Spin-orbit interactions in inversion-asymmetric two-dimensional hole systems: A variational analysis

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We present an in-depth study of the spin-orbit (SO) interactions occurring in inversion-asymmetric twodimensional hole gases at semiconductor heterointerfaces. We focus on common semiconductors such as GaAs, InAs, InSb, Ge, and Si. We develop a semianalytical variational method to quantify SO interactions, accounting for both structure inversion asymmetry (SIA) and bulk inversion asymmetry (BIA). Under certain circumstances, using the Schrieffer-Wolff (SW) transformation, the dispersion of the ground state heavy hole subbands can be written as $E(k) = Ak^2 - Bk^4 \pm Ck^3$ where A, B, and C are material- and structure-dependent coefficients. We provide a simple method of calculating the parameters *A*, *B*, and *C*, yet demonstrate that the simple SW approximation leading to a SIA (Rashba) spin splitting ∝ *k*³ frequently breaks down. We determine the parameter regimes at which this happens for the materials above and discuss a convenient semianalytical method to obtain the correct spin splitting, effective masses, Fermi level, and subband occupancy, together with their dependence on the charge density, and dopant type, for both inversion and accumulation layers. Our results are in good agreement with fully numerical calculations as well as with experimental findings. They suggest that a naive application of the simple cubic Rashba model is of limited use in either common heterostructures or quantum dots. Finally, we find that for the single heterojunctions studied here the magnitudes of BIA terms are always much smaller than those of SIA terms.

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I. INTRODUCTION

The ability to harness the spin degree of freedom is essential for the development of practical semiconductor spintronic devices [\[1,2\]](#page-12-0) and quantum information processing [\[3–5\]](#page-12-0). All-electrical spin control may be possible by exploiting the coupling of the spin and orbital degrees of freedom brought about by the strong spin-orbit interactions in certain semiconductor systems [\[1,2,6\]](#page-12-0). This could lead to faster spin rotations and lower power consumption, as well as the convenience of using solely electric fields, which are easier to apply and localize than magnetic fields. The search for semiconductor systems with strong spin-orbit coupling has led naturally to low-dimensional hole systems [\[7–9\]](#page-12-0). Despite the promising advances of recent years, in particular in the experimental state of the art $[10-22]$, functional hole spin-based devices are yet to be realized. In particular, a comprehensive understanding of the interaction between a hole's spin and its solid-state environment is far from complete.

In group IV and III-V semiconductors the uppermost valence band is described by wave functions originating from bonding atomic *p* orbitals with orbital angular momentum $L = 1$. Together with the spin $S = 1/2$ this results in an effective spin $J = 3/2$ [\[6,23,24\]](#page-12-0), which brings about spin properties of hole systems that are distinct from those of electron systems [\[25,26\]](#page-12-0). The bulk valence band eigenstates are heavy holes (HH) with angular momentum projection $m_J = \pm 3/2$ in the direction of the wave vector and light holes (LH) with $m_J = \pm 1/2$. These are degenerate at the zone center but split by a finite energy at nonzero wave vectors. Confinement to the interface of a heterojunction fixes the hole spin quantization axis in the growth direction \hat{z} , which here we take to be parallel to (001), and lifts the degeneracy of the HH and LH states at the subband edge (in-plane wave vector $k = 0$). At this point the eigenstates remain pure HH and LH, while at finite k , the $k \cdot p$ interaction causes the HH and LH states to mix. However, for typical Fermi wave vectors k_F , the subband states with $k \lesssim k_F$ can still be regarded as approximately HH- or LH-like [\[27\]](#page-12-0).

Inversion asymmetry further lifts the spin degeneracies of HH and LH states [\[6\]](#page-12-0), resulting in additional *k*-dependent energy level splittings. Inversion asymmetry in semiconductor heterostructures may stem from the asymmetry of the confining potential [structural inversion asymmetry (SIA) or Rashba terms [\[28\]](#page-12-0)] or from the asymmetry of the underlying crystal structure [bulk inversion asymmetry (BIA) or Dresselhaus terms [\[29\]](#page-12-0)]. BIA and SIA spin splittings are proportional to odd powers of *k*, with *k*-linear [\[6,30–](#page-12-0)[35\]](#page-13-0) and k^3 terms $[4,6,7,27,29,32,35-41]$ $[4,6,7,27,29,32,35-41]$ frequently representing the dominant contributions. They result in a high degree of nonparabolicity of the hole energy bands. Accounting for the complex couplings between the hole bands is vital if one is to capture all aspects of hole spin dynamics correctly. So far, theoretical studies of two-dimensional hole gases (2DHGs) in group IV and III-V structures have been predominantly numerical and always material specific [\[30](#page-12-0)[,36,42–46\]](#page-13-0). Calculations for single heterojunctions are more complex due to the nature of the confining potential. Unlike quantum wells, where the potential shape is approximately fixed, the shape of the confining potential in single heterojunctions is highly density dependent, thus requiring self-consistent wave functions that produce the potential for a given density.

In this paper, we develop a variational method that enables us to gain a transparent insight into spin-orbit interactions in 2DHGs in III-V and Si-based heterojunctions. We use the Luttinger Hamiltonian [\[24\]](#page-12-0) and the standard envelope function approximation [\[6\]](#page-12-0), combined with a simple self-consistent variational approach [\[47–49\]](#page-13-0), to calculate the spin splitting and effective masses for semiconductors with zincblende and diamond lattices. The variational approach allows one to easily solve for the confinement potential $V(z)$. Once the Luttinger Hamiltonian for the 2DHG is constructed, one can use a Schrieffer-Wolff (SW) transformation [\[50\]](#page-13-0) (in the context of semiconductors also known as Löwdin perturbation theory [\[51\]](#page-13-0)) to derive analytical expressions for the spin-dependent dispersion of 2D hole systems, which subsequently yields the Fermi level, spin splitting, effective masses, and subband occupancy. We apply this method to common semiconductors such as GaAs, InAs, InSb, Ge, and Si, and compare inversion and accumulation layers.

In the axial approximation, i.e., ignoring anisotropic corrections including warping and BIA, the SW transformation enables one to write the dispersion relation in a rather simple form $E_{\pm}(k) = Ak^2 - Bk^4 \pm Ck^3$ where *A*, *B*, and *C* are materialand structure-dependent coefficients, and the $Ck³$ term represents the Rashba spin splitting [\[4,6,7,27](#page-12-0)[,32,36–41\]](#page-13-0). However, we demonstrate that the validity of the SW-transformed model is limited to a relatively narrow range of parameters, indicating that in general the HH spin splitting contains higher-order terms in the wave vector, which are frequently sizeable. The limited applicability of the simple dispersion relation to realistic heterostructures is relevant to the current understanding of the spin-Hall conductivity [\[32,37,39,52\]](#page-13-0), hole spin helix [\[40\]](#page-13-0), and *Zitterbewegung* [\[38,41\]](#page-13-0), all of which have been derived based on the assumption that the HH spin splitting is proportional to $k³$. Our work can be used to determine spin densities and spin-Hall currents in the same way as for electrons [\[53,54\]](#page-13-0). It is also highly relevant to the burgeoning field of hole quantum dots, which are actively researched at present with a view to applications in quantum computing, in particular via electric dipole spin resonance (EDSR) [\[55\]](#page-13-0). The areal number densities of existing single-hole quantum dots are contained in the parameter ranges we study in this work in 2DHGs.

Apart from fully numerical calculations using an unconstrained basis set $[42, 49, 56]$, two approaches can be adopted in regimes in which the cubic spin splitting approximation is inadequate: One can evaluate higher-order terms in perturbation theory, which quickly becomes cumbersome and intractable, or one can perform a numerical diagonalization of the effective Luttinger Hamiltonian restricted to a certain subspace spanned by, e.g., the first and second HH and LH subbands. In this work, we rely on the latter approach when the perturbative methods fail. Our methods yield good agreement with fully numerical results for GaAs holes [\[42,49,56\]](#page-13-0).

Recently the importance of surface termination effects on spin-orbit interactions and the HH spin splitting has been pointed out by Durnev *et al.* [\[34\]](#page-13-0), who focused on the case of quantum wells. Inclusion of these effects is beyond the scope of our present work. Whereas in heterojunctions, where the wave functions vanish near the interface, surface termination effects are expected to be weaker [\[57\]](#page-13-0), a complete description will require the systematic inclusion of these contributions.

This paper is structured as follows: In Sec. II, we introduce the variational method employed for 2DHGs in quasitriangular wells. We apply this method in Sec. [III](#page-4-0) to calculate the spin splitting and effective masses for 2DHGs in inversion and accumulation layers in various semiconductors. The effects of the Dresselhaus spin-orbit interaction terms are outlined in Sec. [IV.](#page-7-0) We discuss our results in Sec. [V,](#page-9-0) while Sec. [VI](#page-10-0) contains a summary and conclusions.

II. THEORETICAL FORMALISM

A. Luttinger Hamiltonian for 2DHGs

Our formulation takes as its starting point the bulk 4×4 Luttinger Hamiltonian [\[24\]](#page-12-0) describing holes in the uppermost valence band with an effective spin $J = 3/2$

$$
H_L = \begin{pmatrix} P+Q & 0 & L & M \\ 0 & P+Q & M^* & -L^* \\ L^* & M & P-Q & 0 \\ M^* & -L & 0 & P-Q \end{pmatrix}, \quad (1)
$$

where

$$
P = \frac{\mu}{2} \gamma_1 (k^2 + k_z^2), \quad Q = -\frac{\mu}{2} \gamma_2 (2k_z^2 - k^2), \tag{2a}
$$

$$
L = -\sqrt{3}\mu \gamma_3 k_- k_z, \quad M = -\frac{\sqrt{3}\mu}{2} (\bar{\gamma} k_-^2 - \zeta k_+^2), \quad (2b)
$$

$$
\bar{\gamma} = \frac{1}{2}(\gamma_2 + \gamma_3), \quad \zeta = \frac{1}{2}(\gamma_3 - \gamma_2), \tag{2c}
$$

$$
k^2 = k_x^2 + k_y^2, \quad k_{\pm} = k_x \pm ik_y,
$$
 (2d)

 $\mu \equiv \hbar^2 / m_0$ with bare electron mass m_0 , and γ_1 , γ_2 , and γ_3 are the Luttinger parameters, see Table I. The wave vector components are defined by the crystallographic orientation. In this work, we consider holes grown on a (001) surface so that $k_x \parallel (100)$, $k_y \parallel (010)$, and $k_z \parallel (001)$. We have expressed H_L in the basis of *J_z* eigenstates $\{ |+\frac{3}{2}\rangle, |-\frac{3}{2}\rangle, |+\frac{1}{2}\rangle, |-\frac{1}{2}\rangle\}$, where \hat{z} is the unit vector perpendicular to the plane of the interface.

The Luttinger Hamiltonian is further simplified in the axial approximation, where the terms proportional to *ζ* are neglected [\[6](#page-12-0)[,42\]](#page-13-0). The axial approximation is appropriate for GaAs, InAs, InSb, and Ge, while for Si *ζ* is significant (Table I) and gives rise to a highly anisotropic Fermi contour, as Sec. [III C](#page-7-0) will show.

The 4×4 Luttinger Hamiltonian (1) is accurate as long as the spin-orbit split-off band is far away from the HH and LH bands. The energy gap Δ_{SO} separating the split-off band from the HH-LH manifold is of the order of 300–800 meV for

TABLE I. Luttinger parameters and bulk Dresselhaus coefficients $[6]$ used in this work, where B_{D1} , B_{D2} , B_{D3} , and B_{D4} are in eV \AA^3 and C_D in eV \AA .

	GaAs	InAs	InSb	Si	Ge
γ_1	6.85	20.40	37.10	4.28	13.38
γ_2	2.10	8.30	16.50	0.34	4.24
γ_3	2.90	9.10	17.70	1.45	5.69
C_D	-0.0034	-0.0112	-0.0082		
B_{D1}	-81.93	-50.18	-934.8		
B_{D2}	1.47	1.26	41.73		
B_{D3}	0.49	0.42	13.91		
B_{D4}	-0.98	-0.84	-27.82		

GaAs, InAs, InSb, and Ge [\[6\]](#page-12-0). For Si, $\Delta_{SO} = 44$ meV, and thus the couplings to the split-off band must be taken into account.

B. Poisson and Schrödinger equations

We consider a single heterojunction with the interface at $z = 0$. We assume that the wave functions vanish at the interface so that in the following we restrict ourselves to $z \ge 0$ (see end of Sec. [V C](#page-10-0) on why this is a reasonable approximation for the heterojunctions studied here). Our variational calculation is based on two steps, each of which is variational in nature. First we construct the self-consistent potential $V(z)$ characterizing the heterojunction assuming a parabolic, spin degenerate, 2DHG dispersion. Then we solve H_L for $V(z)$ in a second variational calculation.

The one-dimensional charge distribution giving rise to the confinement potential $V(z)$, consists of two contributions: The hole density is $p|\psi_h(z)|^2$, where p denotes the number density of 2D holes, and $\psi_h(z)$ is the zero-node HH wave function, assuming that only the lowest subband, labeled HH1, is occupied, which is the most common case. The second contribution is the net donor concentration N_D (Ref. [\[58\]](#page-13-0)). The corresponding Poisson equation is thus:

$$
\frac{d^2}{dz^2}V(z) = -\frac{e^2}{\epsilon_s \epsilon_0} [p|\psi_h(z)|^2 + N_D],\tag{3}
$$

where ϵ_s is the dielectric constant of the semiconductor and ϵ_0 is the vacuum permittivity.

The Poisson equation (3) is solved with three boundary conditions. Firstly, the zero of energy is chosen to be at $z = 0$:

$$
V(z=0) = 0.\t\t(4)
$$

Secondly, the potential $V(z)$ is flat in the bulk, i.e., at $z \geq w$ we have

$$
\frac{d}{dz}V(z \geqslant w) = 0,\tag{5}
$$

where *w* is the width of the space charge layer [\[59\]](#page-13-0). Thirdly, the Fermi energy E_F is given by the HH1 energy at the Fermi wave vector k_F :

$$
E_F \equiv E_{H1} + E_{F1},\tag{6a}
$$

where E_{H1} denotes the subband edge $k = 0$ and E_{F1} is the in-plane kinetic energy of the holes in the HH1 subband at E_F [i.e., $E_{F1} = \hbar^2 k_F^2/(2m^*)$ in a system with a simple parabolic dispersion characterized by an effective mass *m*[∗]]. Furthermore, as the system is in equilibrium, E_F is constant throughout the system. Thus we have:

$$
V(z \ge w) = E_F + \Phi = E_{H1} + E_{F1} + \Phi, \qquad (6b)
$$

where the band bending Φ is the energy difference between the valence band edge and the Fermi energy in the bulk past the space charge layer.

The solution of the Poisson equation (3), in the Hartree approximation $[59,60]$, is given by:

$$
V(z) = V_{2DHG}(z) + V_D(z),
$$
 (7)

where $V_{2DHG}(z)$ is the electrostatic potential due to the 2D holes occupying the HH1 subband and $V_D(z)$ is due to the charged donors in the space charge layer [\[42,59\]](#page-13-0). Assuming that the wave functions $\psi_h(z)$ vanish at the interface $z = 0$, the term $V_{2DHG}(z)$ becomes

$$
V_{2DHG}(z) = \frac{pe^2}{\epsilon_s \epsilon_0} \bigg[z - \int_0^z dz' \int_0^{z'} dz'' |\psi_h(z'')|^2 \bigg], \quad (8)
$$

whereas the contribution from the space charge layer becomes

$$
V_D(z) = \frac{e^2}{\epsilon_s \epsilon_0} N_D(wz - z^2/2),
$$
\n(9)

where

$$
w \equiv \sqrt{\frac{2\epsilon_s \epsilon_0}{e^2 N_D}} \bigg[\Phi + E_{H1} + E_{F1} - \frac{e^2}{\epsilon_s \epsilon_0} p \langle z \rangle \bigg], \qquad (10)
$$

and the expectation value $\langle z \rangle$ is defined via the wave function $\psi_h(z)$.

The wave function $\psi_h(z)$ entering the Poisson equation (3) is the solution of the Schrödinger equation. Thus one usually solves the Poisson and Schrödinger equations in a selfconsistent iterative scheme $[36,42-44,46,61]$. Using $V(z)$ from the Poisson equation, the 2D hole density $p|\psi_h(z)|^2$ is obtained from the Hamiltonian, from which one then constructs a new $V(z)$ by solving the Poisson equation. The process is iterated until $V(z)$ converges. In this work, we instead employ a simplified procedure based on the self-consistent variational scheme presented in Ref. [\[49\]](#page-13-0), which yields good agreement with fully self-consistent numerical calculations. We approximate the wave functions using Fang-Howard variational wave functions, which take the form [\[47\]](#page-13-0)

$$
\psi_v(z) = 2\lambda_v^{3/2} z \exp(-\lambda_v z),\tag{11}
$$

where $v = h$ represents the zero-node HH1 wave function. For the wave functions entering $V(z)$ we neglect *k*-dependent band mixing, which has only a small effect on $V(z)$. The variational parameter λ_h is obtained by minimizing the $k = 0$ ground state HH energy E_{H1} , which, neglecting band mixing, is the sum of the diagonal matrix element of the Luttinger Hamiltonian in Eq. [\(1\)](#page-1-0) for the HH subspace and the expectation value of $V(z)$ in Eq. (7), taking into account that E_{H1} also appears in Eq. (10) .

In the second variational step, we obtain the k and spin dependent eigenfunctions $\Psi_{hk}(z)$ of the total Hamiltonian \tilde{H} = $H_L + V(z)$ for the HH1 subband by expanding $\Psi_{hk}(z)$ in terms of the lowest eigenstates of \hat{H} for $\hat{k} = 0$, when \hat{H} becomes diagonal. For the $k = 0$ HH1 and LH1 states we use the zeronode wave functions (11) with $v = h, l$. For the one-node HH2, LH₂ states we use the form [\[48,62\]](#page-13-0)

$$
\psi_w(z) = \sqrt{12}\lambda_w^{3/2}z[1 - (\lambda_v + \lambda_w)z/3]
$$

$$
\times e^{-\lambda_w z}/\sqrt{1 - \lambda_v/\lambda_w + \lambda_v^2/\lambda_w^2}.
$$
 (12)

By construction, these wave functions are orthogonal to the zero-node wave functions (11). The quantities λ_w with $w = H, L$ denote additional variational parameters. The introduction of these additional variational parameters improves the accuracy of the method $[48]$ as compared to using a single variational parameter $\lambda_v = \lambda_w$. The eigenvalues $E(\mathbf{k})$ and the corresponding *k* dependent expansion coefficients are obtained by diagonalizing the matrix \tilde{H} , whose elements are given as

$$
\tilde{H}_{\nu\nu'} = \langle \nu | H_L + V(z) | \nu' \rangle, \tag{13}
$$

where $|\nu\rangle$ denotes the wave functions [\(11\)](#page-2-0) and [\(12\)](#page-2-0). The two lowest eigenenergies of the 8×8 matrix [\(13\)](#page-2-0) correspond to the dispersion of the spin-split $HH1_±$ subband. In certain regimes these eigenenergies can also be obtained analytically to a good approximation, as shown in the next section.

C. Schrieffer-Wolff transformation and Rashba spin splitting

It is well established that, under certain circumstances, the HH1 subbands may be described by an effective 2×2 Hamiltonian formulated as an expansion in powers of the wave vector *k* [\[6\]](#page-12-0). The general kinematic structure of such a reduced Hamiltonian can be found from the theory of invariants [\[63\]](#page-13-0). Retaining terms up to fourth order in k and postponing the discussion of Dresselhaus terms until Sec. [IV,](#page-7-0) the effective 2×2 Hamiltonian for the subspace spanned by the spin-split HH1 subbands takes the form [\[6\]](#page-12-0)

$$
H_{2\times 2} = [Ak^{2} - Bk^{4} - d(k_{+}^{2} - k_{-}^{2})^{2}]1_{2\times 2}
$$

+ $i\alpha_{R1}(k_{+}\sigma_{+} - k_{-}\sigma_{-}) + i\alpha_{R2}(k_{+}^{3}\sigma_{-} - k_{-}^{3}\sigma_{+})$
+ $i\alpha_{R3}(k_{+}\sigma_{+} - k_{-}\sigma_{-})k^{2}$. (14)

Up to fourth order in k , the Hamiltonian $H_{2\times 2}$ includes all possible terms. The terms weighted by *A*,*B*, and *d* describe the orbital motion, whereas the terms weighted by α_{Ri} represent the Rashba spin-orbit coupling. More specifically, the term Ak^{2} describes the usual parabolic component of the dispersion, while the term $Bk⁴$ represents the lowest-order nonparabolic correction, which remains isotropic and often has a sizable effect on the dispersion of hole systems, as discussed below. The term $d(k_+^2 - k_-^2)^2$] characterizes the warping of the energy contours. As can be seen from Eq. (14) , the most general form of the Rashba SO coupling includes a term linear in *k* proportional to α_{R1} and two terms cubic in *k* weighted by α_{R2} and α_{R3} . The terms weighted by *d*, α_{R1} , and α_{R3} break axial symmetry, which implies that these prefactors are zero when the axial approximation $\zeta = 0$ is employed in Eq. [\(1\)](#page-1-0). Moreover, to lowest order the prefactor α_{R1} stems from the $\mathbf{k} \cdot \mathbf{p}$ coupling between the bonding and antibonding atomic *p* orbitals (the latter give rise to the first excited conduction band) [\[6\]](#page-12-0). For the systems discussed here, the *k*-linear Rashba term thus contributes not more than ∼1% of the total spin splitting, consequently this term is not considered further.

To obtain analytical expressions for the prefactors appearing in Eq. (14), one can apply a Schrieffer-Wolff transformation $[50]$ (Löwdin perturbation theory $[51]$ $[51]$) to Eq. [\(13\)](#page-2-0). The coefficients *A*, *B*, *d*, α_{R2} , and α_{R3} are evaluated to lowest order in the perturbation expansion: Explicit expressions are given in the Appendix. In the axial approximation, with $d = \alpha_{R1}$ $\alpha_{R3} = 0$, the dispersion relation for the HH1 subband takes the simple form

$$
E_{\pm}(k) = Ak^2 - Bk^4 \pm Ck^3,
$$
 (15)

where $C \equiv |\alpha_{R2}|$. Numerical values for *A, B,* and $C \equiv |\alpha_{R2}|$ for typical experimental densities are given in Table II. The resulting subbands, denoted as HH1+ and HH1−, have unequal subband populations p_{\pm} and density of states (DOS) effective masses m_{\pm} .

TABLE II. Values for the material- and structure-dependent coefficients *A* (in 10^{-16} meV m²), *B* (in 10^{-32} meV m⁴), and $C \equiv |\alpha_{R2}|$ (in 10⁻²⁴ meV m³) in the dispersion relation $E(k) =$ $Ak^{2} - Bk^{4} \pm Ck^{3}$, the energies E_{H1} , Δ_{11}^{HL} , $E_{F1} \equiv E_{F} - E_{H1}$ (in meV), and spin splitting Δp for GaAs inversion and accumulation layers with $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$. The densities *p* and Δp are in multiples of 10¹⁵ m[−]2. For inversion layers, the Schrieffer-Wolff approximation fails when the density exceeds 2.5×10^{15} m⁻². For accumulation layers it is valid up to a density of 0.5×10^{15} m⁻².

Density p	A	B	C	E_{H1}	Δ_{11}^{HL}	\mathcal{E}_{F1}	Δp^a	Δp^b
				Inversion layer				
0.5	2.99	0.27	0.53	16.12	8.32	0.91	0.10	0.14
1.0	2.92	0.25	0.58	19.24	8.73	1.67	0.18	0.23
1.5	2.86	0.23	0.61	22.12	9.04	2.30	0.25	0.30
2.0	2.80	0.22	0.64	24.83	9.27	2.83	0.31	0.35
2.5	2.74	0.20	0.67	27.40	9.46	3.26	0.36	0.39
3.0				29.85	9.62	3.64	0.41	0.42
				Accumulation layer				
0.5	2.55	0.69	1.15	8.84	2.68	0.70	0.31	0.33
1.0				12.85	2.87	1.06	0.48	0.43
1.5				16.30	2.97	1.36	0.55	0.47
2.0				19.42	3.04	1.66	0.58	0.49
2.5				22.31	3.08	1.96	0.60	0.50
3.0				25.02	3.12	2.28	0.60	0.51

^aCalculated using the variational method introduced in Sec. [II B](#page-2-0) and a numerical diagonalization of Eq. [\(13\)](#page-2-0).

bCalculated using the fully numerical method devised in Refs. [\[49,56\]](#page-13-0).

To characterize the strength of the Rashba spin-orbit interaction, we use the dimensionless quantity

$$
\Delta p \equiv \frac{|p_+ - p_-|}{p},\tag{16}
$$

where p_{\pm} denotes the spin subband densities with $p = p_{+} + p_{+}$ *p*−. Experimentally, the quantity Δp is usually inferred by analyzing the beating pattern of Shubnikov-de-Haas (SdH) oscillations [\[6,](#page-12-0)[61,64–66\]](#page-13-0). For the single heterojunctions studied here, it can be manipulated by tuning the density *p*. At low temperature, p_{\pm} is given by

$$
p_{\pm} = \int \frac{d^2k}{(2\pi)^2} \theta[E_F - E_{\pm}(k)],\tag{17}
$$

where θ is the Heaviside step function. For hole systems with isotropic Fermi contours, Eq. (17) becomes

$$
p_{\pm} = \frac{k_{F\pm}^2}{4\pi},
$$
 (18)

where $k_{F\pm}$ denotes the Fermi wave vectors for the spin-split bands. Using the coefficients in Eq. (15) , this translates to

$$
p_{\pm} = \frac{p}{2} \pm \frac{pc}{\sqrt{2}X} \sqrt{p\pi \left(6 - \frac{4}{X}(1 - 4p\pi b)\right)},
$$
 (19)

where

$$
X \equiv 1 - 4p\pi b + \sqrt{1 - 4p\pi (2b + c^2 - 4b^2 p\pi)} \tag{20a}
$$

$$
b \equiv B/A \quad c \equiv C/A \tag{20b}
$$

so that the Rashba spin splitting Δp is given by

$$
\Delta p = \frac{\sqrt{2}c}{X} \sqrt{p\pi \left(6 - \frac{4}{X}(1 - 4p\pi b)\right)}.
$$
 (21)

The DOS effective masses $m_±$ of the spin-split subbands at the Fermi energy E_F takes the form

$$
\frac{m_{\pm}}{m_0} = \frac{\mu}{2\pi} \int d^2k \,\delta[E_F - E_{\pm}(k)],\tag{22}
$$

where δ is the Dirac δ function. For isotropic bands, this becomes

$$
\frac{m_{\pm}}{m_0} = \mu \left(\frac{1}{k} \frac{dE_{\pm}(k)}{dk} \right)_{k=k_{F_{\pm}}}^{-1},\tag{23a}
$$

which can be further evaluated for the dispersion [\(15\)](#page-3-0)

$$
\frac{m_{\pm}}{m_0} = \frac{\mu}{2A - 4Bk_{F\pm}^2 \pm 3Ck_{F\pm}},
$$
 (23b)

where the Fermi wave vectors $k_{F\pm}^2$ for a given total density *p* are evaluated using Eqs. [\(18\)](#page-3-0) and [\(19\)](#page-3-0).

D. Comparison with numerical results

To illustrate our approach, Fig. 1 shows a comparison between the results obtained using our variational calculation and those obtained in Ref. [\[42\]](#page-13-0) using an iterative Fang-Howard and Luttinger Hamiltonian scheme. The 2DHG considered in Ref. [\[42\]](#page-13-0) is a hole GaAs single heterojunction with a density of 5×10^{15} m⁻² and a net dopant concentration of $N_D - N_A =$ 1×10^{21} m⁻³ [\[67\]](#page-13-0). The confinement potential *V*(*z*) obtained from solving Eq. [\(3\)](#page-2-0) is shown in the inset of Fig. 1. Our variational approach slightly overestimates the ground state heavy hole energy E_{H1} compared to the numerical results, as expected for a variational calculation. The corresponding DOS effective masses, obtained from Eqs. [\(13\)](#page-2-0) and (23), are $m_+ = 0.52$ *m*₀ and $m_- = 0.12$ *m*₀. These numbers are in close agreement with the full-numerical calculations in Ref. [\[42\]](#page-13-0), where $m_{+} = 0.46$ m_{0} and $m_{-} = 0.12$ m_{0} . We also find good agreement between our variational results for the Rashba spin splitting Δp and those obtained using the numerical method described in Refs. [\[49,56\]](#page-13-0) (see Table [II\)](#page-3-0).

While the simple dispersion in Eq. (15) affords a convenient way to calculate DOS properties such as effective masses and spin splittings, the validity of the Schrieffer-Wolff transformation is limited to certain sets of densities and dopant concentrations. When the separation between subband energies is small, the coefficients *A,B*, and *C* are overestimated (refer to the Appendix for the dependence of *A,B*, and *C* on the energy separations) and one can no longer describe the HH1+ and HH1– bands by Eq. (15) . For example, using the values in Table [II,](#page-3-0) for a GaAs inversion layer with $p =$ 2.5×10^{15} m⁻² and $N_D - N_A = 3 \times 10^{20}$ m⁻³, the Rashba term Ck^3 becomes larger than $Ak^2 - Bk^4$ so that the heavier HH1+ band bends down for $k > k_{\text{bend}} = 1.7 \times 10^8 \text{ m}^{-1}$. As $k_{F+} = 1.5 \times 10^8$ m⁻¹ is very close to HH1+ turning point k_{bend} , the HH1+ dispersion is almost flat at k_{F+} , which means that the $HH1+$ effective mass is overestimated. The Schrieffer-Wolff results for GaAs inversion layers with $N_D - N_A$ = 3×10^{20} m⁻³ and $p \gtrsim 2.5 \times 10^{15}$ m⁻² are invalid, and we

FIG. 1. Comparison between the variational method adopted in this work (solid lines) and the numerical results in Ref. [\[42\]](#page-13-0) (dotted lines), showing the inversion-asymmetric potential (inset) and the spin-split HH1+ and HH1− subbands. The dispersion was obtained by a numerical diagonalization of Eq. [\(13\)](#page-2-0). The system is a GaAs single heterojunction (SHJ) with a 2D hole density $p = 5 \times 10^{15}$ m⁻² and a net dopant concentration of $N_D - N_A = 1 \times 10^{21}$ m⁻³.

resort to a numerical diagonalization of Eq. [\(13\)](#page-2-0) to obtain the energy dispersion. Note that unlike Eq. [\(15\)](#page-3-0), dispersion curves obtained from a numerical diagonalization of Eq. [\(13\)](#page-2-0) include spin splitting to all orders in *k*. Therefore, the dispersions from Eq. [\(13\)](#page-2-0) are a reasonable approximation for any wave number *k* (and hence for any density *p*), so long as only the zero-node heavy hole subbands $HH1\pm$ are occupied. Indeed, in all the structures discussed in this paper, at the densities we consider, only the HH1+ and HH1− bands are occupied (Tables [II,](#page-3-0) [IV,](#page-7-0) and V).

III. FULL RASHBA SPIN SPLITTINGS AND EFFECTIVE MASSES

Generally, spin splittings for holes are much larger than for electrons: For electron inversion layers with the same doping concentration, the spin splittings are about two orders of magnitude smaller [\[6\]](#page-12-0). Below we focus first on GaAs inversion and accumulation layers, which are distinguished by the position of the Fermi level E_F towards the substrate: In an inversion (accumulation) layer, E_F is pinned near the conduction (valence) band. Finally, we discuss the spin splitting and effective masses in InAs, InSb, Ge, and Si hole inversion layers.

FIG. 2. Comparison of the (a) band structure and (b) confinement potential for a GaAs 2D hole system in an inversion and an accumulation layer with a density of 1×10^{15} m⁻², where the net dopant concentration is $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$ and $N_D = 3 \times 10^{20} \text{ m}^{-3}$, respectively. Due to the magnitude of the band bending, the potential for an inversion layer is steeper than that of an accumulation layer.

A. GaAs inversion and accumulation layers

The type of background dopant determines the location of the Fermi level E_F in the substrate and hence the amount of valence band bending Φ at the heterojunction interface [Fig. $2(a)$]. For an accumulation layer, the band bending is less pronounced than for an inversion layer. Consequently, in accordance to Gauss' law, for the same density, the confinement potential for an accumulation layer is less steep than in an inversion layer [Fig. $2(b)$]. This means that the spacing between subbands is smaller (Fig. 3 and Table [II\)](#page-3-0), hence, according to the expressions in the Appendix, the Rashba SO interaction is stronger in an accumulation layer.

The variational approach followed in this work is designed to yield the energies of the $HH1\pm$ subbands. Although it is not intended to give reliable values for the higher subbands, some qualitative observations can be made concerning the HH1-LH1 spacing in inversion and accumulation layers. In Table [II](#page-3-0) and Fig. 3, the HH1-LH1 separation $\Delta_{11}^{HL} \equiv |E_{H1} - E_{L1}|$ at $k = 0$ is smaller in an accumulation layer than in an inversion layer at the same density (Table II). Since the Rashba coefficient increases as the subbands get closer together, it is larger in an accumulation than an inversion layer at a given density (see Appendix for the dependence of the Rashba coefficient on the subband energy separations). The Rashba coefficient increases with density, which is consistent with the experimental results reported in Ref. [\[65\]](#page-13-0) (Fig. 4).

We compare the different trends in the Rashba spin splitting Δp as a function of density in inversion and accumulation layers in Figs. $5(a)$ and $5(b)$. As expected, Δp increases with density in both inversion and accumulation layers, consistent with the experimental observations of Ref. [\[66\]](#page-13-0). However, there is a difference between the dependence of Δp on density for the inversion and accumulation layers. For inversion layers with $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$ and *p* ranging from 5×10^{14} m⁻² to 3×10^{15} m⁻², the Rashba spin splitting increases with density in an almost linear fashion. In the accumulation layer counterparts, however, the spin splitting increases in an almost linear fashion at lower densities but

FIG. 3. Dispersion for 2D holes in a GaAs inversion layer (black lines) and a GaAs accumulation layer (red lines) with a density of 1×10^{15} m⁻². The doping concentration is $N_D - N_A =$ 3×10^{20} m⁻³ for the inversion layer and $N_D = 3 \times 10^{20}$ m⁻³ for the accumulation layer. The SW dispersion closely matches the numerical diagonalization results (black solid lines) up to k_F for the inversion layer (black dash-dotted lines). In the accumulation layer, the HH1-LH1 separation is closer than in the inversion layer, so that the HH1-LH1 anticrossing occurs at a lower *k* compared to the inversion layer, and the Schrieffer-Wolff method fails at this density (red dash-dotted lines).

saturates at higher densities. This feature can be attributed to the fact that, in an accumulation layer, the HH1-LH1 separation is smaller than in an inversion layer (see Appendix) such that the HH1− band anticrosses with the next highest energy

FIG. 4. The ratio of *C* to *A* in the dispersion relation of $E(k) = Ak^2 - Bk^4 \pm Ck^3$ for GaAs inversion layers with a dopant concentration of 3×10^{20} m⁻³. The trend we predict agrees with the experimental results in Ref. [\[65\]](#page-13-0).

FIG. 5. Rashba spin splitting Δp for holes in GaAs (a) inversion layers and (b) accumulation layers compared to the experimental results reported in Ref. [\[66\]](#page-13-0). The saturation of the spin splitting in accumulation layers is due to the HH1-LH1 anticrossing. In (c) we show the effective masses m_{\pm} for GaAs inversion layers and in (d) for accumulation layers. The heavier mass m_+ increases with density whereas *m*[−] is nearly density independent. In accumulation layers *m*⁺ saturates with density due to the proximity of k_F to the HH1-LH1 anticrossing. The doping concentration is $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$ for the inversion layers and $N_D = 3 \times 10^{20}$ m⁻³ for the accumulation layers.

subband (LH1) at a lower *k* than in an inversion layer at the same density (Fig. [3\)](#page-5-0). Consequently, for an accumulation layer with a higher density, *kF*[−] can be near the HH1-LH1 anticrossing. In the anticrossing region the HH1− band is pushed down in energy, hence the separation between the HH1− and HH1+ bands is reduced.

The strength of the Rashba SOI is also evident in the difference between the HH1+ and HH1− effective masses. Figures $5(c)$ and $5(d)$ show the variation of the effective HH1+ and HH− masses with density. As the figures show, there is a remarkable distinction in the dependence of the *m*⁺ and *m*[−] on density for the inversion and accumulation layers we consider. For the inversion layers studied here, the $HH1+$ effective mass also increases in an almost linear fashion with density, whereas *m*_− is essentially constant. The increase in the difference between the effective masses *m*⁺ and *m*[−] with density again implies that the strength of the Rashba SO interactions increase with density. Comparison between numerical and SW results for inversion layers with densities up to $p = 2 \times 10^{15}$ m⁻² shows the SW approach works well. However, at a density $p = 2.5 \times 10^{15}$ m⁻², the HH1+ effective mass, calculated using the values in Table [II,](#page-3-0) is overestimated [Fig. $5(c)$], which signifies that SW breaks down. For the accumulation layers considered here, however, with densities ranging from 5×10^{14} m⁻² to 3×10^{15} m⁻², the HH1+ effective mass increases, then saturates, and becomes increasingly lighter as the density increases. The SW approach is only valid for densities up to \sim 5 × 10¹⁴ m⁻² for the accumulation layers.

That SW fails at lower densities in accumulation layers is expected as the separation between subband energies in an accumulation layer is smaller than in an inversion layer at a given density. On the other hand, the HH1− effective mass is essentially constant at lower densities but increases slightly at a higher density of $p = 3 \times 10^{15}$ m⁻² [Fig. 5(d)]. The dependence of the HH1+ effective mass m_+ on density can be explained by examining a typical accumulation layer band structure (Fig. [3\)](#page-5-0). As the density increases, the Fermi energy increases, and the fact that the HH1+ band curvature becomes slightly steeper at $k > 1 \times 10^8$ m⁻¹ means that the HH1+ effective mass at the Fermi energy decreases slightly with increasing density. The behavior of HH1−, on the other hand, reflects the anticrossing between the HH1-LH1 bands. As explained above, at a sufficiently high density, k_F can be very close to the HH1-LH1 anticrossing. In this region, the HH1− band becomes flatter, hence the corresponding effective mass *m*[−] increases slightly.

B. Zincblende materials and Ge inversion layers

Figure 6 shows the spin splitting Δp and effective masses m_{+} as functions of density for inversion layers in GaAs, Ge, InSb, and InAs. The strength of the SO interaction is reflected in the difference between the subband populations and effective masses of the HH1+ and HH1– subbands. In Ge, InAs, and InSb, the Rashba spin splitting and $HH1+$ effective mass m_+ saturate at a lower density compared to GaAs. The saturation of the spin splitting and m_{+} indicates that HH1 and LH1 are close enough such that the HH1-LH1 anticrossing occurs near E_F . This indicates that Ge, InSb, and InAs exhibit a stronger SO interaction than GaAs, which is consistent with the known fact that SO coupling is stronger in compounds containing elements with larger atomic numbers *Z*. The corollary of this is that SW breaks down at different densities for various materials, which implies that Eq. [\(15\)](#page-3-0) is valid only for low densities for heavy materials. For example, using the values for *A*, *B*, and *C* listed in Table [III,](#page-7-0) one can get a reasonable estimate for the HH1+ and HH1− effective masses of InAs and Ge at a density of 5×10^{14} m⁻². The SW effective masses of InAs are obtained as $m_{+} = 0.094$ m_{0} and $m_0 = 0.036$ *m*₀, which are close to $m_+ = 0.113$ *m*₀ and $m_- =$ 0.036 m_0 obtained by numerically diagonalizing Eq. [\(13\)](#page-2-0). For Ge, the SW effective masses are $m_{+} = 0.108 m_0$ and

FIG. 6. (a) Rashba spin splitting Δp and (b) effective masses m_{\pm} for various inversion layers. The results are obtained from a numerical diagonalization of Eq. [\(13\)](#page-2-0), with $N_D - N_A = 3 \times 10^{20}$ m⁻³.

TABLE III. Values for material-dependent constants *A* (in 10[−]¹⁶ meV m2), *B* (in 10[−]³² meV m4), and *C* (in 10[−]²⁴ meV m3) in the dispersion relation $E(k) = Ak^2 - Bk^4 \pm Ck^3$ for GaAs, InAs, and Ge inversion layers with $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$. All densities below are in 1015 m[−]2. The Schrieffer-Wolff approach breaks down for densities exceeding 0.5×10^{15} m⁻² for InAs as well as for Ge.

		GaAs			InAs			Ge		
Density	A		$B \quad C$	$A \quad B$		C	\overline{A}	B	C	
0.5	2.99					0.27 0.53 9.02 1.94 3.60 5.84		1.13	- 1.60	
1.0		2.92 0.25 0.58								
1.5		2.86 0.23 0.61								
2.0	2.80		$0.22 \quad 0.64$							
2.5	2.74		$0.20 \quad 0.67$							
3.0										

 $m_$ = 0.059 m_0 , which are in excellent agreement with the numerical diagonalization results $m_{+} = 0.118 m_0$ and $m_{-} =$ 0.059 m_0 . For InSb, the SW results are valid at densities \lesssim 4 \times 10^{14} m⁻². Using the values in Table V for 3.3×10^{14} m⁻², the SW effective masses are $m_{+} = 0.087$ m_0 and $m_{-} = 0.019$ m_0 , which also compare well with the numerical diagonalization results of $m_{+} = 0.095$ m_{0} and $m_{-} = 0.020$ m_{0} .

C. Silicon inversion layers

In Si, the hole density is usually of the order of 10^{16} m⁻². so that $k_F \sim 5 \times 10^{-8}$ m⁻¹ [\[68\]](#page-13-0). In this regime, the term Ck^3 in Eq. [\(15\)](#page-3-0) becomes dominant, SW breaks down, and only numerical diagonalization gives reliable results. As pointed out above, due to the large cubic terms, it is necessary to include the spin-orbit split-off band, hence the Luttinger Hamiltonian is now projected onto a 12×12 subspace. In this work, we consider densities up to 2×10^{16} m⁻² (Table VI), since for larger densities the higher subband will start to populate, violating our initial assumption.

Due to the significant size of the prefactor ζ in Eq. [\(1\)](#page-1-0) the SO interaction and hence spin splitting [\(17\)](#page-3-0) is anisotropic: It has a minimum along the (100) direction and a maximum along (110). The anisotropy of the SO strength is shown in Fig. $7(a)$: The difference in the Fermi wave vectors k_{F+} and k_{F-} is

TABLE IV. The energy spacing (in meV) between the HH1 and LH1 levels, as well as the Fermi energy for GaAs, InAs, and Ge hole inversion layers at various densities (in 10^{15} m⁻²) and $N_D - N_A$ = 3×10^{20} m⁻³. In all cases considered below LH1 is far above the Fermi energy, thus validating the assumption that only the HH1 band is occupied.

GaAs			InAs	Ge.			
Density E_{H1} Δ_{11}^{HL} E_{F1} E_{H1} Δ_{11}^{HL} E_{F1} E_{H1} Δ_{11}^{HL} E_{F1}							
0.5					16.12 8.32 0.91 12.87 10.87 2.27 14.72 7.74 1.63		
1.0					19.24 8.73 1.67 16.38 11.42 3.47 18.04 8.16 2.72		
1.5					22.12 9.04 2.30 19.54 11.79 4.33 21.08 8.46 3.50		
2.0					24.83 9.27 2.83 22.47 12.06 5.12 23.91 8.69 4.18		
2.5					27.40 9.46 3.26 25.21 12.28 5.88 26.58 8.87 4.80		
3.0					29.85 9.62 3.64 27.82 12.45 6.61 29.13 9.02 5.43		

TABLE V. The energy spacing (in meV) between the HH1 and LH1 levels, as well as the Fermi energy E_{F1} for an InSb hole inversion layer with $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$. At the densities below (in 10^{15} m⁻²), only the HH1+ and HH1- subbands are occupied. Here the material- and structure-dependent coefficients *A*, *B*, and *C* are listed for $p = 0.33 \times 10^{15} \text{ m}^{-2}$. *A* is given in multiples of 10^{-16} meV m², *B* in multiples of 10^{-32} meV m⁴, and *C* in multiples of 10^{-24} meV m³.

		InSb								
Density	E_{H1}	Δ_{11}^{HL}	E_{F1}	A	B	C				
0.33	9.69	11.72	2.26	15.76	5.84	10.06				
0.50	10.87	12.00	2.87							
0.67	12.02	12.23	3.33							
0.83	13.05	12.42	3.74							
1.00	14.11	12.59	4.15							

 6×10^6 m⁻¹ along (100) and 1×10^8 m⁻¹ along (110). The equienergy lines for Si ground state HHs [Figs. $7(b)$ and $7(c)$] show that Si valence bands are in general warped, and the warping becomes more pronounced as the density increases.

The spin splitting in Si [Fig. $7(d)$] also increases with density, which again reflects the fact that the strength of spin-orbit interaction increases with density. However, spin splittings in Si holes are smaller than in both zincblende materials and Ge, discussed in the previous section. This is expected since amongst all the materials considered in this paper, Si is the lightest element with atomic number $Z = 14$. Nevertheless, the spin splitting in Si hole systems is considerably larger than in Si electron systems.

IV. DRESSELHAUS SPIN-ORBIT INTERACTION

In semiconductor crystals which lack a center of inversion, such as GaAs, InAs, and InSb, Dresselhaus spin-orbit interaction terms are present [\[29\]](#page-12-0). In this section, we discuss the spin splitting as a function of density and material and material parameters in GaAs, InAs, and InSb accounting for both the Rashba and the Dresselhaus interactions. The Dresselhaus spin-orbit coupling in bulk semiconductor hole systems is characterized by a variety of terms, whose relative importance depends on the parameter regime under study. In what follows we establish a hierarchy among these terms in heterojunctions at realistic experimental densities.

TABLE VI. Si hole inversion layer energy level spacings (in meV) for various densities (in 10^{16} m⁻²), showing that only the HH1+ and HH1− bands are occupied. The dopant concentration $N_D - N_A$ for the inversion layer is 3×10^{23} m⁻³.

Density	E_{H1}	Δ_{11}^{HL}	E_{F1}
0.83	152.81	15.36	6.62
1.00	157.13	15.51	7.64
1.33	165.36	15.75	9.36
1.67	173.63	15.96	10.87
2.00	181.48	16.12	12.17

FIG. 7. Panel (a) shows the dispersion for Si holes along (100) and (110), indicating that the strength of the spin-orbit interaction is a maximum along (110) and minimal along (100). The inset shows that the spin splitting is only 6×10^6 m⁻² along (100), compared to 1×10^8 m⁻² along (110). Here, the density is 2×10^{16} m⁻² and the doping concentration is $N_D - N_A = 3 \times 10^{23}$ m⁻³. The equienergy contours (separated by 5 meV) for the (b) HH1+ and (c) HH1− subbands in the inversion layer discussed in panel (a) show a strong valence band warping. Panel (d) shows the spin splitting for Si hole inversion layers at various densities with $N_D - N_A = 3 \times 10^{23} \text{ m}^{-3}$. Here, the dispersion is obtained by numerically diagonalizing Eq. [\(13\)](#page-2-0).

Up to third order in k , the Dresselhaus SO interaction in bulk hole systems is described by the following invariants [\[6,29\]](#page-12-0)

$$
H_D = -\frac{2}{\sqrt{3}} C_D [k_x \{J_x, J_y^2 - J_z^2\} + cp]
$$

\n
$$
- B_{D1} [k_x (k_y^2 - k_z^2) J_x + cp]
$$

\n
$$
- B_{D2} [k_x (k_y^2 - k_z^2) J_x^3 + cp]
$$

\n
$$
- B_{D3} [k_x (k_y^2 + k_z^2) \{J_x, J_y^2 - J_z^2\} + cp]
$$

\n
$$
- B_{D4} [k_x^3 \{J_x, J_y^2 - J_z^2\} + cp],
$$
\n(24)

where J_i , $i = x, y, z$ represent the spin-3/2 matrices, and cp denotes cyclic permutation. We have one invariant linear in k , quantified by the coefficient C_D , while the remaining four invariants are cubic in k . Note that the coefficient C_D is to be distinguished from the coefficient *C* resulting from the SW transformation. The values of C_D , B_{D1} , B_{D2} , B_{D3} used in this work are given in Table [I.](#page-1-0) The Dresselhaus spin-orbit coupling of bulk hole systems typically reflects the competition between the terms with prefactors C_D and B_{D1} , while the contributions due to the remaining terms are approximately two orders of magnitude smaller in magnitude (see Table [I\)](#page-1-0) and can often be neglected [\[6\]](#page-12-0).

According to the theory of invariants [\[63\]](#page-13-0), the effective 2×2 Hamiltonian representing the Dresselhaus spin-orbit coupling in the HH1 \pm subspace up to third order in *k* takes the form

$$
H_{D,2\times 2} = -\gamma_D(k_+ \sigma_- + k_- \sigma_+)
$$

- $\beta_{D1}(k_+ k_- k_+ \sigma_- + k_- k_+ k_- \sigma_+)$
- $\beta_{D2}(k_+^3 \sigma_+ + k_-^3 \sigma_-).$ (25)

To evaluate the prefactors in Eq. (25) , we first project Eq. (24) onto the subspace spanned by the zero- and one-node hole states. Then, by means of the SW transformation, we obtain an effective 2×2 Hamiltonian describing the Dresselhaus interactions in the $HH1\pm$ subspace, so that the prefactors can be obtained by comparison with Eq. (25). Analytical expressions for the prefactors γ_D , β_{D1} , and β_{D2} are listed and discussed in the Appendix, complemented by typical numerical values.

Strictly speaking, our treatment of Dresselhaus SO coupling relies implicitly on a hierarchy of steps. The first step, carried out in detail in Ref. [\[6\]](#page-12-0), is the projection of the bulk 14×14 Hamiltonian yielding the bulk 4×4 Luttinger Hamiltonian, which includes all the possible bulk Dresselhaus terms (24) with the prefactors listed in Table [I.](#page-1-0) In the second step the bulk Luttinger Hamiltonian is projected onto the subspace spanned by the (spin-split) ground state HH1 subband. In this sense our approach is comparable to that of Ref. [\[34\]](#page-13-0) which directly projects from the bulk 14×14 Hamiltonian to the HH1 subspace.

When only the dominant Rashba and Dresselhaus terms are retained the energy dispersion relation becomes

$$
E(k) = Ak^2 - Bk^4 + 2dk^4(1 - \cos 4\phi) \pm \sqrt{((\alpha_{R2})^2 + \beta_{D1}^2)k^6 + 2\beta_{D1}\gamma_D k^4 + \gamma_D^2 k^2 - 2|\alpha_R^{(2)}|(k^2\beta_{D1} + \gamma_D)k^4\sin 2\phi}
$$
 (26)

where $\phi = \arctan (k_y / k_x)$. We find that, for the single heterojunctions studied here, the Dresselhaus terms contribute at most one order of magnitude less than the Rashba term to the total spin splitting at the Fermi energy. Nevertheless, the contribution due to the Dresselhaus coupling is visible in the term proportional to sin 2 ϕ , which causes the difference $\Delta k_F \equiv$ $|k_{F+} - k_{F-}|$ to be anisotropic in *k*. The anisotropic terms in Eq. (14) are likewise sizable. In accordance to Eq. (25) , the extrema of Δk_F occur when $\phi = \pi/4$ or $\phi = 3\pi/4$. The anisotropy of the spin splitting, which we define here as the ratio $\kappa \equiv \Delta k_F(\phi = 3\pi/4)/\Delta k_F(\phi = \pi/4)$, depends on the density as well as on material-specific parameters, as Fig. 8 shows. For example, one can infer from Fig. $8(a)$ that $\kappa = 2.24$ for a GaAs inversion layer with $p = 5 \times 10^{14}$ m⁻² and $\kappa = 1.77$ with $p = 3 \times 10^{15}$ m⁻². The fact that the anisotropy *κ* decreases with density implies that the Rashba coefficient increases faster with density than the cubic and linear Dresselhaus coefficients combined (see Appendix). Comparing different materials [Fig. $8(b)$], one can deduce that $\kappa = 1.89$, $\kappa =$ 1.42, $\kappa = 1.48$ for GaAs, InAs, and InSb inversion layers with $p = 1 \times 10^{15}$ m⁻², respectively. This implies that, amongst the materials considered here, the effect of Dresselhaus SOI is weakest in InAs but strongest in GaAs, an observation that is rather counterintuitive considering the relative values of the bulk Dresselhaus prefactors in these materials. Indeed, this reflects the fact that the relative importance of specific spin-orbit interaction terms is determined by their ratio to the spin-independent terms in the dispersion relation, rather than the absolute magnitude of their corresponding numerical prefactors.

V. DISCUSSION

A. Breakdown of the SW method

The analysis presented in this work is consistent with the known general trend that SO coupling in 2DHGs increases with the carrier number density. We have characterized the strength of the Rashba SO interaction using both the relative spin-split population difference Δp and the effective masses m_+ . We have found that Δp and m_+ are expected to increase almost linearly with density whereas *m*[−] is approximately constant. While the simple dispersion found using the SW transformation provides a good approximation for certain systems, e.g., GaAs inversion layers with $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$ and densities $p < 2.5 \times 10^{15} \text{ m}^{-2}$, there are many parameter regimes in which SW breaks down. The fact that the simple dispersion can fail for realistic parameters challenges the current theories on electric dipole spin resonance (EDSR) [\[55\]](#page-13-0), spin conductivity $[32,37,39,52]$, hole spin helix $[40]$, and *Zitterbewegung* [\[38,41\]](#page-13-0), which assume that hole spin-orbit coupling is simply cubic in wave vector.

B. Inversion vs accumulation layers

Comparing GaAs inversion and accumulation layers, the strength of the Rashba SO interaction is very sensitive to the type of background dopant. Owing to the fact that the confinement potential for inversion layers is steeper than that for accumulation layers, the HH1-LH1 splitting for inversion layers is larger than that for accumulation layers.

FIG. 8. Polar plot of spin splitting $\Delta k_F \equiv |k_{F+} - k_{F-}|$, which is anisotropic in k , for (a) GaAs inversion layers at various densities and (b) GaAs, InAs, InSb inversion layers at $p = 1 \times 10^{15}$ m⁻² in the presence of both Rashba and Dresselhaus interactions. Here, the dopant concentration is $N_D - N_A = 3 \times 10^{20} \text{ m}^{-3}$. The angle ϕ is defined as $\phi = \arctan (k_y / k_x)$. The results here are obtained from a numerical diagonalization of Eq. [\(13\)](#page-2-0).

Consequently, the strength of SO interaction is stronger in an accumulation layer than in an inversion layer (see Appendix for the relation between Rashba or Dresselhaus coefficients and the HH1-LH1 splitting).

In both inversion and accumulation layers, the spin splitting Δp increases with density, yet the dependence of the spin splitting and effective mass on density is quite different. In an accumulation layer, both Δp and m_+ increase almost linearly with *p* at low densities, and both saturate at higher densities. These trends can be attributed to the fact that in an accumulation layer, the HH1-LH1 anticrossing occurs at a lower *k*, so that the spin splitting near the anticrossing is reduced. Correspondingly, in accumulation layers SW breaks down at much lower densities than in inversion layers.

We compare our results for GaAs, InAs, InSb, Ge, and Si inversion layers. The spin splitting and effective mass profiles show that the Rashba SO interaction is stronger in Ge, InAs, and InSb than in GaAs, though the Fermi contour remains isotropic in the absence of Dresselhaus terms. The Si dispersion, on the other hand, is very anisotropic owing to the large cubic contribution stemming from the bulk Luttinger Hamiltonian [\[6\]](#page-12-0). The term $\alpha \zeta$ in the Luttinger Hamiltonian Eq. [\(1\)](#page-1-0) is sizable and is responsible for the highly anisotropic Fermi contour. The spin splitting is a maximum along the (110) and $(1\bar{1}0)$ directions but a minimum along (100) and (010) .

We have considered the Dresselhaus spin-orbit interaction in zincblende semiconductors GaAs, InAs, and InSb. While the magnitude of BIA is always much smaller than SIA, it causes the spin splitting to be anisotropic in k . Figure 8 shows that the spin splitting is a maximum along (110) but a minimum along (110), in accordance with Eq. (26) (26) .

C. Relative importance of the *k***-linear and** *k***-cubic spin-orbit interaction terms**

In general, spin-orbit interaction terms in the system Hamiltonian are characterized by invariants, with each invariant associated with a specific power of *k* and having a particular functional form. The fact that spin-orbit interactions preserve time reversal symmetry [\[6\]](#page-12-0) implies that only odd powers of the wave vector are admissible. To date spinorbit interaction terms of orders linear [\[6,30](#page-12-0)[,32–35\]](#page-13-0) and cubic in k $[4,6,7,27,29,32,35-41]$ $[4,6,7,27,29,32,35-41]$ have received the most attention. The identification of the relevant powers of the wave vector characterizing a specific form of the spin-orbit interaction depends on the underlying mechanism (BIA, SIA, and interface asymmetry), on the dimensionality of the system, and on whether the carriers are electrons or holes. For example, in bulk electron systems the leading-order Dresselhaus term is $\propto k^3$, while in bulk hole systems as well as in 2D electron and hole systems Dresselhaus terms linear in *k* are present. Broadly speaking, one expects terms involving smaller powers of *k* to dominate at low densities, while terms involving higher powers of *k* are expected to dominate at higher densities.

In the structures we have studied, although *k*-linear Rashba terms of the form $i\alpha_{R1}(k_{+}\sigma_{+}-k_{-}\sigma_{-})\equiv -2\alpha_{R1}(k_{y}\sigma_{x}+$ $k_x \sigma_y$ are allowed by symmetry [\[6\]](#page-12-0), we find the *k*-linear Rashba coefficient to be zero since our model does not include excited bulk conduction bands, as discussed in Sec. [II C.](#page-3-0) Correspondingly, *k*-linear Rashba terms have not been included in our analytical (SW) results. Thus, when the SW transformation is applicable, the Rashba spin splitting is described by Eq. [\(15\)](#page-3-0), which is consistent with Ref. [\[6\]](#page-12-0). In contrast, the Dresselhaus spin-orbit interaction is characterized by terms proportional to both k and k^3 which are comparable in magnitude. Therefore, we have included both *k*-linear and *k*-cubed terms in Eq. [\(25\)](#page-8-0).

We would like to comment on the relative importance of the *k*-linear [\[6](#page-12-0)[,32,33,35\]](#page-13-0) and *k*-cubic [\[4,6,29](#page-12-0)[,32,35,39,41\]](#page-13-0) Dresselhaus terms for the systems studied in this work. For GaAs inversion layers with the 2DHG densities considered here, the linear-*k* and cubic-*k* Dresselhaus terms are of the same order of magnitude (see Appendix). For example, for a GaAs inversion layer with $p = 5 \times 10^{14}$ m⁻² and $k_F =$ 5.6×10^7 m⁻¹, we find $\beta_{D1} = 0.080 \times 10^{-24}$ meV m³, and $\gamma_D = 0.030 \times 10^{-8}$ meV m, so that the ratio $\xi_D \equiv \frac{\gamma_D k_F}{\beta_D k_F^3}$ of the *k*-linear and *k*-cubic Dresselhaus terms is $\xi_D = 1.19$. As the density increases, the relative importance of the linear-*k* term decreases. For example, for $p = 2 \times 10^{15}$ m⁻² corresponding to $k_F \approx 1.12 \times 10^8 \text{ m}^{-1}$, we find $\beta_{D1} = 0.084 \times 10^{-24} \text{ meV}$ m³, and $\gamma_D = 0.029 \times 10^{-8}$ meV m, yielding $\xi_D = 0.32$, which is four times less than when $p = 5 \times 10^{14}$ m⁻². Since β_{D1} and γ_D are almost constant at the densities considered in this work, we can easily estimate the density p_D above which the cubic-*k* Dresselhaus term dominates. Taking $\beta_{D1} =$ 0.08×10^{-24} meV m³ and $\gamma_D \approx 0.03 \times 10^{-8}$ meV m, we find that $\xi_D \ll 1$ when $k_F \gg 6.1 \times 10^7$ m⁻¹, which corresponds to $p \gg 6 \times 10^{14}$ m⁻².

We have performed a similar analysis of the *k*-linear and *k*-cubic Dresselhaus terms for InAs and InSb inversion layers. For an InAs inversion layer with $p = 5 \times 10^{14}$ m⁻² we obtain $\xi_D = 5.2$, which implies that the linear-*k* term is much more important than the cubic-*k* Dresselhaus term for the densities considered here. For InAs we find $p_D = 2.6 \times 10^{15}$ m⁻². On the other hand, for an InSb inversion layer with $p = 3.3 \times$ 10^{14} m⁻², $\xi_D = 0.16$. We obtain $p_D = 5.4 \times 10^{13}$ m⁻², so that the linear-*k* contribution is essentially negligible for our range of densities.

The Dresselhaus interaction is absent in semiconductors with a diamond structure such as Si and Ge. At the same time terms of the same symmetry may appear depending on the termination at the interface [\[34\]](#page-13-0). The interface-induced spin splitting depends on the coupling to the conduction and split-off bands. The dominant contribution from this mechanism is linear in *k* and has the same symmetry as the *k*-linear Dresselhaus term, i.e., $H_{int} \propto k_-\sigma_+ + k_+\sigma_$ $k_x \sigma_x + k_y \sigma_y$ [\[34\]](#page-13-0). A quantitative description of interface asymmetry terms is beyond the scope of the present work. We may expect that for the heterojunctions studied here the effect of interface asymmetry is weaker than in quantum wells, since the wave functions experience a strong confinement at the heterointerface, but only a weak confinement towards the substrate. Thus even in a refined model permitting the wave functions to tunnel into the barrier, the wave functions are less pushed into the barrier than in a quantum well. The smaller probability of finding the carriers at the interface then implies a smaller effect of interface asymmetry on spin splitting. Consistent with this qualitative reasoning, initial studies reveal that in heterojunctions the terms due to interface asymmetry are considerably smaller than the Rashba spin-orbit interaction [\[57\]](#page-13-0).

VI. SUMMARY AND OUTLOOK

We have performed a variational analysis of spin-orbit interactions in 2D hole gases in inversion-asymmetric heterojunctions in a number of cubic semiconductors. We have quantified our findings in terms of experimentally accessible quantities: carrier number density, effective masses, and spin splitting. We find that for a broad range of experimentally relevant parameters the frequently used lowest-order expansion of the dispersion breaks down. To address this shortcoming we have provided a simple quasianalytical scheme for calculating spin-orbit related quantities that is in good agreement with numerical studies and experimental data. We recover the known general trend that the spin splitting and $HH1+$ effective mass m_+ increase as functions of density. We have found that in heterojunctions the Rashba SO coupling is in general much stronger than the Dresselhaus SO coupling. More specifically, Rashba SO coupling is much stronger in accumulation layers than in inversion layers, and it is very sensitive to the density of background dopants. Finally, in Si, due to the strong cubic terms already present in the bulk, the Fermi contour is strongly anisotropic.

The approach presented in this work can be extended to study spin-orbit interactions on surfaces other than (001), while the choice of basis functions can be tailored to the system under consideration. For example, the bulk k^3 -Dresselhaus interaction [\(24\)](#page-8-0) applied to a low-symmetry surface includes terms $\propto k_z^3$, which can be problematic depending on one's choice of basis functions. Reference [\[34\]](#page-13-0) thus avoids dealing directly with the Dresselhaus terms [\(24\)](#page-8-0) in determining the spin splitting for 2D hole systems. Instead, this work resorts to the bulk 14×14 extended Kane Hamiltonian, which avoids powers of *kz* higher than second, offering considerable computational flexibility. Nevertheless, although higher powers of k_z require some care, they can easily be treated if the basis functions are sufficiently smooth (e.g., plane waves [\[69\]](#page-13-0) or Fock-Darwin states). This consideration can become important if in certain systems terms $\propto k_z^3$ turn out to be significant, which is not the case in the present work.

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APPENDIX: THE MATERIAL-AND STRUCTURE-DEPENDENT PARAMETERS

In the following we evaluate the prefactors up to second order perturbation theory. The first-order terms are given by the expectation values of the diagonal elements in the 2×2 submatrix spanned by the states $\{|\pm 3/2\rangle\}$ in the Luttinger Hamiltonian [\(13\)](#page-2-0). The second-order terms are mediated by couplings to intermediate states in the usual way [\[6\]](#page-12-0). In all formulas in this section, the index *i* implies a summation over $i = 1, 2.$

The prefactors for the orbital terms take the form

$$
A = \frac{\mu}{2}(\gamma_1 + \gamma_2) + 3\mu^2 \gamma_3^2 \frac{|\langle H_1 | k_z | L_i \rangle|^2}{\Delta_{1i}^{HL}} \tag{A1}
$$

$$
B = -\frac{3}{4}\mu^2(\bar{y} - \zeta)^2 \frac{|\langle H_1 | L_i \rangle|^2}{\Delta_{1i}^{HL}} \tag{A2}
$$

$$
d = \frac{3}{4} \bar{\gamma} \zeta \frac{|\langle H_1 | L_i \rangle|^2}{\Delta_{1i}^{HL}},
$$
 (A3)

where $\Delta_{pq}^{rs} \equiv E_p^r - E_q^s$. Typical values for *A*, *B*, and *d* for GaAs inversion layers are given in Table VII.

The Rashba coefficient α_{R2} reads

$$
\alpha_{R2} = -\frac{3}{2} \mu^2 \gamma_3 \bar{\gamma} \frac{\langle H_1 | L_i \rangle \langle L_i | k_z | H_1 \rangle - \langle H_1 | k_z | L_i \rangle \langle L_i | H_1 \rangle}{\Delta_{1i}^{HL}},
$$
\n(A4)

while the Rashba coefficient α_{R3} is given by

$$
\alpha_{R3} = \frac{3}{2} \mu^2 \gamma_3 \zeta \frac{\langle H_1 | L_i \rangle \langle L_i | k_z | H_1 \rangle - \langle H_1 | k_z | L_i \rangle \langle L_i | H_1 \rangle}{\Delta_{1i}^{HL}} \tag{A5}
$$

implying $\alpha_{R2}/\alpha_{R3} = -\bar{\gamma}/\zeta$. Note that these expressions emerge in second order in the SW transformation. This is in contrast to Ref. [\[6\]](#page-12-0), where these prefactors arose only in third order. This difference is due to the fact that in Ref. [\[6\]](#page-12-0), the prefactors were expressed in terms of basis functions for inversion symmetric systems such as the infinite square well and the simple harmonic oscillator, treating the asymmetric component of the confining potential $V(z)$ explicitly as a perturbation. In this work, the asymmetry of $V(z)$ is encoded in the asymmetric Fang-Howard functions [\(11\)](#page-2-0) and [\(12\)](#page-2-0), so that the expressions $(A4)$ and $(A5)$ for Rashba prefactors depend implicitly on the asymmetry of $V(z)$.

The *k*-linear Dresselhaus coefficient has the form:

$$
\gamma_D = -\frac{\sqrt{3}}{2}C_D + 2\sqrt{3}\mu \gamma_3 C_D \frac{|\langle H_1|k_z|L_i\rangle|^2}{\Delta_{1i}^{HL}} - \frac{3}{4}(B_{D2} + B_{D3})\langle H_1|k_z^2|H_1\rangle.
$$
 (A6)

TABLE VII. Typical values for *A* (in 10⁻¹⁶ meV m²), *B*, *d* (both in 10⁻³² meV m⁴), α_{R2} , α_{R3} (both in 10⁻²⁴ meV m³), γ_D (in 10⁻⁸ meV m), β_{D1} and β_{D2} (both in 10⁻²⁴ meV m³), for 2DHG GaAs hole inversion layers with $N_D - N_A = 3 \times 10^{20}$ m⁻³. Superscripts (1) and (2) distinguish contributions from first- and second-order perturbation theory to the respective coefficients. The density is given in multiples of 10^{15} m⁻².

Density	$\Delta^{(1)}$	$A^{(2)}$		a	α_{R2}	α_{R3}	D	(2 $\scriptstyle\prime$ D	β_{D1}	β_{D2}
0.5	3.41	-0.42	0.27	-0.70	-0.53	0.084	0.022	0.008	0.083	-0.013
2.0	3.41	-0.61	0.22	-0.56	-0.64	0.103	0.017	0.010	0.086	-0.014

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The prefactors of the cubic Dresselhaus terms become

$$
\beta_{D1} = \frac{\sqrt{3}}{2} \mu \bar{\gamma} C_D \frac{|\langle H_1 | L_i \rangle|^2}{\Delta_{1i}^{HL}} - \frac{3}{16} (B_{D2} - B_{D3} + 3B_{D4}) + \frac{3}{4} \mu \bar{\gamma} B_{D1} \frac{\langle H_1 | L_i \rangle \langle L_i | k_z^2 | H_1 \rangle + \langle H_1 | k_z^2 | L_i \rangle \langle L_i | H_1 \rangle}{\Delta_{1i}^{HL}} + \mu \bar{\gamma} \left(\frac{21}{16} B_{D2} + \frac{3}{8} B_{D3} \right) \frac{\langle H_1 | k_z^2 | L_i \rangle \langle L_i | H_1 \rangle + \langle H_1 | L_i \rangle \langle L_i | k_z^2 | H_1 \rangle}{\Delta_{1i}^{HL}} + 3 \mu \gamma_3 B_{D3} \frac{|\langle H_1 | k_z | L_i \rangle|^2}{\Delta_{1i}^{HL}} \tag{A7}
$$

and

$$
\beta_{D2} = -\frac{\sqrt{3}}{2} \mu \zeta C_D \frac{|\langle H_1 | L_i \rangle|^2}{\Delta_{1i}^{HL}} + \frac{3}{16} (B_{D2} - B_{D3} + B_{D4}) - \frac{3}{4} \mu \zeta B_{D1} \frac{\langle H_1 | L_i \rangle \langle L_i | k_z^2 | H_1 \rangle + \langle H_1 | k_z^2 | L_i \rangle \langle L_i | H_1 \rangle}{\Delta_{1i}^{HL}} - \mu \zeta \left(\frac{21}{16} B_{D2} + \frac{3}{8} B_{D3} \right) \frac{\langle H_1 | k_z^2 | L_i \rangle \langle L_i | H_1 \rangle + \langle H_1 | L_i \rangle \langle L_i | k_z^2 | H_1 \rangle}{\Delta_{1i}^{HL}}.
$$
\n(A8)

For the *k*-linear coefficient γ_D , the bulk prefactors B_{D2} and B_{D3} give rise to sizable first-order contributions at low densities. On the other hand, interestingly, there are no contributions proportional to the bulk prefactors B_{D2} , B_{D3} , and B_{D4} from second order in SW perturbation theory. The smallness of B_{D2} , B_{D3} , and B_{D4} implies approximately $\beta_{D1}/\beta_{D2} \simeq -\bar{\gamma}/\zeta$. Nonetheless, for the prefactors β_{D1} and β_{D2} , we retained the contributions up to second order in SW perturbation theory. Here we note that within the extended Kane model we have the relation $B_{D4} = B_{D3} - B_{D2}$ (see Table [I](#page-1-0) and Ref. [6]), so that the first-order contributions from the coefficients B_{D2} , B_{D3} , and B_{D4} to the coefficient β_{D2} cancel. While the contributions from the bulk Dresselhaus terms proportional to B_{D2} , B_{D3} , and B_{D4} are generally small, the terms obtained in first and second-order SW perturbation theory are similar in magnitude. This reflects the fact that the second order yields mixed terms still linear in B_{D2} and B_{D3} , but also proportional to the large Luttinger coefficients. Typical values for the Rashba and Dresselhaus coefficients obtained from the above equations are given in Table [VII.](#page-11-0)

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