{y})$

4. **Diamond Structure**: The diamond crystal structure consists of a fcc Bravais lattice with a 2-atom basis,

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The Diamond structure is a generalization of the graphene lattice in 3D. Many materials include C (diamond), Si, Ge and Sn crystallize in this diamond structure. The nearest-neighbor hopping Hamiltonian is as follows,

$$H_1 = -\sum_{\mathbf{R}} (t|\mathbf{R}, 1)\langle \mathbf{R}, 2| + t|\mathbf{R} + \mathbf{a}_1, 1\rangle\langle \mathbf{R}, 2| + t|\mathbf{R} + \mathbf{a}_2, 1\rangle\langle \mathbf{R}, 2| + t|\mathbf{R} + \mathbf{a}_3, 1\rangle\langle \mathbf{R}, 2| + \text{h.c.})$$
(2.64)

While the second-nearest neighbor terms are numerous. In fact there are 24 terms written below,

$$H_{2} = i \frac{8\Delta_{SO}}{a^{2}} \sum_{\alpha \in \{1,2\}} \sum_{\mathbf{R}} \left[\cdots \right]$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{1}, \alpha\rangle \langle \mathbf{R}, \alpha | \mathbf{S} \cdot (\mathbf{r} \times \mathbf{a}_{1}) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{2}, \alpha\rangle \langle \mathbf{R}, \alpha | \mathbf{S} \cdot (\mathbf{r} \times \mathbf{a}_{2}) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{3}, \alpha\rangle \langle \mathbf{R}, \alpha | \mathbf{S} \cdot (\mathbf{r} \times \mathbf{a}_{3}) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{2} - \mathbf{a}_{1}, \alpha\rangle \langle \mathbf{R}, \alpha | \mathbf{S} \cdot (\mathbf{r} - \mathbf{a}_{1}) \times \left(\mathbf{r} - \frac{a}{2}\hat{\mathbf{y}}\right) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{3} - \mathbf{a}_{1}, \alpha\rangle \langle \mathbf{R}, \alpha | \mathbf{S} \cdot (\mathbf{r} - \mathbf{a}_{1}) \times \left(\mathbf{r} - \frac{a}{2}\hat{\mathbf{z}}\right) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{2} - \mathbf{a}_{3}, \alpha\rangle \langle \mathbf{R}, \alpha | \mathbf{S} \cdot (\mathbf{r} - \mathbf{a}_{3}) \times \left(\mathbf{r} - \frac{a}{2}\hat{\mathbf{y}}\right) + \cdots$$

$$\cdots + |\mathbf{h.c.}|$$

$$\cdots + |\mathbf{h.c.}|$$

$$(2.65)$$

We can explicitly calculate the cross products,

$$r \times a_{1} = \frac{a^{2}}{8}(\hat{x} + \hat{y} + \hat{z}) \times (\hat{y} + \hat{z}) = \frac{a^{2}}{8}(\hat{z} - \hat{y})$$

$$r \times a_{2} = \frac{a^{2}}{8}(\hat{x} + \hat{y} + \hat{z}) \times (\hat{z} + \hat{x}) = \frac{a^{2}}{8}(\hat{x} - \hat{z})$$

$$r \times a_{3} = \frac{a^{2}}{8}(\hat{x} + \hat{y} + \hat{z}) \times (\hat{x} + \hat{y}) = \frac{a^{2}}{8}(\hat{y} - \hat{x})$$

$$(r - a_{1}) \times (r - \frac{a}{2}\hat{y}) = \frac{a^{2}}{16}(\hat{x} - \hat{y} - \hat{z}) \times (\hat{x} - \hat{y} + \hat{z}) = -\frac{a^{2}}{8}(\hat{x} + \hat{y})$$

$$(r - a_{1}) \times (r - \frac{a}{2}\hat{z}) = \frac{a^{2}}{16}(\hat{x} - \hat{y} - \hat{z}) \times (\hat{x} + \hat{y} - \hat{z}) = +\frac{a^{2}}{8}(\hat{x} + \hat{z})$$

$$(r - a_{3}) \times (r - \frac{a}{2}\hat{y}) = \frac{a^{2}}{16}(-\hat{x} - \hat{y} + \hat{z}) \times (\hat{x} - \hat{y} + \hat{z}) = +\frac{a^{2}}{8}(\hat{y} + \hat{z})$$

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Therefore eq. (2.65) becomes,

$$H_{2} = i\Delta_{SO} \sum_{\alpha \in \{1,2\}} \sum_{\mathbf{R}} \left[\cdots \right]$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{1}, \alpha\rangle \langle \mathbf{R}, \alpha| (S^{z} - S^{y}) + \cdots$$

$$\cdots - |\mathbf{R} + \mathbf{a}_{2}, \alpha\rangle \langle \mathbf{R}, \alpha| (-S^{x} + S^{z}) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{3}, \alpha\rangle \langle \mathbf{R}, \alpha| (S^{y} - S^{x}) + \cdots$$

$$\cdots - |\mathbf{R} + \mathbf{a}_{2} - \mathbf{a}_{1}, \alpha\rangle \langle \mathbf{R}, \alpha| (S^{x} + S^{y}) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{3} - \mathbf{a}_{1}, \alpha\rangle \langle \mathbf{R}, \alpha| (S^{x} + S^{z}) + \cdots$$

$$\cdots + |\mathbf{R} + \mathbf{a}_{2} - \mathbf{a}_{3}, \alpha\rangle \langle \mathbf{R}, \alpha| (S^{y} + S^{z}) + \cdots$$

$$\cdots + |\mathbf{h.c.}|$$

$$\cdots + \text{h.c.}$$

The nearest neighbors Hamiltonian eq. (2.64) can be Fourier transformed and partially diagonalized,

$$H_1 = -\sum_{\mathbf{k}} \{ |\mathbf{k}, 1\rangle \langle \mathbf{k}, 2| \left((t + \delta t) + te^{-i\mathbf{k}\cdot\mathbf{a}_1} + te^{-i\mathbf{k}\cdot\mathbf{a}_2} + te^{-i\mathbf{k}\cdot\mathbf{a}_3} \right) + \text{h.c.} \}$$

Similarly for eq. (2.66),

$$H_{2} = i\Delta_{SO} \sum_{\alpha \in \{1,2\}} \sum_{\mathbf{k}} |\mathbf{k}, \alpha\rangle \langle \mathbf{k}, \alpha| \left[\cdots \right]$$

$$\cdots + e^{-i\mathbf{k} \cdot \mathbf{a}_{1}} (S^{z} - S^{y}) + \cdots$$

$$\cdots + e^{-i\mathbf{k} \cdot \mathbf{a}_{2}} (S^{x} - S^{z}) + \cdots$$

$$\cdots + e^{-i\mathbf{k} \cdot \mathbf{a}_{3}} (S^{y} - S^{x}) + \cdots$$

$$\cdots - e^{-i\mathbf{k} \cdot (\mathbf{a}_{2} - \mathbf{a}_{1})} (S^{x} + S^{y}) + \cdots$$

$$\cdots + e^{-i\mathbf{k} \cdot (\mathbf{a}_{3} - \mathbf{a}_{1})} (S^{x} + S^{z}) + \cdots$$

$$\cdots + e^{-i\mathbf{k} \cdot (\mathbf{a}_{2} - \mathbf{a}_{3})} (S^{y} + S^{z}) + \cdots$$

$$\cdots + h.c.$$

$$(2.67)$$

Introducing pseudo-spin $\boldsymbol{\sigma} = (\sigma^x, \sigma^y, \sigma^z)$

$$\sigma^{x} = \begin{pmatrix} 0 & 1 \\ 1 & 0 \end{pmatrix} = |1\rangle\langle 2| + |2\rangle\langle 1|$$

$$\sigma^{y} = \begin{pmatrix} 0 & -i \\ i & 0 \end{pmatrix} = -i(|1\rangle\langle 2| - |2\rangle\langle 1|)$$

$$\sigma^{z} = \begin{pmatrix} 1 & 0 \\ 0 & -1 \end{pmatrix} = |1\rangle\langle 1| - |2\rangle\langle 2|$$

The first and second nearest-neighbor Hamiltonian can be modeled when the tunneling amplitudes in eq. (2.64) are perturbed a small amount $t \mapsto t + \delta t$ using the 5-component vector $\mathbf{d}(\mathbf{k}) = (d_1(\mathbf{k}), d_2(\mathbf{k}), d_3(\mathbf{k}), d_4(\mathbf{k}), d_5(\mathbf{k}))$,

$$\begin{aligned} d_1(\boldsymbol{k}) &= -(t+\delta t) - t[\cos \boldsymbol{k} \cdot \boldsymbol{a}_1 + \cos \boldsymbol{k} \cdot \boldsymbol{a}_2 + \cos \boldsymbol{k} \cdot \boldsymbol{a}_3] \\ d_2(\boldsymbol{k}) &= -t[\sin \boldsymbol{k} \cdot \boldsymbol{a}_1 + \sin \boldsymbol{k} \cdot \boldsymbol{a}_2 + \sin \boldsymbol{k} \cdot \boldsymbol{a}_3] \\ d_3(\boldsymbol{k}) &= \Delta_{\mathrm{SO}}[\sin \boldsymbol{k} \cdot \boldsymbol{a}_2 - \sin \boldsymbol{k} \cdot \boldsymbol{a}_3 - \sin \boldsymbol{k} \cdot (\boldsymbol{a}_2 - \boldsymbol{a}_1) + \sin \boldsymbol{k} \cdot (\boldsymbol{a}_3 - \boldsymbol{a}_1)] \\ d_4(\boldsymbol{k}) &= \Delta_{\mathrm{SO}}[\sin \boldsymbol{k} \cdot \boldsymbol{a}_3 - \sin \boldsymbol{k} \cdot \boldsymbol{a}_1 + \sin \boldsymbol{k} \cdot (\boldsymbol{a}_1 - \boldsymbol{a}_2) - \sin \boldsymbol{k} \cdot (\boldsymbol{a}_3 - \boldsymbol{a}_2)] \\ d_5(\boldsymbol{k}) &= \Delta_{\mathrm{SO}}[\sin \boldsymbol{k} \cdot \boldsymbol{a}_1 - \sin \boldsymbol{k} \cdot \boldsymbol{a}_2 - \sin \boldsymbol{k} \cdot (\boldsymbol{a}_1 - \boldsymbol{a}_3) + \sin \boldsymbol{k} \cdot (\boldsymbol{a}_2 - \boldsymbol{a}_3)] \end{aligned}$$

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Using this the full Hamiltonian can be written as,

$$H(\mathbf{k}) = d_1(\mathbf{k})\sigma^x + d_2(\mathbf{k})\sigma^y + d_3(\mathbf{k})\sigma^z S^x + d_4(\mathbf{k})\sigma^z S^y + d_5(\mathbf{k})\sigma^z S^z$$

Where the matrices on the right of each $d_i(\mathbf{k})$ term are the usual 4 gamma matrices γ^{μ} from the Dirac equation.

$$\begin{split} \gamma^0 &= \sigma^x \quad \gamma^0 \gamma^1 = \sigma^y \quad \gamma^0 \gamma^2 = \sigma^z S^x \quad \gamma^0 \gamma^3 = \sigma^z S^y \quad \gamma^0 \gamma^5 = \sigma^z S^z \\ \gamma^1 &= \sigma^x \sigma^y = i \sigma^z \quad \gamma^2 = i \sigma^y S^x \gamma^3 = i \sigma^y S^y \\ i \gamma^0 \gamma^1 \gamma^2 \gamma^3 &= i \sigma^x i \sigma^z i \sigma^y S^x = \sigma^y S^z \equiv \gamma^5 \end{split}$$

It is convenient to rename these matrices using capital gammas,

$$\begin{split} \Gamma^1 &= \gamma^0 &= \sigma^x \\ \Gamma^2 &= \gamma^0 \gamma^1 &= \sigma^y \\ \Gamma^3 &= \gamma^0 \gamma^2 = \sigma^z S^x \\ \Gamma^4 &= \gamma^0 \gamma^3 = \sigma^z S^y \\ \Gamma^4 &= \gamma^0 \gamma^5 = \sigma^z S^z \end{split}$$

It is easy to see that these Γ matrices obey the Clifford Algebra,

$$\{\Gamma^a, \Gamma^b\} = 2\delta_{ab}$$

Thus we have that,

$$H(\mathbf{k}) = \sum_{a=1}^{5} d_a(\mathbf{k}) \Gamma^a \equiv d_a(\mathbf{k}) \Gamma^a$$
(2.68)

Just like in graphene, we see that the parity transformation $P = \sigma^x = \gamma^0$ leaves Γ^1 even under parity and $\Gamma^{2,3,4,5}$ odd under parity. One reason motivating why we should write eq. (2.68) in terms of the Γ matrices is that we can now easily find the eigenvalues by squaring the Hamiltonian,

$$H^2(\mathbf{k}) = d_a(\mathbf{k})\Gamma^a d_b(\mathbf{k})\Gamma^b = d_a(\mathbf{k})d_a(\mathbf{k})$$

Therefore,

$$\varepsilon_{\pm}(\mathbf{k}) = \pm \sqrt{d_a(\mathbf{k})d_a(\mathbf{k})} = \pm |\mathbf{d}(\mathbf{k})|$$

We now elect to compute the reciprocal basis vectors. Recall that,

$$m{a}_1 = rac{a}{2}(\hat{m{y}} + \hat{m{z}}) \quad m{a}_2 = rac{a}{2}(\hat{m{x}} + \hat{m{z}}) \quad m{a}_3 = rac{a}{2}(\hat{m{x}} + \hat{m{y}})$$

Define the scalar v_c such that,

$$v_c = \boldsymbol{a}_1 \cdot (\boldsymbol{a}_2 \times \boldsymbol{a}_3) = \dots = \frac{a^3}{4}$$

The reciprocal vectors become,

$$\boldsymbol{b}_1 = \frac{2\pi}{v_c} \boldsymbol{a}_2 \times \boldsymbol{a}_3 = \frac{2\pi}{a} (\hat{\boldsymbol{y}} - \hat{\boldsymbol{x}} + \hat{\boldsymbol{z}})$$

Similarly,

$$oldsymbol{b}_2 = rac{2\pi}{v_c}(oldsymbol{\hat{z}} - oldsymbol{\hat{y}} + oldsymbol{\hat{x}})$$

$$oldsymbol{b}_3 = rac{2\pi}{v_c}(oldsymbol{\hat{x}} - oldsymbol{\hat{z}} + oldsymbol{\hat{y}})$$

Therefore the reciprocal lattice of fcc is the bcc lattice vectors and vice versa. When $\delta t = 0$ the 5-component $d(\mathbf{k})$ vanishes at 3 inequivalent band-touching point in the first Brillouin zone.

$$X^{x} = \frac{2\pi}{a}(1,0,0)$$
 $X^{y} = \frac{2\pi}{a}(0,1,0)$ $X^{z} = \frac{2\pi}{a}(0,0,1)$

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Considering the X^z point as an illustrative example we have $k_1 = k_2 = \frac{1}{2}$ and $k_3 = 0$. If we expand in small deviations from the X^z point where $\mathbf{k} \simeq X^z + \mathbf{q} = \left(q_x, q_y, \frac{2\pi}{a} + q_z\right)$ we obtain the following,

$$\begin{aligned} d_1(\mathbf{k}) &\simeq -t - \delta t - t(-1) = -\delta t \\ d_2(\mathbf{k}) &\simeq taq_z \\ d_3(\mathbf{k}) &\simeq -2\Delta_{\mathrm{SO}}aq_x \\ d_4(\mathbf{k}) &\simeq 2\Delta_{\mathrm{SO}}aq_y \\ d_5(\mathbf{k}) &\simeq \frac{\Delta_{\mathrm{SO}}}{8}a^3 (q_y^2 - q_x^2)q_z \end{aligned}$$

Thus we obtain (replacing q with k again),

$$H_{X^z} = -\delta t \sigma^x + tak_z \sigma^y - 2\Delta_{SO} a \sigma^z S^x k_x + 2\Delta_{SO} a \sigma^z S^y k_y$$

Analogously expanding about X^y and X^x ,

$$H_{X^y} = -\delta t \sigma^x + tak_y \sigma^y + 2\Delta_{SO} ak_x \sigma^z S^x - 2\Delta_{SO} ak_z \sigma^z S^z$$

$$H_{X^x} = -\delta t \sigma^x + tak_x \sigma^y - 2\Delta_{SO}ak_y \sigma^z S^y + 2\Delta_{SO}ak_z \sigma^z S^z$$

Notice that $H_{X^{x,y,z}}$ look like anisotropic Dirac Hamiltonians where δt plays the role of the Dirac mass.

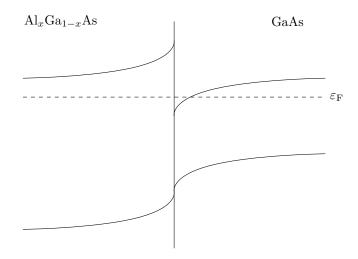
$$\varepsilon_{\pm}(\boldsymbol{k}) = \pm \sqrt{t^2 a^2 k_z^2 + 4 \Delta_{\mathrm{SO}}^2 a^2 \left(k_x^2 + k_y^2\right) + \delta t^2}$$

Changing the sign of δt corresponds to transitions between topological and ordinary insulators.

- $\delta t > 0$: topological
- $\delta t < 0$: ordinary
- $\delta t = 0$: Dirac semimetal

2.7 Quantum Hall Effect

In this subsection we will discuss the Quantum Hall Effect. Specifically we will look at the Quantum Hall Effect in a 2-dimensional electron gas (2DEG) when subjected to a magnetic field. 2DEG arise at heterointerfaces; interfaces between two different semiconductor materials. Semiconductors are insulators with a small band gap $< 1\,\mathrm{eV}$.



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Modeling a 2DEG corresponds to subjecting the electrons in state $|\psi\rangle$ to a potential V(r) and making use of the Schrödinger equation,

$$\left(-\frac{\hbar^2}{2m}\nabla^2 + V(\mathbf{r})\right)\psi(\mathbf{r}) = E\psi(\mathbf{r})$$
(2.69)

Where $\psi(\mathbf{r}) \equiv \langle \mathbf{r} | \psi \rangle$ is the position space wavefunction. In order to model electrons confined to two dimensions (say x, y) use the following potential,

$$V(\mathbf{r}) = \begin{cases} 0 & \mathbf{r} \cdot \hat{\mathbf{z}} = 0 \\ \infty & \mathbf{r} \cdot \hat{\mathbf{z}} \neq 0 \end{cases}$$
 (2.70)

By doing so, $\psi(r)$ is confined to be non-zero within the xy-plane.

$$\psi(\mathbf{r}) = \psi(x\hat{\mathbf{x}} + y\hat{\mathbf{y}}) \equiv \psi(x, y)$$

Furthermore let L_x and L_y denote the side lengths of rectangle space accessible by the electrons equipped with periodic boundary conditions,

$$\psi(x + L_x, y) = \psi(x, y + L_y) = \psi(x, y)$$
(2.71)

The electrons inside the accessible region are subject to no external potential. Therefore the electrons will exhibit plane-wave behaviour with z=0.

$$\psi(\mathbf{r}) = \gamma e^{i\mathbf{k}\cdot\mathbf{r}} = \gamma e^{i(k_x x + k_y y)} \tag{2.72}$$

Where $\mathbf{k} = k_x \hat{\mathbf{x}} + k_y \hat{\mathbf{y}}$ denotes the momentum wave vector of the electron and $\gamma \in \mathbb{R}$ is a normalization factor. The normalization factor comes from the desired normalization of the state $\psi(\mathbf{r})$,

$$1 = \langle \psi | \psi \rangle = \int_{0}^{L_x} dx \int_{0}^{L_y} dy \langle \psi | \boldsymbol{r} \rangle \langle \boldsymbol{r} | \psi \rangle = \int_{0}^{L_x} dx \int_{0}^{L_y} dy \gamma^2 \left| e^{i(k_x x + k_y y)} \right|^2 = L_x L_y \gamma^2$$

Therefore the normalization is simply,

$$\gamma = \frac{1}{\sqrt{L_x L_y}}$$

Consequently eq. (2.72) becomes,

$$\psi(\mathbf{r}) = \frac{1}{\sqrt{L_x L_y}} e^{i(k_x x + k_y y)}$$

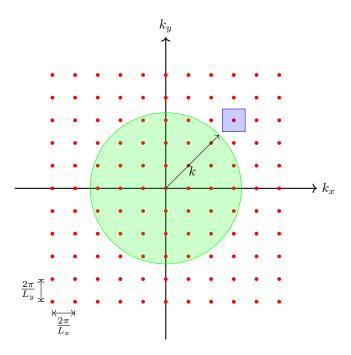
Moreover the boundary conditions eq. (2.71) quantize the accessible values of k_x, k_y ,

$$e^{i(k_x(x+L_x)+k_yy)} = e^{i(k_xx+k_y(y+L_y))} = e^{i(k_xx+k_yy)}$$

$$e^{ik_x L_x} = e^{ik_y L_y} = 1 \implies k_x = \frac{2\pi m_x}{L_x} \quad k_y = \frac{2\pi m_y}{L_y} \quad m_x, m_y \in \{0, \pm 1, \pm 2, \ldots\}$$

Therefore the accessible points in k-space form a square lattice which is illustrated below,

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Clearly, each accessible momentum state occupies an area (shaded in blue) equal to,

$$\left(\frac{2\pi}{L_x}\right)\left(\frac{2\pi}{L_y}\right) = \frac{\left(2\pi\right)^2}{\mathcal{A}}$$

Where $A = L_x L_y$ is the total area. In the inverse of this statement is that there are,

$$\frac{\mathcal{A}}{\left(2\pi\right)^2}\tag{2.73}$$

grid points per unit area of k-space. Using eq. (2.69), we can now calculate the energy for the state $\psi(r_x, r_y) \equiv \langle r | \psi \rangle$.

$$E\psi(\boldsymbol{r}) = -\frac{\hbar^2}{2m}\nabla^2\psi(\boldsymbol{r}) = -\frac{\hbar^2}{2m}\big(\partial_x^2 + \partial_y^2\big)\psi(\boldsymbol{r}) = -\frac{\hbar^2}{2m}\big(-k_x^2 - k_y^2\big)\psi(\boldsymbol{r}) = \frac{\hbar^2}{2m}k^2\psi(\boldsymbol{r})$$

Where $k = \sqrt{k_x^2 + k_y^2}$ is a radius in k-space. Therefore the energy levels are,

$$E = \frac{\hbar^2 k^2}{2m} = \frac{\hbar^2}{2m} (2\pi)^2 \left(\frac{m_x^2}{L_x^2} + \frac{m_y^2}{L_y^2} \right) \qquad m_x, m_y \in \{0, \pm 1, \pm 2, \dots\}$$
 (2.74)

Let $N(\varepsilon)$ denote the number of states accessible to a single electron with energy ε . Since this function is generally non-trivial, we approximate $N(\varepsilon)$ as the infinitesimal amount $\delta N(\varepsilon)$ as follows,

$$\delta N(\varepsilon) \simeq N_{<}(\varepsilon + \delta \varepsilon/2) - N_{<}(\varepsilon - \delta \varepsilon/2)$$

Where $N_{<}(\varepsilon)$ is the number of states accessible to any electron with at most energy ε . Evidently this quantity is proportional to an area in k-space. Namely, the area of k space associated with $N_{<}$ is a circle centered at $(k_x, k_y) = (0, 0)$. Therefore,

$$N_{<}(\varepsilon) \propto \pi k^2 = \pi \frac{2m\varepsilon}{\hbar^2}$$

The proportionality constant is exactly eq. (2.73),

$$N_{<}(\varepsilon) = \frac{\mathcal{A}}{(2\pi)^2} \frac{2m\pi\varepsilon}{\hbar^2}$$

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Therefore,

$$\delta N(\varepsilon) = \frac{\mathcal{A}}{\left(2\pi\right)^2} \frac{2m\pi}{\hbar^2} \delta \varepsilon$$

Forming a differential form,

$$\frac{\mathrm{d}N}{\mathrm{d}\varepsilon} = \frac{\mathcal{A}}{\left(2\pi\right)^2} \frac{2m\pi}{\hbar^2}$$

The density of states is a constant. Normalizing for area,

$$\frac{1}{\mathcal{A}} \frac{\mathrm{d}N}{\mathrm{d}\varepsilon} = \frac{2m\pi}{\left(2\pi\right)^2 \hbar^2}$$

This however, does not account for the fact that each electron has two accessible spin states for each k-space lattice point. Therefore, there are truly twice as many states available to each electron,

$$\frac{1}{\mathcal{A}} \frac{\mathrm{d}N}{\mathrm{d}\varepsilon} = \frac{4m\pi}{\left(2\pi\right)^2 \hbar^2}$$

Therefore the density of states is,

$$g(\varepsilon) \equiv \frac{1}{\mathcal{A}} \frac{\mathrm{d}N}{\mathrm{d}\varepsilon} = \frac{m}{\pi\hbar^2} \tag{2.75}$$

The electron density for a given Fermi wavevector $k_{\rm F}$ is the number of electrons per unit area in r-space. At zero temperature and Fermi energy $\varepsilon_{\rm F}$ the states in k-space are occupied at lowest energies first. Therefore the total number of electrons $N(\varepsilon_{\rm F})$ is given by,

$$N(\varepsilon_{\rm F}) = \mathcal{A} \int_0^{\varepsilon_{\rm F}} \mathrm{d}\varepsilon g(\varepsilon) = \mathcal{A} \int_0^{\varepsilon_{\rm F}} \mathrm{d}\varepsilon \frac{m}{\pi \hbar^2} = \frac{m}{\pi \hbar^2} \varepsilon_{\rm F} \mathcal{A}$$

Given that the Fermi energy and Fermi wavevector are related via eq. (2.74), the electron density is,

$$n(k_{\rm F}) = \frac{1}{\mathcal{A}} N(k_{\rm F}) = \frac{m}{\pi \hbar^2} \frac{\hbar^2 k_{\rm F}^2}{2m}$$

$$n(k_{\rm F}) = \frac{k_{\rm F}^2}{2\pi}$$
(2.76)

Now, we place the 2DEG in the xy-plane and subject it to a magnetic field in the z-direction $\mathbf{B} = B\hat{\mathbf{z}}$. If the electrons are classical particles of charge -e and mass m,

$$\frac{\mathrm{d} \boldsymbol{p}}{\mathrm{d} t} = \boldsymbol{F}_{\mathrm{L}} = -\frac{e}{c} \boldsymbol{v} \times \boldsymbol{B} \qquad \boldsymbol{v} = \frac{\boldsymbol{p}}{m}, \boldsymbol{v} = \frac{\mathrm{d} \boldsymbol{r}}{\mathrm{d} t}$$

Solving these equations yields,

$$m\frac{\mathrm{d}^2 \boldsymbol{r}}{\mathrm{d}t^2} = -\frac{e}{c}\frac{\mathrm{d}\boldsymbol{r}}{\mathrm{d}t} \times \boldsymbol{B}$$

Therefore we obtain a two dimensional system of coupled ordinary differentiable equations,

$$\frac{\mathrm{d}^2 x}{\mathrm{d}t^2} = -\frac{eB}{mc} \frac{\mathrm{d}y}{\mathrm{d}t}$$
$$\frac{\mathrm{d}^2 y}{\mathrm{d}t^2} = \frac{eB}{mc} \frac{\mathrm{d}x}{\mathrm{d}t}$$

Where the frequency $\omega_c = \frac{eB}{mc}$ is the cyclotron frequency. In order to solve this system, introduce a complex parameter z = x + iy such that,

$$\frac{\mathrm{d}^2 x}{\mathrm{d}t^2} = i\omega_x \frac{\mathrm{d}z}{\mathrm{d}t}$$
$$z(t) = z_0 - \frac{iv_0}{\omega_c} e^{i\omega_c t}$$

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Which is circular motion with frequency ω_c . Therefore if we need to model the average time between collisions with impurities, the simplest description involves a decrease in momentum with characteristic time τ .

$$\frac{\mathrm{d}\boldsymbol{p}}{\mathrm{d}t} = -e\left(\boldsymbol{E} + \frac{1}{c}\boldsymbol{v} \times \boldsymbol{B}\right) - \frac{\boldsymbol{p}}{\tau}$$

Therefore without a magnetic field we have,

$$\frac{\mathrm{d}\boldsymbol{p}}{\mathrm{d}t} = \frac{-\boldsymbol{p}}{\tau} \implies \boldsymbol{p}(t) = \boldsymbol{p}(0)e^{-t/\tau}$$

In general,

$$m\frac{\mathrm{d}\boldsymbol{v}}{\mathrm{d}t} = -\frac{m\boldsymbol{v}}{\tau} - e\bigg(\boldsymbol{E} + \frac{1}{c}\boldsymbol{v} \times \boldsymbol{B}\bigg)$$

Assuming that E and B are time-independent, we can look for the time-independent steady state solution corresponds to a fixed v,

$$\frac{\mathrm{d}\boldsymbol{v}}{\mathrm{d}t} = 0 \implies \boldsymbol{v} - \frac{e\tau}{m} \left(\boldsymbol{E} + \frac{1}{c} \boldsymbol{v} \times \boldsymbol{B} \right)$$

The electron v is connect to the current density j through the following equation we have seen before,

$$\mathbf{j} = -en\mathbf{v}$$

Where n is the electron density and e is the electric charge. Therefore,

$$oldsymbol{j} = rac{e^2 n au}{m} \Big(oldsymbol{E} - rac{e au}{m c} oldsymbol{j} imes oldsymbol{B} \Big)$$

In the absence of a magnetic field B, we recover Ohm's law,

$$oldsymbol{j} = rac{e^2n au}{m}oldsymbol{E}$$

Where the proportionality is the conductivity σ ,

$$\sigma = \frac{e^2 n\tau}{m}$$

The electric field is,

$$oldsymbol{E} = rac{oldsymbol{j}}{\sigma} - rac{1}{nec} oldsymbol{B} imes oldsymbol{j}$$

Component-wise,

$$E_x = \frac{1}{\sigma} j_x + \frac{B}{nec} j_y$$
$$E_y = \frac{1}{\sigma} j_y - \frac{B}{nec} j_x$$

If there is no current in the y-direction $(j_y = 0)$ we have,

$$E_y = -\frac{B}{nec}j_x$$
 $\frac{E_y}{Bj_x} = -\frac{1}{nec} = -R_H$

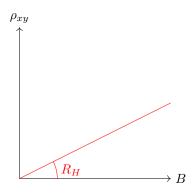
The last equality is a definition of the **Hall coefficient** R_H

$$R_H = \frac{1}{nec}$$

Notice that the Hall coefficient only depends on the density of the electrons and other fundamental constants. Now we introduce the **resistivity tensor**,

$$\rho = \begin{pmatrix} 1/\sigma & BR_H \\ -BR_H & 1/\sigma \end{pmatrix}$$

We refer to this as the **classical Hall effect**.



In reality, the classical Hall effect occurs at weak magnetic fields or in dirty samples. If we let $1/\tau$ characterize the impurity scattering rate, our classical Hall effect is relevant in the domain where,

$$\omega_C \tau \ll 1$$

Recall that the cyclotron frequency is ω_c ,

$$\omega_c = \frac{eB}{mc}$$

If $\omega_C \tau \ll 1$, the electron will be scatter many times before being able to complete a single circular orbit; i.e. if the period of the circular motion of the electron is greater than the mean-free time, then we can model the electrons as a classical particle.

In we have $\omega_C \tau \gg 1$, then the classical Hall effect is no longer the dominant feature. If the circular motion of the electrons becomes quantized, we need to consider the **quantum Hall effect**. The quantum aspects of the electron need to be considered in really clean materials or in high magnetic fields.

To study the quantum Hall effect, consider free quantum particles of charge -e in the xy-plane in a magnetic field $\mathbf{B} = B\hat{z}$ pointing in the \hat{z} direction. In the absence of a magnetic field, we have the relation $\mathbf{p} = m\mathbf{v}$ which makes the Hamiltonian,

$$H = -\frac{\hbar^2 \nabla^2}{2m}$$

However, a magnetic field $\boldsymbol{B} = \boldsymbol{\nabla} \times \boldsymbol{A}$ we have,

$$m\mathbf{v} = \mathbf{p} + \frac{e}{c}\mathbf{A}$$

Therefore the Hamiltonian for a charged particle in a magnetic field is,

$$H = \frac{1}{2m} \left(-i\hbar \nabla + \frac{e}{c} \mathbf{A} \right)^2 - \boldsymbol{\mu} \cdot \mathbf{B}$$

Where μ is the magnetic dipole moment. For simplicity, we neglect μ for now. Therefore, the Hamiltonian of interest is,

$$H = \frac{1}{2m} \left(-i\hbar \nabla + \frac{e}{c} \mathbf{A} \right)^2 \tag{2.77}$$

Where we wish to find the energy eigenstates ψ where $H\psi = E\psi$. Moreover, the magnetic field is determined by $\mathbf{B} = \nabla \times \mathbf{A}$ leaving a gauge invariance in the choice of \mathbf{A} ,

$$A \rightarrow A + \nabla f$$

The gauge choice we will used is called the **Landau gauge**.

$$\mathbf{A} = xB\hat{\mathbf{y}}$$

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Therefore,

$$(\mathbf{\nabla} \times \mathbf{A})_z = \frac{\partial A_y}{\partial x} - \frac{\partial A_x}{\partial y} = B$$

Analogously the x, y components of $\nabla \times \mathbf{A}$ are as follows,

$$(\mathbf{\nabla} \times \mathbf{A})_x = \frac{\partial A_z}{\partial y} - \frac{\partial A_y}{\partial z} = 0$$

$$(\mathbf{\nabla} \times \mathbf{A})_y = \frac{\partial A_x}{\partial z} - \frac{\partial A_z}{\partial x} = 0$$

Therefore eq. (2.77) becomes,

$$H = \frac{1}{2m} \left(-i\hbar \nabla + \frac{e}{c} \mathbf{A} \right)^{2}$$

$$= \frac{1}{2m} \left(-i\hbar \nabla + \frac{e\mathbf{B}}{c} \times \hat{\mathbf{y}} \right)^{2}$$

$$= -\frac{\hbar^{2}}{2m} \frac{\partial^{2}}{\partial x^{2}} + \frac{1}{2m} \left(-i\hbar \frac{\partial}{\partial y} + \frac{e\mathbf{B}}{c} x \right)^{2}$$
(2.78)

Notice that H is not invariant with respect to translations along the x-direction but is invariant with respect to translations along the y-direction. Translation symmetries give rise to wave functions that are also momentum eigenstates. If the energy eigenstates $\psi(x,y)$ are thus functions of x,y and have the following form,

$$H\psi(x,y) = E\psi(x,y)$$

Symmetry analysis suggests we should expect ψ to the separable into a plane wave solution with non-trivial x-dependence $\varphi(x)$.

$$\psi(x,y) = e^{ik_y y} \varphi(x)$$

With k_y to be determined. Substituting this ansatz into eq. (2.78) yields,

$$E\psi(x,y) = -\frac{\hbar^2}{2m} \frac{\partial^2 \varphi}{\partial x^2} e^{ik_y y} + \frac{1}{2m} \left(\hbar k_y + \frac{eB}{c} x \right)^2 \varphi e^{ik_y y}$$

Canceling φ ,

$$E\varphi = -\frac{\hbar^2}{2m}\frac{\partial^2\varphi}{\partial x^2} + \frac{1}{2m}\left(\hbar k_y + \frac{eB}{c}x\right)^2\varphi \tag{2.79}$$

Gives us a 1D differential equation. For convenience of notation we introduce the magnetic length ℓ ,

$$\ell = \sqrt{\frac{\hbar c}{eB}}$$

Additionally re-use the classical cyclotron frequency,

$$\omega_C = \frac{eB}{mc}$$

Using these constants, eq. (2.79) can we re-expressed as,

$$-\frac{\hbar^2}{2m}\frac{\partial^2 \varphi}{\partial x^2} + \frac{m\omega_C^2}{2} (x + k_y \ell^2)^2 \varphi = E\varphi$$

At this stage, it should be clear that this Hamiltonian is nothing more than a 1D Harmonic oscillator with frequency ω_C and centered at $x = -k_y \ell^2$. The solution for the quantum harmonic oscillator can be found in a number of undergraduate quantum theory textbooks.

$$\psi_{nk_y}(x) = \frac{e^{ik_y y}}{\sqrt{\ell \sqrt{\pi 2^n n!}}} e^{-\frac{1}{2\ell^2} (x + k_y \ell^2)^2} H_n\left(\frac{x + k_y \ell^2}{\ell}\right)$$

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Where H_n is the *n*-th Hermite polynomial. The wave function solutions are given by the Hermite polynomials, but the important feature are the quantized energy levels,

$$E_{n,k_y} = \hbar\omega_c \left(n + \frac{1}{2} \right) \qquad n \in \{0, 1, 2, \ldots\}$$

Where n are referred to as the Landau levels. This energy level is completely independent of k_y but k_y is independently quantized like a particle in a periodic potential,

$$k_y = \frac{2\pi}{L_y} m_y$$
 $m_y \in \{0, \pm 1, \pm 2, \ldots\}$

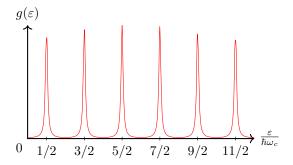
Since the effective harmonic oscillator is centered at $-k_y\ell^2$, it must be that L_x is twice the position of $k_y\ell^2$,

$$-L_x \le -k_y \ell^2 \le 0 \implies 0 \le k_y \le \frac{L_x}{\ell^2}$$

Therefore the total number of states for a fixed Landau level $n \in \{0, 1, 2, ...\}$ is given by the m_y value such that k_y is bounded by L_x/ℓ^2 ,

$$N_{\phi} = \max m_y = \max k_y \frac{L_y}{2\pi} = \frac{L_y L_x}{2\pi \ell^2} = \frac{L_x L_y eB}{2\pi \hbar c}$$
 (2.80)

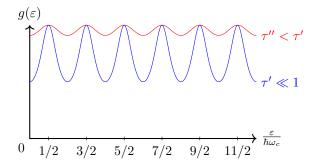
Which is independent of n. Therefore, the number of electrons in each Landau level is the same for each Landau level. The density of states $g(\varepsilon)$ forms a Dirac comb structure. In reality, however, the uncertainty principle smooths this plot due to uncertainty broadening. For low scattering rates (pure materials) we have the following profile.



Broadening increases due to impurities for the following reason. Suppose that an electron exists in a quantum state for some time τ which corresponds to the mean free time between collisions. The uncertainty in energy time is,

$$\Delta \varepsilon \Delta t \sim \Delta \varepsilon \tau \sim \hbar \implies \Delta \varepsilon \sim \frac{\hbar}{\tau}$$

In regimes where τ is very small, the broadening results in the following,



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Which more closely approximates the classical regime when $\omega_c \tau \ll 1$ and the density $g(\varepsilon)$ becomes more uniform.

$$g(\varepsilon) \approx g_0 = \frac{m}{\pi \hbar^2}$$

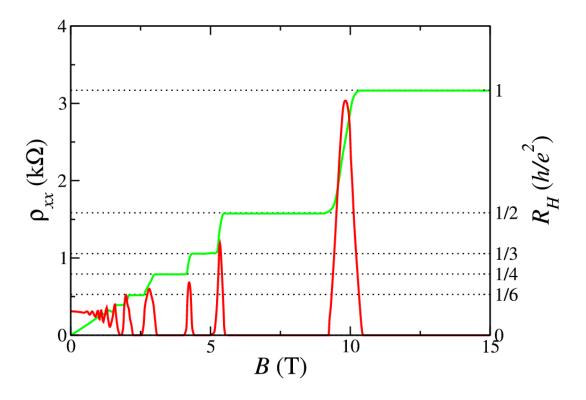
Which is identically the 2D electron density of states in the absence of a magnetic field eq. (2.75).

We refer to these energy levels as Landau levels. Every Landau level is degenerate with respect to all the different values of k. Since E_{nk} is independent of k, we say that E_{nk} has degeneracies in k. Because of this, Landau levels are analogous to bands in a crystal.

An alternative interpretation of eq. (2.80) is to define the following: The total magnetic flux through the sample area $BL_xL_y\equiv\phi$ and the magnetic flux quantum $hc/e\equiv\phi_0$. Then the Landau occupancy number becomes a flux ratio,

$$N_{\phi} = \frac{\phi}{\phi_0}$$

Combining the Landau level description at high magnetic fields B gives a completely new description of the resistivity ρ_{xy} and ρ_{xx} ,



It can be seen that ρ_{xy} develops perfectly flat plateaus exactly when $\rho_{xy} = h/e^2\nu$ where ν is an integer (this is why we refer to this phenomena as the **integer quantum hall effect**; this contrasts the **fractional quantum hall effect** where ν is permitted to be any fraction). In fact ν is called the integer Landau level filling factor corresponding to the number of completely filled Landau levels.

The states at the center of each Landau level are delocalized and can contribute to transport. States in the tails of each Landau level are localized and can not contribute to transport. When considering $k_B T \ll \hbar c$,

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the temperature of the system can be treated as effectively zero. In these cases, the Fermi-Dirac distribution $f(\varepsilon)$

$$f(\varepsilon) = \frac{1}{e^{(\varepsilon - \mu)/kT} + 1}$$

Tends towards the following step function,

$$\lim_{T \to 0} f(\varepsilon) = \begin{cases} 1 & \varepsilon < \mu \\ 0 & \varepsilon > \mu \end{cases}$$

At this temperature scale, the chemical potential is effectively the Fermi energy $\varepsilon_{\rm F}$. This makes sense; the typically energy μ to add/remove an electron would be the equivalent to the edge state fermions $\varepsilon_{\rm F}$. As the magnetic field increases the Landau occupancy N_{ϕ} increases letting $\varepsilon_{\rm F}$ tend to zero at higher and higher energies. For a fixed total number of particles N,

$$\nu = \left\lfloor \frac{N}{N_{\phi}} \right\rfloor \tag{2.81}$$

Gives the total number of *completely* filled Landau levels while the Fermi energy $\varepsilon_{\rm F}$ is simply the energy of the maximal Landau level that is at least partially occupied.

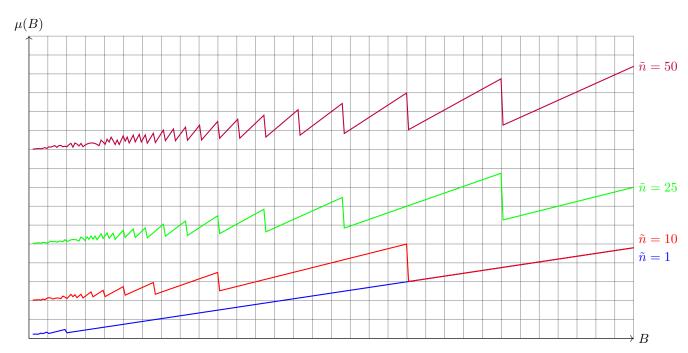
$$\varepsilon_{\rm F} = E_{\nu+1} = \hbar \omega_c \left(\nu + \frac{3}{2} \right) \tag{2.82}$$

Therefore $\varepsilon_{\rm F} = \mu$ becomes a function of B for fixed N,

$$\mu(B) = \hbar \frac{eB}{mc} \bigg(\bigg\lfloor \frac{N2\pi\hbar c}{L_x L_y eB} \bigg\rfloor + \frac{1}{2} \bigg)$$

For fixed N as $B\to\infty$, the floor function eventually reaches zero and the chemical potential tends as a straight line with slope $\frac{\hbar e}{2mc}$. In order to plot this function consider setting $\hbar=c=e=m=1$ and defining fixed $\tilde{n}=2\pi N/(L_xL_y)$,

$$\mu(B) = B\left(\left\lfloor \frac{\tilde{n}}{B} \right\rfloor + \frac{3}{2}\right)$$



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As the magnetic field increases, $N_{\phi} \propto B$ increases and $\varepsilon_{\rm F}$ moves towards zero through the Landau levels. When $\varepsilon_{\rm F}$ is in the mobility gap, the system behaves like an insulator while when it passes through the Landau level centers, it becomes metallic. This explains why nothing changes at the plateaus of ρ_{xy} but doesn't explain why ρ_{xy} is non-zero and quantized to precise values. This has to do with the existence of edge states. Consider a sample 2DEG that is finite in the x-direction admitting zero potential in regions between $\pm L_x/2$ but approaches an infinite barrier at $V(\pm L_x/2) = \infty$. Then the Schrödinger equation yields the familiar shifted harmonic oscillator with potential V(x),

$$\frac{\hbar^2}{2m} \frac{\mathrm{d}^2 \varphi}{\mathrm{d}x^2} + \frac{m\omega_c^2}{2} (x + k_y \ell^2) \varphi + V(x) \varphi = E \varphi$$

At V=0, $\varphi\sim e^{-\frac{1}{2\ell^2}\left(x+k_y\ell^2\right)^2}$ varies on the scale of the magnetic length ℓ . Suppose that V(x) is very slowly varying on this scale. In this limit, we can solve the Schrödinger equation by simply replacing $V(x)\approx V\left(k_y\ell^2\right)$. The eigenvalues then become,

$$E_{nk_y} = \hbar\omega_c \left(n + \frac{1}{2}\right) + V(k_y \ell^2)$$

Thus when $\varepsilon_{\rm F}$ is between the Landau levels in the bulk, there are still states at the Fermi energy of the edges of the system (called the edge states). Their number is equal to the number of filled Landau levels (edges are populated for each Landau level last). Moreover, near the intersection with $\varepsilon_{\rm F}$ the dispersion is *chiral*,

$$E_{nk_y} \simeq \varepsilon_{\rm F} \pm \hbar v_n (k_y \pm k_{y,n})$$

As we have seen before in eq. (2.60), the existence of the chiral edge states implies the Hall effect,

$$I_y = -e \int \frac{\mathrm{d}k}{2\pi} v n_{\mathrm{F}}(\varepsilon_k)$$

Since $dk = d\varepsilon/\hbar v$,

$$I_R = -\frac{e}{2\pi\hbar} \int_{-\infty}^{\varepsilon_{\rm FR}} d\varepsilon n_{\rm F}(\varepsilon)$$

$$I_L = +\frac{e}{2\pi\hbar} \int_{-\infty}^{\varepsilon_{\rm FL}} \mathrm{d}\varepsilon n_{\rm F}(\varepsilon)$$

Therefore,

$$I_y = I_R + I_L = -\frac{e}{h}(\varepsilon_{\rm FR} - \varepsilon_{\rm FL}) = -\frac{e^2}{h}V_x$$

Such each filled Landau level contributes $\sigma_{yx} = -e^2/h$ to the total Hall conductivity. Using the Landau level filling factor ν ,

$$\sigma_{yx} = -\frac{e^2}{h}\nu$$

Analogously,

$$\sigma_{xy} = +\frac{e^2}{h}\nu$$

As long as $\varepsilon_{\rm F}$ is in the *mobility gap* between Landau levels, σ_{xy} stays pinned at this value since the number of edge states doesn't change and the bulk states don't contribute. Therefore,

$$\sigma = \begin{pmatrix} 0 & +\frac{e^2}{h}\nu \\ -\frac{e^2}{h}\nu & 0 \end{pmatrix}$$

The resistivity tensor is defined such that,

$$\rho_{xx} = \frac{\sigma_{xx}}{\sigma_{xx}^2 + \sigma_{xy}^2} = 0$$

$$\rho_{xy} = -\frac{\sigma_{xy}}{\sigma_{xx}^2 + \sigma_{xy}^2} = -\frac{1}{\sigma_{xy}} = -\frac{h}{e^2}\nu$$

Amazingly, ρ_{xx} and σ_{xx} are zero simultaneously. Therefore a 2DEG in the quantum Hall regime behaves both as perfect insulator and a perfect conductor (in the appropriate directions).

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2.7.1 Conclusion

The topological features of electronic structures we have considered thus far are genuine manifestations of quantum mechanics on macroscopic scales. For example, resistivity in the Quantum Hall effect,

$$\rho_{xy} = \frac{h}{e^2 \nu}$$

Has dependence on Planck's constant h which indicates that it has no classical counterpart. This is in contrast with the Classical Hall effect,

$$\rho_{xy} = \frac{B}{nec}$$

which does not rely on quantum mechanics whatsoever. This concludes the first focus of this course and one of the major ideas in condensed matter physics: the effects of the electronic structure topology on macroscopic scales.

3 Electron-electron Interactions

Heretofore, we have consider materials where electrons are modeled as moving in a atomic lattice in *isolation*. In essence, the electrons are treated as being non-interacting. This approximation was well-founded as many material properties can be seen as consequences of non-interacting electrons and in most cases, electron interactions are negligible.

However, in specific cases the electron interactions yield interesting physics. For example Landau Fermi liquid theory, magnetism and superconductivity all require a description of electronic interactions.

In particular, ferro-magnetism is the appearance of spontaneous macroscopic magnetic momentum in certain materials at sufficiently low temperature T. For a material magnetic field H, the magnetization M is

$$M = \chi H \tag{3.1}$$

Where χ is the magnetic susceptibility.

- $\chi > 0$ para-magnets
- $\chi < 0$ dia-magnets

Ferromagnets are distinct from both para and dia-magnets in that they possess magnetic fields $M \neq 0$ even in the absence of H. It is this feature we are referring to when we say that ferromagnets produce magnetic fields *spontaneously*.

Unlike materials with completely filled shells (which have zero spin), materials that have incompletely filled d-shells (such as iron) have non-zero spin,

$$Fe: 3d^64s^2$$

Hund's rule states that atomic orbitals are filled in such a way as to *maximize* the total spin. Equivalently, any incompletely filled atomic orbital will always have non-zero total spin. If any incompletely filled shell has non-zero spin why are d-shells important? d-shells are important to magnetism because the electrons in higher orbital shells are spaced farther apart. This permits the electrons to have a stronger notion of localization for specific positions around atoms.

The combination of incompletely filled d-shells and Hund's rule implies that there is a non-zero atomic magnetic moments μ ,

$$\boldsymbol{\mu} = g\mu_B \boldsymbol{S}$$

Where μ_B is the **Bohr magneton**,

$$\mu_B = \frac{e\hbar}{2mc}$$

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And $g \approx 2$. Therefore,

$$\mu \sim g\mu_B \sim \frac{e\hbar}{mc}$$

The most obvious source interactions are magnetic dipole interactions

$$U_{\text{dip}} = \frac{1}{r^3} (\boldsymbol{\mu}_1 \cdot \boldsymbol{\mu}_2 - 3(\boldsymbol{\mu}_1 \cdot \hat{\boldsymbol{r}})(\boldsymbol{\mu}_2 \cdot \hat{\boldsymbol{r}}))$$

Recalling the Bohr radius a_0 ,

$$a_0 = \frac{\hbar^2}{mc^2}$$

We can write U_{dip} in terms of a distance r,

$$U_{\rm dip} \sim \frac{(g\mu_B)^3}{r^3} = \left(\frac{e\hbar}{mc}\right)^2 \frac{1}{r^3} = \left(\frac{e^2}{\hbar c}\right)^2 \frac{e^2}{a_0} \left(\frac{a_0}{r}\right)^3$$

Where the coefficients are related to the dimensionless Fine structure constant α ,

$$\alpha = \frac{e^2}{\hbar c} \simeq \frac{1}{137}$$

Using this constant,

$$U_{\rm sip} \sim \alpha^2 \frac{e^2}{a_0} \left(\frac{a_0}{r}\right)^3$$

For typical atomic separations $r \sim a_0$, we have $U_{\rm dip} \sim \alpha^2 \frac{e^2}{a_0} \sim 10^{-4} \, {\rm eV}$ which corresponds to temperatures $T \sim U_{\rm dip}/k_B \sim 1 \, {\rm K}$. This result is odd because we know that the critical temperature for iron is,

Fe:
$$T_c = 1043 \,\text{K}$$

Therefore dipole-dipole interactions can *not* be the source of ferro-magnetism.

For ordinary electron-electron Coulomb interactions e^2/a_0 is the typical Coulomb interaction energy between electrons on neighboring atoms.

$$\frac{e^2}{a_0} \sim 1 \,\text{eV}$$
 $T \sim 10^4 \,\text{K}$

This is much closer to the critical temperature for iron, therefore we conclude that ordinary Coulumb interactions are responsible for ferro-magnetism. This result is surprising because the Coulumb potential does not know about magnetic fields; instead the effects of spin (in combination with V(r)) will give rise to ferro-magnetism.

Consider 2 electrons on two atomic orbitals $\varphi_1(\mathbf{r}_1)$ and $\varphi_2(\mathbf{r}_2)$ with the same energy. The Coulumb potential is,

$$V(\boldsymbol{r}_1 - \boldsymbol{r}_2) = \frac{e^2}{|\boldsymbol{r}_1 - \boldsymbol{r}_2|}$$

In general the wave-function for the two electrons will depend on positions r_1, r_2 for the two electrons and spin values $\sigma_i \in \{\uparrow, \downarrow\}$. This position and spin wave-functions are separable,

$$\psi(\boldsymbol{r}_1\sigma_1,\boldsymbol{r}_2\sigma_2) = \psi(\boldsymbol{r}_1,\boldsymbol{r}_2)\chi(\sigma_1,\sigma_2)$$

Since the electrons are Fermions the wave-function must be antisymmetric with respect to the exchange of the two particles,

$$\psi(\mathbf{r}_2\sigma_2,\mathbf{r}_1\sigma_1) = -\psi(\mathbf{r}_1\sigma_1,\mathbf{r}_2\sigma_2)$$

Therefore we have two possibilities for the antisymmetry of ψ ,

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1. $\psi(\mathbf{r}_1, \mathbf{r}_2)$ is antisymmetric:

$$\psi(\boldsymbol{r}_1, \boldsymbol{r}_2) = -\psi(\boldsymbol{r}_2, \boldsymbol{r}_1)$$

$$\chi(\sigma_1, \sigma_2) = +\chi(\sigma_2, \sigma_1)$$

2. $\chi(\sigma_1, \sigma_2)$ is antisymmetric:

$$\psi(\boldsymbol{r}_1,\boldsymbol{r}_2) = +\psi(\boldsymbol{r}_2,\boldsymbol{r}_1)$$

$$\chi(\sigma_1, \sigma_2) = -\chi(\sigma_2, \sigma_1)$$

Since the potential $V(\mathbf{r}_1 - \mathbf{r}_2)$ is independent of the spin, the total spin of two electrons is a conserved quantity. In using this observation, it is convenient to not label the individual spins σ_1, σ_2 (which are not conserved) but instead the total spin S and its projection onto the z-axis M.

$$|\sigma_1,\sigma_2\rangle\mapsto|S,M\rangle$$

If the total number states has to be conserved $(|\sigma_1, \sigma_2\rangle \in \mathcal{H}^2 \otimes \mathcal{H}^2 \implies |S, M\rangle \in \mathcal{H}^2 \otimes \mathcal{H}^2)$, then the four spin states,

$$|\uparrow,\uparrow\rangle$$
 $|\uparrow,\downarrow\rangle$ $|\downarrow,\uparrow\rangle$ $|\downarrow,\downarrow\rangle$

Map to 4 possible values for S, M. The z component of the spin can take on any value from 1, 0, -1 but the total magnitude of the spin must be either 0, 1.

$$|0,0\rangle$$
 $|1,1\rangle$ $|1,0\rangle$ $|1,-1\rangle$

Notice that these four states form and orthonormal basis for $\mathcal{H}^2 \otimes \mathcal{H}^2$. Explicitly these $|S, M\rangle$ states can be written as superpositions of the $|\sigma_1, \sigma_2\rangle$ states,

$$\begin{aligned} |1,1\rangle &= |\uparrow,\uparrow\rangle \\ |1,-1\rangle &= |\downarrow,\downarrow\rangle \\ |1,0\rangle &= \frac{1}{\sqrt{2}}(|\uparrow,\downarrow\rangle + |\uparrow,\downarrow\rangle) \\ |0,0\rangle &= \frac{1}{\sqrt{2}}(|\uparrow,\downarrow\rangle - |\uparrow,\downarrow\rangle) \end{aligned}$$

Importantly, each state $|S, M\rangle$ where the total spin S is S=1 is symmetric with respect to the interchange of two electrons and the spin S=0 state is antisymmetric. This anti-symmetry couples the spin states to the orbital components of the wave function. Since the spin S=1 states are symmetric, the wavefunction in position space with S=1 must be anti-symmetric,

$$\psi_{1M}(\boldsymbol{r}_1, \boldsymbol{r}_2) = \frac{1}{\sqrt{2}} (\varphi_1(\boldsymbol{r}_1) \varphi_2(\boldsymbol{r}_2) - \varphi_2(\boldsymbol{r}_1) \varphi_1(\boldsymbol{r}_2)) |1, M\rangle$$

Analogously,

$$\psi_{00}(\mathbf{r}_1, \mathbf{r}_2) = \frac{1}{\sqrt{2}} (\varphi_1(\mathbf{r}_1) \varphi_2(\mathbf{r}_2) + \varphi_2(\mathbf{r}_1) \varphi_1(\mathbf{r}_2)) |0, 0\rangle$$

In order to determine the correction to the energy values due to the Coulumb interactions, use perturbation theory. Specifically, the ψ_{1M} states are 3-fold degenerate in position space but are distinct when spin states are included. Recalling that the corrections can be computed by calculating the matrix elements of the corrections V,

$$\langle 1, M|V|1, M \rangle = \iint d^3r_1 d^3r_2 \frac{1}{\sqrt{2}} (\varphi_1^*(\boldsymbol{r}_1) \varphi_2^*(\boldsymbol{r}_2) - \varphi_2^*(\boldsymbol{r}_1) \varphi_1^*(\boldsymbol{r}_2)) V(\boldsymbol{r}_1 - \boldsymbol{r}_2) \frac{1}{\sqrt{2}} (\varphi_1(\boldsymbol{r}_1) \varphi_2(\boldsymbol{r}_2) - \varphi_2(\boldsymbol{r}_1) \varphi_1(\boldsymbol{r}_2))$$

$$= \frac{1}{2} \iint d^3r_1 d^3r_2 V(\boldsymbol{r}_1 - \boldsymbol{r}_2) \left[|\varphi_1(\boldsymbol{r}_1)|^2 |\varphi_2(\boldsymbol{r}_2)|^2 + |\varphi_1(\boldsymbol{r}_2)|^2 |\varphi_2(\boldsymbol{r}_1)|^2 + \cdots \right]$$

$$\cdots - \varphi_1^*(\boldsymbol{r}_1) \varphi_2^*(\boldsymbol{r}_2) \varphi_2(\boldsymbol{r}_1) \varphi_1(\boldsymbol{r}_2) - \varphi_2^*(\boldsymbol{r}_1) \varphi_1^*(\boldsymbol{r}_2) \varphi_1(\boldsymbol{r}_1) \varphi_2(\boldsymbol{r}_2) \right]$$

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Recognize that since the potential only depends on the distance $|r_1, r_2|$, our integral over r_1, r_2 counts each term twice. Therefore,

$$\langle 1, M|V|1, M\rangle = \iint d^3r_1 d^3r_2 V(\boldsymbol{r}_1 - \boldsymbol{r}_2) \left[\left| \varphi_1(\boldsymbol{r}_1) \right|^2 \left| \varphi_2(\boldsymbol{r}_2) \right|^2 - \varphi_1^*(\boldsymbol{r}_1) \varphi_2^*(\boldsymbol{r}_2) \varphi_2(\boldsymbol{r}_1) \varphi_1(\boldsymbol{r}_2) \right]$$

Similarly, we can calculate the matrix element for $|0,0\rangle$ (the elements $\langle 0,0|V|1,M\rangle$ are zero),

$$\langle 0, 0 | V | 0, 0 \rangle = \iint d^3 r_1 d^3 r_2 V(\mathbf{r}_1 - \mathbf{r}_2) \left[\left| \varphi_1(\mathbf{r}_1) \right|^2 \left| \varphi_2(\mathbf{r}_2) \right|^2 + \varphi_1^*(\mathbf{r}_1) \varphi_2^*(\mathbf{r}_2) \varphi_2(\mathbf{r}_1) \varphi_1(\mathbf{r}_2) \right]$$

Notice that $\langle 0,0|V|0,0\rangle$ differs from $\langle 1,M|V|1,M\rangle$ by the sign in the second set of terms. To facilitate discussions, let \mathcal{C} define the **Coulomb integral** \mathcal{C} ,

$$C = \iint d^3r_1 d^3r_2 V(\boldsymbol{r}_1 - \boldsymbol{r}_2) \left[\left| \varphi_1(\boldsymbol{r}_1) \right|^2 \left| \varphi_2(\boldsymbol{r}_2) \right|^2 \right]$$

Which is shared by both matrix elements and is a equivalent to a classical measurement of a Coulomb energy,

$$C = \iint d^3r_1 d^3r_2 V(\boldsymbol{r}_1 - \boldsymbol{r}_2) \rho_1(\boldsymbol{r}_1) \rho_2(\boldsymbol{r}_2)$$

The differing integral is called the **exchange integral** \mathcal{J} ,

$$\mathcal{J} = \iint d^3r_1 d^3r_2 V(\boldsymbol{r}_1 - \boldsymbol{r}_2) [\varphi_1^*(\boldsymbol{r}_1) \varphi_2^*(\boldsymbol{r}_2) \varphi_2(\boldsymbol{r}_1) \varphi_1(\boldsymbol{r}_2)]$$

In conclusion,

$$\langle 0, 0|V|0, 0\rangle = \mathcal{C} + \mathcal{J}$$
 $\langle 1, M|V|1, M\rangle = \mathcal{C} - \mathcal{J}$

The state with higher (or lower) energy correction will depend on the sign of \mathcal{J} . If $\mathcal{J} > 0$, this leads to alignment of the individual spins and to ferro-magnetism. Recalling $\psi_{1M}(\mathbf{r}_1, \mathbf{r}_2)$,

$$\psi_{1M}(\boldsymbol{r}_1, \boldsymbol{r}_2) = \frac{1}{\sqrt{2}} (\varphi_1(\boldsymbol{r}_1)\varphi_2(\boldsymbol{r}_2) - \varphi_2(\boldsymbol{r}_1)\varphi_1(\boldsymbol{r}_2)) |1, M\rangle$$

It can be seen that if we let $r_1 = r_2$ the antisymmetry forces,

$$\psi_{1M}(\boldsymbol{r}_1,\boldsymbol{r}_1)=0$$

Therefore the probability of both electrons being found in the same place is zero and therefore the Coulomb interaction (which would normally tend to ∞ as $r_1 \to r_2$). Therefore the electrons tend toward their minimal energy state which corresponds to aligning their spins S = 1. An alternative way of seeing this is to consider \mathcal{J} in the case where $r_1 = r_2$, \mathcal{J} tends toward \mathcal{C} which is positive.

Let $S = S_1 + S_2$ be the total spin operator for the two electrons. Therefore,

$$S^2 = (S_1 + S_2)^2 = S_1^2 + S_2^2 + 2S_1 \cdot S_2$$

Therefore the corrective energy is proportional to the product $S_1 \cdot S_2$,

$$H = -\mathcal{J} S_1 \cdot S_2$$

Which is minimized for aligned spins $S_1 \cdot S_2 > 0$. This result can be generalized to any number of electron spins using the **Heisenberg model**,

$$H = -\frac{1}{2} \mathcal{J} \sum_{\langle ij \rangle} \mathbf{S}_i \cdot \mathbf{S}_j - g\mu_B \mathbf{B} \cdot \sum_i \mathbf{S}_i$$
(3.2)

Where $\langle ij \rangle$ denotes nearest neighbor terms, the factor of 1/2 avoids double-counting the interactions, and \mathcal{J} forms a coupling constant proportional to,

$$\mathcal{J} \sim V \psi^2 \sim \frac{e^2}{a_0}$$

And the μ_B terms is simply the ordinary magnetic moment coupling. It is important to note that the nearest-neighbor approximation permits one to assume that \mathcal{J} is a constant (depending only on the distance between nearest neighbors). The Heisenberg model eq. (3.2) is one of the simplest examples of a quantum many-body problem which are notoriously difficult to solve and are typically only approximated at best. The Heisenberg model cannot be solved exactly; one such approximation is called mean-field theory.

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3.1 Mean-field Theory of the Heisenberg Model

Begin with the Heisenberg model for many interacting spins $\{S_i\}$ $(i \in \{1, ..., N\})$ subjected to an external magnetic field B,

$$H = -\frac{1}{2} \mathcal{J} \sum_{\langle ij \rangle} \mathbf{S}_i \cdot \mathbf{S}_j - g\mu_B \mathbf{B} \cdot \sum_i \mathbf{S}_i$$
(3.3)

where $\langle ij \rangle$ indicates the sum is over nearest neighbor spins,

$$\langle ij \rangle \equiv \left\{ (i,j) \in \left\{ 1, \dots, N \right\}^2 \mid i \text{ and } j \text{ are nearest neighbors} \right\}$$

the factor of 1/2 avoids double-counting the interactions, the exchange integral \mathcal{J} is treated as a constant, g is the spin g-factor of the electrons, and μ_B is the magnetic moment coupling constant.

Performing a mean-field approximation, we assert that the spins S_i variation (specifically $S_i - \langle S_i \rangle$) is small.

$$oldsymbol{S}_i = \langle oldsymbol{S}_i
angle + \mathcal{O}ig(oldsymbol{S}_i^2ig)$$

This permits us to write $S_i \cdot S_j$ in a form that exploits this mean-field theory approximation,

$$egin{aligned} m{S}_i \cdot m{S}_j &= (m{S}_i - \langle m{S}_i
angle + \langle m{S}_i
angle) \cdot (m{S}_j - \langle m{S}_j
angle + \langle m{S}_j
angle) \ &= (m{S}_i - \langle m{S}_i
angle) \cdot (m{S}_j - \langle m{S}_j
angle) + m{S}_i \cdot \langle m{S}_j
angle + m{S}_j \cdot \langle m{S}_i
angle - \langle m{S}_i
angle \cdot \langle m{S}_j
angle \end{aligned}$$

Neglecting $(S_i - \langle S_i \rangle) \cdot (S_j - \langle S_j \rangle)$ as it is quadratic in the assumed small spin deviation, we obtain,

$$H = -\frac{1}{2} \mathcal{J} \sum_{\langle ij \rangle} \{ \mathbf{S}_i \cdot \langle \mathbf{S}_j \rangle + \mathbf{S}_j \cdot \langle \mathbf{S}_i \rangle - \langle \mathbf{S}_i \rangle \cdot \langle \mathbf{S}_j \rangle \} - g \mu_B \mathbf{B} \cdot \sum_i \mathbf{S}_i$$
 (3.4)

Finally, make the approximation that each of the spins shared the same expectation,

$$\langle S_i \rangle = \langle S_i \rangle = \langle S \rangle$$

$$H = -\frac{1}{2} \mathcal{J} \sum_{\langle ij \rangle} \{ \mathbf{S}_i \cdot \langle \mathbf{S} \rangle + \mathbf{S}_j \cdot \langle \mathbf{S} \rangle - \langle \mathbf{S} \rangle \cdot \langle \mathbf{S} \rangle \} - g \mu_B \mathbf{B} \cdot \sum_i \mathbf{S}_i$$
 (3.5)

Noticing that each term in the summand of the \mathcal{J} term only depends on at most one index (either i or j or neither), we define the z-coordination number as the number of nearest neighbors for each atomic site i,

$$z_i = \#\{(k,j) \in \{1,\ldots,N\}^2 \mid (k,j) \in \langle ij \rangle\}$$

and then assume that the lattice structure is such that $z_i = z$ is independent of i. Therefore eq. (3.5) becomes,

$$H = -\frac{1}{2} \mathcal{J} z \langle \mathbf{S} \rangle \cdot \sum_{i} \{ 2\mathbf{S}_{i} - \langle \mathbf{S} \rangle \} - g\mu_{B} \mathbf{B} \cdot \sum_{i} \mathbf{S}_{i}$$
(3.6)

At this stage it is notationally convenient to re-label a number of terms in this expression. The first is to renormalize the overall Hamiltonian by a constant,

$$H \mapsto H - \frac{1}{2} \mathcal{J}zN\langle \mathbf{S} \rangle^2$$

The second is to modify the units of **B** in order to eliminate the constants $g\mu_B$,

$$m{B}\mapsto rac{1}{g\mu_B}m{B}$$

Therefore our Heisenberg model (eq. (3.6)) under these modifications becomes,

$$H = -(\mathcal{J}z\langle \mathbf{S}\rangle + \mathbf{B}) \cdot \sum_{i} \mathbf{S}_{i}$$
(3.7)

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Where $B_m \equiv \mathcal{J}z\langle S \rangle$ can be regarded as the molecular magnetic field generated by the spins of the material. Without loss of generality, fix the external magnetic field in the \hat{z} direction $(B = B\hat{z})$. Subject to this field, the energy of the system is minimized when B_m aligns in the same direction as $B(\langle S \rangle \to \langle S^z \rangle \hat{z})$. Therefore we obtain,

$$H = -(\mathcal{J}z\langle S^z \rangle + B) \sum_{i} S_i^z$$

Recall that in eq. (3.2) we have the following terms,

$$H \sim -rac{1}{2} {\cal J} \sum_{\langle ij
angle} {m S}_i \cdot {m S}_j$$

which possesses a symmetry with respect to an arbitrary rotations of all the spins. Therefore there is no preferred direction to the spin system in the absence of magnetic fields. However a particular direction is chosen subjected to the magnetic field B. This is an example of **spontaneously broken symmetry** in which the arbitrary direction is chosen spontaneously. In regard to eq. (3.7), minimizing the energy of the system also forces $\sum_i S_i^z$ to become maximized (i.e. the spins all aligning in the same direction $S_i^z = \frac{1}{2}, \forall i$). However the minimal energy configuration is only accessible when the temperature T is exactly zero. At finite temperature, each spin has a variety of accessible spin states $S_i^z \in \mathcal{S} \equiv \left\{-\frac{1}{2}, +\frac{1}{2}\right\}$. Since $\langle S^z \rangle$ has be assumed to be the same for each site i, a finite temperature T modifies S_i^z such that $H_i \equiv -(\mathcal{J}z\langle S^z \rangle + B)S_i^z$ is increased (by decreasing S_i^z) in accordance with the increase T in a manner independent of i. Therefore one can construct a partition function for the accessible spins states \mathcal{S} which will take on the same form for each S_i^z ,

$$Z \equiv \sum_{S^z \in \mathcal{S}} \exp\left[-\frac{H}{k_B T}\right] = \sum_{S^z \in \mathcal{S}} \exp\left[\frac{(B_m + B)S^z}{k_B T}\right]$$
(3.8)

Therefore the spin expectation will be,

$$\langle S^z \rangle = \frac{1}{Z} \sum_{S^z \in S} S^z \exp\left[\frac{(B_m + B)S^z}{k_B T}\right]$$
 (3.9)

Explicitly,

$$Z = \exp\left[-\frac{B_m + B}{2k_B T}\right] + \exp\left[\frac{B_m + B}{2k_B T}\right] = 2\cosh\left[\frac{B_m + B}{2k_B T}\right]$$

$$\langle S^z \rangle = \frac{1}{Z} \left\{ \frac{1}{2} \exp\left[\frac{B_m + B}{2k_B T}\right] - \frac{1}{2} \exp\left[-\frac{B_m + B}{2k_B T}\right] \right\} = \frac{1}{2Z} 2 \sinh\left[\frac{B_m + B}{2k_B T}\right] = \frac{1}{2} \tanh\left[\frac{B_m + B}{2k_B T}\right]$$

Therefore the macroscopic magnetization per atom $M \equiv \langle S^z \rangle$ must satisfy the following implicit formula,

$$M = \frac{1}{2} \tanh \left[\frac{\mathcal{J}zM + B}{2k_B T} \right] \tag{3.10}$$

In order to study spontaneous magnetization, we set the external field B. In the absence of an external magnetic field B = 0, we assume that there material magnetizes spontaneously (and spontaneously breaks rotation symmetry) if and only if eq. (3.10) has a solution for the given temperature T.

$$M = \frac{1}{2} \tanh \left[\frac{\mathcal{J}zM}{2k_B T} \right] \tag{3.11}$$

For visualization, consider a re-scaling of the temperature parameter $\mathcal{T} \equiv \frac{2k_BT}{\mathcal{I}z}$ such that eq. (3.11) becomes,

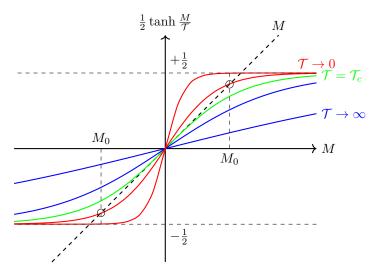
$$M = \frac{1}{2} \tanh \left[\frac{M}{\mathcal{T}} \right] \tag{3.12}$$

When plotted, the solutions (or absence of solutions becomes clearer). Since $\tanh(0) = 0$, there is always a solution of M = 0 at all temperatures. If the slope of the RHS of eq. (3.12) is ever greater than a slope of 1,

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there are two non-trivial solutions $M = \pm M_0$. This transition between no magnetization and magnetization is a phase transition and occurs at a **critical temperature** of $\mathcal{T}_c = 1$ which corresponds to,

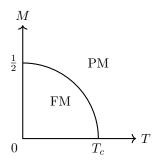
$$1 = \frac{\mathrm{d}}{\mathrm{d}M} \left\{ \frac{1}{2} \tanh \left[\frac{M}{\mathcal{T}} \right] \right\} \Big|_{M=0} = \frac{1}{2\mathcal{T}} \mathrm{sech}^2 \left[\frac{M}{\mathcal{T}} \right] \Big|_{M=0} = \frac{1}{2\mathcal{T}} \implies \mathcal{T}_c = \frac{1}{2} \implies \mathcal{T}_c = \frac{\mathcal{J}z}{4k_B}$$



The critical temperature distinguishes two different phases of the material,

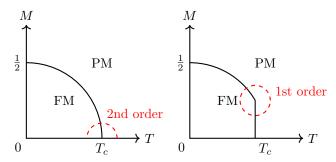
- $T < T_c$ ferro-magnetism
- $T < T_c$ no ferro-magnetism

For iron we have $\mathcal{J} \sim 1 \, \text{eV}$ and $T \sim 10^4 \, \text{K}$. This is an example of a **phase transition**.



We refer to M as the M-order parameter which allows us to formally distinguish between ferro-magnets and para-magnets. We have $M \neq 0$ in the FM phase but M = 0 in the PM phase.

Depending on the nature of the phase transition boundary, we can have two types of phase translations: first order phase translations which are discontinuous or second order phase transitions which are continuous (as is the case for this phase transition).



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Given that M vanishes continuously as T increases, we can Taylor series eq. (3.11) near $T = T_c$,

$$M = \frac{1}{2} \left[\frac{\mathcal{J}zM}{2k_BT} - \frac{1}{3} \left(\frac{\mathcal{J}zM}{2k_BT} \right)^3 \right]$$

Recalling the form of T_c we cancel M and obtain a quadratic equation for M,

$$1 = \frac{T_c}{T} - \frac{1}{3} \left(\frac{T_c}{T}\right)^3 M^2$$

Therefore M^2 as a function of temperature (at least near $T \simeq T_c$) can be written as,

$$M^2 = \frac{3}{4} \left(\frac{T}{T_c}\right)^3 \left(\frac{T_c}{T} - 1\right)$$

$$M = \pm \frac{1}{2} \frac{T}{T_c} \sqrt{3 \left(1 - \frac{T}{T_c}\right)}$$

Since $T \simeq T_c$ we can neglect the prefactor,

$$M(T) = \pm \frac{1}{2} \sqrt{3\left(1 - \frac{T}{T_c}\right)}$$

Therefore near the critical temperature we have,

$$M(T) \sim \sqrt{T_c - T}$$

Defining the **reduced temperature** t,

$$t = \frac{T - T_c}{T_c}$$

We have that M(t) behaves as follows,

$$M(t) \simeq (-t)^{\beta} \tag{3.13}$$

Where $\beta = \frac{1}{2}$ is called the **critical exponent**. In general (and as in this example), the critical exponent is typically some number $\beta \in \mathbb{R}$ that is independent of the microscopic details of the material (like \mathcal{J} for example). Since β entirely determines the nature of the phase transition, we consider β to be a universal behaviour and is of particular interest. This universality is analogous to the Quantum Hall effect in which the conductivity is only dependent on universal constants.

The magnetic susceptibility χ (first introduced in eq. (3.1)) in general is defined as,

$$\chi = \frac{\mathrm{d}M}{\mathrm{d}B}$$

For the mean-field theory approximations of the Heisenberg model we have obtained,

$$M = \frac{1}{2} \tanh\left(\frac{B + \mathcal{J}zM}{2k_B T}\right)$$

Which for $T > T_c$ if $B \approx 0$ is small, the magnetization is also small $M \approx 0$. Expanding in Taylor series once again,

$$M = \frac{1}{2} \frac{B + \mathcal{J}zM}{2k_B T}$$

Solving for M as a response function of B,

$$M = \frac{B}{4k_BT} + M \frac{\mathcal{J}z}{4k_BT}$$

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$$M\left(1 - \frac{\mathcal{J}z}{4k_BT}\right) = \frac{B}{4k_BT}$$
$$M\left(1 - \frac{T_c}{T}\right) = \frac{B}{4k_BT}$$
$$M = \left(1 - \frac{T_c}{T}\right)^{-1} \frac{B}{4k_BT}$$

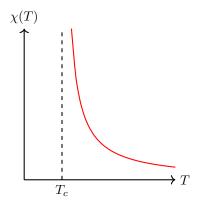
Therefore the magnetic susceptibility is χ ,

$$\chi = \frac{M}{B} = \frac{1}{4k_B(T - T_c)}$$

This is known as the Cure-Weiss law. Notice that this is consistent with our classification of paramagnets (which have $\chi > 0$ and $T > T_c$). Notice that χ diverges when $T \to T_c$; this is an alternative form of critical behaviour with critical exponent $\gamma = -1$,

$$\chi \sim (t)^{\gamma} \tag{3.14}$$

The divergence of susceptibility is an example of non-analytic behaviour in a real physical system. The physical meaning of χ tells you how strongly M responds to B.



The susceptibility χ diverges at $T = T_c$ because when $T < T_c$ we have spontaneous magnetization $(M \neq 0)$ even in the absence of B = 0.

3.2 Arbitrary Spin Systems

In , we performed mean-field theory analysis on the Heisenberg model eq. (3.3) where the accessible spin states for the electrons formed the smallest possible set $S = \{-\frac{1}{2}, +\frac{1}{2}\}$ which were the possible S^z values for a particle with intrinsic spin of $S = \frac{1}{2}$. Instead we consider a particle with intrinsic spin $S \in \{0, \frac{1}{2}, 1, \frac{3}{2}, 2, \ldots\}$ such that,

$$\mathcal{S} = \{-S, -S+1, \dots, S-1, S\}$$

Of course we can immediately disregard the trivial case of S=0 whereby there is no magnetization because the Heisenberg model is trivially zero.

Returning to the partition function eq. (3.8),

$$Z = \sum_{S^z \in S} \exp\left[-\frac{H}{k_B T}\right] = \sum_{S^z \in S} \exp\left[\frac{(B_m + B)S^z}{k_B T}\right]$$

Explicitly if we let k be an index associated with the sum over spins defined such that k = S + S,

$$k \in \{0, 1, \dots, 2S - 1, 2S\}$$

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Also let $\eta \equiv \exp\left[\frac{(B_m+B)}{k_BT}\right]$,

$$Z = \sum_{S^z \in \mathcal{S}} \exp\left[\frac{(B_m + B)S^z}{k_B T}\right]$$

$$= \sum_{k=0}^{2S} \eta^{(k-S)}$$

$$= \sum_{k=0}^{2S} \eta^k \eta^{-S}$$

$$= \frac{1}{\eta^S} \sum_{k=0}^{2S} \eta^k$$

$$= \frac{1}{\eta^S} \left(\sum_{k=0}^{\infty} \eta^k - \sum_{k=2S}^{\infty} \eta^k\right)$$

$$= \frac{1}{\eta^S} \left(\sum_{k=0}^{\infty} \eta^k - \eta^{2S} \sum_{k=2S+1}^{\infty} \eta^{k-2S}\right)$$

$$= \frac{1}{\eta^S} \left(\sum_{k=0}^{\infty} \eta^k - \eta^{2S+1} \sum_{k=0}^{\infty} \eta^k\right)$$

$$= \frac{1 - \eta^{2S+1}}{\eta^S} \sum_{k=0}^{\infty} \eta^k$$

Ignoring the fact that $|\eta| = \left| \exp\left[\frac{(B_m + B)}{k_B T}\right] \right| \not< 1$, we can make use of the geometric series (confident that the expression is convergent because the original sum was finite),

$$Z = \frac{1 - \eta^{2S+1}}{\eta^S (1 - \eta)} \tag{3.15}$$

Additionally we may make use of an arithmetic series to derive the expectation value $\langle S^z \rangle$ (analogous to eq. (3.9)),

$$\langle S^z \rangle = \frac{1}{Z} \sum_{S^z \in \mathcal{S}} S^z \exp\left[\frac{(B_m + B)S^z}{k_B T}\right] = \frac{1}{Z} \sum_{k=0}^{2S} (k - S)\eta^{k-S}$$

$$\langle S^z \rangle = \frac{1}{Z} \sum_{k=0}^{2S} (k - S)\eta^{k-S}$$

$$= \frac{1}{Z\eta^S} \left\{ \sum_{k=0}^{2S} k\eta^k - S \sum_{k=0}^{2S} \eta^k \right\}$$

$$= \frac{1}{Z\eta^S} \left\{ \sum_{k=0}^{2S} k\eta^k - S\eta^S Z \right\}$$

$$= \frac{1}{Z\eta^S} \left\{ \eta \sum_{k=0}^{2S} k\eta^{k-1} - S\eta^S Z \right\}$$

$$= \frac{1}{Z\eta^S} \left\{ \eta \sum_{k=0}^{2S} \frac{d\eta^k}{d\eta} - S\eta^S Z \right\}$$

$$= \frac{1}{Z\eta^S} \left\{ \eta \frac{d}{d\eta} \sum_{k=0}^{2S} \eta^k - S\eta^S Z \right\}$$

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$$= \frac{1}{Z\eta^S} \left\{ \eta \frac{\mathrm{d}}{\mathrm{d}\eta} (\eta^S Z) - S\eta^S Z \right\}$$

Taking the derivative of eq. (3.15),

$$\begin{split} \frac{\mathrm{d}}{\mathrm{d}\eta} \left(\eta^S Z \right) &= \frac{\mathrm{d}}{\mathrm{d}\eta} \left\{ \eta^S \frac{1 - \eta^{2S+1}}{\eta^S (1 - \eta)} \right\} \\ &= \frac{\mathrm{d}}{\mathrm{d}\eta} \left\{ \left(1 - \eta^{2S+1} \right) (1 - \eta)^{-1} \right\} \\ &= -(2S + 1)\eta^{2S} (1 - \eta)^{-1} + \left(1 - \eta^{2S+1} \right) (1 - \eta)^{-2} \end{split}$$

Therefore,

$$\begin{split} \eta \frac{\mathrm{d}}{\mathrm{d}\eta} (\eta^S Z) - S \eta^S Z &= -(2S+1) \eta^{2S+1} (1-\eta)^{-1} + \eta (1-\eta^{2S+1}) (1-\eta)^{-2} - S (1-\eta^{2S+1}) (1-\eta)^{-1} \\ \langle S^z \rangle &= \frac{1}{Z \eta^S} \left\{ -(2S+1) \eta^{2S+1} (1-\eta)^{-1} + \eta (1-\eta^{2S+1}) (1-\eta)^{-2} - S (1-\eta^{2S+1}) (1-\eta)^{-1} \right\} \\ \langle S^z \rangle &= \frac{1}{Z \eta^S (1-\eta)^2} \left\{ -(2S+1) \eta^{2S+1} (1-\eta) + \eta (1-\eta^{2S+1}) - S (1-\eta^{2S+1}) (1-\eta) \right\} \\ \langle S^z \rangle &= \frac{1}{Z \eta^S (1-\eta)^2} \left\{ 2S \eta^{2S+1} (\eta-1) - \eta^{2S+1} (1-\eta) + \eta (1-\eta^{2S+1}) + S (1-\eta^{2S+1}) (\eta-1) \right\} \\ \langle S^z \rangle &= \frac{1}{Z \eta^S} \left\{ -(\eta^{2S+1} - \eta^{2S+2}) - (\eta^{2S+2} - \eta) + S (1+\eta^{2S+1}) (\eta-1) \right\} \\ \langle S^z \rangle &= \frac{1}{Z \eta^S} \frac{S (\eta-1) (\eta^{2S+1} + 1) - \eta (\eta^{2S} - 1)}{(1-\eta)^2} \\ \langle S^z \rangle &= \frac{1}{Z \eta^S} \frac{S (\eta^{2S+2} + \eta - \eta^{2S+1} - 1) + (\eta - \eta^{2S+1})}{(1-\eta)^2} \\ \langle S^z \rangle &= \frac{S (\eta^{2S+2} + \eta - \eta^{2S+1} - 1) + (\eta - \eta^{2S+1})}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{S (\eta^{2S+2} + \eta - \eta^{2S+1} - 1) - \frac{1}{2} (\eta^{2S+1} - 1 - \eta^{2S+2} + \eta - \eta^{2S+1} + 1)}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{\frac{2S+1}{2} (\eta^{2S+2} + \eta - \eta^{2S+1} - 1) - \frac{1}{2} (\eta^{2S+2} - \eta + \eta^{2S+1} - 1)}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{(\eta^{2S+1} + 1) (\eta - 1) - \frac{1}{2} (\eta^{2S+1} - 1) (\eta+1)}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{(\eta^{2S+1} + 1) (\eta-1)}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{(\eta^{2S+1} - 1) (\eta+1)}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} - 1}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} - 1}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} - 1}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} - 1}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} - 1}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} - 1}{(\eta^{2S+1} - 1) (\eta-1)} \\ \langle S^z \rangle &= \frac{2S+1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} - \frac{1}{2} \frac{\eta^{2S+1} + 1}{(\eta^{2S+1} - 1) (\eta-1)} \\$$

Defining χ such that $\eta = e^{\chi}$,

$$\begin{split} \langle S^z \rangle &= \frac{2S+1}{2} \frac{e^{(2S+1)\chi}+1}{e^{(2S+1)\chi}-1} - \frac{1}{2} \frac{e^{\chi}+1}{e^{\chi}-1} \\ \langle S^z \rangle &= \frac{2S+1}{2} \coth \left(\frac{2S+1}{2}\chi\right) - \frac{1}{2} \coth \left(\frac{1}{2}\chi\right) \end{split}$$

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$$\langle S^z \rangle = SB_S(S\chi)$$

Where $B_S(x)$ is the Brillouin function,

$$B_S(x) \equiv \frac{2S+1}{2S} \coth\left(\frac{2S+1}{2S}x\right) - \frac{1}{2S} \coth\left(\frac{1}{2S}x\right)$$

Therefore analogously to eq. (3.10) the magnetization $M \equiv \langle S^z \rangle$ must satisfy the following equation,

$$M = SB_S \left(\frac{S(\mathcal{J}zM + B)}{k_B T} \right)$$

In the absence of an external magnetic field we have (letting $\mathcal{T} = k_B T/\mathcal{J}z$),

$$M = SB_S \left(\frac{SM}{\mathcal{T}}\right)$$

Since $\coth(x) = -\coth(-x)$ is odd symmetric, we have that

$$SB_S\left(\frac{SM}{\mathcal{T}}\right) = -SB_S\left(\frac{S(-M)}{\mathcal{T}}\right)$$

Therefore we can restrict our focus to potential solutions where $M \geq 0$ being confident that a negative M solution exists as well. Next,

$$SB_S\left(\frac{SM}{\mathcal{T}}\right) = \frac{2S+1}{2} \coth\left(\frac{2S+1}{2S}\frac{M}{\mathcal{T}}\right) - \frac{1}{2} \coth\left(\frac{1}{2S}\frac{M}{\mathcal{T}}\right)$$

For $M \geq 0$,

$$SB_{S}\left(\frac{SM}{\mathcal{T}}\right) \ge \frac{1}{2} \coth\left(\frac{2S+1}{2S}\frac{M}{\mathcal{T}}\right) - \frac{1}{2} \coth\left(\frac{1}{2S}\frac{M}{\mathcal{T}}\right)$$
$$SB_{S}\left(\frac{SM}{\mathcal{T}}\right) \ge \frac{1}{2} \coth\left(\frac{1}{2S}\frac{M}{\mathcal{T}}\right) - \frac{1}{2} \coth\left(\frac{1}{2S}\frac{M}{\mathcal{T}}\right)$$
$$SB_{S}\left(\frac{SM}{\mathcal{T}}\right) \ge 0$$

Finally,

$$\frac{\mathrm{d}}{\mathrm{d}\mathcal{M}}\left(SB_S\left(\frac{SM}{\mathcal{T}}\right)\right) = \frac{\mathrm{d}}{\mathrm{d}\mathcal{M}}\left\{\frac{2S+1}{2}\coth\left(\frac{2S+1}{2S}\frac{M}{\mathcal{T}}\right) - \frac{1}{2}\coth\left(\frac{1}{2S}\frac{M}{\mathcal{T}}\right)\right\}$$

$$\frac{\mathrm{d}}{\mathrm{d}\mathcal{M}}\left(SB_S\left(\frac{SM}{\mathcal{T}}\right)\right) = -\frac{2S+1}{2}\frac{2S+1}{2S\mathcal{T}}\operatorname{csch}^2\left(\frac{2S+1}{2S}\frac{M}{\mathcal{T}}\right) + \frac{1}{2}\frac{1}{2S\mathcal{T}}\operatorname{csch}^2\left(\frac{1}{2S}\frac{M}{\mathcal{T}}\right)$$

Since $\frac{\mathrm{d}}{\mathrm{d}\mathcal{M}}\left(SB_S\left(\frac{SM}{\mathcal{T}}\right)\right)$ as positive M>0 solution exists if and only if $SB_S\left(\frac{SM}{\mathcal{T}}\right)$ has slope greater than 1 at M=0. This defines the critical temperature.

$$\frac{\mathrm{d}}{\mathrm{d}\mathcal{M}} \left(SB_S \left(\frac{SM}{\mathcal{T}} \right) \right) \Big|_{M=0} = \lim_{M \to 0^+} \left\{ -\frac{2S+1}{2} \frac{2S+1}{2S\mathcal{T}} \operatorname{csch}^2 \left(\frac{2S+1}{2S} \frac{M}{\mathcal{T}} \right) + \frac{1}{2} \frac{1}{2S\mathcal{T}} \operatorname{csch}^2 \left(\frac{1}{2S} \frac{M}{\mathcal{T}} \right) \right\}
\frac{\mathrm{d}}{\mathrm{d}\mathcal{M}} \left(SB_S \left(\frac{SM}{\mathcal{T}} \right) \right) \Big|_{M=0} = \frac{S(S+1)}{3\mathcal{T}}
\mathcal{T}_c = \frac{S(S+1)}{3} \implies \mathcal{T}_c = \frac{\mathcal{J}z}{k_B} \frac{S(S+1)}{3}$$

Which for S = 1/2 we recover the usual $T_c = \frac{\mathcal{J}z}{4k_B}$.

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3.3 Mean-field Theory and Helmholtz Free Energy

In the previous section (section 3.2), we analyzed the mean-field theory approximation to the Heisenberg model H,

$$H = -rac{1}{2}\mathcal{J}\sum_{\langle ij
angle}oldsymbol{S}_i\cdotoldsymbol{S}_j$$

Although very useful, the mean-field theory approximations fail an predicting all phenomena that are manifest in the Heisenberg model. Here we will make the same first approximations,

$$S_i = S_i - \langle S_i \rangle + \langle S_i \rangle \tag{3.16}$$

And obtain the following,

$$H = -\frac{1}{2} \mathcal{J} \sum_{\langle ij \rangle} [\mathbf{S}_i \cdot \langle \mathbf{S}_j \rangle + \mathbf{S}_j \cdot \langle \mathbf{S}_i \rangle - \langle \mathbf{S}_i \rangle \cdot \langle \mathbf{S}_j \rangle] - g\mu_B \mathbf{B} \cdot \sum_i \mathbf{S}_i$$
(3.17)

Letting $\mathbf{B} = B\hat{\mathbf{z}}$ and assuming that the expectation of each S_i is magnetized as follows,

$$\langle \mathbf{S}_i \rangle = M \hat{\mathbf{z}} \qquad \forall i$$

To obtain the following Hamiltonian,

$$H = \frac{\mathcal{J}zN}{2}M^2 - (\mathcal{J}zM + B)\sum_{i} S_i^z$$

From this, we can define the partition function Z (as used in statistical mechanics) over the energy states,

$$Z = \left[\sum_{S^z = \pm 1/2} e^{-\frac{\mathcal{J}zM^2}{2k_B T} + \frac{\mathcal{J}zMS^z}{k_B T}} \right]^N$$

Such that the **Helmholtz free energy** F has the following form,

$$F \equiv E - TS = -k_B T \ln Z$$

Simplifying Z,

$$Z = \left[e^{-\frac{\mathcal{J}zM^2}{2k_BT} + \frac{\mathcal{J}zM}{2k_BT}} + e^{-\frac{\mathcal{J}zM^2}{2k_BT} - \frac{\mathcal{J}zM}{2k_BT}} \right]^N$$

$$Z = e^{-\frac{\mathcal{J}zM^2N}{2k_BT}} \left[e^{\frac{\mathcal{J}zM}{2k_BT}} + e^{-\frac{\mathcal{J}ZM}{2k_BT}} \right]^N$$

$$Z = e^{-\frac{\mathcal{J}zM^2N}{2k_BT}} \left[2\cosh\left(\frac{\mathcal{J}ZM}{2k_BT}\right) \right]^N$$

Therefore the free energy is,

$$F = \frac{\mathcal{J}zM^2N}{2} - k_BT \ln\left[2^N \cosh^N\left(\frac{\mathcal{J}zM}{2k_BT}\right)\right]$$
$$F = \frac{\mathcal{J}zM^2N}{2} - Nk_BT \ln\left[2\cosh\left(\frac{\mathcal{J}zM}{2k_BT}\right)\right]$$

In order to find the equilibrium state of this system, we need to minimize the Helmholtz free energy with respect to M,

$$\frac{\partial F}{\partial M} = \mathcal{J}zMN - Nk_BT \frac{2\frac{\mathcal{J}z}{2k_BT}\sinh\left(\frac{\mathcal{J}zM}{2k_BT}\right)}{2\cosh\left(\frac{\mathcal{J}zM}{2k_BT}\right)}$$

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$$\frac{\partial F}{\partial M} = \mathcal{J}zMN - Nk_BT \frac{\mathcal{J}z}{k_BT} \tanh\left(\frac{\mathcal{J}zM}{2k_BT}\right)$$

Minimizing corresponds to setting $\frac{\partial F}{\partial M} = 0$. Doing so yields,

$$M = \frac{1}{2} \tanh \left(\frac{\mathcal{J}zM}{2k_B T} \right)$$

Which is identical to eq. (3.11) as expected. This approach is simply an alternative method for deriving eq. (3.11). Near $T = T_c$, we can expand F(M) in Taylors series with respect to M. Defining f to be F/N,

$$f \equiv \frac{F}{N} = \frac{\mathcal{J}zM^2}{2} - k_B T \ln \left[2 \cosh \left(\frac{\mathcal{J}zM}{2k_B T} \right) \right]$$

As in turns out, we need to expand f(M) to fourth order. Before doing so, notice that f(M) is an even function in M (which is equivalent to saying that f(M) = f(-M)). Therefore within the first four orders, we have two non-zero terms.

$$f(M) = \frac{\mathcal{J}zM^2}{2} - \frac{k_B T}{2} \left(\frac{\mathcal{J}zM}{2k_B T}\right)^2 + \frac{k_B T}{12} \left(\frac{\mathcal{J}zM}{2k_B T}\right)^4 + \mathcal{O}(M^4)$$

For notational convenience, drop $\mathcal{O}(M^4)$ and make use of natural units such that $k_BT \mapsto T$ where T now has units of energy.

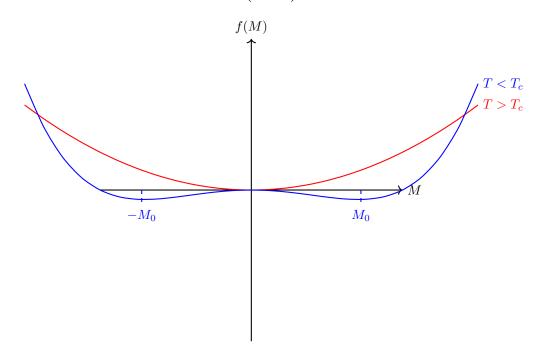
$$f(M) = \frac{\mathcal{J}zM^2}{2} - \frac{T}{2} \left(\frac{\mathcal{J}zM}{2T}\right)^2 + \frac{T}{12} \left(\frac{\mathcal{J}zM}{2T}\right)^4$$

Moreover, making use of the critical temperature T_c ,

$$f(M) = 2T_c M^2 \left(1 - \frac{T_c}{T}\right) + \frac{4}{3} \frac{T_c^4}{T^3} M^4$$

Again since $T \approx T_c$ when $M \approx 0$ we can neglect prefactors,

$$f(M) = 2T_c \left(1 - \frac{T_c}{T}\right) M^2 + \frac{4}{3}T_c M^4$$



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As you can see when T transitions from $T > T_c$ to $T < T_c$, two new free energy minima form at precisely $\pm M_0$.

$$\frac{\partial f}{\partial M} = 0 \implies 0 = \frac{\partial}{\partial M} \left(2T_c \left(1 - \frac{T_c}{T} \right) M^2 + \frac{4}{3} T_c M^4 \right) = 4T_c \left(1 - \frac{T_c}{T} \right) M + \frac{16}{3} T_c M^3$$

For $M \neq 0$ (which corresponds to a maximum of f when $T < T_c$),

$$0 = 4T_c \left(1 - \frac{T_c}{T}\right) + \frac{16}{3}T_c M^2$$
$$4T_c \left(\frac{T_c}{T} - 1\right) = \frac{16}{3}T_c M^2$$
$$M^2 = \frac{3}{4}\left(\frac{T_c}{T} - 1\right)$$
$$M = \pm M_0 \equiv \pm \sqrt{\frac{3}{4}\left(\frac{T_c}{T} - 1\right)}$$

The important scaling feature is that M scales as

$$M \sim \left(\frac{T_c - T}{T_c}\right)^{1/2}$$

For M nearby $\pm M_0$; this is identical to the critical exponent of $\beta = 1/2$ found earlier (eq. (3.13)).

3.4 Landau Theory

Let us now return to out original Heisenberg model,

$$H = -rac{\mathcal{J}}{2} \sum_{\langle ij
angle} oldsymbol{S}_i \cdot oldsymbol{S}_j$$

Notice that H is invariant with respect to arbitrary rotation of all spins because $S_i \cdot S_j$ is preserved under rotations (note that this is a global symmetry of all spins simultaneously). If we define M to be a vector corresponding to all of the mean spins,

$$m{M} \equiv \langle m{S}_i
angle$$

We induce the following free-energy density $f(\mathbf{M})$ which depends on each of the independent spins. $f(\mathbf{M})$ will have the same global symmetry of rotations. Therefore $f(\mathbf{M})$ can only depend on $\mathbf{M} \cdot \mathbf{M} = |\mathbf{M}|^2$ to various powers,

$$f(\mathbf{M}) = a\mathbf{M} \cdot \mathbf{M} + \frac{b}{2}(\mathbf{M} \cdot \mathbf{M})^2 + \cdots$$
(3.18)

Where $a, b = a(t), b(t) \in \mathbb{R}$ are yet unknown coefficients that are parameterized by $t = (T - T_c)/T_c$. This remarkable result require no calculations, just symmetry arguments. When $|t| \ll 1$, T is very close to T_c . We expand out a(t) and b(t) in Taylor series of t,

$$a(t) = a_0 + a_1 t + \cdots$$
 $b(t) = b_0 + b_1 t + \cdots$

We can find a_0, a_1 and b_0, b_1 from the requirement that $|\mathbf{M}| \neq 0$ when t < 0 $(T < T_c)$ and $|\mathbf{M}| = 0$ when t > 0 $(T > T_c)$. Therefore for $f(\mathbf{M})$,

$$f(\boldsymbol{M}) = a|\boldsymbol{M}|^2 + \frac{b}{2}|\boldsymbol{M}|^4$$

The value of f is minimized at thermal equilibrium,

$$\frac{\partial f}{\partial |\boldsymbol{M}|} = 2a|\boldsymbol{M}| + 2b|\boldsymbol{M}|^3 = 0$$

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Which implies that for $|\mathbf{M}| \neq 0$,

$$|m{M}| = \sqrt{-rac{a}{b}}$$

If b < 0, then for sufficiently large $|\mathbf{M}|$ we have that f < 0 regardless of the value of a. Therefore we must limit our expansion of b(t) to the first order in b and demand that b > 0 always,

$$b(t) = b_0 > 0$$

Since b > 0 holds always, a physically realizable solution $\mathbf{M} \in \mathbb{R}^N$ requires,

$$T > T_c \iff a > 0$$
 $T < T_c \iff a < 0 \implies a(t) = a_1 t \quad a_1 > 0$

Therefore we conclude that near the minima of f,

$$|\boldsymbol{M}| \sim (-t)^{1/2}$$

Which is again the universal result we obtained eq. (3.13). The remarkable observation is that the this analysis follows entirely from symmetry arguments. In general, the this can be generalized to second-order phase transitions and the general theory is known as the **Landau Theory of Continuous Phase Transitions**.

If we subject our interacting spin system to an external magnetic field B, we have that f(M) is modified accordingly,

$$f(\mathbf{M}) = a|\mathbf{M}|^2 + \frac{b}{2}|\mathbf{M}|^4 - \mathbf{B} \cdot \mathbf{M}$$

Which is distinct from the above because B singles out a particular direction of M ($f(M) \neq f(|M|)$). Therefore we assume that $M = M\hat{B}$ (at least when considering minimum energy configurations),

$$f(M) = aM^2 + \frac{b}{2}M^4 - BM$$

Once again the minimum energy configuration occurs when $\partial f/\partial M=0$

$$\frac{\partial f}{\partial M} = 2aM + 2bM^4 - B = 0$$

Which yields,

$$2a\frac{\mathrm{d}M}{\mathrm{d}B} + 6bM^2\frac{\mathrm{d}M}{\mathrm{d}B} = 1$$

Re-introduce the magnetic susceptibility defined as,

$$\chi \equiv \frac{\mathrm{d}M}{\mathrm{d}B}$$

$$2a\chi + 6bM^2\chi = 1$$

Therefore $\chi(M)$ has the following form,

$$\chi(M) = \frac{1}{2a + 6bM^2}$$

For large temperatures $T > T_c$, we have M = 0 whenever B = 0,

$$\chi = \frac{1}{2a} = \frac{1}{2a_1 t} \sim (t)^{\gamma} \qquad \gamma = -1$$

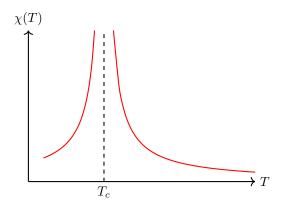
Where $\gamma = -1$ is the susceptibility critical exponent as we have derived earlier (eq. (3.14)). For lower temperatures $T < T_c$ we have that $M^2 = -a/b$.

$$\chi = \frac{1}{2a - 6b\frac{a}{b}} = -\frac{1}{4b} = -\frac{1}{4a_1t} \sim (-t)^{\gamma}$$

With the same critical exponent (the only difference being the leading coefficients and the sign of t). Together we have,

$$\chi \sim |t|^{\gamma}$$

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The values of critical exponents from our mean-field theory approach are wrong in the sense that they do not correspond to something physical. In general, critical exponents depend on,

- 1. Nature of order parameter
- 2. Symmetry
- 3. Dimensionality of the sample

In the language of this more general Landau theory, the mean-field theory approach does not accurately describe the above dependence. First notice that when we first hypothesized eq. (3.18) we assumed that M was a constant vector that was independent of spatial coordinates (i.e. position in the material). There was no good reason as to why we assumed this and in general we had to have M(r) dependent on r. In this case the free energy F would be the functional,

$$F \stackrel{?}{=} \int d^3r \left[a(\boldsymbol{M} \cdot \boldsymbol{M}) + \frac{b}{2} (\boldsymbol{M} \cdot \boldsymbol{M})^4 \right]$$

The '?' above the equals sign is present because this form for F does not capture any ordering structure among the spins. If we re-examine the Heisenberg model,

$$H = -rac{\mathcal{J}}{2}\sum_{\left\langle ij
ight
angle}oldsymbol{S}_{i}\cdotoldsymbol{S}_{j} = -rac{\mathcal{J}}{4}\sum_{\left\langle ij
ight
angle}igg[\left(oldsymbol{S}_{i}-oldsymbol{S}_{j}
ight)^{2}-oldsymbol{S}_{i}^{2}-oldsymbol{S}_{j}^{2}igg]$$

One can see that the S_i^2, S_j^2 are independent of the order or structure of the spin locations but $S_i - S_j$ approximates a spin gradient,

$$(\boldsymbol{S}_i - \boldsymbol{S}_j)^2 \sim \nabla \boldsymbol{S} \cdot \nabla \boldsymbol{S} = (\nabla M^x)^2 + (\nabla M^y)^2 + (\nabla M^z)^2$$

Therefore the free-energy should penalize non-uniformity in M,

$$F = \int d^3r \left[a(\mathbf{M} \cdot \mathbf{M}) + \frac{b}{2} (\mathbf{M} \cdot \mathbf{M})^4 + \frac{1}{2} \rho \nabla \mathbf{M} \cdot \nabla \mathbf{M} \right]$$
(3.19)

Where $\rho > 0$ is some unknown positive constant. This forms the starting point for an *improved* Landau theory. Calculating critical exponents under the above theory is complicated. Even if we could calculate critical exponents using eq. (3.19), this would still be an incomplete theory for ferro-magnetism. Equation (3.19) still implies that any material at sufficiently low temperatures $T < T_c$ would become a ferromagnet. However not all materials become ferromagnets in this way. The failure of eq. (3.19) is that is still does not encapsulate the idea that the electrons which carry the spins (elementary magnetic moments). Therefore we need to take into account the itinerant character of the electrons. The model that accomplishes this is called the **Hubbard model**. The kinetic energy terms of the Hubbard model are,

$$H_{\rm kin} = -t \sum_{\langle ij\rangle\sigma} (|i\sigma\rangle\langle j\sigma| + |j\sigma\rangle\langle i\sigma|) \tag{3.20}$$

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where t is now the usual tunneling constant of an electron traveling through a material (as we invoked frequently in section 2). Additionally we need to characterize the long-range interactions between spins. This however if too difficult so we approximate the interactions to only be short range. Simplifying for one orbital per atom,

$$H_{\rm int} = U \sum_{i} n_{i\uparrow} n_{i\downarrow} \tag{3.21}$$

with $n_{i,\sigma}$ denotes the number operator for the number of atoms localized to atom i with spin σ . Terms of the form $n_{i\uparrow}n_{i\uparrow}$ are not present due to the Pauli exclusion principle. The Hubbard model is simply eq. (3.20) combined with eq. (3.21),

$$H = -t \sum_{\langle ij \rangle \sigma} (|i\sigma\rangle\langle j\sigma| + |j\sigma\rangle\langle i\sigma|) + U \sum_{i} n_{i\uparrow} n_{i\downarrow}$$
(3.22)

Even this extremely simple model cannot be solved exactly (doing so would certainly yield a Nobel prize in physics). We perform mean-field theory approximations in $n_{i,\sigma}$,

$$n_{i\sigma} = \langle n_{i\sigma} \rangle + n_{i\sigma} - \langle n_{i\sigma} \rangle$$

Therefore,

$$n_{i\uparrow}n_{i\downarrow} = (\langle n_{i\uparrow}\rangle + n_{i\uparrow} - \langle n_{i\uparrow}\rangle)(\langle n_{i\downarrow}\rangle + n_{i\downarrow} - \langle n_{i\downarrow}\rangle)$$

$$n_{i\uparrow}n_{i\downarrow} = (n_{i\uparrow} - \langle n_{i\uparrow}\rangle)(n_{i\downarrow} - \langle n_{i\downarrow}\rangle) + n_{i\uparrow}\langle n_{i\downarrow}\rangle + n_{i\downarrow}\langle n_{i\uparrow}\rangle + \langle n_{i\uparrow}\rangle\langle n_{i\downarrow}\rangle - 2\langle n_{i\uparrow}\rangle\langle n_{i\downarrow}\rangle$$

Once again we assume that the fluctuations are small

$$n_{i\uparrow}n_{i\downarrow} = n_{i\uparrow}\langle n_{i\downarrow}\rangle + n_{i\downarrow}\langle n_{i\uparrow}\rangle - \langle n_{i\uparrow}\rangle\langle n_{i\downarrow}\rangle$$

Now we let n denote the average number of electrons per atom (here we assume that this number is uniform $n_i = n$),

$$n = \langle n_{i\uparrow} \rangle + \langle n_{i\downarrow} \rangle$$

We define the degree of spin polarization ξ such that,

$$\xi \equiv \frac{\langle n_{i\uparrow} \rangle - \langle n_{i\downarrow} \rangle}{\langle n_{i\uparrow} \rangle + \langle n_{i\downarrow} \rangle}$$

$$\langle n_{i\uparrow} \rangle = n \left(\frac{1+\xi}{2} \right) \qquad \langle n_{i\downarrow} \rangle = n \left(\frac{1-\xi}{2} \right)$$
(3.23)

Where $\xi_i = \xi$ is uniform as well for simplicity. Now we have eq. (3.22) taking the following form,

$$H = -t \sum_{\langle ij \rangle \sigma} (|i\sigma\rangle\langle j\sigma| + |j\sigma\rangle\langle i\sigma|) + \frac{Un}{2} \sum_{i} (n_{i\uparrow} + n_{i\downarrow}) - \frac{Un\xi}{2} \sum_{i} (n_{i\uparrow} - n_{i\downarrow}) - \frac{Un^2}{4} (1 - \xi^2) N$$
 (3.24)

Performing a Fourier transform on the lattice site basis states $|i\sigma\rangle$,

$$|i\sigma\rangle = \frac{1}{\sqrt{N}} \sum_{k} |k\sigma\rangle\langle k\sigma| i\sigma\rangle = \frac{1}{\sqrt{N}} \sum_{k} |k\sigma\rangle e^{-i\mathbf{k}\cdot\mathbf{R}_{i}}$$

We obtain the usual localization energies,

$$\varepsilon_k = -2t \sum_{\boldsymbol{a}} \cos(\boldsymbol{k} \cdot \boldsymbol{a})$$

Such that eq. (3.24) becomes,

$$H = \sum_{k,\sigma} \varepsilon_k |k\sigma\rangle\langle k\sigma| + \frac{Un}{2} \sum_i (n_{i\uparrow} + n_{i\downarrow}) - \frac{Un\xi}{2} \sum_i (n_{i\uparrow} - n_{i\downarrow}) - \frac{Un^2}{4} (1 - \xi^2) N$$
 (3.25)

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The second term of eq. (3.25) is a constant (it simply adds up all of the electrons which is presumed constant). Therefore we can neglect this term by absorbing it into the definition of H. Splitting the third term into a sum over spin up states and spin down states yields,

$$H = \sum_{k} \left(\varepsilon_{k} - \frac{Un\xi}{2} \right) |k\uparrow\rangle\langle k\uparrow| + \sum_{k} \left(\varepsilon_{k} + \frac{Un\xi}{2} \right) |k\downarrow\rangle\langle k\downarrow| - \frac{Un^{2}}{4} \left(1 - \xi^{2} \right) N$$
 (3.26)

The physics behind the appearance (or lack of appearance) of magnetism in this model is the comparison between the energy cost of polarization and the interaction energy gain. In order to find the interaction energy gain (average per state),

$$\frac{\langle H_{\rm int} \rangle}{N} = -\frac{Un\xi}{2} \frac{1}{N} \sum_{i} (\langle n_{i\uparrow} \rangle - \langle n_{i\downarrow} \rangle) + \frac{Un^2 \xi^2}{4} = -\frac{Un^2 \xi^2}{2} + \frac{Un^2 \xi^2}{4} = -\frac{Un^2 \xi^2}{4}$$

The fact that this interaction energy gain is negative means that the system gains (lowers) its energy due to spin polarization. Now let us calculate the kinetic energy cost. There is a kinetic energy cost because we have to fill the \uparrow and \downarrow spin states to two different energies. Pursuant to eq. (3.26) define this two energies as $\varepsilon_{k\uparrow}, \varepsilon_{k\downarrow}$

$$\varepsilon_{k\uparrow} = \varepsilon_k - \frac{Un\xi}{2}$$
 $\varepsilon_{k\downarrow} = \varepsilon_k + \frac{Un\xi}{2}$

For a given Fermi energy ε_F we can define two effective Fermi energies for either the spin \uparrow or spin \downarrow states.

$$\varepsilon_{k\uparrow} - \varepsilon_F = \varepsilon_k - \frac{Un\xi}{2} - \varepsilon_F \equiv \varepsilon_k - \varepsilon_{F\uparrow} \qquad \varepsilon_{k\downarrow} - \varepsilon_F = \varepsilon_k + \frac{Un\xi}{2} - \varepsilon_F \equiv \varepsilon_k - \varepsilon_{F\downarrow}$$

Where $\varepsilon_{F\uparrow}, \varepsilon_{F\downarrow}$ are defined as follows,

$$\varepsilon_{F\uparrow} = \varepsilon_F + \frac{Un\xi}{2}$$
 $\varepsilon_{F\downarrow} = \varepsilon_F - \frac{Un\xi}{2}$

Therefore the effective Fermi-energy for the spin $\uparrow (\downarrow)$ states is raised (lowered). Therefore the expected occupation for the \uparrow states is equal to half the total $\frac{n}{2}$ plus an extra amount given in terms of the density of states $g(\varepsilon)$,

$$\langle n_{\uparrow} \rangle = \frac{n}{2} + \int_{\varepsilon_F}^{\varepsilon_{F\uparrow}} d\varepsilon g(\varepsilon) \approx \frac{n}{2} + g(\varepsilon_F)(\varepsilon_{F\uparrow} - \varepsilon_F)$$
 (3.27)

Where we approximate $\varepsilon_{F\uparrow} \approx \varepsilon_F$ such that the integral is linearized. Analogously for n_{\downarrow} we have,

$$\langle n_{\downarrow} \rangle = \frac{n}{2} - \int_{\varepsilon_{F\downarrow}}^{\varepsilon_F} d\varepsilon g(\varepsilon) \approx \frac{n}{2} - g(\varepsilon_F)(\varepsilon_F - \varepsilon_{F\downarrow})$$
 (3.28)

Therefore $\langle n_{\uparrow} \rangle - \langle n_{\downarrow} \rangle$ is the difference of eq. (3.27) and eq. (3.28),

$$\langle n_{\uparrow} \rangle - \langle n_{\downarrow} \rangle = g(\varepsilon_F)(\varepsilon_{F\uparrow} - \varepsilon_F) + g(\varepsilon_F)(\varepsilon_F - \varepsilon_{F\downarrow}) = g(\varepsilon_F)(\varepsilon_{F\uparrow} - \varepsilon_{F\downarrow}) = 2g(\varepsilon_F)(\varepsilon_{F\uparrow} - \varepsilon_F)$$

Additionally using eq. (3.23) and our earlier assumption of occupation uniformity yields,

$$\langle n_{\uparrow} \rangle - \langle n_{\perp} \rangle = n\xi$$

Therefore combining these two results yields,

$$\varepsilon_{F\uparrow} - \varepsilon_F = \frac{n\xi}{2g(\varepsilon_F)}$$

Then the kinetic energy cost of spin polarization is given by,

$$E_{\rm kin} = \int_{\varepsilon_F}^{\varepsilon_F \uparrow} \mathrm{d}\varepsilon (\varepsilon g(\varepsilon)) - \int_{\varepsilon_F \downarrow}^{\varepsilon_F} \mathrm{d}\varepsilon (\varepsilon g(\varepsilon))$$

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$$\begin{split} E_{\rm kin} &\simeq \frac{g(\varepsilon_F)}{2} \left(\varepsilon_{F\uparrow}^2 - \varepsilon_F^2 - \varepsilon_F^2 + \varepsilon_{F\downarrow}^2 \right) \\ &= \frac{g(\varepsilon_F)}{2} [(\varepsilon_{F\uparrow} - \varepsilon_F)(\varepsilon_{F\uparrow} + \varepsilon_F) + (\varepsilon_{F\downarrow} - \varepsilon_F)(\varepsilon_{F\downarrow} + \varepsilon_F)] \\ &= \frac{g(\varepsilon_F)}{2} (\varepsilon_{F\uparrow} - \varepsilon_F) [(\varepsilon_{F\uparrow} + \varepsilon_F) - (\varepsilon_{F\downarrow} + \varepsilon_F)] \\ &= \frac{g(\varepsilon_F)}{2} (\varepsilon_{F\uparrow} - \varepsilon_F) [(\varepsilon_{F\uparrow} - \varepsilon_F) - (\varepsilon_{F\downarrow} - \varepsilon_F)] \\ &= g(\varepsilon_F)(\varepsilon_{F\uparrow} - \varepsilon_F)^2 \\ &= \frac{n^2 \xi^2}{4g(\varepsilon_F)} \end{split}$$

Therefore the kinetic energy cost is positive. In total, the energy cost is,

$$E_{\text{tot}} = E_{\text{kin}} + E_{\text{int}} = \frac{n^2 \xi^2}{4g(\varepsilon_F)} - \frac{Un^2 \xi^2}{4} = \frac{n^2 \xi^2}{4g(\varepsilon_F)} (1 - Ug(\varepsilon_F))$$

Therefore the interaction energy gain dominates the kinetic energy cost when the term in brackets is negative or when,

$$Ug(\varepsilon_F) > 1$$

This inequality is known as the **Stoner criterion** named after Edmund Clifton Stoner. The Stoner criterion explains the importance of d-orbitals. In d-orbitals, $g(\varepsilon_F)$ is larger because d-orbitals give rise to more narrow bands. Moreover U is larger since d-electrons are more strongly localized (the Coulomb interaction is stronger).

3.5 Mott Insulators and Anti-Ferromagnetism

Returning to the Hubbard model,

$$H = -t \sum_{\langle ij \rangle \sigma} (|i\sigma\rangle\langle j\sigma| + |j\sigma\rangle\langle i\sigma|) + U \sum_{i} n_{i\uparrow} n_{i\downarrow}$$
(3.29)

Consider a half-filled band where there is one electron per site $(n_{i\uparrow} = 1 \text{ or } n_{i\downarrow} = 1, \text{ not both})$. According to band energy theory, this is a metal. However, for large enough U, it becomes energetically unfavorable for electrons to move, since moving through a half-filled energy and always involves doubly occupying some sites $(n_{i\uparrow} = 1 \text{ and } n_{i\downarrow} = 1)$ which by eq. (3.29) costs energy U. Therefore at roughly $U \sim zt$ there is transition from metal to insulator, which is called a **Mott insulator**. An important question in a Mott insulator is what happens to the spins of the localized electrons?

Supposing U is sufficiently high that $U \gg 2t$, therefore we can set t = 0 in eq. (3.29) and analyze,

$$H = U \sum_{i} n_{i\uparrow} n_{i\downarrow}$$

The ground state of a Mott insulator is where there is exactly one electron at every site, but the spin can either be \uparrow or \downarrow . Therefore the ground state is 2^N -fold degenerate. This degeneracy is lifted when $t \neq 0$. The question then becomes, how is this degeneracy lifted? To see how this happens, consider the two-site problem where there are N=2 electrons,

$$H = -t \sum_{\sigma} (|1\sigma\rangle\langle 2\sigma| + |2\sigma\rangle\langle 1\sigma|) + U \sum_{i \in \{1,2\}} n_{i\uparrow} n_{i\downarrow}$$

Since there are 2 electrons (1 for each site), there are 4 single-electron states,

$$|1\uparrow\rangle$$
 $|1\downarrow\rangle$ $|2\uparrow\rangle$ $|2\downarrow\rangle$

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As discussed before, the wavefunction for 2 electrons must be anti-symmetric. Therefore there are 6 possible states of 2 electrons (the notation $|1\downarrow,2\uparrow\rangle$ reads: on site 1 there is a spin \downarrow electron and on site 2 there is a spin \uparrow electron),

$$\begin{split} |1\uparrow,1\downarrow\rangle &= -|1\downarrow,1\uparrow\rangle \\ |1\uparrow,2\uparrow\rangle &= -|2\uparrow,1\uparrow\rangle \\ |1\uparrow,2\downarrow\rangle &= -|2\downarrow,1\uparrow\rangle \\ |1\downarrow,2\uparrow\rangle &= -|2\uparrow,1\downarrow\rangle \\ |1\downarrow,2\downarrow\rangle &= -|2\downarrow,1\downarrow\rangle \\ |2\uparrow,2\downarrow\rangle &= -|2\downarrow,2\uparrow\rangle \end{split}$$

Therefore we need to find the matrix representation of H in terms of this basis. The matrix-element of such a representation is only nonzero when there are 2 electrons on either site 1 or site 2 (and not one on each),

$$\langle 1 \uparrow, 1 \downarrow | H_{\text{int}} | 1 \uparrow, 1 \downarrow \rangle = \langle 2 \uparrow, 2 \downarrow | H_{\text{int}} | 2 \uparrow, 2 \downarrow \rangle = U$$

The remaining elements are all zero. The kinetic energy term turns $|1\uparrow\rangle$ into $|2\uparrow\rangle$ and $|1\downarrow\rangle$ into $|2\downarrow\rangle$ and vice versa. Thus we have,

$$\langle 1 \downarrow, 2 \uparrow | H_{\text{kin}} | 1 \downarrow, 1 \uparrow \rangle = -t$$
$$\langle 1 \downarrow, 2 \uparrow | H_{\text{kin}} | 1 \uparrow, 1 \downarrow \rangle = t$$

Therefore the complete Hamiltonian $(H_{kin} + H_{int})$ can be written as,

$$H = \begin{pmatrix} U & 0 & -t & t & 0 & 0 \\ 0 & 0 & 0 & 0 & 0 & 0 \\ -t & 0 & 0 & 0 & 0 & -t \\ t & 0 & 0 & 0 & 0 & t \\ 0 & 0 & 0 & 0 & 0 & 0 \\ 0 & 0 & -t & t & 0 & U \end{pmatrix}$$

Clearly the states $|1\uparrow, 2\uparrow\rangle$ and $|1\downarrow, 2\downarrow\rangle$ correspond to the 2nd and 5th rows and columns of this matrix and are decoupled from the other states and have eigenenergy 0. The remaining Hamiltonian, indexed by states $|1\uparrow, 2\downarrow\rangle, |1\uparrow, 2\downarrow\rangle, |1\downarrow, 2\uparrow\rangle$ and $|2\uparrow, 2\downarrow\rangle$ (respectively) can be written as a 4×4 matrix,

$$H = \begin{pmatrix} U & -t & t & 0 \\ -t & 0 & 0 & -t \\ t & 0 & 0 & t \\ 0 & -t & t & U \end{pmatrix}$$

Its eigenvalues are,

$$E_1 = 0$$
 $E_2 = U$ $E_3 = \frac{1}{2} \left(U + \sqrt{U^2 + 16t^2} \right)$ $E_4 = \frac{1}{2} \left(U - \sqrt{U^2 + 16t^2} \right)$

The ground states of the system correspond to the lowest energy eigenvalues. Clearly this is E_4 ,

$$E_g = E_4 = \frac{1}{2} \left(U - \sqrt{U^2 + 16t^2} \right)$$

The corresponding (unnormalized) eigenvector is,

$$|z\rangle = \left(1, \frac{U + \sqrt{U^2 + 16t^2}}{4t}, -\frac{U + \sqrt{U^2 + 16t^2}}{4t}, 1\right)$$

Specifically for Mott insulators we are interested in the case where $U \gg t$. In this case,

$$|z\rangle pprox \left(0, \frac{U}{2t}, -\frac{U}{2t}, 0
ight)$$

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This ground state wavefunction is,

$$|\psi\rangle = \frac{1}{\sqrt{2}}(|1\uparrow,2\downarrow\rangle - |1\downarrow,2\uparrow\rangle)$$

Where $|\psi\rangle$ corresponded to a total spin of $\mathbf{S}=0$ and a ground state energy of,

$$E_g = \frac{1}{2} \left(U - \sqrt{U^2 + 16t^2} \right) \approx \frac{1}{2} \left(U - U \left(1 + 8 \frac{t^2}{U^2} \right) \right) = -\frac{4t^2}{U}$$

Therefore $\mathcal{J} = \frac{4t^2}{U}$. Using the two spin model $H = \mathcal{J}S_1 \cdot S_2$ we can immediately generalize to N interacting spin systems where the spin terms are restricted to nearest neighbor interactions,

$$H = rac{\mathcal{J}}{2} \sum_{\langle ij
angle} oldsymbol{S}_i \cdot oldsymbol{S}_j$$

Which is a good description of an anti-ferromagnet because anti-ferromagnets are Mott insulators.

- 1. Ferromagnetism arises from lowering of the interaction energy between the electrons, when the spins are aligned. Ferromagnets are metals. (e.g. Fe, Co, Ni, Mn)
- 2. Anti-Ferromagnetism arises from lowering of kinetic energy when spins are anti-aligned. Anti-Ferromagnets are Mott insulators. (e.g. MnO, NiO, La_2CnO_4)

In is important to understand the qualitative differences between ferromagnets and anti-ferromagnets given their similar models:

Ferromagnets: For ferromagnets our interacting model (with two spins) is,

$$H = -\mathcal{J}S_1 \cdot S_2$$

Where the spins become aligned or anti-aligned in the ground state.

$$|\psi\rangle \in \{|\uparrow,\uparrow\rangle,|\downarrow,\downarrow\rangle\}$$

Where the components of the spin $S = \frac{\hbar}{2}\sigma$ can be written as

$$S_1 \cdot S_2 = S_1^x S_1^x + S_1^y S_1^y + S_1^z S_1^z = \frac{1}{2} (S_1^+ S_2^- + S_1^- S_2^+) + S_1^z S_2^z$$

Therefore we see that.

$$\begin{split} S^z|\uparrow\rangle &= \frac{\hbar}{2}\sigma^z|\uparrow\rangle = \frac{\hbar}{2}|\uparrow\rangle \\ S^z|\downarrow\rangle &= \frac{\hbar}{2}\sigma^z|\downarrow\rangle = \frac{\hbar}{2}|\downarrow\rangle \\ S^+|\uparrow\rangle &= \frac{\hbar}{2}\sigma^+|\uparrow\rangle = \frac{\hbar}{2}\begin{pmatrix} 0 & 2 \\ 0 & 0 \end{pmatrix}\begin{pmatrix} 1 \\ 0 \end{pmatrix} = 0 \\ S^-|\uparrow\rangle &= \frac{\hbar}{2}\sigma^+|\uparrow\rangle = \frac{\hbar}{2}\begin{pmatrix} 0 & 0 \\ 2 & 0 \end{pmatrix}\begin{pmatrix} 1 \\ 0 \end{pmatrix} = \hbar|\downarrow\rangle \ S^+|\downarrow\rangle \\ S^-|\downarrow\rangle &= \frac{\hbar}{2}\sigma^+|\downarrow\rangle = \frac{\hbar}{2}\begin{pmatrix} 0 & 0 \\ 2 & 0 \end{pmatrix}\begin{pmatrix} 0 \\ 1 \end{pmatrix} = 0 \end{split}$$

Therefore $|\uparrow,\uparrow\rangle$ is an eigenstate of H with eigenenergy $-\mathcal{J}\hbar^2/4$,

$$H|\psi\rangle = -\frac{\mathcal{J}\hbar^2}{4} \left[\sigma_1^z \sigma_2^z + \frac{1}{2} \left(\sigma_1^+ \sigma_2^- + \sigma_1^- \sigma_2^+ \right) \right] |\uparrow,\uparrow\rangle = -\frac{\mathcal{J}\hbar^2}{4} |\uparrow,\uparrow\rangle$$

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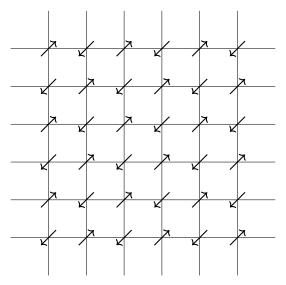
This analysis seems trivial, but becomes illuminating when we consider the analogy for anti-ferromagnets. **Ferromagnets:** For anti-ferromagnets our interacting model (with two spins) is,

$$H = +\mathcal{J}S_1 \cdot S_2$$

A simple guess for the ground state would reasonably be $|\uparrow,\downarrow\rangle$ (or $|\downarrow,\uparrow\rangle$). This however does not work,

$$H|\psi\rangle = \frac{\mathcal{I}\hbar^2}{4} \left[\sigma_1^z \sigma_2^z + \frac{1}{2} \left(\sigma_1^+ \sigma_2^- + \sigma_1^- \sigma_2^+ \right) \right] |\uparrow,\downarrow\rangle \not\propto |\uparrow,\downarrow\rangle$$

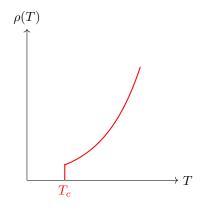
Therefore $|\uparrow,\downarrow\rangle$ is *not* the ground state of for anti-ferromagnets. This state, and the analogous multi-spin state, where neighboring spin states have opposite spin, is called the **Neel state**.



4 Superconductivity

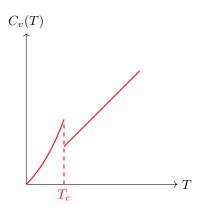
4.1 Ginzburg-Landau Free Energy

Most metallic elements (except ferromagnets) and even some non-metals undergo a transition into a superconducting state as the temperature is lowered. Superconductivity is the phenomena where resistivity suddenly drops to zero below a critical temperature T_c ,



A related feature is that that specific heat C_v of the material jumps at the critical temperature as well from $C_v \sim T$ to $C_v \sim e^{-\frac{\Delta}{k_B T}}$.

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There are two possible ways one could interpret the features of superconductivity. The first is that superconductors are simply ideal metals in which below T_c the electrons do not experience scattering. Alternatively, it could be that a superconductor is a new state of matter. If was the **Meissner effect** in the 1930s involving liquefied Helium that first demonstrated the latter; superconductivity is genuinely a new state of matter. In an ideal metal with $\rho = 0$ when $T < T_c$ if we were to start from $T > T_c$ we have $\nabla \times E = -\frac{1}{c} \frac{\partial B}{\partial t}$. When $T < T_c$ we have that $\frac{\partial B}{\partial t} = 0$ (for an ideal metal). Therefore $\beta E = \rho j \implies E = 0$ inside the sample when $T < T_c$. This is fundamentally different than a superconductor which expels all magnetic fields below T_c .

- Ideal metal: $T < T_c$, $\rho = 0$, $\boldsymbol{B} = \text{const}$
- Superconductor: $T < T_c$, $\rho = 0$, B = 0

Typical superconductors like Al, Sn, Hg have very low critical temperatures ($T_c \sim 1\,\mathrm{K} - 10\,\mathrm{K}$) which are non-ideal for use in practical electronics. High temperature super conductors like $\mathrm{La_{2-x}Sr_xCnO_4}$ have $T_c \sim 34\,\mathrm{K}$. The current record holder is $\mathrm{HgBa_2Ca_2CN_3O_8}$ with $T_c \sim 133\,\mathrm{K}$.

Microscopically, superconductivity originates from the pairing of electrons. In metallic superconductors this is due to the exchange of phonons: electrons moving through the crystal lattice, polarizes the crystal lattice locally, which attracts another electron. In cuprate high- T_c superconductors, pairing is thought to be due to the anti-ferromagnetic spin-spin interaction of the parent (undoped) Mott insulator.

The most important qualitative properties of superconductors can be described using semi-phenomenological **Landau-Ginzburg theory**.

Pairs of electrons form bosonic states and are considered bosons. Bosons are not constrained by the Pauli exclusion principle; any number of bosons may occupy the ground state. Therefore at zero or low temperatures, all the bosons will want to *condense* into the lowest energy state; the state with zero momentum. In this state, all of the bosons are essentially described by a single wavefunction $\varphi(r)$. Landau and Ginzburg realized that one may think of $\varphi(r)$ as an order parameter, just like magnetization in a ferromagnet. Near T_c we may then write down a Taylor expansion of the free energy F in powers of φ . Since the free energy should not depend on the phase of $\varphi(r)$, it may only depend on $|\varphi|^2$, $|\varphi|^4$, . . . and so on. Thus,

$$F = F_0 + Va|\varphi|^2 + \frac{1}{2}Vb|\varphi|^4 + \cdots$$

Where F_0 is the free energy of the normal metallic state which we will omit henceforth. V is the total volume. The ground state occurs when F is minimized,

$$\frac{1}{V}\frac{\partial F}{\partial |\varphi|} = 2a|\varphi| + 2b|\varphi|^3 = 0$$

Therefore we have a ground state,

$$|\varphi| = \sqrt{-\frac{a}{b}}$$

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Which is possible when a < 0 (b > 0 always) when $T < T_c$. Therefore,

$$a \sim \frac{T - T_c}{T_c}$$

Therefore,

$$|\varphi| \sim \sqrt{T_c - T} \sim (-t)^{\beta} \qquad \beta = \frac{1}{2}$$

To describe properties of superconductors it turns out to be very important to describe situations when $\varphi(r)$ is not uniform. In this case we need to add extra terms to the free energy which depend on the non-uniformity of $\varphi(r)$; specifically $\nabla \varphi(r)$. Naively, the following should work by symmetry,

$$F = \kappa \int \mathrm{d}^3 r |\boldsymbol{\nabla} \varphi|^2$$

Where $\kappa > 0$. However, it does not really work since this violates a very important physical principle: gauge invariance. When we talked about electrons in a magnetic field, we saw that the momentum operator in a magnetic field was,

$$\hat{\boldsymbol{p}} = -i\hbar \boldsymbol{\nabla} + \frac{e}{c} \boldsymbol{A}$$

which was a result of demanding gauge invariance. Consider a wavefunction $\psi(\mathbf{r})$. Let us multiply it by an arbitrary position-dependent phase factor $\mathcal{X}(\mathbf{r})$.

$$\psi(\mathbf{r}) \mapsto e^{i\mathcal{X}(\mathbf{r})}\psi(\mathbf{r})$$

Acting on this new wave-function with the momentum operator yields,

$$\left(-i\hbar\boldsymbol{\nabla} + \frac{e}{c}\boldsymbol{A}\right)\left(e^{i\mathcal{X}(\boldsymbol{r})}\psi(\boldsymbol{r})\right) = e^{i\mathcal{X}(\boldsymbol{r})}\left[-i\hbar\boldsymbol{\nabla} + \frac{e}{c}\boldsymbol{A} + \hbar\boldsymbol{\nabla}\mathcal{X}\right]\psi(\boldsymbol{r})$$

Now recall that \boldsymbol{A} is defined only up to a gradient $\boldsymbol{A} \mapsto \boldsymbol{A} + \boldsymbol{\nabla} f$, thus we can make use of the following gauge transformation of \boldsymbol{A} ,

$$m{A} \mapsto m{A} - rac{\hbar c}{e} m{
abla} \mathcal{X}$$

and obtain,

$$\left(-i\hbar\boldsymbol{\nabla} + \frac{e}{c}\boldsymbol{A}\right)\left(e^{i\mathcal{X}(\boldsymbol{r})}\psi(\boldsymbol{r})\right) = e^{i\mathcal{X}(\boldsymbol{r})}\left[-i\hbar\boldsymbol{\nabla} + \frac{e}{c}\boldsymbol{A}\right]\psi(\boldsymbol{r})$$

This means that the Schrodinger equation for a particle of charge -e in an electromagnetic field is invariant under the following transformation,

$$\psi \mapsto e^{i\mathcal{X}}\psi \qquad \boldsymbol{A} \mapsto \boldsymbol{A} - \frac{\hbar c}{e} \boldsymbol{\nabla} \mathcal{X}$$

In our case, $\varphi(r)$ is a wave-function of particles with charge -2e (one for each electron). Thus the free energy must be invariant under the following,

$$\varphi \mapsto e^{i\mathcal{X}} \varphi \qquad \mathbf{A} \mapsto \mathbf{A} - \frac{\hbar c}{2e} \mathbf{\nabla} \mathcal{X}$$

This means that the free energy must have the following form,

$$F = \int d^3r \left[\frac{1}{2m^*} \left| \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi \right|^2 + a|\varphi|^2 + \frac{b}{2}|\varphi|^4 + \frac{|\mathbf{B}|^2}{8\pi} - \frac{\mathbf{B} \cdot \mathbf{H}}{4\pi} \right]$$
(4.1)

Where $e^* = 2e$ and $m^* = 2m$ and H is an external magnetic field. Equation (4.1) is called the **Ginzburg-Landau free energy** of a superconductor.

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The thermodynamic equilibrium of a state of a superconductor is determined by minimizing F with respect to φ . Determining the equilibrium points requires some variational calculus. Varying F with respect to $\varphi^*(r)$,

$$\delta_{\varphi^*} F = \int d\mathbf{r} \left[a\phi \delta \varphi^* + b\varphi |\varphi|^2 \delta \varphi^* + \frac{1}{2m^*} \left(i\hbar \nabla \delta \varphi^* + \frac{e^*}{c} \mathbf{A} \delta \varphi^* \right) \cdot \left(-i\hbar \nabla \varphi + \frac{e^*}{c} \mathbf{A} \varphi \right) \right]$$
(4.2)

Consider the first term in the product expansion,

$$\int d\mathbf{r} \nabla \delta \varphi^* \cdot \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi$$

$$= -\int d\mathbf{r} \delta \varphi^* \nabla \cdot \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi + \int d\mathbf{r} \nabla \cdot \left[\delta \varphi^* \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi \right]$$

$$= -\int d\mathbf{r} \delta \varphi^* \nabla \cdot \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi + \oint \delta \varphi^* \left(-i\hbar \nabla \varphi + \frac{e^*}{c} \mathbf{A} \varphi \right) \cdot d\mathbf{S}$$

Here S is the surface of the sample. It is possible to neglect the surface term since it only determines the boundary conditions for φ in a finite sample (i.e. $\delta \varphi^*|_S = 0$). Therefore eq. (4.2) becomes,

$$\delta_{\varphi^*} F = \int d\mathbf{r} \left[a\varphi + b\varphi |\varphi|^2 + \frac{1}{2m^*} \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right)^2 \varphi \right] \delta\varphi^* = 0$$

The thermodynamic equilibrium occurs when $\delta_{\varphi^*}F = 0$. $\delta_{\varphi^*}F$ can only equal zero if the expression in the square brackets vanishes at every point in \boldsymbol{r} (this is due to the Raymond-Dubois lemma). Thus we obtain the first Ginzburg-Landau equation.

$$\frac{1}{2m^*} \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right)^2 \varphi + a\varphi + b|\varphi|^2 \varphi = 0$$
(4.3)

Equation (4.3) looks like a Schrodinger equation for a particle in a magnetic field, but it is clearly non-linear in φ . Note however, that the magnetic field $\mathbf{B} = \nabla \times \mathbf{A}$ inside the sample is also a thermodynamic variable and we must also vary F with respect to \mathbf{A} analogous to how we varied F with respect to φ^* in eq. (4.2).

$$\delta_{\mathbf{A}}F = \int d\mathbf{r} \left[\frac{1}{2m^*} \frac{e^*}{c} \delta \mathbf{A} \varphi^* \cdot \left(-i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi + \frac{1}{2m^*} \left(i\hbar \nabla + \frac{e^*}{c} \mathbf{A} \right) \varphi^* \cdot \frac{e^*}{c} \delta \mathbf{A} \varphi + \cdots \right]$$
$$\cdots + \frac{1}{4\pi} (\nabla \times \mathbf{A}) \cdot (\nabla \times \delta \mathbf{A}) - \frac{\mathbf{H}}{4\pi} \cdot (\nabla \times \delta \mathbf{A})$$

Consider the last two terms, we make use of the following identity: $a \cdot (\nabla \times b) = b \cdot (\nabla \times a) - \nabla \cdot (a \times b)$

$$\frac{1}{4\pi} \int d\boldsymbol{r} [(\boldsymbol{\nabla} \times \boldsymbol{A}) - \boldsymbol{H}] \cdot (\boldsymbol{\nabla} \times \delta \boldsymbol{A}) = \frac{1}{4\pi} \int d\boldsymbol{r} \delta \boldsymbol{A} \cdot \boldsymbol{\nabla} \times (\boldsymbol{\nabla} \times \boldsymbol{A}) - \frac{1}{4\pi} \oint d\boldsymbol{S} \cdot [\delta \boldsymbol{A} \times (\boldsymbol{\nabla} \times \boldsymbol{A} - \boldsymbol{H})]$$

The surface integral vanishes because the magnetic field at the surface of the conductor is fixed by boundary conditions. Thus $\delta A|_S = 0$. Thus we obtain,

$$\delta_{\mathbf{A}}F = \int d\mathbf{r} \left[\frac{-i\hbar e^*}{2m^*c} (\varphi^* \nabla \varphi - \varphi \nabla \varphi^*) + \frac{e^{*2}}{m^*c^2} |\varphi|^2 \mathbf{A} + \frac{1}{4\pi} \nabla \times (\nabla \times \mathbf{A}) \right] \cdot \delta \mathbf{A} = 0$$

Setting the integrand to be zero and using the fact that $\nabla \times (\nabla \times A) = \nabla \times B = \frac{4\pi}{c} j$ we obtain an expression for j,

$$\mathbf{j} = \frac{i\hbar e^*}{2m^*} (\varphi^* \nabla \varphi - \varphi \nabla \varphi^*) - \frac{e^{*2}}{m^* c} |\varphi|^2 \mathbf{A}$$
(4.4)

This is referred to as the **second Ginzburg-Landau equation**. We will now analyze the consequences of eqs. (4.3) and (4.4). We begin by assuming that $T < T_c$ so that we are in the superconducting state. If we assume that φ is uniform throughout the material, then eq. (4.3) becomes,

$$a\varphi^* + b|\varphi|^2\varphi = 0$$

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Therefore we have obtained a non-trivial solution to eq. (4.3),

$$|\varphi| = \sqrt{\frac{|a|}{b}}$$

If we interpret $\varphi(\mathbf{r})$ as the wavefunction of the pair condensate, then $|\varphi|^2$ should have the meaning of the number density of pairs in the material,

$$|\varphi|^2 = n_s^*$$

Letting $\varphi(\mathbf{r}) = \sqrt{n_s^*}e^{i\theta(\mathbf{r})}$, then in a superconducting sample, we can expect n_s^* to be uniform and only $\theta(\mathbf{r})$ to be depending on \mathbf{r} . Calculating the current via eq. (4.4) and we obtain,

$$oldsymbol{j} = -rac{\hbar e^* n_s^*}{m^*}igg(oldsymbol{
abla} heta + rac{e^*}{\hbar c}oldsymbol{A}igg)$$

Therefore the superconducting current j is directly related to the gradient of the phase of the macroscopic condensate wavefunction $\nabla \theta(r)$. This explains both the Meissner effect and the effect of zero resistance. First consider the Meissner effect. It is possible to remove the phase by a gauge transformation $(A \mapsto A - \frac{\hbar c}{e^*} \nabla \theta)$,

$$\boldsymbol{j} = -\frac{e^{*2}n_s^*}{m^*c}\boldsymbol{A} \tag{4.5}$$

Then take the curl of both sides of eq. (4.5). If $\nabla \times B = \frac{4\pi}{c} j$,

$$\nabla \times (\nabla \times \boldsymbol{B}) = \frac{4\pi}{c} (\nabla \times \boldsymbol{j}) = \nabla (\nabla \cdot \boldsymbol{B}) - \nabla^2 \boldsymbol{B} = -\nabla^2 \boldsymbol{B}$$

$$oldsymbol{
abla} imes oldsymbol{j} = -rac{e^{*2}n_s^*}{m^*c}oldsymbol{
abla} imes oldsymbol{A} = -rac{e^{*2}n_s^*}{m^*c}oldsymbol{B}$$

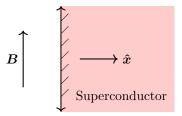
Thus we obtain the following equation for B,

$$\nabla^2 \mathbf{B} = \frac{4\pi e^{*2} n_s^*}{m^* c^2} \mathbf{B} = \frac{1}{\lambda^2} \mathbf{B}$$
 (4.6)

Where λ can be interpreted as the penetration depth of the magnetic field into the superconductor,

$$\lambda = \sqrt{\frac{m^* c^2}{4\pi e^{*2} n_s^*}} \tag{4.7}$$

If, for example, we place the magnetic field parallel to the superconductor with surface normal co-linear to the \hat{x} -axis,



then eq. (4.6) determines that **B** only penetrates a distance of λ into the superconductor,

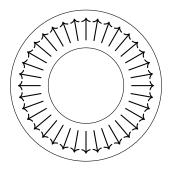
$$\frac{\mathrm{d}^2 B}{\mathrm{d}x^2} = \frac{1}{\lambda^2} B \implies B(x) = B(0)e^{-x/\lambda}$$

In simple, metallic superconductors, $\lambda \sim 100\,\text{Å}$ which is much shorter than the characteristic size of the superconducting object. This gives the effect that the magnetic field is expelled from the superconductor and explains that eq. (4.6) is the origin of the Meissner effect.

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4.2 Persistent Currents

What about the effect of virtually zero resistivity and **persistent currents** in superconducting materials? This can be accomplished by setting up gradients of the phase $\theta(\mathbf{r})$ inside a superconducting ring.



If we recall some important terms of eq. (4.1),

$$F = \int d^3r \frac{\hbar^2 n_s^{*2}}{2m^*} \left(\boldsymbol{\nabla} \theta + \frac{e^*}{\hbar c} \boldsymbol{A} \right)^2 + \cdots$$

In order to minimize F, we roughly want the following to hold inside the ring.

$$\nabla \theta = -\frac{e^*}{\hbar c} A$$

This implies that the current must flow at the edge of the ring creating the corresponding magnetic flux,

$$oldsymbol{j} = rac{c}{4\pi} oldsymbol{
abla} imes oldsymbol{B}$$

and therefore we must have,

$$\oint \nabla \theta \cdot d\ell = 2\pi n$$

Where n is some integer. Going from the n = 1 to n = 0 state requires unwinding the phase, which costs a huge amount of energy. The above also implies the quantization of magnetic flux inside the superconducting ring.

$$\oint \nabla \theta \cdot d\boldsymbol{\ell} = 2\pi n = -\frac{e^*}{\hbar c} \oint \nabla A \cdot d\boldsymbol{\ell} = -\frac{e^*}{\hbar c} \varphi$$

Thus the magnetic field is quantized in units of φ_0 ,

$$\varphi_0 = \frac{hc}{e^*} = \frac{hc}{2e}$$

and a measurement of magnetic flux quantization proves that electrons pairing in superconductors is a valid model.

4.3 Type I & Type II Superconductors

One important length scale for a superconductor is the penetration depth λ of the magnetic field (eq. (4.7)) derived above via eq. (4.6).

$$\lambda = \sqrt{\frac{m^*c^2}{4\pi e^{*2}n_s^*}}$$

There is another length scale, the coherence length ξ , which is related to how the superconducting order parameter $\varphi(\mathbf{r})$ varies spatially. Consider the first Ginzburg-Landau equation eq. (4.3),

$$\frac{1}{2m^*} \biggl(-i\hbar \boldsymbol{\nabla} + \frac{e^*}{c} \boldsymbol{A} \biggr)^2 \varphi + a\varphi + b |\varphi|^2 \varphi = 0$$

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Suppose there was no magnetic field A = 0 and look for a solution that satisfies the following boundary conditions,

$$\varphi(x=0) = 0 \qquad \varphi(x \to \infty) = \sqrt{-\frac{a}{b}}$$
 (4.8)

This solution represents a boundary between a normal and superconducting material,

$$-\frac{\hbar^2}{2m^*}\frac{\mathrm{d}^2\varphi}{\mathrm{d}x^2} + a\varphi + b|\varphi^*|^3\varphi = 0 \tag{4.9}$$

The solution to eq. (4.9) that satisfies the relevant boundary conditions eq. (4.8) is,

$$\varphi(x) = \sqrt{-\frac{a}{b}} \tanh\left(\frac{x}{\sqrt{2}\xi}\right)$$

Where ξ is the **coherence length**,

$$\xi = \sqrt{\frac{\hbar^2}{-2m^*a}}$$

The coherence length ξ determines the extent to which the superconducting wavefunction will penetrate into the region associated with the normal material. The ratio of λ and ξ , $\mathcal{X} = \lambda/\xi$ is an important parameter. Superconductors with $\mathcal{X} < 1/\sqrt{2}$ are called **type I superconductors**, while $\mathcal{X} > 1/\sqrt{2}$ are **type II superconductors**.

Consider a superconductor in a magnetic field. When the field is expelled, the free energy density is given by,

$$f = a|\varphi|^2 + \frac{b}{2}|\varphi|^4$$

The non-trivial solution occurs at $|\varphi|^2 = -\frac{a}{b}$ which means that the free energy density becomes,

$$f = -\frac{a^2}{b} + \frac{a^2}{2b} = -\frac{a^2}{2b}$$

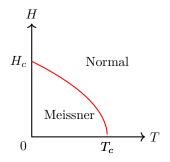
Which is referred to as the **condensation energy**. The system gains condensation energy by transitioning into the superconducting state. However, magnetic energy is lost when the field is expelled from the superconductor:

$$f_m = \frac{B^2}{8\pi} - \frac{BH}{4\pi} = -\frac{H^2}{8\pi}$$

When the magnetic field is large enough, the loss of magnetic energy will become larger than the gain in condensation energy and the superconductor will be destroyed and will allow the field to penetrate. This happens at the thermodynamic critical field H_c ,

$$\frac{H_c^2}{8\pi} = \frac{a^2}{2b} \implies H_c = \sqrt{\frac{4\pi a^2}{b}}$$

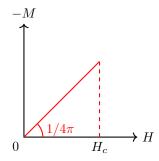
At this critical magnetic field, $a \sim T - T_c$ which implies $H_c \sim |T - T_c|$.



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Additionally the magnetic susceptibility χ produces perfect diamagnetism,

$$B = H + 4\pi M = 0 \implies H = -4\pi M \implies \chi = \frac{\partial M}{\partial H} = -\frac{1}{4\pi}$$



In reality, this only holds true for type I superconductors.

Suppose we have constructed a large field H, much greater than H_c , and begin decreasing the field strength. We can expect the superconducting state to appear when the first Ginzburg-Landau equation (eq. (4.3)) has non-trivial solution. The field, at which this happens, is different from H_c , and is called H_{c2} . Near H_{c2} we can expect $|\varphi|$ to be small, enabling us to neglect the non-linear term in eq. (4.3). In this regime, we have,

$$\frac{1}{2m^*} \biggl(-i\hbar \boldsymbol{\nabla} + \frac{e^*}{c} \boldsymbol{A} \biggr)^2 \varphi = -a \varphi$$

This is simply the Schrödinger equation for a charged particle in a magnetic field with energy -a. If we choose the Landau gauge $\mathbf{A} = xB\hat{\mathbf{y}}$, then eq. (4.3) becomes,

$$-\frac{\hbar^2}{2m^*} \left(\frac{\partial^2}{\partial x^2} + \frac{\partial^2}{\partial z^2} \right) \varphi + \frac{1}{2m^*} \left(-i\hbar \frac{\partial}{\partial y} + \frac{e^*}{c} Bx \right)^2 \varphi = -a\varphi$$

Conjecturing an ansatz of the following form $\varphi(\mathbf{r}) = e^{i(k_y y + k_z z)} \phi(x)$, we obtain the auxiliary equation,

$$-\frac{\hbar^2}{2m^*}\frac{\mathrm{d}^2\phi}{\mathrm{d}x^2} + \left(\frac{e^*B}{c}\right)^2x^2\phi + \frac{\hbar k_y e^*B}{c}x\phi = \left(-a - \frac{\hbar^2k_y^2}{2m^*} - \frac{\hbar^2k_z^2}{2m^*}\right)\phi$$

This is simply a Schrodinger equation for a harmonic oscillator with the minimum of the potential energy shifted from x = 0 to $x = x_0 = -k_y \ell^2$ where have defined the usual **magnetic length**,

$$\ell = \sqrt{\frac{\hbar c}{e^* B}}$$

Completing the square we obtain,

$$-\frac{\hbar^2}{m^*} \frac{d^2 \phi}{dx^2} + \frac{m^* \omega_c^2}{2} (x - x_0)^2 \phi = E \phi$$

Where we have the defined the following constants.

$$E = -a - \frac{\hbar^2 k_z^2}{2m^*} \qquad \omega_c = \frac{e^* B}{m^* c}$$

The eigenenergies of this Schrodinger equation will have the following form,

$$E_n = \hbar\omega_c \left(n + \frac{1}{2} \right)$$

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For this equation to have a solution, we must have $E = E_n$ for some $n \in \{1, 2, ...\}$. The largest value of B for which a solution exists is found from,

$$E_0 = \frac{\hbar \omega_c}{2} = E(k_z = 0) = -a \implies \frac{\hbar e^* B_{\text{max}}}{2m^* c} = -a \implies B_{\text{max}} = H_{c2} = -\frac{2am^* c}{\hbar e^*}$$

 H_{c2} can be expressed in terms of \mathcal{X} and H_c ,

$$H_{c2} = -\frac{2am^*c}{\hbar e^*} = \frac{H_c}{-a\sqrt{\frac{4\pi}{b}}} = \sqrt{2}\frac{m^*c}{\hbar e^*}\sqrt{\frac{b}{2\pi}}H_c$$

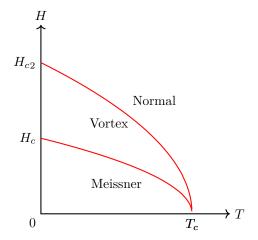
Using $n_s = |\varphi|^2 = -\frac{a}{b}$,

$$\mathcal{X} = \frac{\lambda}{\xi} = \sqrt{\frac{m^*c^2}{4\pi e^{*2}n_s}} \cdot \sqrt{\frac{-2m^*a}{\hbar^2}} = \sqrt{\frac{m^*c^2b}{4\pi e^{*2}(-a)}} \cdot \sqrt{\frac{-2m^*a}{\hbar^2}} = \frac{m^*c}{\hbar e^*} \sqrt{\frac{b}{2\pi}}$$

Thus we obtain,

$$H_{c2} = \sqrt{2} \mathcal{X} H_c$$

Which justifies the different behaviour of type I and II superconductors. When $\mathcal{X} > 1/\sqrt{2}$ we have $H_{c2} > H_c$.



In type II superconductors, there exists a superconducting state at fields $H > H_c$. In this state the field penetrates inside the superconductor, yet $|\varphi|$ is still non-zero and the sample is superconducting.

$$H_{c2} = -\frac{2am^*c}{\hbar e^*} = -\frac{2m^*a}{\hbar^2}\frac{\hbar c}{e^*} = \frac{\varphi_0}{2\pi\xi^2}$$

Where $\varphi_0 = hc/e^*$ is the flux quantum. Thus H_{c2} corresponds to one flux quantum per area $2\pi\xi^2$.

4.4 Josephson Effects

Let us imagine we have two pieces of a superconducting material that are initially detached from each other and each are described by their own wavefunctions $\phi_1(\mathbf{r}), \phi_2(\mathbf{r})$.

$$oxed{\phi_1(m{r})}$$

Furthermore assume that $|\phi_1| = |\phi_2| = \sqrt{n_s}$ (i.e. the two wavefunctions have the same norm but might having differing phases that are spatially dependent).

$$\phi_1(\mathbf{r}) = \sqrt{n_s}e^{i\theta_1(\mathbf{r})}$$
 $\phi_2(\mathbf{r}) = \sqrt{n_s}e^{i\theta_2(\mathbf{r})}$

Afterwards, suppose we bring the two materials together such that they become adjacent. It is possible to model this composite system simply by considering the tunneling between the two regions with amplitude t,

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$$oxed{\phi_1(m{r})} \phi_2(m{r})$$

Electrons in the first material $|1\rangle$ can tunnel to the second material $|2\rangle$ and vice versa via the following Hamiltonian,

$$H = -t(|1\rangle\langle 2| + |2\rangle\langle 1|) = \begin{pmatrix} 0 & -t \\ -t & 0 \end{pmatrix}$$

The composite wavefunction at the interface is $\phi(\mathbf{r}) = \frac{1}{\sqrt{2}}(\phi_1(\mathbf{r}) + \phi_2(\mathbf{r}))$ whose dynamics is modeled by the Schrodinger equation $H\phi = E\phi$. Because of ϕ , we have the following **supercurrent**,

$$\boldsymbol{j} = \frac{i\hbar e^*}{2m^*} (\phi^* \boldsymbol{\nabla} \phi - \phi \boldsymbol{\nabla} \phi^*)$$

Computing the gradient would be difficult in general because the phases θ_1, θ_2 are position dependent. Instead, we can liberally approximate the gradient across the interface as,

$$oldsymbol{
abla}\phipproxrac{\phi_2-\phi_1}{L}oldsymbol{\hat{x}}$$

Where L is the characteristic length of the interface in the \hat{x} -direction. Then we have,

$$\phi^* \nabla \phi - \phi \nabla \phi^* \propto (\phi_1^* + \phi_2^*)(\phi_2 - \phi_1) - (\phi_2 + \phi_1)(\phi_2^* - \phi_1^*)$$

$$= \phi_1^* \phi_2 - |\phi_1|^2 + |\phi_2|^* - \phi_2^* \phi_1 - |\phi_2|^2 + |\phi_1|^* + \phi_1^* \phi_2$$

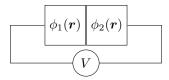
$$= 2(\phi_1^* \phi_2 - \phi_2^* \phi_1)$$

$$= 2n_s (e^{-i\theta_1} e^{i\theta_2} - e^{i\theta_1} e^{-i\theta_2})$$

$$= -4in_s \sin \theta \qquad \theta \equiv \theta_1 - \theta_2$$

This supercurrent is present even in the absence of a external electric field. If the two adjacent materials are superconductors, this current always flows across the junction on the absence of voltage. This current where $I = I_c \sin \theta$ is called the **DC-Josephson effect/current**.

Instead if we subject the two materials to an external potential V,



Then the system Hamiltonian becomes,

$$H = \begin{pmatrix} eV & -t \\ -t & -eV \end{pmatrix} \tag{4.10}$$

Where the dynamics of ϕ are governed by the Schrodinger equation $i\hbar \frac{\partial \phi}{\partial t} = H\phi$ which is explicitly,

$$i\hbar \frac{\partial \phi_1}{\partial t} = eV\phi_1 - t\phi_2$$

$$i\hbar \frac{\partial \phi_2}{\partial t} = -t\phi_1 - eV\phi_2$$
(4.11)

Instead of solving this set of equations directly, we can make use of the conjugate forms of eq. (4.11),

$$-i\hbar \frac{\partial \phi_1^*}{\partial t} = eV\phi_1^* - t\phi_2^*$$

$$-i\hbar \frac{\partial \phi_2^*}{\partial t} = -t\phi_1^* - eV\phi_2^*$$
(4.12)

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Multiplying eqs. (4.11) and (4.12) by factors of ϕ_1, ϕ_2 , we can sum the results and reconstruct the supercurrent,

$$i\hbar(\phi_2^*\phi_1 - \phi_1^*\phi_2) = i\hbar\left(\phi_1\frac{\partial\phi_2^*}{\partial t} + \phi_1^*\frac{\partial\phi_1}{\partial t} - \phi_2\frac{\partial\phi_1^*}{\partial t} - \phi_1^*\frac{\partial\phi_2}{\partial t}\right)$$

Substituting eq. (4.11),

$$i\hbar(\phi_{2}^{*}\phi_{1} - \phi_{1}^{*}\phi_{2}) = eV\phi_{2}^{*}\phi_{1} - t|\phi_{2}|^{2} + eV\phi_{2}^{*}\phi_{1} + t|\phi_{1}|^{2} + eV\phi_{1}^{*}\phi_{2} - t|\phi_{1}|^{2} + eV\phi_{1}^{*}\phi_{2} + t|\phi_{2}|^{2}$$

$$= 2eV(\phi_{2}^{*}\phi_{1} - \phi_{1}^{*}\phi_{2})$$

$$= 4eVn_{s}\cos\theta$$

$$= i\hbar\frac{\partial}{\partial t}(-2in_{s}\sin\theta)$$

$$= 2\hbar n_{s}\cos\theta\frac{\partial\theta}{\partial t}$$

Therefore we obtain the **Josephson relation** for θ ,

$$\frac{\partial \theta}{\partial t} = \frac{2eV}{\hbar}$$

If V is a constant, i.e. we apply a external DC voltage and the phase response becomes linear in time,

$$\phi(t) = \phi(0) + \frac{2eV}{\hbar}t$$

The responding current however is oscillatory,

$$I(t) = I_c \sin\left[\phi(0) + \frac{2eV}{\hbar}t\right]$$

This is referred to as the AC-Josephson effect. It is surprising to see an AC response from the application of a DC external voltage. The fact that the frequency of oscillation depends on the voltage V and fundamental constants $\frac{2e}{\hbar}$, measurements of this frequency represent our most precise measurement of $\frac{e}{\hbar}$. Analyzing the Josephson effect in this way is incomplete; we have failed incorporate the effects of resistance of the junction. If we subject the junction to a larger current than I_c we will have some residual effects,

- $I < I_c$: the current flows without resistance, $\implies V = 0$
- $I > I_c$: part of the current will become non-superfluid

This is referred to as a **resistively shunted Josephson junction**. The total current can be decomposed into the two effects,

$$I = I_c \sin \theta + \frac{V}{R}$$

Where $I_c \sin \theta$ is the supercurrent through the junction and $\frac{V}{R}$ is the dissipative current which is modeled as a current passing through a resistor with resistance R.

Therefore given that the time-derivative of the phase $\frac{\partial \theta}{\partial t}$ is governed by the external potential V,

$$V = \frac{\hbar}{2e} \frac{\partial \theta}{\partial t} \tag{4.13}$$

We conclude that,

$$I = I_c \sin \theta + \frac{\hbar}{2eR} \frac{\mathrm{d}\theta}{\mathrm{d}t}$$

which is a differential equation for $\theta(t)$ that can be solved.

$$\frac{\mathrm{d}\theta}{\mathrm{d}t} = \frac{2eR}{\hbar}(I - I_c \sin \theta) = \frac{2IeR}{\hbar} \left(1 - \frac{I_c}{I} \sin \theta\right)$$

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Let λ and ω_0 be the following parameters,

$$\lambda \equiv \frac{I_c}{I} < 1 \qquad \omega_0 = \frac{2IeR}{\hbar}$$

Where λ is dimensionless and ω_0 as units of frequency. Therefore we obtain the following non-linear differential equation,

$$\frac{\mathrm{d}\theta}{\mathrm{d}t} = \omega_0 (1 - \lambda \sin \theta)$$

Integrating this separable equation,

$$\omega_0 \cdot (t - t_0) = \int \frac{\mathrm{d}\theta}{1 - \lambda \sin \theta}$$

Since $\lambda < 1$ we have that the integrand never diverges and moreover, it can be solved analytically,

$$\tan\frac{\theta}{2} = \lambda + \sqrt{1 - \lambda^2} \tan\left[\frac{1}{2}\sqrt{1 - \lambda^2}\omega_0(t - t_0)\right]$$

Therefore,

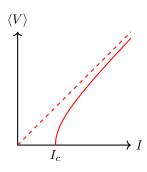
$$\theta(t) = 2 \tan^{-1} \left\{ \lambda + \sqrt{1 - \lambda^2} \tan \left[\frac{1}{2} \sqrt{1 - \lambda^2} \omega_0(t - t_0) \right] \right\}$$

Using eq. (4.13) we obtain a time-dependent voltage V(t) in response to a DC current,

$$V(t) = R \frac{I^2 - I_c^2}{I - I_c \cos \omega t} \qquad \omega = \frac{2e}{\hbar} R \sqrt{I^2 - I_c^2}$$

If instead we are interested in the time averaged voltage $\langle V \rangle$, we can calculate that as well,

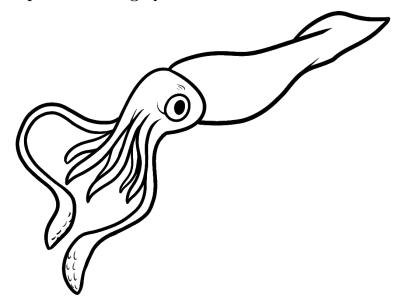
$$\langle V \rangle = \frac{w}{2\pi} \int_0^{\frac{2\pi}{\omega}} V(t) dt = \frac{\hbar\omega}{2e} = R\sqrt{I^2 - I_c^2}$$



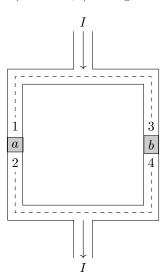
Evidently at high current I, the junction behaves classically.

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4.5 SQUID - Superconducting Quantum Interference Device



Suppose we had two Josephson junctions (labeled a, b) arranged in the following configuration,



It is possible to calculate the relative phase differences between the labeled points 1, 2, 3, 4 by integrating along the dashed paths,

$$egin{aligned} oldsymbol{j} &= -rac{\hbar e^* n_s}{m^*} igg(oldsymbol{
abla} heta + rac{e^*}{\hbar c} oldsymbol{A} igg) \ oldsymbol{
abla} heta &= -rac{e^*}{\hbar c} oldsymbol{A} \ heta_3 - heta_1 + heta_2 - heta_4 &= -rac{2e}{\hbar c} \oint oldsymbol{A} \cdot \mathrm{d} oldsymbol{\ell} \ arphi_a &= heta_2 - heta_1 & arphi_b &= heta_4 - heta_3 \end{aligned}$$

Therefore the difference of the two phase differences across the two junctions is,

$$\varphi_a - \varphi_b = -\frac{2e}{\hbar c}\phi = -2\pi\frac{\phi}{\phi_0}$$
 $\phi_0 \equiv \frac{\hbar c}{2e}$

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Therefore the currents across the junctions are,

$$I_a = I_c \sin \varphi_a$$
 $I_b = I_c \sin \varphi_b$

And the total interference becomes,

$$I = I_a + I_b = I_c(\sin\varphi_A + \sin\varphi_b) = 2I_c\sin\left(\frac{\varphi_a + \varphi_b}{2}\right)\cos\left(\frac{\varphi_a - \varphi_b}{2}\right)$$

Which corresponds to a macroscopic interference between the two junctions. But we also have that,

$$\varphi_b = \varphi_a + 2\pi \frac{\phi}{\phi_0}$$

Therefore,

$$I = 2I_c \sin\left(\frac{\pi\phi}{\phi_0}\right) \cos\left(\varphi_a + \frac{\pi\phi}{\phi_0}\right)$$

The maximal current through the SQUID for a given ϕ is therefore,

$$I_{\text{max}} = 2I_c \left| \cos \left(\frac{\pi \phi}{\phi_0} \right) \right|$$

This current acts as very sensitive magnetometer.

4.6 Josephson Junction Qubit

A Josephson junction already has a two-level behave depending on whether $I > I_c$ or $I < I_c$. We have already consider the effect of the supercurrent $I_c \sin \theta$ and the residual ordinary current $\frac{V}{R}$ but here is a final term that heretofore we have neglected; namely that the Josephson junction also acts as a capacitor,

$$I = I_c \sin \theta + \frac{V}{R} + C \frac{\mathrm{d}V}{\mathrm{d}t} \tag{4.14}$$

Where $V = \frac{\hbar}{2e} \frac{\mathrm{d}\theta}{\mathrm{d}t}$ still holds. Therefore eq. (4.14) can be written in terms of θ ,

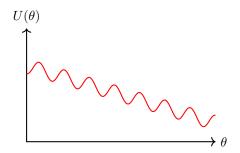
$$I = I_c \sin \theta + \frac{\hbar}{2eR} \frac{d\theta}{dt} + \frac{\hbar C}{2e} \frac{d^2\theta}{dt^2}$$

$$\frac{d^2\theta}{dt^2} + \frac{1}{RC} \frac{d\theta}{dt} + \frac{2eI_c}{\hbar C} \sin \theta = \frac{2eI}{\hbar C}$$
(4.15)

Above we considered the over-damped limit of the above equation where $\frac{1}{RC}$ is very large and we could neglect the second derivative term. Classically we can interpret θ as the coordinate of a classical particle. Doing so, we can correct assign a physical interpretation to the terms in eq. (4.15). Doing so means that eq. (4.15) acts as an equation of motion for a particle in a potential,

$$U(\theta) \sim -I\theta - I_c \cos \theta$$

Which is known as a washboard potential and can be visualized as a tilted cosine curve,



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When $I < I_c$, the particle is trapped in one of the minimum where $\frac{d\theta}{dt} = 0 \implies V = 0$. Alternatively if $I > I_c$ then the particle is moving down this potential and its motion gives rise to a finite voltage. For each of the potential wells, there will be a number of quantized energy levels that the particle can have. For sufficiently small junctions, it is possible to create a superposition between the lowest energy states of neighboring wells. This is called a **current-biased qubit**.

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