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Three-dimensional topological insulator quantum dot for optically controlled quantum memory and quantum computing

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We present the model of a quantum dot (QD) consisting of a spherical core-bulk heterostructure made of three-dimensional (3D) topological insulator (TI) materials, such as PbTe/Pb$_{0.31}$Sn$_{0.69}$Te, with bound massless and helical Weyl states existing at the interface and being confined in all three dimensions. The number of bound states can be controlled by tuning the size of the QD and the magnitude of the core and bulk energy gaps, which determine the confining potential. We demonstrate that such bound Weyl states can be realized for QD sizes of few nanometers. We identify the spin locking and the Kramers pairs, both hallmarks of 3D TIs. In contrast to topologically trivial semiconductor QDs, the confined massless Weyl states in 3D TI QDs are localized at the interface of the QD and exhibit a mirror symmetry in the energy spectrum. We find strict optical selection rules satisfied by both interband and intraband transitions that depend on the polarization of electron-hole pairs and therefore give rise to the Faraday effect due to the Pauli exclusion principle. We show that the semiclassical Faraday effect can be used to read out spin quantum memory. When a 3D TI QD is embedded inside a cavity, the single-photon Faraday rotation provides the possibility to implement optically mediated quantum teleportation and quantum information processing with 3D TI QDs, where the qubit is defined by either an electron-hole pair, a single electron spin, or a single hole spin in a 3D TI QD. Remarkably, the combination of interband and intraband transition gives rise to a large dipole moment of up to 450 Debye. Therefore, the strong-coupling regime can be reached for a cavity quality factor of $Q \approx 10^4$ in the infrared wavelength regime of around 10 $\mu$m.

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I. INTRODUCTION

Three-dimensional (3D) topological insulators (TIs) are narrow-bandgap materials with topologically protected gapless surface/interface states that are characterized by the linear spectrum of massless Weyl fermions. In such materials, the spins of the Kramers pairs are locked at a right angle to their momenta on the Fermi surface due to spin-orbit coupling, leading to suppression of backscattering from edges and nonmagnetic impurities. Such states are of great importance in low-power opto-spintronics.

Decoherence can be circumvented by highly polarized spin states with helical spin texture, leading to a phase coherence length of several hundred nanometers in nanostructures.

In 3D TI nanostructures the special properties of topologically protected surface states of TIs are amplified because of the large surface-to-volume ratio. In addition, the chemical potential can be electrically tuned using a gate voltage. For example, the coherent propagation of the Weyl electrons around the perimeter of a nanoribbon provides excellent evidence of the topological nature of the surface states in TI nanostructures. Experiments on both the physical and chemical synthesis of TI nanostructures have been done recently to understand their transport properties at the nanoscale.

Recently, in a TI quantum dot (QD) with tunable barriers based on ultrathin Bi$_2$Se$_3$ films, Coulomb blockade with around 5 meV charging energy was observed.

So far, a theoretical study of electronic properties of 2D helical states occurring at the nanoscale of 3D TIs, such as in QDs, is still lacking. In this article, we present the study of bound Weyl states that are confined at the interface of a spherical core-bulk heterostructure QD made of 3D TI materials such as Pb$_{1-x}$Sn$_x$Te. We show that at the interface massless Weyl fermions are confined in all three dimensions. The directions of spin and momentum are tangent to the surface of the QD. Remarkably, their inherent spin-momentum locking property exists even in a QD. Because of the linear dispersion there is a mirror symmetry in the energy spectrum between positive and negative energy states, in contrast to topologically trivial semiconductors. We demonstrate that this symmetry in energy spectrum is preserved for the QD spectrum.

Several methods have been proposed to implement optically controlled quantum memory and optically mediated quantum computing with topologically trivial QDs. Quantum memories have been recently reviewed in Ref. A recent review on optically controlled quantum computing with electron spins can be found in Ref. Optical control of single-electron spin memory has been experimentally demonstrated using GaAs QDs and InGaAs QDs. Exciton memory has been implemented experimentally in a semiconductor nanopost. For the purpose of using a hole spin as quantum memory or qubit, high coherence of hole spins in InGaAs QDs has been experimentally shown. Reference demonstrates experimentally that a single spin can be read out using Faraday rotation. Schemes for optically controlled two-qubit interaction have been proposed that are based on the exchange of virtual photons inside a cavity, the optical RKKY interaction, and dipole-dipole interaction. Substantial experimental progress has been made to implement optically controlled electron spin-state preparation, hole spin-state preparation, single-spin readout, dephasing protection, two-qubit gate, two-QD spin entanglement, and spin-photon entanglement.
In Refs. 43 and 44 we developed the method of the Faraday rotation of a single photon due to the Pauli exclusion principle occurring on a topologically trivial QD. Our proposed method can be used for entangling remote excitons, electron spins, and hole spins. We showed that this entanglement can be used for the implementation of optically mediated quantum teleportation and quantum computing. Our ideas and methods have been plagiarized in Ref. 45.

Here we show that classical and single-photon Faraday rotation due to the Pauli exclusion principle in a 3D TI QD occur due to strict optical selection rules satisfied by both interband and intraband transitions that depend on the polarization of electron-hole (e-h) pairs. Based on this finding we propose that 3D TI QDs can be used as quantum memory and for the implementation of optically mediated quantum teleportation and quantum computing. First, we propose that a single e-h pair in a 3D TI QD can be used as a quantum memory. The information is stored in form of the polarization state of the e-h pair. In order to be able to read out this information multiple times, we develop the method of Faraday rotation due to the Pauli exclusion principle in a 3D TI QD. This entanglement is the resource for the implementation of quantum teleportation and quantum computing.

In wide band gap semiconductor QDs optical interband and intraband transitions are energetically separated because the band gap is typically much larger than the QD level spacing. In contrast to that, we show that in 3D TI QDs interband and intraband transitions combine because of the vanishing band gap at band crossing. The resulting large dipole moment of a single e-h pair in a 3D TI QD can be used as a qubit to implement optically mediated quantum teleportation and quantum computing. We develop the method of Faraday rotation of a classical electromagnetic field due to Pauli exclusion principle in a 3D TI QD. Second, we propose that the polarization of a single e-h pair, a single electron spin, or a single hole spin can be used as a qubit in a 3D TI QD for the implementation of optically mediated quantum teleportation and quantum computing. We develop the method of single-photon Faraday rotation in a 3D TI QD, which creates the entanglement between a single photon and a qubit on the 3D TI QD. This entanglement is the resource for the implementation of quantum teleportation and quantum computing.

II. MODEL BASED ON DIRAC EQUATION

In Fig. 1 we show the model of our spherically symmetric 3D TI QD of a core-bulk structure with a single interface at radius \( r = r_0 \). This core-bulk structure consists, for example, of an inner core of PbTe and an outer bulk of Pb_{0.31}Sn_{0.69}Te with band gaps of 0.187 and 0.187 eV, respectively, or vice versa, so that Weyl fermions are generated at the interface. Here we used the band gap formula provided in Ref. 24 for determining \( x \). Note that the band crossing happens in Pb_{1-x}Sn_{x}Te at \( x = 0.35 \) at 4 K. The Weyl fermions are subjected to the spherically symmetric potential \( \Delta(r) \) [Fig. 1(b)].

To understand the properties of a 3D TI QD, we start with the Dirac Hamiltonian within the \( \mathbf{k} \cdot \mathbf{p} \) approximation. Neglecting the far band terms, we have

\[
H = v_1 \mathbf{\alpha} \cdot \hat{p} + v_1 \mathbf{\alpha} \cdot \hat{p} + \beta \Delta, \tag{1}
\]

where \( \mathbf{\alpha} = \begin{pmatrix} 0 & \sigma_1 \\ \sigma_2 & 0 \end{pmatrix} \) are the Dirac \( \mathbf{\alpha} \) matrices, \( \sigma \) are the Pauli matrices, \( \beta = \begin{pmatrix} 1 \\ 0 \end{pmatrix} \) is the Dirac \( \beta \) matrix, and \( \hat{p} \) is the momentum operator. The Fermi velocities \( v_1 \) and \( v_1 \) in angular and radial direction are determined by the \( v_1 = P_L / m_0 \) and \( v_1 = P_\perp / m_0 \) respectively, where \( P_L \) and \( P_\perp \) are the interband matrix elements. \( m_0 = 9.10938188 \times 10^{-31} \) kg is the free electron mass. \( \Delta(r) = E_r / 2 \) is the gap energy parameter.

Assuming spherical symmetry for the 3D TI QD, \( \Delta(r) \) depends on the radial coordinate only, which breaks the crystal symmetry in radial direction, and has the symmetry \( \Delta(r - r_0) = -\Delta(r - r_0) \), where \( r_0 \) is the radius of the QD. Therefore, the angular parts are separated from the radial part of the Dirac Hamiltonian (1). Thus, we can follow the derivation of the solution for the central-force problem of a hydrogen atom in relativistic quantum mechanics. The eigenfunctions of \( H \) are four-component spinors \( \Phi = [ \phi_+ \phi_- ]^{\top} \), where \( f_- \) and \( f_+ \) are the radial functions and \( \phi_+^{\top} \) and \( \phi_-^{\top} \) are the normalized spin-angular functions corresponding to the \( L_- \) and \( L_+ \) band, respectively, such as in Pb_{1-x}Sn_{x}Te. After eliminating the
angular parts, the radial part of the Dirac Hamiltonian (1) takes the form

\[ H = \left( \begin{array}{cc} \Delta(r) & -v_{\gamma}\left( \frac{d}{dr} + \frac{\gamma}{r} \right) \\ v_{\gamma}\left( \frac{d}{dr} + \frac{\gamma}{r} \right) & -\Delta(r) \end{array} \right), \tag{2} \]

where \( v_{\gamma} = 2.24 \times 10^5 \) m/s for \( \text{Pb}_{1-x}\text{Sn}_x\text{Te} \) and \( \kappa = \pm (j + \frac{1}{2}) \) is a nonzero positive or negative integer, \( j \) being the total angular momentum quantum number. For given \( \kappa \), it is known from relativistic quantum mechanics that the angular momenta \( l_+ \) and \( l_- \) for \( \phi_+ \) and \( \phi_- \) are determined by the relations \(-\kappa = j(j + 1) - l_-(l_+ + 1) + 1/4 \) and \( \kappa = j(j + 1) - l_+(l_- + 1) + 1/4 \), respectively. By solving \( H^2\Phi = \epsilon^2\Phi \), we obtain

\[ \left( r^2 \frac{d^2}{dr^2} + 2r \frac{d}{dr} \right) F_{\pm} = (\lambda^2 r^2 + \kappa(\kappa \pm 1)) F_{\pm} - \beta r^2 \frac{d\Delta}{dr} F_{\pm}, \tag{3} \]

where \( F_{\pm} = r f_{\pm} \), \( \beta = 1/v_{\gamma} \), and \( \lambda = \beta \sqrt{(\Delta_0^2 - \epsilon^2)} \). \( \lambda \) behaves like a wave vector \( k \) whose allowed quantized values determine the particle’s energy levels. In a flat geometry of a thin layer of a 3D TI, \( \Delta(z) \) can be chosen to be \( \Delta(z) = \Delta(\infty)\tanh(z/H) \). We adopt a similar potential along the radial direction of the form \( \Delta(r') = \Delta_0\text{sgn}(r' - r_0) \). Hence, the source term in Eq. (3) is \( f_{\pm}(r') = 2\Delta_0\beta f_{\pm}(r_0)r'\delta(r' - r_0) \). Equations (3) can be solved by using the corresponding differential equation for the Green’s function, i.e.,

\[ \left[ \frac{d}{dr} \left( r^2 \frac{d}{dr} \right) - (\lambda^2 r^2 + \kappa(\kappa \pm 1)) \right] G_{\pm} = \delta(r - r'). \tag{4} \]

The solutions regular at \( r = 0 \) with outgoing wave behavior at \( r \to \infty \) are the product of spherical modified Bessel functions of the order \( \kappa \) for \( G_- \) and of the order \( \kappa - 1 \) for \( G_+ \), i.e., \( G_-(r, r', \lambda) = C_- T_{\kappa}(\lambda r) K_{\kappa-1}(\lambda r) \), \( G_+(r, r', \lambda) = C_+ T_{\kappa}(\lambda r) K_{\kappa}(\lambda r) \), where \( r_\text{c} \) (\( r_+ \)) is the smaller (larger) of \( r \) and \( r' \). The functions \( T(\lambda r) \) and \( K(\lambda r) \) are, respectively, the first and the second kind of modified spherical Bessel functions, and \( C_{\pm} \) are the normalization constants. These constants are determined by the discontinuity in slope implied by the \( \delta \) function in Eq. (4). Integration is performed at the interface of the QD along the radial direction: \( [r \frac{dG_{\pm}}{dr}]_{r_+}^{r_\text{c}} = 0 \), where \( \eta \) is an infinitesimal quantity with \( \eta > 0 \). For \( r = r' + \eta, r_\text{c} = r, r_+ = r' \), and for \( r = r' - \eta, r_\text{c} = r', r_+ = r \). Consequently, the normalization constants are \( C_- = 1/\lambda \sqrt{\frac{\kappa}{\kappa^2 - 1}} W_{-\kappa} \) and \( C_+ = 1/\lambda \sqrt{\frac{\kappa}{\kappa^2 - 1}} W_{+\kappa} \), where \( W_{\pm\kappa} = [T_{\kappa}(\lambda r')K_{\kappa}(\lambda r) - T_{\kappa}(\lambda r)K_{\kappa}(\lambda r')]_{r_\text{c}}^{r_+} \) and \( W_{\pm\kappa} = [T_{\kappa}(\lambda r')K_{\kappa-1}(\lambda r) - T_{\kappa}(\lambda r)K_{\kappa-1}(\lambda r')]_{r_\text{c}}^{r_+} \) are the Wronskians of \( T(\lambda r) \) and \( K(\lambda r) \), respectively, for \( \kappa \) and \( \kappa - 1 \) order, and \( T(\lambda r) \) and \( K(\lambda r) \) are derivatives of the Bessel functions. The Wronskian of two linearly independent functions is proportional to \( 1/r^2 \) for Sturm-Liouville-type equations such as Eq. (4) (see the Appendix A). The solutions of Eqs. (3) are \( F_{\pm} = \int G_{\pm}(r, r', \lambda) F_{\pm}(r')dr' = 2\Delta_0\beta \int G_{\pm}(r, r', \lambda) F_{\pm}(r_0)r'\delta(r' - r_0)dr' \), i.e.,

\[ F_-(r) = 2\Delta_0\beta F_+(r_0)I_{\kappa}(\lambda r_0)K_{\kappa-1}(\lambda r_0)/\lambda W_{\kappa}, \tag{5} \]

\[ F_+(r) = 2\Delta_0\beta F_-(r_0)I_{\kappa-1}(\lambda r_0)K_{\kappa}(\lambda r_0)/\lambda W_{\kappa-1}, \tag{6} \]

where \( r_\text{c} \) (\( r_+ \)) is now the smaller (larger) of \( r \) and \( r_0 \). A transcendental equation is obtained by solving Eqs. (5) and (6). Evaluating at \( r = r_0 \) we obtain

\[ [zT_{\kappa}(z)K_{\kappa}(z)]/[zT_{\kappa-1}(z)K_{\kappa-1}(z)] = 1/4\Delta_0^2\beta^2 r_0^2, \tag{7} \]

where \( z = \lambda r_0 \). In Fig. 2, we show the plot of Eq. (7) where the function \( F(z) \) is defined as \( F(z) = [zT_{\kappa}(z)K_{\kappa}(z)]/[zT_{\kappa-1}(z)K_{\kappa-1}(z)] \).

### III. Bound States of the Weyl Fermions

Each term in the square bracket on the left-hand side of Eq. (7) is a monotonically decreasing function of \( z \) (for \( z > 0 \)), with maximum value of \( 1/(2\kappa + 1) \) for \( \kappa \) order term and \( 1/(2\kappa - 1) \) for \( \kappa - 1 \)th order term occurring at \( z = 0 \) (see the Appendix B). Therefore, their product has a maximum value of \( 1/(4\kappa^2 - 1) \) at \( z = 0 \) and is equal to \( 1/(4\Delta_0^2\beta^2 r_0^2) \). Since \( F(z) \) is a monotonically decreasing function, for each \( \kappa \), there is at most a single solution given by the intersection of \( F(z) \) with the constant \( 1/(4\Delta_0^2\beta^2 r_0^2) \) (dashed line and solid line in Fig. 2). The critical limit for having a single solution is determined by the intersection at the maximum value of \( F(z) \), which occurs at \( z = 0 \). This means that there exists a single solution of Eq. (7) for each \( \kappa \) as long as the condition \( 1/(4\Delta_0^2\beta^2 r_0^2) \leq 1/(4\kappa^2 - 1) \) is satisfied. Figure 2 shows the plot of the first three different values of \( \kappa \), \( \kappa = 1 \) (red), \( 2 \) (blue), and \( 3 \) (pink), each a monotonically decreasing line (solid line).

![FIG. 2. (Color online) Plot of Eq. (7) showing the intersections of the monotonically decreasing \( F(z) \) (solid lines) with the constants (dashed lines). Intersection at \( z = 0 \) gives the minimum threshold of the size of a QD to have two bound states, one positive- and one negative-energy state, for a given confining potential. For a larger QD, multiple bound states exist, corresponding to multiple intersection points. The intersection points A, B, and C are example points where we evaluate the wave functions. The energy of the bound states are determined by the relation \( z = \lambda r_0 \).](image-url)
cut by a horizontal line (dashed line) at most one time. Since 
\( \lambda = \beta \sqrt{\Delta_0^2 - \varepsilon^2} \), each single solution gives rise to two bound states with same magnitude but opposite sign of energy, giving rise to the mirror symmetry in the energy spectrum. Indeed, this makes sense since Weyl fermions are massless at zero band gap with the linear dispersion relation. Note that there is no radial quantum number because in general a Dirac potential allows only for a single positive-energy and a single negative-energy solution in radial direction.

As the size of the QD grows, it is filled with more and more bound states (see Fig. 2), where for smaller value of \( F(z) \) a horizontal dashed line makes multiple cuts at different values of the energy (i.e., \( z \)) for different \( \kappa \). For negative \( \kappa \), the solutions diverge at the origin and are therefore physically not valid. This result has profound implications because the sign of \( \kappa \) determines whether \( j \) is parallel or antiparallel to the spin \( s \). Since \( \kappa \) is only allowed to be positive, only one spin orientation with respect to \( j \) is permitted. This corresponds to the spin locking effect, which is a hallmark of 3D TIs. This allows us to write the solutions in the equation of spin-angular functions, i.e.,

\[
\varphi_{\beta,\gamma}^m = \sqrt{\frac{l_+ - m + \frac{1}{2}}{2l_+ + 1}} \gamma_{l_+}^m \begin{bmatrix} 1 \\ 0 \\ \end{bmatrix}
\]

\[
\varphi_{\beta,\gamma}^m = \sqrt{\frac{l_+ + m + \frac{1}{2}}{2l_+ + 1}} \gamma_{l_+}^m \begin{bmatrix} 1 \\ 0 \\ \end{bmatrix}
\]

(8)

where \( l_- = j + \frac{1}{2} \) and \( l_+ = j - \frac{1}{2} \).

The condition \( 1/4\Delta^2 \beta^2 r_o^2 = 1/(4\kappa^2 - 1) \) determines the lower limit of the size of the QD to hold two bound interface states, a positive- and a negative-energy state, for a given value of the confining potential strength. The critical QD size depends on the Fermi velocities and band gaps of the 3D TI materials. In Pb_{1-x}Sn_xTe, \( \Delta_v = 0.0935 \) eV, half of the band gap of PbTe. Choosing \( v_1 = 2.24 \times 10^6 \) m/s\(^5\) results in a critical QD size of \( r_0 = 1.4 \) nm for \( \kappa = 1 \) at \( z = 0 \). Similarly for \( \kappa = 2 \) at \( z = 0 \), the critical QD size for Pb_{1-x}Sn_xTe is \( r_0 = 3 \) nm. The energies of the bound states are determined from \( z = \lambda r_o \), which gives very shallow energy levels of \( \varepsilon = \pm \Delta_0 \) for \( z = 0 \).

For a given value of \( \kappa \), quantum numbers characterizing the wave functions \( \phi_- \) and \( \phi_+ \) can be determined. For \( \kappa = 1, 2, 3 \), and 4, the possible combinations of the quantum numbers are shown in Table I for both spinors \( \phi_- \) and \( \phi_+ \). Here we observe that the \( \phi_- \) component is characterized by the spin being antiparallel to its angular momentum, whereas the \( \phi_+ \) component is characterized by the spin being parallel to its angular momentum. We show now how to identify the Kramers pairs. According to Kramers theorem, which applies to a time-reversal invariant system, a spin 1/2 state is at least twofold degenerate on the surface of a 3D TI. Hence, we obtain the following examples of Kramers pairs. For \( \kappa = 1 \), the four-spinor state with \( m_z = \frac{1}{2} \),

\[
\Phi_{\frac{1}{2};\frac{1}{2}} = \begin{bmatrix} f_-(r)Y_{1}^{0}[0] \\ if_+(r)Y_{0}^{0}[1] \\ \end{bmatrix}
\]

(10)

has as Kramers partner the four-spinor state with \( m_z = -\frac{1}{2} \),

\[
\Phi_{\frac{1}{2};-\frac{1}{2}} = \begin{bmatrix} f_-(r)Y_{1}^{-1}[0] \\ if_+(r)Y_{0}^{0}[1] \\ \end{bmatrix}
\]

(11)

For \( \kappa = 2 \), the four-spinor state with \( m_z = \frac{3}{2} \),

\[
\Phi_{\frac{3}{2};\frac{3}{2}} = \begin{bmatrix} f_-(r)Y_{2}^{0}[0] \\ if_+(r)Y_{1}^{1}[1] \\ \end{bmatrix}
\]

(12)

has as Kramers partner the four-spinor state with \( m_z = -\frac{3}{2} \),

\[
\Phi_{\frac{3}{2};-\frac{3}{2}} = \begin{bmatrix} f_-(r)Y_{2}^{-2}[0] \\ if_+(r)Y_{1}^{-1}[1] \\ \end{bmatrix}
\]

(13)
For $\kappa = 2$, the four-spinor state with $m_{f} = \frac{1}{2}$,

\[
\Phi_{\frac{1}{2}}^{m_{f}} = \begin{bmatrix}
  f_{-}(r)Y_{\frac{1}{2}}^{1}
  \\
  if_{+}(r)Y_{\frac{1}{2}}^{-1}
\end{bmatrix},
\]

is given by

\[
\begin{align*}
 f_{-}(r)(-\sqrt{\frac{2}{3}}Y_{2}^{-1}[1] & + \sqrt{\frac{2}{3}}Y_{0}^{-1}[0]) \\
 if_{+}(r)(\sqrt{\frac{2}{3}}Y_{1}^{-1}[1] & + \sqrt{\frac{2}{3}}Y_{0}^{-1}[1])
\end{align*}
\] (14)

has as Kramers partner the four-spinor with $m_{f} = -\frac{1}{2}$.

\[
\Phi_{-\frac{1}{2}}^{m_{f}} = \begin{bmatrix}
  f_{-}(r)Y_{\frac{1}{2}}^{-1}
  \\
  if_{+}(r)Y_{\frac{1}{2}}^{1}
\end{bmatrix},
\]

is given by

\[
\begin{align*}
 f_{-}(r)(-\sqrt{\frac{2}{3}}Y_{2}^{-1}[1] & + \sqrt{\frac{2}{3}}Y_{0}^{-1}[0]) \\
 if_{+}(r)(\sqrt{\frac{2}{3}}Y_{1}^{-1}[1] & + \sqrt{\frac{2}{3}}Y_{0}^{-1}[1])
\end{align*}
\] (15)

In general, the number of Kramers pairs is determined by the spin multiplicity for each $m_{j}$ value.

In Figs. 3 and 4 we show the spatial wave functions of the $f_{-}$ and $f_{+}$ components inside and outside the QD made of the core-bulk heterostructure PbTe/Pb$_{0.31}$Sn$_{0.69}$Te. Figure 3 shows the example of the intersection point A and Fig. 4 shows the example of the intersection points B and C (points A, B, and C are shown in Fig. 2). Since the four-spinors must be continuous at the boundary, also each of the two-spinor components must be continuous, i.e., $f_{-}^{\text{in}} = f_{-}^{\text{out}}$ and $f_{+}^{\text{in}} = f_{+}^{\text{out}}$ at the QD surface. The horizontal solid and short dashed lines in the figures represent the energy eigenvalues, respectively, at the intersection point A, corresponding to $r_{0} = 2$ nm, and at the intersection point B and C, corresponding to $r_{0} = 3.5$ nm. Eigenvalues are $\epsilon_{\pm} = \pm 0.80\Delta_{0}$ at point A, $\epsilon_{\pm} = \pm 0.91\Delta_{0}$ at point B, and $\epsilon_{\pm} = \pm 0.48\Delta_{0}$ at point C.

In order to show that the solutions correspond to Weyl fermions, we perform an expansion of Eq. (7) for large $z$ to obtain the energy eigenvalues in the continuum limit. Using

\[
\frac{1}{2z} \left[ 1 - \frac{2\kappa(\kappa + 1)}{(2z)^{2}} \right] \times \frac{1}{2z} \left[ 1 - \frac{2\kappa(\kappa - 1)}{(2z)^{2}} \right] = 1/4\Delta_{0}^{2}\beta^{2}r_{0}^{2}.
\] (16)

This can be written as

\[
\epsilon^{4} - \epsilon^{2}\Delta_{0}^{2} + \frac{\Delta_{0}^{2}\kappa^{2}}{\beta^{2}r_{0}^{2}} = 0,
\] (17)

which results in the energy eigenvalues for the electron and hole,

\[
\epsilon_{\pm} = \pm \kappa \sqrt{\Delta_{0}^{2}r_{0}^{2}}.
\] (18)

This corresponds to the linear spectrum of free massless Dirac fermions, i.e., free Weyl fermions on a sphere. This means that the energy splittings between the trapped Weyl states in the quantum dot result from the confinement of the Weyl fermions on a sphere. The solution in Eq. (18) corresponds to the eigenspectra found in Ref. 51 for zero magnetic field and without quantum confinement effects.

In the continuum limit, the Nielsen-Ninomiya fermion doubling theorem is satisfied by the pairs of Dirac cones positioned at antipodal points of the sphere defined by the surface of the QD (see Appendix C for details). However, for a general finite QD radius $r_{0}$ the eigenstates are bound and have a discrete energy spectrum. Since the Nielsen-Ninomiya fermion doubling theorem is valid only for continuum states, it does not apply to the bound Weyl fermions in a 3D TI QD with finite radius $r_{0}$. 
IV. OPTICAL EXCITATIONS

The $k \cdot p$ Hamiltonian contains also a quadratic term in the momenta, namely

$$H_q = \begin{pmatrix} \frac{(p_+ + eA_\perp)^2}{2m_\parallel} + \frac{(p_+ + eA_\parallel)^2}{2m_\parallel} & 0 \\ 0 & \frac{(p_+ + eA_\perp)^2}{2m_\perp} + \frac{(p_+ + eA_\parallel)^2}{2m_\perp} \end{pmatrix}.$$  \hspace{1cm} (19)

where $A = (A_\perp, A_\parallel)$ is the vector potential, $E = \partial A/\partial t$ in the Coulomb gauge, and we made use of the equivalence between $(e/m)A \cdot p$ and $-e\vec{r} \cdot \vec{E}$. We identify the interaction Hamiltonian as

$$H_{\text{int}} = ev_1\sigma_{zA} + ev_1\sigma_{\perp A} \cdot A_\perp - e\vec{r} \cdot \vec{E}$$

$$= \begin{pmatrix} -e\vec{r} \cdot \vec{E} & ev_1\sigma_{zA} + ev_1\sigma_{\perp A} \cdot A_\perp \\ ev_1\sigma_{\perp A} + ev_1\sigma_{\parallel A} \cdot A_\parallel & -e\vec{r} \cdot \vec{E} \end{pmatrix}.$$  \hspace{1cm} (20)

It will turn out that both interband and intraband transitions contribute. It is important to note that $v_1 = P_1/m_0$ and $v_1 = P_2/m_0$ include the Kane interband matrix elements $P = (u_{k_1}^\dagger, P u_{k_0}^\dagger)$, where $u_{k_i}^\dagger$ are the Bloch’s functions for the $L^\parallel$ bands. This means that the interband transitions are governed by the interband Hamiltonian $H_{\text{inter}} = ev_1\sigma_{zA} + ev_1\sigma_{\perp A} \cdot A_\perp$, where the Dirac $\alpha$ matrices couple the $L^\perp$ band with the $L^\parallel$ band. The Hamiltonian $H_{\text{intra}} = -e\vec{r} \cdot \vec{E}$ accounts for intraband transitions with $\vec{P}$ operating on the envelope wave functions only. $H_{\text{intra}}$ is proportional to the identity in four-spinor space and therefore couples the $L^\perp$ band to itself and the $L^\parallel$ band to itself. Thus the interband Hamiltonian $H_{\text{inter}}$ and the intraband Hamiltonian $H_{\text{intra}}$ are not equivalent in this description. On the one hand, $H_{\text{inter}}$ gives rise to interband transitions because it contains the Kane interband matrix elements $P_1$ and $P_2$. On the other hand, $H_{\text{intra}}$ gives rise to intraband transitions because the electric dipole operator $e\vec{r}$ operates on the envelope wave functions. Remarkably, both terms lead to the same strict optical selection rules and add up to a combined optical matrix element, as shown below. This enhancement of the optical matrix element is a feature of the 3D TI QD. In contrast, in a wide-band-gap semiconductor QD the interband and intraband transitions are energetically separated, i.e., interband transitions occur typically around the band-gap energy, whereas intraband transitions occur around the energy level separation due to the confinement of the QD.

Figure 5 shows the possible transitions between the states $\kappa = 1$ and $\kappa = 2$. It is to be noted that there is a complete symmetry in the solutions in the sense that a $\kappa$ state can be chosen from either the positive- or the negative-energy solutions. The optical matrix elements are given by

$$\langle \phi_f | H_{\text{inter}} | \phi_i \rangle = ev_1 \langle \phi_f | \sigma_{zA} | \phi_i \rangle + ev_1 \langle \phi_f | \sigma_{\perp A} \cdot A_\perp | \phi_i \rangle \cdot A_\perp$$

$$- e \langle \phi_f | \vec{r} | \phi_i \rangle \cdot \vec{E}.$$  \hspace{1cm} (22)

The incoming photon’s wavelength is much larger than the dot size. Therefore, the transitions are vertical, which means $A = (A_{x0}, A_{y0}, A_{z0})e^{i\vec{q} \cdot \vec{r}} \approx (A_{x0}, A_{y0}, A_{z0})$ can be used, yielding the electric dipole approximation. The transition energies $\hbar \omega_0 = \epsilon_{k=2} - \epsilon_{k=1}$ are large compared with the room temperature $k_B T = 25$ meV and the Coulomb charging energy of about 5 meV. For the control of the number of electrons and holes in the 3D TI QD it is necessary to work at low temperatures of around 1 K.

As an example, here we consider transitions between the states $\kappa = 1$ (at point C) and $\kappa = 2$ (at point B). The matrix elements of the Dirac-$\alpha$ matrix are given by

$$\langle \Phi_f | \alpha | \Phi_i \rangle = \langle \phi_{\kappa=2}^+ | \sigma^+ | \phi_{\kappa=1}^- \rangle + \langle \phi_{\kappa=2}^- | \sigma^- | \phi_{\kappa=1}^+ \rangle.$$  \hspace{1cm} (23)

The matrix elements of $\vec{r}$ are given by

$$\langle \phi_f | \vec{r} | \phi_i \rangle = \langle \phi_{\kappa=2}^+ | \vec{r} | \phi_{\kappa=1}^- \rangle + \langle \phi_{\kappa=2}^- | \vec{r} | \phi_{\kappa=1}^+ \rangle.$$  \hspace{1cm} (24)

The spherical harmonics can be determined using the Table I. In order to obtain optical selection rules for circular polarizations, it is useful to express the scalar products of the interband and the intraband Hamiltonian in the form...
\[ e \cdot \alpha = e, \alpha_x + e, \alpha_x + e, \alpha_- \text{ and } e \cdot \hat{r} = e, \hat{r}_x + e, \hat{r}_x + e, \hat{r}_- \]

respectively, where \( e_\pm = (e_x \pm ie_y) / \sqrt{2} \) are the unit vectors of circular polarizations, \( \alpha_\pm = (\alpha_x \pm i \alpha_y) / \sqrt{2} \), and \( \hat{r}_\pm = (\hat{r}_x \pm i \hat{r}_y) / \sqrt{2} \). Using our spinor states \( | \phi_\pm \rangle \) and radial wave function functions \( f_r \), we obtain the following nonzero matrix elements for \( \alpha \):

\[
\langle \phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha_+ \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = 2 \sqrt{2} \left( f_r + f_r^* \right),
\]

(25)

\[
\langle \phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha_- \phi^{\pm 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\pm 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(26)

For \( \hat{r} \) we obtain the following nonzero matrix elements:

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r} \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(27)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha_- \phi^{\pm 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\pm 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = -2 f_r f_r^*.
\]

(28)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha_+ \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(29)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r} \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(30)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(31)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r} \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(32)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(33)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r} \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \alpha \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(34)

\[
\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle \equiv \langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle = \frac{2}{3} \left( f_r + f_r^* \right),
\]

(35)

where \( \sigma_+ = (\alpha_x - i \alpha_y) / \sqrt{2} = (0, \sqrt{2}) \), \( \sigma_- = (\alpha_x + i \alpha_y) / \sqrt{2} = (0, \sqrt{2}) \), and the normalization and orthogonality condition

\[
\langle \Omega | \Phi^{\pm 2} | \phi^{\mp 1} \rangle = \delta_{\Omega} \delta_{\phi \Omega} \text{have been used. All other matrix elements are zero.}
\]

The transition energy difference between the states \( \kappa = 1 \) (at point C) and \( \kappa = 2 \) (at point B) is 0.43\( \Delta_0 \) and 1.39\( \Delta_0 \) within the same energy solution and between the negative- and positive-energy solutions, respectively (see Fig. 4). For \( \Delta_0 = 93.5 \) meV (half of the band gap of PbTe), the corresponding wavelengths are 31 \( \mu \)m and 9.5 \( \mu \)m. Consider the transitions as shown in Fig. 5. We find that the \( z \) component of the matrix element gives rise to \( \pi \) transitions with \( \kappa = 2, m_j = 1/2 \leftrightarrow \kappa = 2, m_j = -1/2 \) and with \( \kappa = 2, m_j = -1/2 \leftrightarrow \kappa = 1, m_j = -1/2 \). Thus, these \( \pi \) transitions are coupled to light polarized linearly in \( z \) direction. The \( x - iy \) and \( x + iy \) components of the matrix element give rise to the \( \sigma^+ \) transition with \( \kappa = 2, m_j = 1/2 \leftrightarrow \kappa = 1, m_j = 1/2 \) and with \( \kappa = 2, m_j = -1/2 \leftrightarrow \kappa = 1, m_j = -1/2 \). Thus, these \( \sigma^+ \) transitions are coupled to the components of the corresponding circular polarization of light. We can take advantage of these strict optical selection rules to implement the semiclassical and quantum Faraday effect showed below.

The overlap integrals \( \langle f_r, \kappa = 2 | f_r, \kappa = 1 \rangle \) and \( \langle f_r, \kappa = 2 | f_r, \kappa = 1 \rangle \) for the transitions between the points B and C are evaluated to be 0.31 and 0.24, respectively. The Kane energy, \( E_p = 2 P_k^2 / m_0 \), is calculated to be 7.3 eV, which is about three times smaller than the Kane energy value of 22.7 eV for GaAs.\(^{53,54}\) The smaller Kane energy here is due to the fact that the Fermi velocity is an order of magnitude smaller than the Fermi velocity in GaAs. The polarization matrix elements of \( \hat{r} \) accounts for the strength of the in-plane intraband transitions at the band crossing. We calculate the magnitude of the matrix elements for \( \sigma^+ \) transitions and find that \( e |\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r}_- \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle | = 128 \) Debye and \( e |\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r}_+ \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle | = 221 \) Debye. For the \( \pi \) transitions we find the magnitude of the matrix elements as, \( e |\langle \Phi^{\pm 2}|_{\frac{i}{2} \frac{1}{2}} | \hat{r}_- \phi^{\mp 1}|_{\frac{i}{2} \frac{1}{2}} \rangle | = 181 \) Debye.
V. FARADAY EFFECT FOR 3D TI QDS

In Refs. 43,44, and 55–57 we showed that the single-photon Faraday rotation cannot only be used for quantum spin memory but also for quantum teleportation and quantum computing with wide-band-gap semiconductor QDs. In Ref. 58 we showed that the conditional Faraday rotation can be used for optical switching of classical information. In Ref. 59 we proposed a single-photon Mach-Zehnder interferometer for quantum networks based on the single-photon Faraday effect. In Ref. 60 a single spin in a wide-band-gap semiconductor QD was detected using the Faraday rotation. In order to implement these applications with 3D TI QDs, we need strict optical selection rules for the circular polarization of the photons. Since, indeed, for 3D TI QDs we obtain strict optical selection rules for circular polarization of photons, we suggest that it is possible to implement quantum memory, quantum teleportation, and quantum computing using the single-photon Faraday rotation in 3D TI QDs. In order to prove this conjecture, we derive the Faraday effect for 3D TI QDs. For the derivation of the Faraday effect for a classical laser beam due to Pauli exclusion principle we are going to follow in Sec. VII we are going to derive also the Faraday effect for 3D TI QDs.

In order to simplify the notation, we write the light-matter interaction Hamiltonian as

\[ H_{\text{int}} = \frac{e P E_0}{\hbar (\epsilon_0 - \epsilon r)} \left( e^{-i (\epsilon_0 - \epsilon r) t} + e^{-i (\epsilon_0 - \epsilon r) t} \right) e \cdot \alpha. \]

where \( P = m_0 V \) is the Kane interband matrix element. The transition rate for a single 3D TI QD can then be calculated using Fermi’s golden rule,

\[ W_{fi} = \frac{2\pi}{\hbar} \left( \epsilon E_0 \right)^2 \left| \langle \Phi_f | \frac{P}{i m_0 \omega} e \cdot \alpha + e \cdot \hat{r} | \Phi_i \rangle \right|^2 \times f(\epsilon_f) [1 - f(\epsilon_f)] \delta(\epsilon_f - \epsilon_i - \hbar \omega). \]  

where \( f(\epsilon) = \exp(\frac{-\epsilon}{k T}) + 1 \) is the Fermi-Dirac distribution function, \( \epsilon_f \) is the Fermi energy, \( |\Phi_i\rangle \) denotes the initial Weyl state, \( |\Phi_f\rangle \) denotes the final Weyl state, and the minus (−) sign in front of \( \hbar \omega \) corresponds to absorption and the plus (+) sign to emission. Thus, the absorption energy of photon per spin state is \( P = \hbar \omega \sum_{i,f} W_{fi} \). Comparing with the total power \( P = 2\sigma V E_0^2 \) dissipated in the system of volume \( V \), where \( \sigma = \sigma_1 + i \sigma_2 \) is the complex conductivity, and including absorption and emission, it follows that the real part of the conductivity is

\[ \sigma_1 = \frac{\pi e^2 \omega}{V} \sum_{i,f} \left| \langle \Phi_f | \frac{P}{i m_0 \omega} e \cdot \alpha + e \cdot \hat{r} | \Phi_i \rangle \right|^2 \times [f(\epsilon_f) - f(\epsilon_f)] \delta(\epsilon_f - \epsilon_i - \hbar \omega), \]

which can be written in terms of the oscillator strengths \( f_{ji} = \frac{(2 m_0 \omega f_j / \hbar)|\langle \Phi_f | \frac{P}{i m_0 \omega} e \cdot \alpha + e \cdot \hat{r} | \Phi_i \rangle|^2}{\hbar \omega (\omega^2 - \alpha_j^2)} \).

\[ \sigma_1(\omega) = \frac{\pi e^2}{2 m_0 V} \sum_{i,f} f_{ji} f(\epsilon_j) - f(\epsilon_f) \delta(\epsilon_f - \epsilon_i - \hbar \omega). \]  

(39)

Using the relation \( \epsilon_\sigma = 1 + \frac{\epsilon_0}{\epsilon_\sigma} \), where \( \epsilon_0 \) is the free-space permittivity, between the complex conductivity and the complex dielectric function \( \epsilon_\sigma = \epsilon_1 + i \epsilon_2 \) and taking advantage of the Kramers-Kronig relations the complex dielectric function is given by

\[ \epsilon_\sigma(\omega) = 1 - \frac{\epsilon_0^2 e^2}{\epsilon_0 m_0 V} \sum_{i,j} f_{ji} [f(\epsilon_j) - f(\epsilon_f)] \left( \omega^2 - \alpha_j^2 \right) + i \gamma \omega. \]  

(40)

In order to describe the Faraday rotation, we need to consider only the states \(|\Phi_{\pi+}^{\pm1}\rangle\), \(|\Phi_{\pi-}^{\pm1}\rangle\), and \(|\Phi_{\sigma+}^{\pm2}\rangle\) coupled by circularly polarized light (see Fig. 5). We denote their energy difference by \( \hbar \omega_0 = \epsilon_{\pi+2} - \epsilon_{\pi+1} \). Defining the the quantity

\[ M_{f,i} = \left| \langle \Phi_{\sigma+}^{\pm2} | \frac{P}{i m_0 \omega} e \cdot \alpha + e \cdot \hat{r} | \Phi_{\pi+}^{\pm1} \rangle \right|^2 \]  

(41)

we can rewrite the complex dielectric function as

\[ \epsilon_\sigma(\omega) = \epsilon_{QD}(\omega) - \frac{2 e^2 \omega_0}{\epsilon_0 \hbar} \left( M_{\pi+;\pi-}^{\pm1} + M_{\pi-;\pi+}^{\pm1} \right) \left( \omega^2 - \omega_0^2 \right) + i \gamma \omega \]  

(42)

where \( \Delta_S \) is the Stark energy shift (see below) and \( \gamma \) is the line broadening. Summation over the other states is included in \( \epsilon_{QD}(\omega) \), which is the dielectric function of Pb0.63Sn0.37Te, corresponding to the material at the interface. \( \rho = 1 / V \) is the 3D TI QD density. This expression can be split into a component of the dielectric function for the \( \sigma^+ \) polarization,

\[ \epsilon_+(\omega) = \epsilon_{QD}(\omega) - \frac{2 e^2 \omega_0}{\epsilon_0 \hbar} \left( M_{\pi+;\pi-}^{\pm1} + M_{\pi-;\pi+}^{\pm1} \right) \left( \omega^2 - \omega_0^2 \right) + i \gamma \omega \]  

(43)

and a component of the dielectric function for the \( \sigma^- \) polarization,

\[ \epsilon_-(\omega) = \epsilon_{QD}(\omega) - \frac{2 e^2 \omega_0}{\epsilon_0 \hbar} \left( M_{\pi+;\pi-}^{\pm1} + M_{\pi-;\pi+}^{\pm1} \right) \left( \omega^2 - \omega_0^2 \right) + i \gamma \omega \]  

(44)

Consequently, the indices of refraction for \( \sigma^\pm \) polarization are given by \( n_{\pm} = \sqrt{\epsilon_{\pm}} \). Assuming that the length of the material is \( L \), the Faraday rotation can now be understood by considering the electric component of the plane wave after passing through...
the material at position \( z = L \),
\[
\mathbf{E}(z = L) = \frac{E_0}{\sqrt{2}} (e^{i k_x t} \mathbf{e}_x + e^{i k_y t} \mathbf{e}_y) e^{-i \omega t}
\]
\[
= E_0 \left( \cos \frac{\Delta n \omega L}{c} \mathbf{e}_x + \sin \frac{\Delta n \omega L}{c} \mathbf{e}_y \right) \times e^{i (k_x L - \omega t + (a - 1) \frac{\pi}{4})},
\]  
(45) 
where \( \mathbf{e}_x = (e_x + i e_y)/\sqrt{2} \) are the circular polarization unit vectors, \( n = (n_+ + n_-)/2 \) is the average index of refraction, \( c \) is the speed of light in vacuum, and \( \Delta n = n_+ - n_- \) is the difference in index of refraction between \( \sigma^+ \) and \( \sigma^- \) polarization. Thus, the Faraday rotation angle is given by
\[
\vartheta = \frac{\Delta n \omega L}{2c}.
\]  
(46) 
This formula shows that the Faraday rotation angle depends on the populations of the states \( |\Phi^{x=2}_{m=\pm 1}\rangle \), \( |\Phi^{x=2}_{m=\pm 2}\rangle \), and \( |\Phi^{x=2}_{m=\pm \frac{3}{2}}\rangle \), as determined by the Fermi functions, which can be used in the quasiequilibrium, i.e., when the time is much smaller than the electron-hole recombination time. A similar Faraday effect has already been successfully used to experimentally detect a single spin inside a GaAs QD.

VI. QUANTUM MEMORY WITH 3D TI QDS

Let us first describe the quantum memory with 3D TI QDs. In order to obtain the maximum Faraday effect, it is possible to apply an oscillating electric field \( \mathbf{E}(t) \) pointing in \( z \) direction, which splits the \( |\Phi^{x=2}_{m=\pm 1}\rangle \) states from the \( |\Phi^{x=2}_{m=\pm 2}\rangle \) states due to the optical Stark effect (see Fig. 6). The coupling to the electric field is described by the relativistic Stark Hamiltonian
\[
H_S = \begin{pmatrix}
-ezE_z e^{i \omega t} & ev_1 \sigma_z A_z \\
ev_1 \sigma_z A_z & -ezE_z e^{-i \omega t}
\end{pmatrix},
\]  
(47) 
where \( E_z(t) = E_S (e^{i \omega t} + e^{-i \omega t}) \) and thus \( A_z(t) = \frac{E_S \omega}{m_e} (e^{i \omega t} - e^{-i \omega t}) \). In second-order perturbation theory we obtain the quadratic Stark effect. The only nonzero contributions come from the matrix element coupling the \( |\Phi^{x=2}_{m=\pm 1}\rangle \) state to the \( |\Phi^{x=2}_{m=\pm 2}\rangle \) state, and from the matrix element coupling the \( |\Phi^{x=2}_{m=\pm \frac{3}{2}}\rangle \) state to the \( |\Phi^{x=2}_{m=\pm 1}\rangle \) state. This yields the Stark energy shift
\[
\Delta_S = e^2 E_S^2 \frac{P_{m=\pm 1} |\Phi^{x=2}_{m=\pm 1}\rangle (\sigma_z + z |\Phi^{x=2}_{m=\pm 2}\rangle)}{\hbar (\omega - \omega_S)}
\]
\[
= e^2 E_S^2 \frac{P_{m=\pm 1} |\Phi^{x=2}_{m=\pm 1}\rangle (\sigma_z + z |\Phi^{x=2}_{m=\pm 2}\rangle)}{\hbar (\omega - \omega_S)}.
\]  
(48) 
The Stark energy shift can be determined by applying an oscillating electric field whose amplitude is measured along \( z \) direction. The amplitude of the electric field can be calculated as \( |E_S| = \sqrt{2 S n}/\lambda e_0 c \), where \( S \) is the power of the laser, \( n \) is the index of refraction of the medium through which the light propagates and \( A \) is the area of the aperture of the laser source. A laser power of 50 mW with energy \( \hbar \omega_S = 30 \text{ meV} \) and area of the aperture of 1 \( \mu \text{m}^2 \) in a medium with \( n = 5.7 \) (for \( \text{Pb}_{0.68}\text{Sn}_{0.32}\text{Te} \) at room temperature) can produce an electric field of 1.46 \( \times 10^7 \text{ V/m} \). Using the Fermi velocity of \( v_1 = 2.24 \times 10^5 \text{ m/s} \) to calculate \( P_c \), our calculations show that the matrix element in Eq. (48) is \( e^2 |\langle \Phi^{x=2}_{m=\pm 1} | P_{m=\pm 1} (\sigma_z + z |\Phi^{x=2}_{m=\pm 2}\rangle)| = 410 \text{ Debye} \). With the transition energy difference \( \hbar \omega_0 = 130 \text{ meV} \) we get a Stark energy shift of \( \Delta_S = 14 \text{ meV} \).

It has already been shown experimentally that single-electron loading is possible in 3D TI QDs. We focus on two possible level configurations due to the electron-hole symmetry in 3D TI QDs:

(i) Figure 6 shows the first level configuration where the electron states are given by the \( |\Phi^{x=2}_{m=\pm 1}\rangle \) and the hole states are given by the \( |\Phi^{x=2}_{m=\pm 2}\rangle \). Figure 7 shows the second level configuration where the electron states are given by the \( |\Phi^{x=2}_{m=\pm \frac{3}{2}}\rangle \) and the hole states are given by the \( |\Phi^{x=2}_{m=\pm 1}\rangle \).

Only due to the symmetry between positive- and negative-energy solutions in a 3D TI QD it is possible to choose either of these two level configurations.

Then, using the optical selection rules shown in Fig. 5, we can use \( \sigma^+ \) polarized light to create an e-h pair with polarization +1, as shown in Fig. 6. This corresponds to writing the information +1 on the 3D TI QD. Alternatively,
we can use \( \sigma^- \) polarized light to create an e-h pair with polarization \(-1\), as shown in Fig. 6. This corresponds to writing the information \(-1\) on the 3D TI QD.

If we want to read out the information several times before the electron-hole recombination, we can take advantage of the Faraday effect due to the Pauli exclusion principle. For this method, we apply a \( \pi \) pulse of circularly polarized light, thereby writing the information \(+1\) or \(-1\), respectively, as shown in Fig. 8. For \(+1\) polarization, the Fermi functions, corresponding to populations in quasiequilibrium, are \( f(\epsilon_{\uparrow \downarrow} = 1) \), \( f(\epsilon_{\uparrow \uparrow} = 1) = 1 \), \( f(\epsilon_{\downarrow \uparrow} = 1) = 1 \), and \( f(\epsilon_{\downarrow \downarrow} = 0) = 0 \). For \(-1\) polarization, the Fermi functions, corresponding to populations in quasiequilibrium, are \( f(\epsilon_{\uparrow \downarrow} = 1) = 1 \), \( f(\epsilon_{\uparrow \uparrow} = 0) = 0 \), \( f(\epsilon_{\downarrow \uparrow} = 0) = 0 \), \( f(\epsilon_{\downarrow \downarrow} = 1) = 1 \), and \( f(\epsilon_{\downarrow \downarrow} = 0) = 0 \). Since the off-resonant interaction does not destroy the quantum state on the 3D TI QD, the information can be read out several times before recombination. These results are in complete agreement with the quantum-optical calculations shown below.

Let us assume a \( \sigma^+ \) polarized pump pulse of energy \( \hbar \omega_0 \) excites an e-h pair with polarization \(+1\) due to the \( \sigma^+ \) transition from the state \( |\Phi_{\epsilon^+} = \uparrow \rangle \) to the state \( |\Phi_{\epsilon^+} = \downarrow \rangle \) in the level configuration shown in Fig. 6. Then a linearly polarized probe pulse of energy \( \hbar \omega \) with certain detuning energy is applied to read it out. There are three virtual transitions that can occur while probing, one \( \sigma^- \) transition: \( |\Phi_{\epsilon^+} = \uparrow \rangle \leftrightarrow |\Phi_{\epsilon^+} = \downarrow \rangle \) and two \( \sigma^+ \) transitions: \( |\Phi_{\epsilon^+} = \downarrow \rangle \leftrightarrow |\Phi_{\epsilon^+} = \uparrow \rangle \) and \( |\Phi_{\epsilon^+} = \downarrow \rangle \leftrightarrow |\Phi_{\epsilon^+} = \uparrow \rangle \). The matrix elements are evaluated to be \( M_{\epsilon^+ \downarrow , \epsilon^+ \uparrow } = \mp 8.07 \times 10^{-18} \text{ m}^2 \) and \( M_{\epsilon^+ \downarrow , \epsilon^+ \downarrow } = -2.67 \times 10^{-18} \text{ m}^2 \). The sign of the matrix elements \( M_{\epsilon^+ \downarrow , \epsilon^+ \uparrow } \) is determined by the Fermi functions. The corresponding dipole moments are 454 Debye and 261 Debye. For a quantitative estimate, we choose a transition energy gap between the negative- and positive-energy solutions of \( \hbar \omega_0 = 130 \text{ meV} \), a linearly polarized probe pulse with detuning energy of \( \hbar \delta = 1 \text{ meV} \) and a cavity photon with a bandwidth of \( \hbar \gamma = 100 \text{ \mu eV} \). We further assume that there is a single QD in a slab material of length \( L = 0.1 \mu \text{m} \). With these values for our 3D TI QD of size 3.5 nm we obtain the real part of the Faraday rotation angle of \( \delta = -624 \mu \text{rad} \). This Faraday rotation angle is well above the angle value that has been measured for the experimental detection of a single spin in GaAs QDs. A similar calculation can be done for a \( \sigma^- \) polarized pump pulse that excites an e-h pair with polarization \(-1\) from the state \( |\Phi_{\epsilon^+} = \uparrow \rangle \) to the state \( |\Phi_{\epsilon^+} = \downarrow \rangle \). Due to the symmetry of the positive- and negative-energy solutions in 3D TI QDs, a large variety of level configurations can be considered to achieve the Faraday effect.

The largest dipole moment of 452 Debye is one order of magnitude larger than the typical value of 75 Debye for GaAs QDs, and two orders of magnitude larger than the typical value of a few Debye for atoms. This large strength of the coupling of infrared light to 3D TI QDs can partially compensate the weak overlap of the photon with the 3D TI QD, which is due to the wavelength of the infrared light being so much larger than the size of the 3D TI QD.

VII. SINGLE-PHOTON FARADAY EFFECT FOR 3D TI QDs

Let us consider a 3D TI QD in the level configuration shown in Fig. 6 inside a cavity. We define \( c_{\uparrow \pm} \), \( c_{\downarrow \pm} \) and \( c_{3 \pm} \) as the annihilation operators of the states \( |\Phi_{\epsilon^+} = \uparrow \rangle \), \( |\Phi_{\epsilon^+} = \downarrow \rangle \), and \( |\Phi_{\epsilon^+} = \uparrow \rangle \), respectively. Then the Jaynes-Cummings model gives rise to the Hamiltonian \( H = H_p + H_QD + H_{p-QD} \), where

\[
H_p = \hbar \omega_0 (c_{\uparrow \downarrow}^\dagger c_{\uparrow \downarrow} + c_{\downarrow \uparrow}^\dagger c_{\downarrow \uparrow}),
\]

\[
H_QD = \sum_{j=1}^3 \hbar \omega_j (c_{j \downarrow}^\dagger c_{j \uparrow} + c_{j \uparrow}^\dagger c_{j \downarrow}),
\]

\[
H_{p-QD} = \hbar g_1 (a_{\uparrow \downarrow} c_{\uparrow \downarrow} + a_{\downarrow \uparrow} c_{\downarrow \uparrow} + \text{H.c.}) + \hbar g_2 (a_{\uparrow \downarrow} c_{\uparrow \downarrow} + a_{\downarrow \uparrow} c_{\downarrow \uparrow} + \text{H.c.}) + \text{H.c.,}
\]

are the cavity photon Hamiltonian, the QD Hamiltonian describing the Weyl states, and the interaction Hamiltonian describing the photon-QD interaction, respectively. We can safely neglect the vacuum energy \( \hbar \omega_0/2 \) per mode. The photon-QD coupling constants are given by \( \hbar g_1 = \sqrt{\hbar \omega_0/2} V_0 \langle \Phi_{\epsilon^+} = \uparrow \rangle |\langle \Phi_{\epsilon^+} = \uparrow \rangle\rangle_{\text{mod}} \cdot \alpha + \epsilon \cdot \hat{r} |\Phi_{\epsilon^+} = \uparrow \rangle \rangle_{\text{mod}} \) and \( \hbar g_2 = \sqrt{\hbar \omega_0/2} V_0 \langle \Phi_{\epsilon^+} = \uparrow \rangle |\langle \Phi_{\epsilon^+} = \uparrow \rangle\rangle_{\text{mod}} \cdot \alpha + \epsilon \cdot \hat{r} |\Phi_{\epsilon^+} = \uparrow \rangle \rangle_{\text{mod}} \), where \( V_0 \) is the modal volume. After switching to the electron-hole picture using the new electron and hole operators \( c_{\pm} = c_{3 \pm} \).
and $v_{j^+}^j = c_{j^+}$ for $j = 1, 2$, we obtain
\begin{align}
H_p &= \hbar \omega_\perp (a_+^\dagger a_+ + a_-^\dagger a_-), \\
H_{QD} &= \hbar \omega_\parallel (c_+^\dagger c_+ + c_-^\dagger c_-) + \sum_{j=1}^2 \hbar \omega_j (v_{j^+}^j v_{j^+}^j + v_{j_-}^j v_{j_-}^j), \\
H_{int} &= \hbar g_1 (a_+^\dagger c_+^\dagger v_{j^+}^j + a_-^\dagger c_-^\dagger v_{j_-}^j) + \text{H.c.} \\
&+ \hbar g_2 (a_+^\dagger c_-^\dagger v_{j_+}^j + a_-^\dagger c_+^\dagger v_{j_-}^j) + \text{H.c.}
\end{align}
(52)
(53)
(54)

where $\hbar \omega_\parallel - \hbar \omega_\perp = \hbar \omega_\perp + \hbar \delta$ and $\hbar \omega_\perp - \hbar \omega_\perp = \Delta_S$. Since the interaction between the EM fields and the QDs is off-resonant, we can apply an adiabatic approximation. For that, we omit the time derivatives $\delta \frac{\partial}{\partial t}$ in the Heisenberg equations to zero, i.e., taking the stationary limit. Then we obtain
\begin{align}
p_{1\pm\pm} &= -g_1 \langle a_\pm (1 - c_{\pm}^\dagger c_{\pm} - v_{1\pm}^\dagger v_{1\pm}) \rangle, \\
p_{2\pm\pm} &= -g_2 \langle a_\pm (1 - c_{\mp}^\dagger c_{\mp} - v_{2\pm}^\dagger v_{2\pm}) \rangle.
\end{align}
(62)
(63)

Inserting this result into the interaction Hamiltonian leads to an effective interaction Hamiltonian of the form
\begin{align}
H_{\text{int}}^{\text{eff}} &= -\frac{\hbar g_1^2}{\delta + \frac{\Delta_S}{\hbar}} \sum_\sigma (2a_\sigma^\dagger a_\sigma + 1) (1 - c_{\pm}^\dagger c_{\pm} - v_{1\sigma}^\dagger v_{1\sigma}) \\
&- \frac{\hbar g_2^2}{\delta} \sum_\sigma (2a_\sigma^\dagger a_\sigma + 1) (1 - c_{\mp}^\dagger c_{\mp} - v_{2\sigma}^\dagger v_{2\sigma}).
\end{align}
(64)

where $\delta$ has the opposite sign of $\sigma$. It becomes obvious that if electrons or holes are present, the effective interaction can be suppressed. Most importantly, this suppression of interaction depends on the spin of the present electrons or holes. This is exactly the mechanism for the Faraday effect due to Pauli exclusion principle. Let us now calculate the time evolution of the photon operator in the rotating frame under the effective interaction Hamiltonian, i.e.,
\begin{align}
\frac{i\hbar}{\delta} \frac{\partial a_\pm}{\partial t} &= [a_\pm, H_{\text{int}}^{\text{eff}}] \\
&= -2g_1 \left[ \left( \frac{\hbar g_1^2}{\delta + \frac{\Delta_S}{\hbar}} (1 - c_{\pm}^\dagger c_{\pm} - v_{1\pm}^\dagger v_{1\pm}) \right) \\
&+ \left( \frac{\hbar g_2^2}{\delta} (1 - c_{\mp}^\dagger c_{\mp} - v_{2\pm}^\dagger v_{2\pm}) \right) \right],
\end{align}
(65)

resulting in the solution
\begin{align}
a_\pm(t) &= a_\pm(0) \exp \left\{ -i \left[ \left( \frac{2g_1^2}{\delta + \frac{\Delta_S}{\hbar}} (1 - c_{\pm}^\dagger c_{\pm} - v_{1\pm}^\dagger v_{1\pm}) \right) \\
&+ \frac{2g_2^2}{\delta} (1 - c_{\mp}^\dagger c_{\mp} - v_{2\pm}^\dagger v_{2\pm}) \right] t \right\},
\end{align}
(66)

This formula is the main result of this section. It shows that the Faraday rotation of the linearly polarized light depends strongly on the presence of electrons and holes due to the Pauli exclusion principle.

**VIII. QUANTUM TELEPORTATION AND QUANTUM COMPUTING WITH 3D TI QDs**

Here we show that the single-photon Faraday rotation can be used to entangle a single photon with either a single e-h pair, a single electron, or a single hole. This entanglement can be used as a resource to implement optically mediated quantum teleportation and quantum computing. The quantum-informational methods for the implementation of quantum teleportation and quantum computing are described in Refs. 43 and 44. We describe here the physical methods for creating the entanglement.
A. Photon polarization: e-h pair polarization entanglement

Let us consider now the Faraday effect due to an e-h pair on the QD for the level configuration shown in Fig. 6. The initial state before the photon-QD interaction reads

$$|\psi_+(0)\rangle = \frac{1}{\sqrt{2}}(e^{-i\theta_0}a_+^\dagger + e^{i\theta_0}a_-^\dagger)c_+^\dagger v_{2+}^\dagger |0\rangle,$$  \hspace{1cm} (67)

where the photon is linearly polarized at an angle $\theta_0$ from the $x$ axis. If the initial e-h pair is $-1$ polarized, then the state after time $t$ is given by

$$|\psi_-(t)\rangle = \frac{1}{\sqrt{2}}(e^{-i(\theta_0+\varphi_0)}a_+^\dagger(0) + e^{i(\theta_0+\varphi_0)}a_-^\dagger(0))c_+^\dagger v_{2+}^\dagger |0\rangle$$  

+ $e^{i(\theta_0+\varphi_0)}a_-^\dagger(0))c_+^\dagger v_{2+}^\dagger |0\rangle$  \hspace{1cm} (68)

with $\varphi_0 = -2\frac{\varphi}{\tau}$ and $\varphi_0 = 2\frac{\varphi}{\tau}$, resulting in a Faraday rotation angle of $\psi_-(t) = [\varphi_0(t) + \varphi_0(t)]/2 = -(2\varphi - \pi\theta)$). If the initial e-h pair is $+1$ polarized, then the state after time $t$ is given by

$$|\psi_+(t)\rangle = \frac{1}{\sqrt{2}}(e^{-i(\theta_0+\varphi_0)}a_+^\dagger(0)$$  

+ $e^{i(\theta_0+\varphi_0)}a_-^\dagger(0))c_+^\dagger v_{2+}^\dagger |0\rangle$  \hspace{1cm} (70)

with $\varphi_0 = -2\frac{\varphi}{\tau}$ and $\varphi_0 = 2\frac{\varphi}{\tau}$, resulting in a Faraday rotation angle of $\psi_+(t) = [\varphi_0(t) + \varphi_0(t)]/2 = +\pi(2\varphi - \pi\theta)$). These results are in complete agreement with the result using Fermi’s golden rule above.

In addition, the quantum-optical calculation lets us entangle the photon with the electron-hole state on the 3D TI QD. In particular, if we choose the initial state to be

$$|\psi(0)\rangle = \frac{1}{\sqrt{2}}(a_+^\dagger + a_-^\dagger)(c_+^\dagger v_{2+}^\dagger + c_-^\dagger v_{2-}^\dagger)|0\rangle$$  \hspace{1cm} (72)

the photon and the e-h pair get entangled for $\varphi_0 = \pm \frac{\pi}{2}$, i.e., after a time $\tau = \pi/4(2\varphi - \pi\theta)$, yielding

$$|\psi(t)\rangle = \frac{1}{\sqrt{2}}(e^{-i\varphi}a_+^\dagger(0) + e^{i\varphi}a_-^\dagger(0))c_+^\dagger v_{2-}^\dagger$$  

+ $\frac{1}{\sqrt{2}}(e^{-i(-\varphi)}a_+^\dagger(0) + e^{i(-\varphi)}a_-^\dagger(0))c_-^\dagger v_{2+}^\dagger |0\rangle$  \hspace{1cm} (73)

This state consists of a photon entangled to the e-h pair on the 3D TI QD.

B. Photon polarization: electron spin entanglement

Now let us consider the Faraday effect due to a single electron for the level configuration shown in Fig. 9. Here the electron is in an $s$-like state. If the initial state is

$$|\psi(0)\rangle = \frac{1}{\sqrt{2}}(e^{-i\theta_0}a_+^\dagger + e^{i\theta_0}a_-^\dagger)c_+^\dagger |0\rangle$$  \hspace{1cm} (74)

then state after the interaction is given by

$$|\psi(t)\rangle = \frac{1}{\sqrt{2}}(e^{-i\theta_0-\frac{\varphi}{\tau}}a_+^\dagger(0) + e^{i\theta_0+\frac{\varphi}{\tau}}a_-^\dagger(0))c_+^\dagger |0\rangle$$  \hspace{1cm} (75)
then state after the interaction is fully entangled after a time \( \tau = \pi/(4(\frac{\kappa}{\delta} \pm \frac{\gamma}{\delta})) \), i.e.,

\[
|\psi(\tau)\rangle = \frac{1}{\sqrt{2}} \left( e^{-i\frac{\pi}{2}} a_+^{(0)}(0) + e^{i\frac{\pi}{2}} a_-^{(0)}(0) \right) c_+^{(0)} + \frac{1}{\sqrt{2}} \left( e^{-i\frac{\pi}{2}} a_+^{(0)}(0) + e^{i\frac{\pi}{2}} a_-^{(0)}(0) \right) c_-^{(0)}. \tag{79}
\]

This state is a fully entangled electron-photon state.

The Faraday effect due to a single hole for the level configuration shown in Fig. 10 can be calculated in a similar way. Here the hole is in an s-like state. Other possible configurations include a single electron in a p-like state or a single hole in a p-like state.

Due to the symmetry between positive- and negative-energy solutions, in a 3D TI QD it is possible to define an electron spin qubit in terms of an s-like or a p-like state. At the same time it is possible to define a hole spin qubit in terms of an s-like or a p-like state. This cannot be done in a conventional wide-band-gap semiconductor QD, where the electron is associated with an s-like state and the hole is associated with a p-like state.\(^4^6\)

For a quantitative description, we can assume a single 3D TI QD embedded in a semiconductor microcavity. The strong and weak interaction can occur between the QD e-h pair and discretized cavity modes at resonance, \(\omega_{21} = \omega_e\). The e-h-photon coupling parameter \(g\) is given by

\[
g = (\pi \epsilon_f f)^{1/2}/(4\pi \epsilon_f \epsilon_m V_m \lambda)^{1/2},
\]

where \(\epsilon_f\) is the dielectric constants for the cavity material,\(^6^3\) \(\epsilon_m\) is the free electron mass, and \(V_m\) is the mode volume. The mode volume for a mode of wavelength \(\lambda\) is \(V_m = (\lambda/2n)^3\), where \(n = \sqrt{\epsilon_f}\) (for a GaAs microcavity, \(n = 3.31\)). Using \(\hbar \omega_{21} = 130\) meV, the oscillator strengths for the transition \(\Phi_{s=\pm 1} \leftrightarrow \Phi_{s=\pm 1}^{\mp 1}\) and \(\Phi_{s=\pm 1} \leftrightarrow \Phi_{s=\mp 1}^{\pm 1}\) are obtained, respectively, \(f_1 \approx 9\) and \(f_2 \approx 27\). This gives us an estimate of \(\hbar \Gamma_1 \approx 10\) \(\mu\)eV and \(\hbar \Gamma_2 \approx 17\) \(\mu\)eV. For a detuning energy of \(\hbar \delta = 100\) \(\mu\)eV the time it takes to fully entangle the electron spin and the photon polarization is calculated to be of the order of 180 ps. The necessary condition to be in the strong coupling regime is that \(g\) must be large compared to both spontaneous emission rate and cavity decay loss rate.\(^6^6\) Thus, for \(Q \gtrsim \omega/\Gamma_1 \approx 1.3 \times 10^4\) the 3D TI QD is in the strong coupling regime. For \(Q = 10^5\), the photon decay rate is given by

\[
\kappa = \frac{\omega_0}{Q} = 3.1 \times 10^9\text{ s}^{-1}.
\]

This gives a cavity photon life time of \(3\) ns.

**IX. CONCLUSION**

We have shown that Weyl fermions can be confined in all three dimensions at the spherically shaped interface between two narrow-band-gap semiconductor alloys, such as the core-bulk heterostructure made of PbTe/Pb(0.3)Sn(0.7)Te. This configuration provides us with the model of a spherical 3D TI QD with tunable size \(r_0\) and potential \(\Delta_0\), which allows for complete control over the number of bound interface states. The most important features of 3D TI have been identified in a 3D TI QD, namely the spin locking effect and the Kramers degeneracy. We found that the Weyl states are confined on the surface of the QD, in contrast to the electrons and holes in

![Diagram](Image)
topologically trivial semiconductor QDs. We showed that due to the large dipole moment of 450 Debye it is possible to reach the strong-coupling regime inside a cavity with a quality factor of $Q \approx 10^4$ in the infrared wavelength regime around 10 $\mu$m. Because of the strict optical selection rules, the 3D TI QD gives rise to interesting applications based on the semiclassical and quantum Faraday effect. We found that the 3D TI QD is a good candidate for quantum memory, quantum teleportation, and quantum computing with single spins in 3D TI QDs using infrared light. In particular, a single e-h pair, a single electron, or a single hole can be used as a qubit for the implementation of optically mediated quantum computing with 3D TI QDs. Interestingly, we found that due to the symmetry between positive- and negative-energy solutions, in a 3D TI QD it is possible to define an electron spin qubit in terms of an $s$-like or a $p$-like state. At the same time it is possible to define a hole spin qubit in terms of an $s$-like or a $p$-like state. This cannot be done in a zincblende wide-direct-gap semiconductor QD, where the electron is associated with an $s$-like state and the hole is associated with a $p$-like state.

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APPENDIX A: CALCULATION OF THE WRONSKIAN

The Wronskian of the functions $I_\kappa(z)$ and $K_\kappa(z)$ is defined as

$$W_\kappa [I_\kappa(z), K_\kappa(z)] = I_\kappa(z) K'_\kappa(z) - I'_\kappa(z) K_\kappa(z),$$

where the prime denotes the derivative of the function. For independent solutions, it is to be noted that Wronskian is proportional to $1/p(x)$ in a Sturm-Liouville-type equation $\frac{d}{dx}[p(x)\frac{dy}{dx}] + g(x)y = 0$. Therefore, Wronskians in the text are calculated to be

$$W_\kappa [I_\kappa(z), K_\kappa(z)] = -\frac{1}{z^2};$$

$$W_{\kappa-1} [I_{\kappa-1}(z), K_{\kappa-1}(z)] = -\frac{1}{z^2}.$$  

APPENDIX B: LIMITING FORM OF BESSEL FUNCTIONS

The limiting forms of modified Bessel functions for $z \to 0$ are given by

$$I_\kappa(z) = \frac{1}{\Gamma(\kappa + 1)} \left(\frac{2}{z}\right)^\kappa,$$

$$K_\kappa(z) = \frac{\Gamma(\kappa + 1)}{2} \left(\frac{z}{2}\right)^\kappa.$$

(B1)

The modified spherical Bessel functions can be written in terms of modified Bessel functions as

$$I_\kappa(z) = \left(\frac{\pi}{2z}\right)I_{\kappa + \frac{1}{2}}(z),$$

$$K_\kappa(z) = \left(\frac{2}{\pi z}\right)K_{\kappa + \frac{1}{2}}(z).$$

(B2)

Therefore, the function

$$F(z) = [zI_{\kappa-1}(z)K_\kappa(z)][zI_{\kappa-1}(z)K_\kappa(z)]$$

has the limiting form $F(z) = \frac{1}{z^4}$ as $z \to 0$.

The asymptotic expansion ($z \to \infty$) of the modified Bessel functions are given by

$$I_\kappa(z) = \frac{\exp \left(\frac{z}{2}\right)}{\sqrt{2\pi z}} \times \left\{ 1 - \frac{4\kappa^2 - 1}{8z} + \frac{(4\kappa^2 - 1)(4\kappa^2 - 9)}{2!(8z^2)} - \cdots \right\}$$

$$K_\kappa(z) = \frac{\pi}{\sqrt{2z}} \exp \left(-z\right) \times \left\{ 1 + \frac{4\kappa^2 - 1}{8z} + \frac{(4\kappa^2 - 1)(4\kappa^2 - 9)}{2!(8z^2)} - \cdots \right\}.$$

(B3)

APPENDIX C: FERMION DOUBLING THEOREM

Nielsen and Ninomiya investigated Weyl fermions on a crystal. They formulated a no-go theorem, called the fermion doubling theorem, requiring that Weyl nodes in a crystal always exist in pairs of opposite chirality. The reason for this theorem is that the number of Weyl fermions in the first Brillouin zone must be conserved. This conservation law can be checked by calculating the Berry flux in the first Brillouin zone.

It is important to note that the fermion doubling theorem is only valid for continuum states. Therefore it does not apply to the bound eigenstates of the 3D TI QD, which have a discrete eigenspectrum. Below we give arguments for the validity of the fermion doubling theorem in the continuum limit, which corresponds to the asymptotic limit when the 3D TI QD radius $r_0$ becomes infinite.

A typical calculation of the Berry curvature $B_\kappa(k) = V_k \times A_\kappa(k)$ considers a single band Bloch state $u_{n\kappa}\mathbf{r} e^{ik\mathbf{r}}$, which gives rise to the Berry connection $A_\kappa(k) = i \int_\Omega d^3r \; u^*_{n\kappa}(\mathbf{r}) V_k u_{n\kappa}(\mathbf{r})$. As long as the $r$th band does not touch or cross any other band, the Berry flux is zero, i.e., $\nabla_k \cdot B_\kappa(k) = 0$. However, if there is a band crossing, this situation changes drastically due to the monopole at the crossing point. Using $k \cdot p$ approximation, around the crossing point in the first Brillouin zone the Berry connection becomes $A_\kappa(k) = i \langle \chi_\kappa | \nabla_k | \chi_\kappa \rangle$, where $\chi_\kappa$ is the four-spinor of the solution $\Phi_\kappa = \chi_\kappa F(r)$ of Eq. (1). Assuming a very large QD, where quantum confinement can be neglected, the four-spinor reads

$$\chi_\kappa = \begin{pmatrix} \pm e^{i\frac{\kappa \pi}{4}} \frac{1}{\kappa} \\ \pm e^{i\frac{\kappa \pi}{4}} \frac{1}{\kappa} \\ \pm e^{-i\frac{\kappa \pi}{4}} \frac{1}{\kappa} \\ e^{i\frac{\kappa \pi}{4}} \frac{1}{\kappa} \end{pmatrix},$$

(C1)

where $e^{i\phi} = \frac{k_x + i k_y}{k_z}$ and the position-dependent function is given by

$$F(r) = Ce^{-\frac{1}{\kappa} \int_\Delta \frac{d\mathbf{c}}{\pi} e^{-i k \cdot \mathbf{c}} k | r_k \cdot \mathbf{k} r_k | r_k}.$$
normalization constant. In order to capture the Berry curvature apart from the azimuthal angle $\varphi$ we need to add the dependence on the polar angle $\theta$. At the same time, we perform the gauge transformations $e^{\pm i\varphi}$ to shift the singularity of the Berry curvature to the south pole. This means we calculate the Berry curvature with respect to the normalized four-spinors

$$
\chi_{C,+} = \frac{1}{\sqrt{2}} \begin{pmatrix} e^{-i\frac{\varphi}{2}} \cos \frac{\theta}{2} \\ e^{i\frac{\varphi}{2}} \sin \frac{\theta}{2} \\ e^{i\frac{\varphi}{2}} \cos \frac{\theta}{2} \\ e^{-i\frac{\varphi}{2}} \sin \frac{\theta}{2} \end{pmatrix},
$$

$$
\chi_{C,-} = \frac{1}{\sqrt{2}} \begin{pmatrix} -e^{-i\frac{\varphi}{2}} \sin \frac{\theta}{2} \\ e^{i\frac{\varphi}{2}} \cos \frac{\theta}{2} \\ e^{-i\frac{\varphi}{2}} \sin \frac{\theta}{2} \\ e^{-i}\frac{\varphi}{2} \cos \frac{\theta}{2} \end{pmatrix}.
$$

The Berry connection is then given by

$$
A_{\pm}(k) = i \langle \chi_{C,\pm} | \nabla_k | \chi_{C,\pm} \rangle = \mp \frac{(1 - \cos \theta)}{2k \sin \theta} e_{\varphi}
$$

where $e_{\varphi}$ is the unit vector pointing in $\varphi$ direction. Thus, we obtain the Berry phase

$$
\gamma_{\pm} = \oint A_{\pm}(k) \cdot dk = \mp \pi (1 - \cos \theta)
$$

and the Berry curvature

$$
B_{\pm}(k) = \mp \frac{1}{2k^2} e_{\varphi}.
$$

Note that the Berry curvature for the four-spinor is the same as the Berry curvature of a two-spinor. For a loop on the 2D surface where $\theta = \pi/2$, we get $\gamma_{\pm} = \mp \pi$, which gives rise to the topological phase shift seen in Shubnikov-de Haas oscillations for the surface of 3D topological insulators. From $\Delta(1/k) = \mp 4\pi \delta^{(3)}(k)$ and $\nabla(1/k) = \mp \frac{1}{r} e_k$ it follows that the Berry curvature is the solution of the equation

$$
\nabla_k \cdot B_{\pm}(k) = \mp 4\pi g \delta^{(3)}(k),
$$

where $g = \mp 1/2$ is the strength of the Dirac monopole for positive and negative helicity of the four-spinor, which is identical to the result for two-spinors (see Refs. 69 and 71).

In order to understand the helicity of the Weyl fermions at the interface, we have shown in Ref. 49 that the helicity operator is given by

$$
\hat{H}_{\text{TI}} = (1/|p_\perp|) \begin{pmatrix} \sigma_\perp \times p_\perp \cdot \hat{z} \\ 0 \\ -\sigma_\perp \times p_\perp \cdot \hat{z} \end{pmatrix},
$$

which commutes with the Hamiltonian in Eq. (1) and yields $\hat{H}_{\text{TI}} \Phi_{\pm} = (\pm 1/2) \Phi_{\pm}$, where the $+$ sign denotes the positive helicity of positive-energy solutions and the $-$ sign denotes the negative helicity of negative-energy solutions. This provides the possibility to write an effective 2D Hamiltonian for the Weyl fermions on the surface of 3D topological insulators, i.e.,

$$
H_{2D} = \hbar v \begin{pmatrix} \sigma_\perp \times k_\perp \cdot \hat{z} \\ 0 \\ -\sigma_\perp \times k_\perp \cdot \hat{z} \end{pmatrix}.
$$

This effective 2D Hamiltonian can be reduced to two Weyl Hamiltonians of the form $H_{2D}^\pm = \pm \hbar v (\sigma_\perp \times k_\perp) \cdot \hat{z}$. It is important to note that both two-spinors of $\chi_{\pm}$, the two-spinor $\chi_{\perp}$ of the $L^-$ band and the two-spinor $\chi_{\perp}^L$ of the $L^+$ band have the same helicity, in contrast to the commonly used Weyl Hamiltonians $H_0(k) = \pm \hbar v \sigma \cdot k$. The reason for this is that the two two-spinors are coupled through the mass term $\Delta(\varphi)$ in $\hat{z}$ direction, as given in the 3D Hamiltonian in Eq. (1).72-75

In order to satisfy the fermion doubling theorem, usually the Dirac cones on the opposite side of the slab of a 3D topological insulator are identified as the fermion doublers. In the case of the 3D IT QD, for $r_o \to \infty$, i.e., in the continuum limit, the Berry curvature in $k$ space for a 2D interface, given by Eq. (C5), determines the Weyl nodes that need to satisfy the fermion doubling theorem. Hence, according to Ref. 51, we can adopt the mapping of the two opposite surfaces of a 3D slab of TI onto the northern and southern hemispheres of a sphere. We then identify the pairs of Dirac cones with opposite helicity as the ones located on the antipodal points on the surface of the sphere defined by the QD, as shown in Fig. 11. Note that in both cases, the slab and the QD, the pairs of Dirac cones map into each other through the parity transformation, which in general reverses the helicity. We can identify a current on the surface of the sphere flowing along a latitude. The parity transformation then maps one latitude on the northern hemisphere with one type of helicity to its partner latitude on the southern hemisphere with the opposite helicity. These arguments show that the fermion doubling theorem is satisfied for a 3D TI QD in the continuum limit.

![FIG. 11. (Color online) Two antipodal points on the surface of the sphere defined by the QD are identified as the Dirac cones of opposite helicity. One point lies on the northern hemisphere, while its antipodal point lies on the southern hemisphere. The currents flowing along the latitudes can be imagined as angular momentum states of a 3D TI QD in the continuum limit. At the antipodal points momenta (red arrows) point in opposite $\varphi$ direction to each other while spins (blue arrows) point in the same $\theta$ direction, where $\theta$ and $\varphi$ are the spherical angular unit vectors. Hence, they have opposite chirality. This satisfies the fermion doubling theorem.](085316-15)
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