

Atomic structure calculations of helium with correlated exponential functions

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The technique of quantum electrodynamics (QED) calculations of energy levels in the helium atom is reviewed. The calculations start with the solution of the Schrödinger equation and account for relativistic and QED effects by perturbation expansion in the fine-structure constant α . The nonrelativistic wave function is represented as a linear combination of basis functions depending on all three interparticle radial distances, r_1 , r_2 and $r = |\vec{r}_1 - \vec{r}_2|$. The choice of the exponential basis functions of the form $\exp(-\alpha r_1 - \beta r_2 - \gamma r)$ allows us to construct an accurate and compact representation of the nonrelativistic wave function and to efficiently compute matrix elements of numerous singular operators representing relativistic and QED effects. Calculations of the leading QED effects of order $\alpha^5 m$ (where m is the electron mass) are complemented with the systematic treatment of higher-order $\alpha^6 m$ and $\alpha^7 m$ QED effects.

I. INTRODUCTION

The helium atom is the simplest many-body atomic system in the nature. Since the advent of quantum mechanics, helium was used as a benchmark case for developing and testing various calculational approaches of many-body atomic theory. Today, the nonrelativistic energy of various helium electronic states can be computed with an essentially arbitrary numerical accuracy [1, 2]. The same holds also for the leading-order relativistic correction. Subsequently, the quantum electrodynamics (QED) effects in the atomic structure of helium can be clearly identified and studied by comparison of theoretical predictions with the large body of available experimental data. Experimental investigations of helium spectra have progressed rapidly over the years, recently reaching the precision of a few tens of Hertz [3].

For light atomic systems such as helium, relativistic and QED corrections to energy levels can be systematically accounted for by the perturbation expansion in the fine-structure constant α . The starting point of the expansion is the nonrelativistic energy of order $\alpha^2 m$ ($= 2\text{Ry}$, where m is the electron mass and Ry is the Rydberg energy). The leading relativistic correction is of order $\alpha^4 m$, whereas QED effects enter first in order $\alpha^5 m$. A large body of work has been done in recent years to calculate QED effects in helium spectra. Extensive calculations of helium energies were accomplished by Gordon Drake *et al.* [4–6]. Their calculations are complete through order $\alpha^5 m$ and approximately include some higher-order QED effects. The next-order $\alpha^6 m$ QED correction was for a long time known only for the fine-structure intervals [7, 8]. For individual energy levels, these effects were derived and calculated numerically by one of us (K.P.) [9–11]. The higher-order $\alpha^7 m$ QED effects were evaluated by us first for the fine structure [12–14] and just recently for the triplet $n = 2$ states of helium [15–17].

The purpose of this article is to review and systematize the technique of calculations of the helium atomic structure, developed in numerous investigations over the last three decades. The starting point of the calculations is the Schrödinger equation, which is solved variationally after expanding the wave function into a finite set of explicitly-correlated basis functions depending on all three interparticle radial distances. It has been known for a long time [18] that inclusion of the interelectronic distance explicitly into the basis set is crucially important for constructing an accurate representation of the two-electron wave function. Moreover, it has also long been recognized [19] that an accurate wave-function representation should satisfy the so-called cusp conditions at the two-particle coalescence points $|\vec{r}_i - \vec{r}_j| = 0$. The cusp condition is expressed [20], after averaging over angles and for the singlet states, as

$$\psi^{(S)}(r) \underset{r \rightarrow 0}{=} \psi^{(S)}(0) (1 + \lambda r) + O(r^2), \quad (1)$$

where r is an interparticle distance and the parameter $\lambda = 1/2$ for the electron-electron and $\lambda = -Z$ for the electron-nucleus cusp (where Z is the nuclear charge number).

The two most successful basis sets used in the literature for high-precision calculations of the atomic structure of helium are: the Hylleraas basis set adopted by Drake *et al.* [4–6] and the exponential basis set put forward by Korobov [21, 22] and used in numerous calculations of our group. Both these basis sets are explicitly correlated and are able to reproduce the cusp conditions with great accuracy. In the present work we will concentrate on the exponential basis set, because only this basis has been successfully used in calculations of higher-order QED effects so far.

II. WAVE FUNCTIONS

The spatial wave function ψ_{LM_L} with a specified total angular momentum L and its momentum projection M_L for a two-electron atom is standardly represented as

$$\psi_{LM_L} = \sum_{l_1 l_2} f_{l_1 l_2}(r_1, r_2, r) Y_{LM_L}^{l_1 l_2}(\hat{r}_1, \hat{r}_2), \quad (2)$$

where $f_{l_1 l_2}$ is the radial part of the wave function, $\vec{r} = \vec{r}_1 - \vec{r}_2$, and $\hat{r} = \vec{r}/r$. Furthermore, $Y_{LM_L}^{l_1 l_2}$ are the bipolar spherical harmonics,

$$Y_{LM_L}^{l_1 l_2}(\hat{r}_1, \hat{r}_2) = \sum_{m_1 m_2} \langle l_1 m_1 l_2 m_2 | LM_L \rangle Y_{l_1 m_1}(\hat{r}_1) Y_{l_2 m_2}(\hat{r}_2), \quad (3)$$

where $\langle l_1 m_1 l_2 m_2 | lm \rangle$ is the Clebsch-Gordan coefficient and Y_{lm} are the spherical harmonics. We stress that the radial part of the wave function is assumed to be explicitly correlated, i.e., the function f depends on all interparticle distances, r_1 , r_2 , and r . In this case, the sum over l_1 and l_2 in Eq. (2) is restricted [23] by two conditions

$$(A) : l_1 + l_2 = L, \quad \text{or} \quad (B) : l_1 + l_2 = L + 1, \quad (4)$$

which lead to wave functions of different parities $(-1)^{l_1 + l_2}$. The bipolar spherical harmonics are usually handled in the spherical coordinates using the apparatus of Racah algebra, see, e.g., Ref. [24]. We find, however, that calculations with explicitly correlated functions are more conveniently performed in Cartesian coordinates. One of the reasons is that the action of numerous momentum operators encountered in calculations is most easily evaluated in the Cartesian coordinate system. The corresponding calculations can easily be automatized and performed with the help of systems of symbolic computations.

For this purpose, the expansion of the wave function is more conveniently made in terms of the bipolar solid harmonics. In order to define them, we start with the solid harmonics,

$$\mathcal{Y}_{LM}(\vec{r}) = \sqrt{4\pi} A_L r^L Y_{LM}(\hat{r}), \quad (5)$$

where the normalization coefficient A_L is fixed below. The solid harmonics obey the following summation rule,

$$\begin{aligned} \frac{1}{2L+1} \sum_{M=-L}^L \mathcal{Y}_{LM}^*(\vec{r}') \mathcal{Y}_{LM}(\vec{r}) &= A_L^2 r'^L r^L P_L(\hat{r}' \cdot \hat{r}) \\ &= (r'^i r'^j r'^k \dots)^{(L)} (r^i r^j r^k \dots)^{(L)}, \end{aligned} \quad (6)$$

where $(r^i r^j r^k \dots)^{(L)}$ is a traceless and symmetric tensor of the order L constructed from components of the vector \vec{r} with Cartesian indices i, j, k, \dots and the summation over these Cartesian indices is implicit. The last equation

determines A_L , which is related to the coefficient of x^L in the Legendre polynomial $P_L(x)$, specifically,

$$A_L^{-2} = \frac{1}{2^L} \binom{2L}{L}. \quad (7)$$

We now define the bipolar solid harmonics $\mathcal{Y}_{LM_L}^{l_1 l_2}$ as

$$\mathcal{Y}_{LM_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \Big|_{l_1 + l_2 = L} = \frac{1}{L!} (\vec{r}_1 \cdot \vec{\nabla}_\xi)^{l_1} (\vec{r}_2 \cdot \vec{\nabla}_\xi)^{l_2} \mathcal{Y}_{LM}(\vec{\xi}), \quad (8)$$

$$\begin{aligned} \mathcal{Y}_{LM_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \Big|_{l_1 + l_2 = L+1} &= \frac{1}{L!} (\vec{r}_1 \cdot \vec{\nabla}_\xi)^{l_1-1} (\vec{r}_2 \cdot \vec{\nabla}_\xi)^{l_2-1} \\ &\times (\vec{R} \cdot \vec{\nabla}_\xi) \mathcal{Y}_{LM}(\vec{\xi}), \end{aligned} \quad (9)$$

where $\vec{R} \equiv \vec{r}_1 \times \vec{r}_2$, $\vec{\xi}$ is an arbitrary vector, and the right-hand-side of the above equations does not depend on $\vec{\xi}$ after the L -fold differentiation.

The bipolar solid harmonics are proportional to the corresponding bipolar spherical harmonics with a prefactor that does not depend on angles, so their angular parts are exactly the same. Now, using Eq. (6), we obtain that the bipolar solid harmonics $\mathcal{Y}_{LM_L}^{l_1 l_2}$ obey the analogous summation rule

$$\begin{aligned} \frac{1}{2L+1} \sum_{M_L=-L}^L \mathcal{Y}_{LM_L}^{l_1' l_2'}(\vec{r}_1', \vec{r}_2') \mathcal{Y}_{LM_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \\ = \mathcal{Y}_{i_1 \dots i_L}^{l_1' l_2'}(\vec{r}_1', \vec{r}_2') \mathcal{Y}_{i_1 \dots i_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2), \end{aligned} \quad (10)$$

where $\mathcal{Y}_{i_1 \dots i_L}^{l_1 l_2}$ are the symmetric and traceless tensors of rank L with Cartesian indices $i_1 \dots i_L$,

$$\mathcal{Y}_{i_1 \dots i_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \Big|_{l_1 + l_2 = L} = (r_1^{i_1} \dots r_1^{i_{l_1}} r_2^{i_{l_1+1}} \dots r_2^{i_L})^{(L)}, \quad (11)$$

$$\mathcal{Y}_{i_1 \dots i_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \Big|_{l_1 + l_2 = L-1} = (r_1^{i_1} \dots r_1^{i_{l_1}} r_2^{i_{l_1+1}} \dots r_2^{i_{L-1}} R^{i_L})^{(L)}. \quad (12)$$

The summation formula (10) shows that the matrix elements with the spatial wave function

$$\psi_{LM_L} = \sum_{l_1 l_2} f_{l_1 l_2}(r_1, r_2, r) \mathcal{Y}_{LM_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \quad (13)$$

can be represented in terms of matrix elements with the Cartesian wave function

$$\psi^{i_1 i_2 \dots i_L} = \sum_{l_1 l_2} F_{l_1 l_2}(r_1, r_2, r) \mathcal{Y}_{i_1 i_2 \dots i_L}^{l_1 l_2}(\vec{r}_1, \vec{r}_2) \quad (14)$$

as follows

$$\frac{1}{2L+1} \sum_{M_L} \langle \psi'_{LM_L} | Q | \psi_{LM_L} \rangle = \langle \psi'^{i_1 i_2 \dots i_L} | Q | \psi^{i_1 i_2 \dots i_L} \rangle, \quad (15)$$

where Q is an arbitrary spatial operator. Eq. (14) is the Cartesian representation of the spatial wave function used in the present work.

We now present explicit formulas for the Cartesian wave functions for different values of the angular momentum and parity. For $L = 0$ we have $l_1 = l_2 = 0$ and only even parity. The wave function is just a scalar,

$$\psi({}^{1,3}S^e) = F(r_1, r_2, r) \pm (1 \leftrightarrow 2), \quad (16)$$

where the upper sign in \pm corresponds to the singlet and the lower sign to the triplet state. For $L = 1$, we have $(l_1, l_2) = (0, 1), (1, 0)$ for the odd parity and $(l_1, l_2) = (1, 1)$ for the even parity. The corresponding wave functions are vectors,

$$\vec{\psi}({}^{1,3}P^o) = \vec{r}_1 F(r_1, r_2, r) \pm (1 \leftrightarrow 2), \quad (17)$$

$$\vec{\psi}({}^{1,3}P^e) = (\vec{r}_1 \times \vec{r}_2) F(r_1, r_2, r) \pm (1 \leftrightarrow 2). \quad (18)$$

The $L = 2$ odd and even wave functions are second-rank tensors,

$$\psi^{ij}({}^{1,3}D^o) = [r_1^i (\vec{r}_1 \times \vec{r}_2)^j + r_1^j (\vec{r}_1 \times \vec{r}_2)^i] F \pm (1 \leftrightarrow 2), \quad (19)$$

$$\psi^{ij}({}^{1,3}D^e) = (r_1^i r_1^j)^{(2)} F + (r_1^i r_2^j)^{(2)} G \pm (1 \leftrightarrow 2), \quad (20)$$

where we suppressed arguments of the radial functions F and G and the elementary second-rank tensors are defined as

$$(r_a^i r_b^j)^{(2)} = \frac{1}{2} \left(r_a^i r_b^j + r_b^i r_a^j - \frac{2}{3} \delta^{ij} \vec{r}_a \cdot \vec{r}_b \right). \quad (21)$$

Explicit expressions for the $L = 3$ and $L = 4$ functions can be found in Appendix A of Ref. [25]. The spatial wave functions are normalized by

$$\langle \psi'(S) | \psi(S) \rangle = \langle \psi^i(P) | \psi^j(P) \rangle = \langle \psi^{ij}(D) | \psi^{ij}(D) \rangle = 1. \quad (22)$$

III. EVALUATION OF MATRIX ELEMENTS

The spin-dependent wave function with definite values of the total momentum J , its projection M , the angular momentum L , and the spin S is given by

$$\psi_{JM} = \sum_{M_L M_S} \langle LM_L SM_S | JM \rangle \psi_{LM_L} \chi_{SM_S}, \quad (23)$$

where M_S is the spin projection, χ_{SM_S} is the spin function, and ψ_{LM_L} is the spatial wave function. As described in the previous section, in our calculations we evaluate all matrix elements in Cartesian coordinates. The spatial wave function with the angular momentum L is represented in the form (14); namely, as a traceless tensor of rank L symmetric in all Cartesian indices carried by \vec{r}_1 , \vec{r}_2 , and $\vec{r}_1 \times \vec{r}_2$. In addition, it is assumed that the wave function has a definite symmetry with respect to $\vec{r}_1 \leftrightarrow \vec{r}_2$.

The norm and the expectation value of any spin-independent operator are immediately reduced to the spatial matrix element,

$$\langle \psi_{JM} | Q | \psi_{JM} \rangle = \langle \psi^{i_1 i_2 \dots i_L} | Q | \psi^{i_1 i_2 \dots i_L} \rangle, \quad (24)$$

where the summation over Cartesian indices is implicit. This equation is sufficient for determining the nonrelativistic wave function and the nonrelativistic energy. The relativistic and QED corrections involve operators depending on the electron spin. The expectation value of an arbitrary operator Q on a state with definite J , for the singlet $S = 0$ states, is expressed as

$$\frac{1}{2J+1} \sum_M \langle \psi_{JM} | Q | \psi_{JM} \rangle = \text{Tr} \left[\langle \psi^{i_1 i_2 \dots i_L} | Q | \psi^{i_1 i_2 \dots i_L} \rangle \left(I - \frac{\vec{S}^2}{2} \right) \right], \quad (25)$$

where I is the unity matrix, $\vec{J} = \vec{L} + \vec{S}$, $\vec{S} = (\vec{\sigma}_1 + \vec{\sigma}_2)/2$, and the trace is performed in the 4-dimensional space of two spins. Further evaluation of the matrix element proceeds by performing the trace of the operators in the spin space, with help of the following trace rules,

$$\text{Tr } I = 4, \quad (26)$$

$$\text{Tr } S^i = 0, \quad (27)$$

$$\text{Tr } S^i S^j = 2 \delta^{ij}, \quad (28)$$

$$\text{Tr } S^i S^j S^k = i \epsilon^{ijk}, \quad (29)$$

$$\text{Tr } S^i S^j S^k S^l = \delta^{ij} \delta^{kl} + \delta^{jk} \delta^{il}. \quad (30)$$

In the case of a spin-independent operator Q , Eq. (25) is reduced to Eq. (24).

For the triplet states one considers three values of $J = L-1, L, L+1$. The expectation value then takes the form

$$\begin{aligned} \langle \psi | Q | \psi \rangle_J &= \text{Tr} \left[\langle \psi^{j i_2 \dots i_L} | Q | \psi^{i i_2 \dots i_L} \rangle \right. \\ &\times \left. \left(\frac{\delta^{ij}}{3} \vec{S}^2 \left(\frac{1}{2} - A_{JL} - B_{JL} \right) + S^i S^j A_{JL} + S^j S^i B_{JL} \right) \right]. \end{aligned} \quad (31)$$

For the spin-independent operators, this equation is equivalent to Eq. (24). The coefficients A_{JL} and B_{JL} are obtained by considering two particular cases, $Q = L^i S^i$ and $Q = (L^i L^j)^{(2)} (S^i S^j)^{(2)}$. The left-hand-side of Eq. (31) is then immediately expressed in terms of J and L , whereas the right-hand-side is evaluated by using

$$-i \epsilon^{ijk} \langle \psi^{i i_2 \dots i_L} | L^j | \psi^{k i_2 \dots i_L} \rangle = (L+1) \langle \psi | \psi \rangle, \quad (32)$$

$$\langle \psi^{i i_2 \dots i_L} | (L^i L^j)^{(2)} | \psi^{j i_2 \dots i_L} \rangle = -\frac{(L+1)(2L+3)}{6} \langle \psi | \psi \rangle. \quad (33)$$

This consideration gives

$$A_{JL} = \frac{L}{L+1} \left\{ -\frac{2L+1}{2L+3}, 1, 0 \right\}, \quad (34)$$

$$B_{JL} = \frac{1}{L+1} \left\{ \frac{2L}{2L+3}, L-1, -L-1 \right\}, \quad (35)$$

for $J = L+1, L$, and $L-1$, correspondingly. These are all the formulas needed to factorize out the spin dependence of matrix elements and to express them in terms of spatial integrals.

The expectation values of an arbitrary operator Q for

For the P states, we obtain

$$\langle {}^1P_1|Q|{}^1P_1 \rangle = \text{Tr} \left[\langle {}^1P^j|Q|{}^1P^i \rangle \delta^{ij} \left(I - \frac{S^2}{2} \right) \right], \quad (38)$$

$$\langle {}^3P_0|Q|{}^3P_0 \rangle = \text{Tr} \left[\langle {}^3P^j|Q|{}^3P^i \rangle \left(\delta^{ij} \frac{S^2}{2} - S^j S^i \right) \right], \quad (39)$$

$$\langle {}^3P_1|Q|{}^3P_1 \rangle = \text{Tr} \left[\langle {}^3P^j|Q|{}^3P^i \rangle \frac{1}{2} S^i S^j \right], \quad (40)$$

$$\langle {}^3P_2|Q|{}^3P_2 \rangle = \text{Tr} \left[\langle {}^3P^j|Q|{}^3P^i \rangle \frac{1}{10} \left(2S^2 \delta^{ij} - 3S^i S^j + 2S^j S^i \right) \right]. \quad (41)$$

The results for the D states are

$$\langle {}^1D_2|Q|{}^1D_2 \rangle = \text{Tr} \left[\langle {}^1D^{ij}|Q|{}^1D^{ij} \rangle \left(I - \frac{1}{2} S^2 \right) \right], \quad (42)$$

$$\langle {}^3D_1|Q|{}^3D_1 \rangle = \text{Tr} \left[\langle {}^3D^{jk}|Q|{}^3D^{ik} \rangle \left(\frac{1}{2} \delta^{ij} S^2 - S^j S^i \right) \right], \quad (43)$$

$$\langle {}^3D_2|Q|{}^3D_2 \rangle = \text{Tr} \left[\langle {}^3D^{jk}|Q|{}^3D^{ik} \rangle \left(-\frac{1}{6} \delta^{ij} S^2 + \frac{2}{3} S^i S^j + \frac{1}{3} S^j S^i \right) \right], \quad (44)$$

$$\langle {}^3D_3|Q|{}^3D_3 \rangle = \text{Tr} \left[\langle {}^3D^{jk}|Q|{}^3D^{ik} \rangle \left(\frac{11}{42} \delta^{ij} S^2 - \frac{10}{21} S^i S^j + \frac{4}{21} S^j S^i \right) \right]. \quad (45)$$

IV. INTEGRALS WITH EXPONENTIAL BASIS FUNCTIONS

The radial parts of the wave function (14) are represented as linear combinations of the exponential basis functions,

$$F(r_1, r_2, r) = \sum_{k=1}^N c_k e^{-\alpha_k r_1 - \beta_k r_2 - \gamma_k r}, \quad (46)$$

where c_k are linear coefficients, N is the size of the basis, and α_k, β_k , and γ_k are nonlinear parameters obtained in the process of the basis optimization. One of the great features of the exponential basis functions is that the evaluation of radial integrals is very simple. A calculation of radial matrix elements of various operators with wave functions of the form (46) is reduced to evaluation of the integrals $I(i, j, k)$,

$$I(i, j, k) = \frac{1}{16\pi^2} \int d^3r_1 \int d^3r_2 r_1^{i-1} r_2^{j-1} r^{k-1} e^{-\alpha r_1 - \beta r_2 - \gamma r}. \quad (47)$$

the singlet and triplet wave functions are obtained from Eqs. (25) and (31). We now write explicitly the corresponding expressions. The results for the S states are

$$\langle {}^1S_0|Q|{}^1S_0 \rangle = \text{Tr} \left[\langle {}^1S|Q|{}^1S \rangle \left(I - \frac{S^2}{2} \right) \right], \quad (36)$$

$$\langle {}^3S_1|Q|{}^3S_1 \rangle = \text{Tr} \left[\langle {}^3S|Q|{}^3S \rangle \frac{S^2}{6} \right]. \quad (37)$$

For matrix elements of the nonrelativistic Hamiltonian, only integrals with non-negative values of i, j , and k are required. All such integrals can be obtained by differentiation of the master integral $I(0, 0, 0)$ over the nonlinear parameters,

$$I(n_i, n_j, n_k) = (-1)^{n_i+n_j+n_k} \frac{\partial^{n_i}}{\partial \alpha^{n_i}} \frac{\partial^{n_j}}{\partial \beta^{n_j}} \frac{\partial^{n_k}}{\partial \gamma^{n_k}} I(0, 0, 0), \quad (48)$$

for $n_i, n_j, n_k \geq 0$. The expression for the master integral $I(0, 0, 0)$ is very simple,

$$I(0, 0, 0) = \frac{1}{(\alpha + \beta)(\beta + \gamma)(\gamma + \alpha)}. \quad (49)$$

Matrix elements of relativistic corrections involve integrals with additional inverse powers of r_1, r_2 , and r , whose evaluation requires two additional master integrals. Their expression can be obtained by integrating Eq. (49) with respect to the corresponding nonlinear pa-

TABLE I. Expectation values of singular operators for the 2^3S and 2^3P states of helium, in atomic units. The numerical uncertainty is less than the last significant digit.

	2^3S	2^3P
$1/r^3$	0.038 861	0.047 927
$1/r^4$	0.026 567	0.043 348
$1/r^5$	0.017 580	0.027 240
$1/r_1^3$	-23.022 535	-21.886 142
$1/r_1^4$	25.511 837	24.525 751

rameters. The results are

$$I(0, 0, -1) = \frac{1}{(\alpha + \beta)(\alpha - \beta)} \ln \left(\frac{\alpha + \gamma}{\beta + \gamma} \right), \quad (50)$$

$$I(-1, 0, -1) = \frac{1}{2\beta} \left[\frac{\pi^2}{6} + \frac{1}{2} \ln^2 \left(\frac{\alpha + \beta}{\beta + \gamma} \right) + \text{Li}_2 \left(1 - \frac{\alpha + \gamma}{\alpha + \beta} \right) + \text{Li}_2 \left(1 - \frac{\alpha + \gamma}{\beta + \gamma} \right) \right], \quad (51)$$

where Li_2 is the dilogarithm function [26]. Other integrals for relativistic corrections are obtained by differentiating the above formulas for master integrals.

We note that Eq. (50) contains a spurious singularity at $\alpha = \beta$. The zero in the denominator is compensated by the vanishing logarithm function and thus is not a real singularity but can lead to numerical instabilities. In order to transform Eq. (50) to an explicitly regular form, we introduce a regularized logarithm function $\bar{\ln}_1(x)$ by separating out the first term of the Taylor expansion,

$$\ln(1+x) \equiv x \bar{\ln}_1(x). \quad (52)$$

$$I_{\log}(-1, 0, -1) \underset{\alpha > \beta}{=} \frac{1}{2\beta} \left\{ \frac{1}{2} \ln \left(\frac{\alpha - \beta}{\alpha + \beta} \right) [\ln^2(\alpha + \gamma) - \ln^2(\beta + \gamma)] + \ln(\alpha + \gamma) \left[\text{Li}_2 \left(\frac{-\beta + \gamma}{\alpha + \gamma} \right) - \text{Li}_2 \left(\frac{\beta + \gamma}{\alpha + \gamma} \right) \right] + \text{Li}_3 \left(\frac{-\beta + \gamma}{\alpha + \gamma} \right) - \text{Li}_3 \left(\frac{\beta + \gamma}{\alpha + \gamma} \right) \right\}, \quad (56)$$

where Li_3 is the trilogarithm function [26]. Eq. (56) is valid for $\alpha > \beta$. The corresponding result for $\alpha < \beta$ is obtained by the analytic continuation with help of the following identities [26]

$$\text{Li}_2(-z) + \text{Li}_2(-z^{-1}) = -\frac{\pi^2}{6} - \frac{\ln^2(z)}{2}, \quad (57)$$

$$\text{Li}_3(-z) - \text{Li}_3(-z^{-1}) = -\frac{\pi^2}{6} \ln(z) - \frac{1}{6} \ln^3(z). \quad (58)$$

The result for the case of $\alpha = \beta$ is straightforwardly obtained from Eq. (56).

Introducing $\bar{\ln}_1(x)$ with $x = (\alpha - \beta)/(\beta + \gamma)$ in Eq. (50), we obtain a regular representation of this formula. In practical calculations we encounter more spurious singularities of this kind. They are eliminated with the help of functions $\bar{\ln}_n(x)$, which are introduced analogously to $\bar{\ln}_1(x)$ by separating n first terms of the Taylor expansion of $\ln(1+x)$.

Matrix elements of QED corrections involve several integrals with large negative powers of radial distances, like $1/r^3$, $1/r^4$, and even $1/r^5$. Such integrals are singular and need proper definitions. With the exponential functions, it is possible to obtain simple and numerically stable representations for such integrals. The corresponding procedure is described in Appendix A. Numerical results for basic singular integrals for the 2^3S and 2^3P states of helium are presented in Table I.

In our calculations of the $\alpha^7 m$ QED effects [27], integrals with $\ln r$ were encountered for the first time,

$$I_{\log}(i, j, k) = \frac{1}{16\pi^2} \int d^3 r_1 \int d^3 r_2 r_1^{i-1} r_2^{j-1} r^{k-1} \times (\ln r + \gamma_E) e^{-\alpha r_1 - \beta r_2 - \gamma r}, \quad (53)$$

where γ_E is the Euler gamma constant. Such integrals are evaluated with the help of the following set of master integrals [27]:

$$I_{\log}(0, 0, 0) = \frac{1}{(\alpha - \beta)(\alpha + \beta)} \left[\frac{\ln(\alpha + \gamma)}{\alpha + \gamma} - \frac{\ln(\beta + \gamma)}{\beta + \gamma} \right], \quad (54)$$

$$I_{\log}(0, 0, -1) = \frac{1}{2(\alpha - \beta)(\alpha + \beta)} [\ln^2(\beta + \gamma) - \ln^2(\alpha + \gamma)], \quad (55)$$

V. NONRELATIVISTIC ENERGY AND WAVE FUNCTION

The nonrelativistic Hamiltonian of the helium atom for the infinitely heavy nucleus is

$$H_0 = \frac{\vec{p}_1^2}{2} + \frac{\vec{p}_2^2}{2} - \frac{Z}{r_1} - \frac{Z}{r_2} + \frac{1}{r}, \quad (59)$$

where $\vec{p}_a = -i\vec{\nabla}_a$ is the momentum operator of the electron a and Z is the nuclear charge number ($Z = 2$ for helium). The Schrödinger equation is

$$H_0 \psi(\vec{r}_1, \vec{r}_2) = E_0 \psi(\vec{r}_1, \vec{r}_2). \quad (60)$$

TABLE II. Convergence study of the nonrelativistic energy E_0 of the 2^3P state of He, for the infinitely heavy nucleus, in atomic units. N is the size of the basis.

N	E_0	Increment
100	-2.133 164 189 889 061 228 337 6	
200	-2.133 164 190 766 840 570 131 8	-0.89×10^{-9}
400	-2.133 164 190 779 088 013 045 2	-0.12×10^{-10}
800	-2.133 164 190 779 281 832 163 4	-0.20×10^{-12}
1200	-2.133 164 190 779 283 169 438 0	-0.14×10^{-14}
1600	-2.133 164 190 779 283 201 696 6	-0.33×10^{-16}
2000	-2.133 164 190 779 283 204 908 9	-0.32×10^{-17}
2400	-2.133 164 190 779 283 205 102 6	-0.20×10^{-18}
2800	-2.133 164 190 779 283 205 142 0	-0.39×10^{-19}
3200	-2.133 164 190 779 283 205 145 6	-0.36×10^{-20}
3600	-2.133 164 190 779 283 205 146 4	-0.87×10^{-21}
Ref. [2]	-2.133 164 190 779 283 205 146 992 763 806	

A direct solution of the Schrödinger equation is standardly substituted by the problem of finding the minimum or, generally, a stationary point of the variational functional

$$\Phi(\psi) = \frac{\langle \psi | H_0 | \psi \rangle}{\langle \psi | \psi \rangle}. \quad (61)$$

The variational eigenvalues obtained from this functional are the upper bounds to the true eigenvalues, and the corresponding eigenvectors provide the linear coefficients c_k of the basis-set expansion (46). It is important that the variational principle works equally well for the ground and for the excited states.

The finite nuclear mass correction to the nonrelativistic energy is induced by the nuclear kinetic energy operator

$$\delta_M H = \frac{\vec{P}^2}{2M}, \quad (62)$$

where M is the nuclear mass and $\vec{P} = -\vec{p}_1 - \vec{p}_2$ is the nuclear momentum. There are two ways to incorporate the nuclear mass effect to the nonrelativistic energy: (i) to include the operator $\delta_M H$ into the nonrelativistic Hamiltonian H_0 and solve the nuclear-mass dependent Schrödinger equation and (ii) to solve the Schrödinger equation for the infinitely heavy nucleus and to account for the nuclear mass effects by perturbation theory.

In our calculations with the exponential basis we found that the inclusion of $\delta_M H$ into the nonrelativistic Hamiltonian leads to numerical instabilities for S states (but not for P and higher- L states). So, for S states we account for the nuclear mass effects by perturbation theory (up to the third order in $1/M$ [28]), whereas for the P and D states we usually include $\delta_M H$ into the solution of the Schrödinger equation. We checked that for the P and D states both methods yield equivalent results.

It should be mentioned that in the literature it is customary to split the operator $\delta_M H$ into the mass-scaling and mass-polarization parts,

$$\delta_M H = \frac{\vec{p}_1^2 + \vec{p}_2^2}{2M} + \frac{\vec{p}_1 \cdot \vec{p}_2}{M}. \quad (63)$$

The effect of the mass scaling (caused by the first term in Eq. (63)) can be incorporated into the nonrelativistic Hamiltonian (59) by switching to the reduced-mass atomic units $r \rightarrow \mu r$, where $\mu = 1/(1 + m/M)$ is the reduced mass. As a result, the mass-scaling term leads to the appearance of the reduced mass prefactor in the nonrelativistic energy $E_0 \rightarrow \mu E_0$ and only the mass polarization term needs to be accounted for separately. We find it more convenient to keep the nuclear kinetic energy operator in the closed form (62), because this greatly simplifies consideration of higher-order recoil QED effects.

Because the nonrelativistic Hamiltonian H_0 does not depend on spin, its matrix elements are immediately reduced to radial integrals with the spatial wave functions according to Eq. (24). Computing the action of gradients $\nabla_{1,2}$ on the wave functions (14), we express the matrix elements $\langle \psi | H_0 | \psi \rangle$ as a linear combination of integrals $I(i, j, k)$ with $i, j, k \geq 0$, which are rational functions of the nonlinear parameters α_n, β_n , and γ_n .

The choice of the nonlinear basis parameters α_n, β_n , and γ_n is crucially important for obtaining an accurate and compact representation of the wave function and the energy E_0 . The general approach is to perform the variational optimization of the basis parameters, by searching for a minimum of the eigenvalue of the Hamiltonian matrix corresponding to the desired reference state. Because the optimization of each individual nonlinear parameter is not effective from the computational point of view, we use the approach introduced by Vladimir Korobov [21]. In this method, the (real) nonlinear parameters α, β , and γ are quasirandomly distributed in the intervals

$$\begin{aligned} \alpha &\in [A_1, A_2], \\ \beta &\in [B_1, B_2], \\ \gamma &\in [C_1, C_2], \end{aligned} \quad (64)$$

and the parameters $A_{1,2}, B_{1,2}$, and $C_{1,2}$ are determined by the variational optimization. We note that the nonlinear parameters as well as $A_{1,2}, B_{1,2}$, and $C_{1,2}$ can be both positive and negative. However, in order to ensure the normalizability of the wave function and its physical

behavior at large r_1 , r_2 , and r , we require that

$$\{\alpha + \beta, \alpha + \gamma, \beta + \gamma\} > \epsilon, \quad (65)$$

where $\epsilon \sim \sqrt{2E_{\text{io}}}$, with E_{io} being the ionization energy. The performance of the basis set can be significantly improved if one introduces several sets of intervals $A_{1,2}$, $B_{1,2}$, and $C_{1,2}$ which are optimised variationally. In our calculations we use typically two or three sets of intervals. This can be considered as an analogue of several different exponential scales in the Hylleraas-type calculations by Drake *at al.* [6, 29].

We also note that in calculations for excited $1snl$ states it is advantageous to include several screened hydrogenic wave functions of the type $\phi_{1s}^Z(\vec{r}_1)\phi_{n'l}^{Z-1}(\vec{r}_2)$ with $n' \leq n$ in the basis, whose parameters are excluded from optimization. This ensures that the variational optimization is localized at the local minimum with the desired principal quantum number n and does not collapse to lower n 's.

Our procedure for determination of the nonrelativistic wave function and energy looks as follows. For a given size of the basis N the nonlinear parameters α_n , β_n , and γ_n with $n = 1, \dots, N$ are distributed quasirandomly within the initial set of intervals with parameters A_i , B_i , and C_i . Then, the $N \times N$ matrix of the nonrelativistic Hamiltonian H_0 is computed. The linear coefficients c_n and the desired reference-state eigenvalue E_0 are determined by the inverse iteration method. The inversion of the Hamiltonian matrix is performed by the LDU decomposition method. This procedure is repeated for different sets of the parameters A_i , B_i , and C_i , searching for the minimum value of the energy eigenvalue.

A disadvantage of working with the exponential basis is that the basis quickly degenerates as N is increased (i.e. the determinant of the Hamiltonian matrix becomes very small), which leads to numerical instabilities in linear algebra routines. Because of this the usage of an extended-precision arithmetics is mandatory. In our calculations we used the Fortran 95 libraries for the octuple-precision (about 64 digits) arithmetics written by V. Korobov [30], quad-double routine by D. H. Bailey, and MP-FUN/MPFR library by D. H. Bailey [31].

Table II shows an example of the convergence of numerical results with the exponential basis with increase of the basis size. We observe that with just $N = 200$ basis functions one obtains the nonrelativistic energy with about 10-digit accuracy.

VI. RELATIVISTIC CORRECTION

The relativistic correction splits the nonrelativistic energy levels with quantum numbers $L > 0$ and $S > 0$ into sublevels according to the value of the total momentum J . This effect is known as the fine structure. It is often convenient to consider separately the centroid energy levels obtained by averaging over all J sublevels, and the

fine-structure intervals between individual J sublevels. The centroid energy is defined as

$$E(^nL) = \frac{\sum_{JM} E(^nL_{JM})}{(2L+1)(2S+1)} = \frac{\sum_J (2J+1)E(^nL_J)}{(2L+1)(2S+1)}. \quad (66)$$

The relativistic correction is induced by the Breit Hamiltonian, which is conveniently separated into the spin-independent and the spin-dependent parts,

$$H_{\text{Breit}} = H_A + H_{\text{fs}}. \quad (67)$$

In the leading order of perturbation theory, the spin-independent part H_A contributes only to the centroid energy, whereas the spin-dependent part H_{fs} causes the fine structure splitting.

A. Centroid energy

The spin-independent part of the Breit Hamiltonian is given by

$$\begin{aligned} H_A = & -\frac{1}{8}(\vec{p}_1^4 + \vec{p}_2^4) + \frac{Z\pi}{2}[\delta^3(r_1) + \delta^3(r_2)] \\ & + \pi\delta^3(r) - \frac{1}{2}p_1^i \left(\frac{\delta^{ij}}{r} + \frac{r^i r^j}{r^3} \right) p_2^j \\ & + \frac{Z}{2M} \left[p_1^i \left(\frac{\delta^{ij}}{r_1} + \frac{r_1^i r_1^j}{r_1^3} \right) + p_2^i \left(\frac{\delta^{ij}}{r_2} + \frac{r_2^i r_2^j}{r_2^3} \right) \right] P^j, \end{aligned} \quad (68)$$

where $\vec{P} = -\vec{p}_1 - \vec{p}_2$ is the nuclear momentum. In order to account for the finite nuclear mass effects, the expectation value of the operator H_A should be evaluated with the eigenfunctions ψ_M of the Schrödinger Hamiltonian with the finite nuclear mass (i.e. the sum of Eqs. (59) and (62)). Alternatively, the wave function ψ_M can be constructed by perturbation theory in $1/M$. In our calculations, we include the nuclear recoil effect for the relativistic correction perturbatively for the S states, and nonperturbatively for the $L > 0$ states.

The matrix element of H_A is reduced to the radial integral with the spatial wave functions according to Eq. (24) and can be evaluated numerically. However, the expectation values of the operators \vec{p}_a^4 and $\delta^3(r_a)$ are slowly converging with respect to the size of the basis because these operators are nearly singular. It is possible to significantly improve the speed of convergence if one transforms these operators to a more regular form [32]. Specifically, for a given nearly singular operator H_X we search for another, more regular operator H_{XR} and an additional operator Q_X , which satisfy the following equation

$$H_X = H_{XR} + \{H_0 - E_0, Q_X\}, \quad (69)$$

where $\{.,.\}$ denotes the anticommutator. It is obvious that $\langle H_X \rangle = \langle H_{XR} \rangle$, as long as the expectation value is evaluated with the eigenfunctions of the Hamiltonian H_0 .

TABLE III. Convergence study of the leading relativistic correction, $\langle H_A \rangle$, for the 2^3P state of He, for the infinitely heavy nucleus. Units are $\alpha^4 m = \alpha^2$ a.u.

N	ΔE	Increment
100	-1.967 366 535 960	
200	-1.967 360 971 947	0.55×10^{-5}
400	-1.967 358 035 372	0.29×10^{-5}
600	-1.967 358 371 368	-0.33×10^{-6}
800	-1.967 358 354 920	0.16×10^{-7}
1200	-1.967 358 362 599	-0.76×10^{-8}
1600	-1.967 358 376 018	-0.12×10^{-7}
2000	-1.967 358 374 001	0.20×10^{-8}
2400	-1.967 358 374 197	-0.20×10^{-9}
2800	-1.967 358 374 236	-0.38×10^{-10}
3200	-1.967 358 374 256	-0.26×10^{-10}
3600	-1.967 358 374 254	0.15×10^{-11}

In practice, it is usually possible to find such a pair of operators H_{XR}, Q_X that the most singular part of H_X is absorbed in the anticommutator. The additional operator Q_X is generally a combination of $Z/r_1, Z/r_2$, and $1/r$, with the coefficients in front of these terms determined by requiring the cancellation of all Dirac- δ -like contributions.

Specifically, we find the following regularized form of the operator H_A (without the nuclear recoil) [10]

$$H_{AR} = -\frac{1}{2}(E_0 - V)^2 - p_1^i \frac{1}{2r} \left(\delta^{ij} + \frac{r^i r^j}{r^2} \right) p_2^j + \frac{1}{4} \vec{\nabla}_1^2 \vec{\nabla}_2^2 - \frac{Z}{4} \left(\frac{\vec{r}_1}{r_1^3} \cdot \vec{\nabla}_1 + \frac{\vec{r}_2}{r_2^3} \cdot \vec{\nabla}_2 \right), \quad (70)$$

where $V = -Z/r_1 - Z/r_2 + 1/r$. The operator $\vec{\nabla}_1^2 \vec{\nabla}_2^2$ in the above formula is not self-adjoint and requires an explicit definition. Its action on a trial function ϕ on the right should be understood as plain differentiation (omitting $\delta^3(r)$); no differentiation by parts is allowed in the matrix element. It can be checked that the operators H_A and H_{AR} satisfy the following equation

$$H_A = H_{AR} + \{H_0 - E_0, Q\}, \quad (71)$$

where

$$Q = -\frac{1}{4} \left(\frac{Z}{r_1} + \frac{Z}{r_2} - \frac{2}{r} \right). \quad (72)$$

Formulas with the finite nuclear mass are analogous but more lengthy; they are given by Eqs. (62)-(67) of Ref. [33].

Table III presents numerical results for the leading relativistic correction to the 2^3P centroid energy, performed with different basis sets. We observe that, for the same basis size, the number of correct digits for the matrix element is half as much as for the nonrelativistic energy.

B. Fine structure

The fine structure of energy levels is induced by spin-dependent operators. The spin-dependent part of the Breit Hamiltonian is conveniently written as a sum of three operators with different spin structure,

$$H_{\text{fs}} = H_B + H_C + H_D, \quad (73)$$

with

$$H_B = \left[\frac{Z}{4} \left(\frac{\vec{r}_1}{r_1^3} \times \vec{p}_1 + \frac{\vec{r}_2}{r_2^3} \times \vec{p}_2 \right) (1 + 2\kappa) - \frac{3}{4} \frac{\vec{r}}{r^3} \times (\vec{p}_1 - \vec{p}_2) \left(1 + \frac{4\kappa}{3} \right) - \frac{Z}{2M} \left(\frac{\vec{r}_1}{r_1^3} + \frac{\vec{r}_2}{r_2^3} \right) \times \vec{P} (1 + \kappa) \right] \frac{\vec{\sigma}_1 + \vec{\sigma}_2}{2}, \quad (74)$$

$$H_C = \left[\frac{Z}{4} \left(\frac{\vec{r}_1}{r_1^3} \times \vec{p}_1 - \frac{\vec{r}_2}{r_2^3} \times \vec{p}_2 \right) (1 + 2\kappa) + \frac{1}{4} \frac{\vec{r}}{r^3} \times (\vec{p}_1 + \vec{p}_2) - \frac{Z}{2M} \left(\frac{\vec{r}_1}{r_1^3} - \frac{\vec{r}_2}{r_2^3} \right) \times \vec{P} (1 + \kappa) \right] \frac{\vec{\sigma}_1 - \vec{\sigma}_2}{2}, \quad (75)$$

$$H_D = \frac{1}{4} \left(\frac{\vec{\sigma}_1 \cdot \vec{\sigma}_2}{r^3} - 3 \frac{\vec{\sigma}_1 \cdot \vec{r} \vec{\sigma}_2 \cdot \vec{r}}{r^5} \right) (1 + \kappa)^2, \quad (76)$$

where $\kappa = \alpha/2\pi + O(\alpha^2)$ is the anomalous magnetic moment correction and $\vec{\sigma}_a$ is the vector of Pauli matrices acting on a 'th electron. We note that the operators H_B, H_C , and H_D contain radiative corrections in form of the electron anomalous magnetic moment. In this way we account for the complete QED effects of order $\alpha^5 m$ to the fine structure.

It should be mentioned that the matrix element of H_C is nonzero only if the operator is sandwiched between wave functions with different spin values. Therefore, any symmetrical matrix element of H_C vanishes, and this operator does not contribute in the leading order of perturbation theory. We note, however, that H_C contributes to the second-order perturbation corrections (in the order $\alpha^6 m$).

In order to perform the spin-angular reduction in the matrix elements of H_{fs} , it is convenient to introduce spatial operators Q_B, Q_C , and Q_D , explicitly separating the spatial and the spin degrees of freedom,

$$H_B = \vec{Q}_B \cdot \frac{\vec{\sigma}_1 + \vec{\sigma}_2}{2}, \quad (77)$$

$$H_C = \vec{Q}_C \cdot \frac{\vec{\sigma}_1 - \vec{\sigma}_2}{2}, \quad (78)$$

$$H_D = Q_D^{ij} \frac{1}{2} \sigma_1^i \sigma_2^j. \quad (79)$$

Using Eqs. (31), (38)-(41) and performing traces of the spin operators, we express all matrix elements in terms

of spatial radial integrals. For the 3P states, we obtain

$$\frac{1}{2J+1} \sum_M \langle {}^3P_{JM} | H_B | {}^3P_{JM} \rangle = i\epsilon_{jkl} \langle {}^3P^j | Q_B^k | {}^3P^l \rangle u_J(P), \quad (80)$$

$$\frac{1}{2J+1} \sum_M \langle {}^3P_{JM} | H_D | {}^3P_{JM} \rangle = \langle {}^3P^j | Q_D^{jl} | {}^3P^l \rangle v_J(P), \quad (81)$$

where

$$u_J(P) = (1, 1/2, -1/2), \quad (82)$$

$$v_J(P) = (-1, 1/2, -1/10), \quad (83)$$

for $J = 0, 1$, and 2 , respectively.

For the 3D states, an analogous calculation yields

$$\begin{aligned} \frac{1}{2J+1} \sum_M \langle {}^3D_{JM} | H_B | {}^3D_{JM} \rangle \\ = i\epsilon_{jli} \langle {}^3D^{jk} | Q_B^l | {}^3D^{ik} \rangle u_J(D), \end{aligned} \quad (84)$$

$$\begin{aligned} \frac{1}{2J+1} \sum_M \langle {}^3D_{JM} | H_D | {}^3D_{JM} \rangle \\ = \langle {}^3D^{ik} | Q_D^{ij} | {}^3D^{jk} \rangle v_J(D), \end{aligned} \quad (85)$$

where

$$u_J(D) = (1/2, 1/6, -1/3), \quad (86)$$

$$v_J(D) = (-1, 1, -2/7), \quad (87)$$

for $J = 1, 2$, and 3 , respectively.

VII. LEADING QED CORRECTION

The leading QED contribution is of the order $\alpha^5 m$. For the fine structure, this contribution is already accounted for by the electron anomalous magnetic moment terms in the Breit Hamiltonian, as given by Eqs. (74)-(76). So, we need to examine only the centroid energy.

The spin-independent $m\alpha^5$ Hamiltonian representing the leading QED effects was derived in the 1950s by Araki and Sucher [34, 35]

$$\begin{aligned} H^{(5)} = & \left(\frac{19}{30} + \ln(\alpha^{-2}) - \ln k_0 \right) \frac{4Z}{3} [\delta^3(r_1) + \delta^3(r_2)] \\ & + \left(\frac{164}{15} + \frac{14}{3} \ln \alpha \right) \delta^3(r) - \frac{7}{6\pi} \frac{1}{r_\epsilon^3}, \end{aligned} \quad (88)$$

where $\ln k_0$ is the so-called Bethe logarithm defined as

$$\ln k_0 = \frac{\langle (\vec{p}_1 + \vec{p}_2) (H_0 - E_0) \ln [2(H_0 - E_0)] (\vec{p}_1 + \vec{p}_2) \rangle}{2\pi Z \langle \delta^3(r_1) + \delta^3(r_2) \rangle}, \quad (89)$$

and $1/r_\epsilon^3$ is the regularized $1/r^3$ operator (distribution) defined by its matrix elements with an arbitrary smooth function $f(\vec{r})$ as

$$\int d^3r f(\vec{r}) \frac{1}{r_\epsilon^3} = \lim_{\epsilon \rightarrow 0} \int d^3r f(\vec{r}) \left[\frac{1}{r^3} \Theta(r - \epsilon) \right.$$

$$\left. + 4\pi \delta^3(r) (\gamma_E + \ln \epsilon) \right]. \quad (90)$$

The nuclear recoil correction to the leading QED contribution consists of two parts,

$$E_M^{(5)} = 2 \langle H^{(5)} \frac{1}{(E_0 - H_0)'} \delta_M H \rangle + \langle \delta_M H^{(5)} \rangle, \quad (91)$$

where $\delta_M H$ is defined by Eq. (62), and $\delta_M H^{(5)}$ is the recoil addition to the $\alpha^5 m$ Hamiltonian given by [36]

$$\begin{aligned} \delta_M H^{(5)} = & \frac{1}{M} \left[\left(\frac{62}{3} + \ln(\alpha^{-2}) - 8 \ln k_0 - \frac{4}{Z} \delta_M \ln k_0 \right) \right. \\ & \left. \times \frac{Z^2}{3} [\delta^3(r_1) + \delta^3(r_2)] - \frac{7Z^2}{6\pi} \left(\frac{1}{r_{1,\epsilon}^3} + \frac{1}{r_{2,\epsilon}^3} \right) \right]. \end{aligned} \quad (92)$$

Here, $\delta_M \ln k_0$ is the correction to the Bethe logarithm $\ln k_0$ induced by the nonrelativistic kinetic energy operator $\vec{P}^2/2$, and $1/r_{a,\epsilon}^3$ is the regularized $1/r_a^3$ operator defined analogously to Eq. (90).

The recoil correction to the Bethe logarithm $\delta_M \ln k_0$ is often separated into the mass-scaling and mass-polarization parts,

$$\delta_M \ln k_0 = 1 + \delta_{p_1 p_2} \ln k_0, \quad (93)$$

where $\delta_{p_1 p_2}$ denotes the perturbation due to the mass polarization operator $\vec{p}_1 \cdot \vec{p}_2$. The corresponding separation for the $1/r_\epsilon^3$ matrix element reads

$$\delta_M \left\langle \frac{1}{r_\epsilon^3} \right\rangle = -3 \left\langle \frac{1}{r_\epsilon^3} \right\rangle + \langle 4\pi \delta^3(r) \rangle + \delta_{p_1 p_2} \left\langle \frac{1}{r_\epsilon^3} \right\rangle. \quad (94)$$

From the computational point of view, the numerical evaluation of the QED effects involves two new features, as compared to the relativistic correction: matrix elements of the singular operators $1/r^3$ and $1/r_a^3$ and the Bethe logarithm. Calculation of expectation values of singular operators with exponential basis functions is examined in Appendix A; it does not present any computational difficulties. On the contrary, the computation of the Bethe logarithm is rather nontrivial; it is examined in the next section.

A. Bethe logarithm

There are two different approaches developed for the calculation of the Bethe logarithm in few-electron atoms. The first one starts with the definition (89) and uses the basis-set representation of the Hamiltonian as a sum of the spectrum of the eigenfunctions. The difficulty is that the sum in the numerator is nearly diverging because the dominant contribution comes from the high-energy continuum states of the spectrum. This problem is solved by using a basis set whose spectrum of pseudostates spans a huge range of energies [37].

An alternative approach was first introduced by C. Schwartz [23] and further developed by V. Korobov [38–40]. Within this method, the Bethe logarithm $\ln k_0$ is represented as an integral over the momentum of the virtual photon, with subtracting the ultraviolet asymptotics and performing the limit,

$$\ln k_0 = \frac{1}{D} \lim_{\Lambda \rightarrow \infty} \left[\langle \vec{\nabla}^2 \rangle \Lambda + D \ln 2\Lambda + \int_0^\Lambda dk k J(k) \right], \quad (95)$$

where $D = 2\pi Z \langle \delta^3(r_1) + \delta^3(r_2) \rangle$, $\vec{\nabla} \equiv \vec{\nabla}_1 + \vec{\nabla}_2$, and

$$J(k) = \left\langle \vec{\nabla} \frac{1}{E_0 - H_0 - k} \vec{\nabla} \right\rangle. \quad (96)$$

The asymptotic expansion of $J(k)$ for large k reads

$$J(k) \underset{k \rightarrow \infty}{=} -\frac{1}{k} \langle \vec{\nabla}^2 \rangle - \frac{D}{k^2} + \frac{2\sqrt{2}ZD}{k^{5/2}} - 2Z^2 D \frac{\ln k}{k^3} + \dots \quad (97)$$

Splitting the integration interval $(0, \Lambda)$ into two parts $(0, K)$ and (K, Λ) , where K is an arbitrary cutoff parameter, we can rewrite Eq. (95) as

$$\begin{aligned} \ln k_0 &= \frac{K}{D} \langle \vec{\nabla}^2 \rangle + \ln(2K) + \frac{1}{D} \int_0^K dk k J(k) \\ &+ \frac{1}{D} \int_K^\infty dk k \left[J(k) + \frac{1}{k} \langle \vec{\nabla}^2 \rangle + \frac{D}{k^2} \right]. \end{aligned} \quad (98)$$

The above expression is finite, does not depend on K , and is suitable for a numerical evaluation.

We now address the angular reduction in the second-order matrix element $J(k)$ given by Eq. (96). It is performed in several steps. First, we represent the gradient acting on the reference-state wave function $\nabla^j \psi^{i_1 \dots i_L}$ as a sum of irreducible Cartesian tensors, as described in Appendix B. For example, the gradient acting on a P -state wave function $\nabla^j \psi^i$ is represented as a sum of the $L = 0$, $L = 1$, and $L = 2$ irreducible Cartesian tensors, which induce, correspondingly, the $L = 0$, $L = 1$, and $L = 2$ angular-momentum contributions from the resolvent. The second-order matrix element of an irreducible tensor $\Phi^{i_1 \dots i_L}$ is transformed as

$$\left\langle \Phi^{i_1 \dots i_L} \left| \frac{1}{E_0 - H_0 - k} \right| \Phi^{i_1 \dots i_L} \right\rangle = \langle \Phi^{i_1 \dots i_L} | \tilde{\Phi}^{i_1 \dots i_L} \rangle, \quad (99)$$

where $\tilde{\Phi}^{i_1 \dots i_L}$ is the solution of the inhomogeneous Schrödinger equation

$$(E_0 - H_0 - k) \tilde{\Phi}^{i_1 \dots i_L} = \Phi^{i_1 \dots i_L}. \quad (100)$$

Inserting the explicit representation of $\tilde{\Phi}$ as a sum over the spectrum, we obtain

$$\left\langle \Phi^{i_1 \dots i_L} \left| \frac{1}{E_0 - H_0 - k} \right| \Phi^{i_1 \dots i_L} \right\rangle = \sum_n \frac{|\langle \Phi^{i_1 \dots i_L} | \psi_n^{i_1 \dots i_L} \rangle|^2}{E_0 - E_n - k}. \quad (101)$$

An alternative way to arrive at this expression is to observe that the scalar product $\langle \Phi | \psi \rangle$ includes an integration over the continuous and a summation over the discrete variables, namely $\langle \Phi | \psi \rangle \equiv \langle \Phi^{i_1 \dots i_L} | \psi^{i_1 \dots i_L} \rangle = \sum_{i_1 \dots i_L} \int d^{3n}r \Phi^{i_1 \dots i_L}(r) \psi^{i_1 \dots i_L}(r)$.

The advantage of the integral representation of the Bethe logarithm is that $J(k)$ has a form of the symmetric second-order perturbation correction and thus obeys the variational principle. We therefore can variationally optimize the basis-set representation of the resolvent $1/(E_0 - H_0 - k)$ for different $k > E_0 - E_{(1s)^2}$. For lower values of k , the basis can be variationally optimized if one fixes pre-optimized parameters of the more deeply bound states with $E_n < E_0$.

Our numerical procedure was performed in two steps. First, we optimized the basis for several different scales of the photon momentum, $k = 10^i$, with typical values of $i = 1, \dots, 4$. After that, the computation of the function $J(k)$ was performed with a basis obtained by merging together the optimized sets for the two closest k_i points, thus essentially doubling the size of the basis. In the second step, we perform the integration over k . The integral over $(0, K)$ (with the typical choice of $K = 10$) was calculated analytically, after the full diagonalization of the Hamiltonian matrix. The remaining interval was split into two parts, (K, K_2) and (K_2, ∞) , with the typical choice of $K_2 = 10^4$. The integral over the former was performed with help of Gauss-Legendre quadratures, after the change of variables $t = 1/k^2$. The remaining part of the integral was calculated analytically, after fitting numerical values of $J(k)$ to the known form of the asymptotic expansion,

$$J(k) = \text{pol} \left(\frac{1}{\sqrt{k}} \right) + \frac{\ln k}{k} \text{pol} \left(\frac{1}{k} \right), \quad (102)$$

where $\text{pol}(x)$ denotes a polynomial of x . The first terms of this expansion are given by Eq. (97), whereas the higher-order coefficients are obtained by fitting.

Calculations of the Bethe logarithm for the finite nuclear mass can be performed analogously to the above, or by perturbation theory. The numerical procedure for evaluation of the recoil correction to the Bethe logarithm by perturbation theory is described in Appendix A of Ref. [41].

Table IV presents a comparison of different calculations of the Bethe logarithm for the 2^3P state of helium. The most accurate results for the ground and excited states of helium are obtained by Korobov in Ref. [40]. Results for He-like ions can be found in Refs. [37, 41].

VIII. $\alpha^6 m$ QED EFFECTS

The $\alpha^6 m$ QED corrections to energy levels in atoms are represented by the sum of the expectation value of the effective $\alpha^6 m$ Hamiltonian $H^{(6)}$ and the second-order

TABLE IV. Comparison of different calculations of the Bethe logarithm for the 2^3P state of helium.

Drake and Goldman 1999 [37]	4.369 985 20 (2)
Korobov 2004 [39]	4.369 985 356 (1)
Yerokhin and Pachucki 2010 [41]	4.369 985 364 4 (2)
Korobov 2019 [40]	4.369 985 364 549 (3)

perturbation correction induced by the Breit Hamiltonian,

$$E^{(6)} = \langle H^{(6)} \rangle + \left\langle H_{\text{Breit},R}^{(4)} \frac{1}{(E_0 - H_0)'} H_{\text{Breit},R}^{(4)} \right\rangle, \quad (103)$$

where $H_{\text{Breit},R}^{(4)}$ is the regularized Breit Hamiltonian of the order $\alpha^4 m$,

$$H_{\text{Breit},R}^{(4)} = H_{AR} + H_B^{(4)} + H_C^{(4)} + H_D^{(4)}. \quad (104)$$

We note that in order to avoid admixture of higher-order contributions in $E^{(6)}$, we have to retain only the $\alpha^4 m$ part in the definition of the Breit Hamiltonian, i.e., to set the magnetic moment anomaly $\kappa \rightarrow 0$ in the definitions (74)-(76). This is indicated by the superscript “4” in the corresponding operators.

Formulas for the effective $\alpha^6 m$ Hamiltonian $H^{(6)}$ are rather lengthy and will not be reproduced here. In the case of fine structure, they were first obtained by Douglas and Kroll in 1974 [42] and later re-derived in Refs. [43, 44]. For the energy centroid, the situation is greatly complicated because of the appearance of numerous diverging operators. The corresponding derivation was accomplished by one of us (K.P.), in Ref. [9] for the triplet states and in Ref. [10] for the singlet states of helium. The complete formulas suitable for numerical evaluation can be found in Ref. [25].

The nuclear recoil $\alpha^6 m$ correction has the same structure as the non-recoil one, but the expressions for the operators are much more complicated. This correction was calculated in Ref. [33] for the triplet states and in Ref. [45] for the singlet states of helium.

A. Second-order terms

We now discuss the evaluation of the second-order contributions, represented by the second term in Eq. (103). Such corrections were first calculated for the fine structure by Hambro [46] and by Lewis and Serafino [7]. Later, the fine-structure calculations were greatly improved in Refs. [47–49]. For the centroid energies, the second-order corrections were calculated in Refs. [10, 11] for the $2S$ and $2P$ states and in Refs. [25, 50] for the nD states of helium.

It is convenient to rewrite Eq. (103), expressing the second-order perturbation correction more explicitly,

$$E^{(6)} = \langle H^{(6)} \rangle + \left\langle H_{AR} \frac{1}{(E_0 - H_0)'} H_{AR} \right\rangle$$

$$\begin{aligned} &+ \left\langle H_B^{(4)} \frac{1}{(E_0 - H_0)'} H_B^{(4)} \right\rangle + \left\langle H_C^{(4)} \frac{1}{(E_0 - H_0)'} H_C^{(4)} \right\rangle \\ &+ \left\langle H_D^{(4)} \frac{1}{(E_0 - H_0)'} H_D^{(4)} \right\rangle + 2 \left\langle H_B^{(4)} \frac{1}{(E_0 - H_0)'} H_D^{(4)} \right\rangle \\ &+ 2 \left\langle H_{AR} \frac{1}{(E_0 - H_0)'} [H_B^{(4)} + H_D^{(4)}] \right\rangle. \quad (105) \end{aligned}$$

We note that the non-symmetrical second-order corrections (the last two terms in the above equation) vanish for the centroid energy, but contribute to the fine structure.

The second-order perturbative corrections are calculated as follows. In the first step, we perform traces over the spin degrees of freedom in the matrix elements. Then we decompose the product of a tensor operator Q and the reference-state wave function ψ into the irreducible tensor parts $\tilde{\psi}$, as described in Appendix B. In the last step we calculate the second-order matrix elements induced by the irreducible parts $\tilde{\psi}$ as (see Eq. (101))

$$\left\langle \tilde{\psi} \left| \frac{1}{(E_0 - H_0)'} \right| \tilde{\psi} \right\rangle = \sum_n^{E_n \neq E_0} \frac{|\langle \tilde{\psi}^{i_1 \dots i_J} | \psi_n^{i_1 \dots i_J} \rangle|^2}{E_0 - E_n}. \quad (106)$$

The numerical evaluation of symmetrical second-order contributions was carried out with the variational optimization of the nonlinear parameters of the basis set for the resolvent $1/(E_0 - H_0)$. Convergence of numerical results is often rather slow, especially for contributions with H_{AR} . This is associated with the fact that the effective wave function $|\delta\psi\rangle = 1/(E_0 - H_0)' |H_{AR}\rangle$ has an integrable singularity at $r_a \rightarrow 0$. In order to represent such wave functions with the exponential basis, very large (both positive and negative) exponents are required. In order to effectively span large regions of parameters, we used non-uniform distributions of the nonlinear parameters. E.g., for the nonlinear parameters α_i we used the distributions of the kind [9]

$$\alpha_i = A_1 + (t_i^{-a} - 1) A_2, \quad (107)$$

with $a = 2$ and 3 , where the variable t_i has a uniform quasirandom distribution over the interval $(0, 1)$ and the variables $A_{1,2}$ are subjects of variational optimization. An example of the convergence study of the second-order correction $\langle H_{AR} \frac{1}{(E_0 - H_0)'} H_{AR} \rangle$ is given in Table V. Numerical evaluation of non-symmetrical second-order contributions was carried out with basis sets, optimized for the corresponding symmetrical corrections.

IX. $\alpha^7 m$ QED EFFECTS

The $\alpha^7 m$ QED correction to energy levels in atoms is given [12] by the sum of the relativistic correction to the Bethe logarithm E_L , the expectation value of the effective $\alpha^7 m$ Hamiltonian $H^{(7)}$, and the perturbation of the $\alpha^5 m$ QED operator by the Breit Hamiltonian,

$$E^{(7)} = E_L + \langle H^{(7)} \rangle + 2 \left\langle H_{\text{Breit},R}^{(4)} \frac{1}{(E_0 - H_0)'} H_R^{(5)} \right\rangle. \quad (108)$$

TABLE V. Convergence study of the second-order perturbation correction $\langle H_{AR} \frac{1}{(E_0 - H_0)} H_{AR} \rangle$ for the 2^3P state of helium. Units are $\alpha^6 m = \alpha^4$ a.u.

N	ΔE	Increment
200	-15.847 010 059	
400	-15.848 416 968	-0.14×10^{-2}
800	-15.848 498 832	-0.82×10^{-4}
1200	-15.848 507 251	-0.82×10^{-5}
1600	-15.848 508 295	-0.10×10^{-5}
2000	-15.848 508 667	-0.37×10^{-6}
2400	-15.848 508 705	-0.39×10^{-7}
2800	-15.848 508 781	-0.75×10^{-7}
Ref. [11]	-15.848 510(2)	

The regularized effective $\alpha^5 m$ Hamiltonian is [17]

$$H_R^{(5)} = -\frac{Z}{\pi} \left(\frac{19}{45} + \frac{2}{3} \ln \frac{\alpha^{-2}}{2} \right) \left(\frac{\vec{r}_1 \cdot \vec{\nabla}_1}{r_1^3} + \frac{\vec{r}_2 \cdot \vec{\nabla}_2}{r_2^3} \right) - \frac{7}{6\pi} \frac{1}{r^3} + H_B^{(5)} + H_C^{(5)} + H_D^{(5)}, \quad (109)$$

where $H_{B,C,D}^{(5)}$ are the $O(\alpha)$ parts of the corresponding spin-dependent parts of the Breit Hamiltonian, $H_B = H_B^{(4)} + \alpha H_B^{(5)}$, etc. The operator $H_R^{(5)}$ is non-Hermitian and is assumed to act on a ket trial function ϕ on the right.

The relativistic correction to the Bethe logarithm is rather complicated. We will not discuss its calculation here, but direct the reader to original studies. This correction was first calculated for the fine structure of the 2^3P state; the corresponding calculations for helium and helium-like ions were performed in Refs. [12–14]. In our recent investigation [15] we performed a calculation for the energy centroid of the 2^3S and 2^3P states. For singlet states of helium, this correction has never been calculated so far.

The derivation of the effective $\alpha^7 m$ Hamiltonian $H^{(7)}$ for helium is an extremely difficult problem. It was first accomplished by one of us (K.P.) for the fine structure in Refs. [12, 13]. Recently, we performed [16, 17] the derivation of $H^{(7)}$ for triplet states of helium and calculated [27] the corresponding correction to the energies of the 2^3S and 2^3P states. For singlet states, the effective $\alpha^7 m$ Hamiltonian is unknown.

From the computational point of view, the main difficulty of the evaluation of the $\alpha^7 m$ correction is the calculation of the Bethe-logarithm contribution E_L . The computational scheme is similar to that for the plain Bethe logarithm and is described in Ref. [15]. Conversely, the computation of the expectation value of $H^{(7)}$ and the second-order corrections is very similar to the calculation of the $\alpha^6 m$ corrections.

X. OTHER EFFECTS

The finite nuclear size correction is given by (in relativistic units)

$$E_{\text{fns}} = \frac{2\pi}{3} Z\alpha \langle \delta^{(3)}(r_1) + \delta^{(3)}(r_2) \rangle R^2 \times [1 - (Z\alpha)^2 \ln(mRZ\alpha)], \quad (110)$$

where R is the root-mean-square nuclear charge radius, and the expectation value of the Dirac δ functions is assumed to include the finite-nuclear-mass correction induced by $\delta_M H$.

The higher-order QED effects are approximated on the basis of known results for hydrogenic atoms. Specifically, the hydrogenic one-loop and two-loop corrections for the $2s$ state of He^+ are given by [51]

$$E_{\text{rad1}}^{(8+)}(\text{hydr}) = \frac{Z^7}{8\pi} 83.824, \quad (111)$$

$$E_{\text{rad2}}^{(8+)}(\text{hydr}) = \frac{Z^6}{8\pi^2} \left(-\frac{8}{27} \ln^3[(Z\alpha)^{-2}] + 0.639 \ln^2[(Z\alpha)^{-2}] + 41.387 \ln[(Z\alpha)^{-2}] - 81.1 \pm 10 \right). \quad (112)$$

An approximation for the higher-order $\alpha^8 m$ QED correction to the ionization energies of the helium atom is obtained from the corresponding hydrogenic $2s$ contribution by

$$E^{(8+)} = E^{(8+)}(\text{hydr}) \frac{\langle \delta^3(r_1) + \delta^3(r_2) \rangle - Z^3/\pi}{Z^3/8\pi}. \quad (113)$$

XI. COMPARISON OF THEORY AND EXPERIMENT

In this section we summarize numerical results of QED calculations of energy levels in ^4He and compare theoretical predictions with available experimental results. Table VI presents such a comparison for transitions between states with the principal quantum number $n = 2$. We note that our present theoretical uncertainty for the $2^3S - 2^1S$ transition is increased as compared to our previous work [28]. The reason is an accidental cancelation of the estimated $\alpha^7 m$ term between the 2^3S and 2^1S states in Ref. [28]. Now the $\alpha^7 m$ correction is calculated for the 2^3S state and the theoretical uncertainty is defined by the 2^1S state only. Table VI shows good agreement of theory and experiment for the singlet-singlet and triplet-triplet transitions but some tension for the singlet-triplet transitions. Specifically, we note a 2.3σ deviation from the experimental result [53] for the $2^3S - 2^1P$ transition (with σ denoting the standard deviation).

Of particular importance is the agreement observed for the $2^3P - 2^3S$ transition, because in this case two triplet states are involved, for which the theoretical accuracy

TABLE VI. Comparison of theory and experiment for the intrashell $n = 2$ transitions in ${}^4\text{He}$, in MHz. 2^3P stands for the centroid energy.

Transition	Theory [17, 25, 28]	Experiment	Reference	Difference
$2^1S_0 - 2^1P_1$	145 622 891.6 (2.3)	145 622 892.89 (18)	Luo 2013 [52]	-1.3 (2.3)
$2^3S_1 - 2^1P_1$	338 133 595.8 (0.4)	338 133 594.4 (5)	Notermans 2014 [53]	1.4 (0.6)
$2^3S_1 - 2^1S_0$	192 510 704.2 (1.9)	192 510 702.148 72 (20)	Rengelink 2018 [54]	2.1 (1.9)
$2^3S_1 - 2^3P$	276 736 495.620 (54)	276 736 495.600 0 (14) ^a	Zheng 2017 [55]	0.020 (54)
		276 736 495.649 (2) ^a	Pastor 2004 [56]	-0.029 (54)

^a using theoretical results for the 2^3P fine structure from Table VIII.

TABLE VII. Comparison of theory and experiment for the ionization energy (IE) and n - n' transitions in ${}^4\text{He}$, in MHz.

Transition	Theory [17, 25, 28]	Experiment	Reference	Difference
1^1S_0 (IE)	5 945 204 172 (36)	5 945 204 212 (6)	Kandula 2011 [58]	-40 (36)
2^1S_0 (IE)	960 332 038.0 (1.9)	960 332 041.01 (15)	Lichten 1991 [59]	-3.0 (1.9)
		960 332 040.491 (32)	Clausen 2021 [60]	-2.5 (1.9)
$1^1S_0 - 2^1S_0$	4 984 872 134 (36)	4 984 872 315 (48)	Bergeson 1998 [61]	-181 (60)
$1^1S_0 - 2^1P_1$	5 130 495 026 (36)	5 130 495 083 (45)	Eikema 1997 [62]	-57 (58)
$2^1S_0 - 3^1D_2$	594 414 289.3 (1.9)	594 414 291.803 (13)	Huang 2018 [63]	-2.5 (1.9)
$2^1P_1 - 3^1D_2$	448 791 397.8 (0.4)	448 791 399.11 (27)	Luo 2013 [64]	-1.3 (0.5)
$2^3S_1 - 3^3D_1$	786 823 849.540 (57)	786 823 850.002 (56)	Dorrer 1997 [65]	-0.462 (80)
$2^3P_0 - 3^3D_1$	510 059 754.863 (28)	510 059 755.352 (28)	Luo 2016 [66]	-0.489 (40)

is the highest. Theoretical calculations of energies for the 2^3S and 2^3P states [17] are complete through order $\alpha^7 m$, with resulting theoretical uncertainty below 100 kHz, whereas for the 2^1S and 2^1P states the theory [28] is complete up to order $\alpha^6 m$ only and the theoretical accuracy is on the level of 1 MHz. For the D states, theoretical calculations [25, 50] are also complete at order $\alpha^6 m$, but the absolute theoretical precision is much higher since the QED effects are smaller. In general, we conclude that for the intrashell $n = 2$ transitions there is good agreement for transitions between the states with the same spin multiplicity and some tension for the states of different spin multiplicity.

The situation becomes even more strained when we consider ionization energies and transitions involving states with different n 's. The corresponding comparison is presented in Table VII. We immediately notice that all differences between theory and experiment are of the same sign and that most of them are outside of the theoretical error bars. The largest discrepancies are found for the $2^3S_1 - 3^3D_1$ and the $2^3P_0 - 3^3D_1$ transition, of 6 and 12 σ , correspondingly. These transitions involve the triplet states, for which theoretical uncertainties are the smallest, so that 0.5 MHz differences from the experimental values lead to large relative deviations.

The comparison in Tables VI and VII suggests that there might be a contribution missing in theoretical calculations of energy levels, which weakly depends on L but strongly depends on the principal quantum number n (the latter is natural because the $1/n^3$ scaling is typical for all QED effects). This conjecture was put forward in Ref. [50] and since then strengthened by subsequent

calculations and measurements. Such a missing contribution most likely originates from the $\alpha^6 m$ or $\alpha^7 m$ QED corrections because all other theoretical effects are cross-checked against independent calculations [5].

Table VIII presents the comparison of theoretical and experimental results for the fine-structure intervals of the 2^3P state in ${}^4\text{He}$. Theoretical predictions for these intervals are of greater accuracy than for other intervals of the $n = 2$ manifold. This is both due to the fact that the theory of these intervals [14, 57] is complete at the order $\alpha^7 m$ and due to the smallness of QED effects. We observe a generally good agreement between theory and experiment for the fine-structure intervals. The only tension is a 1.4 σ deviation for the $P_{1,2}$ interval measured in Ref. [3]. We note that all pre-2010 experimental results were to a greater or lesser degree influenced by unaccounted quantum-interference effects and were reevaluated in Refs. [74, 75].

Summarizing, we have reviewed a large amount of work accomplished during the last decades in calculations of QED effects in the atomic structure of the helium atom. The leading-order $\alpha^5 m$ QED effects are nowadays well established by independent calculations and tested by comparison with numerous experiments. However, recent calculations of higher-order $\alpha^6 m$ and $\alpha^7 m$ QED effects revealed some small but systematic deviations from high-precision experimental transition energies. Having in mind the importance of the helium spectroscopy for determination of nuclear properties and fundamental constants, we conclude that further theoretical and experimental efforts are needed in order to find the reasons behind the observed discrepancies.

TABLE VIII. Comparison of theory and experiment for the 2^3P fine-structure intervals in ^4He , in kHz .

	Reference	$2^3P_0 - 2^3P_2$	$2^3P_1 - 2^3P_2$	$2^3P_0 - 2^3P_1$
Theory	Pachucki and Yerokhin 2010 [14]	31 908 131.4 (1.7)	2 291 178.9 (1.7)	29 616 952.3 (1.7)
Experiment	Kato 2018 [3]		2 291 176.590 (25)	
	Zheng 2017 [55]	31 908 130.98 (13)	2 291 177.56 (19)	
	Feng 2015 [67]		2 291 177.69 (36)	
	Smiciklas 2010 [68] ^a	31 908 131.25 (32)		
	Borbely 2009 [69] ^a		2 291 177.55 (35)	
	Zelevinsky 2005 [70] ^a	31 908 126.8 (3.0)	2 291 176.8 (1.1)	29 616 951.7 (3.0)
	Guisfredi 2005 [71] ^a			29 616 953. (10.0)
George 2001 [72] ^a			29 616 950.8 (0.9)	
Castilleja 2000 [73] ^a			2 291 177.1 (1.0)	

^a reevaluated in Refs. [74, 75].

AUTHOR CONTRIBUTIONS

All authors contributed equally. All authors have read and agreed to the published version of the manuscript.

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CONFLICTS OF INTEREST

The authors declare no conflict of interest.

Appendix A: Expectation values of singular operators

In this section we discuss the evaluation of matrix elements of singular operators $1/r^3$, $1/r^4$, and $1/r^5$. The $1/r^3$ operator is standardly defined as

$$\int d^3r \frac{f(\vec{r})}{r^3} \equiv \lim_{\epsilon \rightarrow 0} \left[\int_{\epsilon}^{\infty} dr \frac{f(r)}{r} + f(0) (\gamma_E + \ln \epsilon) \right], \quad (\text{A1})$$

where $f(r) \equiv \int d\Omega f(\vec{r})$ assumed to be a smooth function that allows a Taylor expansion at $r = 0$. Further singular operators are defined [16, 17] as

$$\int d^3r \frac{f(\vec{r})}{r^4} \equiv \lim_{\epsilon \rightarrow 0} \left[\int_{\epsilon}^{\infty} dr \frac{f(r)}{r^2} - \frac{f(0)}{\epsilon} + f'(0) (\gamma_E + \ln \epsilon) \right], \quad (\text{A2})$$

$$\int d^3r \frac{f(\vec{r})}{r^5} \equiv \lim_{\epsilon \rightarrow 0} \left[\int_{\epsilon}^{\infty} dr \frac{f(r)}{r^3} - \frac{f(0)}{2\epsilon^2} - \frac{f'(0)}{\epsilon} \right]$$

$$+ \frac{f''(0)}{2} (\gamma_E + \ln \epsilon) \Big], \quad (\text{A3})$$

and

$$\int d^3r f(\vec{r}) \frac{\ln r}{r^4} \equiv \lim_{\epsilon \rightarrow 0} \left[\int_{\epsilon}^{\infty} dr \frac{f(r) \ln r}{r^2} - f(0) \frac{(1 + \ln \epsilon)}{\epsilon} - f'(0) \frac{\ln^2 \epsilon}{2} \right]. \quad (\text{A4})$$

Here we corrected the sign misprint in Eq. (E3) of Ref. [17]. Note that the definition of the $1/r^5$ operator given by Eqs. (154) and (155) of Ref. [16] is valid for triplet states only.

We now obtain explicit formulas for integrals of singular operators with exponential functions, starting with the operator $1/r^3$,

$$I_{\epsilon}(1, 1, -2) = \lim_{\epsilon \rightarrow 0} \frac{1}{16\pi^2} \int d^3r_1 \int d^3r_2 e^{-\alpha r_1 - \beta r_2 - \gamma r} \times \left[\frac{1}{r^3} \Theta(r - \epsilon) + 4\pi \delta^3(r) (\gamma_E + \ln \epsilon) \right]. \quad (\text{A5})$$

It is evident that

$$-\frac{\partial}{\partial \gamma} I_{\epsilon}(1, 1, -2) = I(1, 1, -1), \quad (\text{A6})$$

where $I(1, 1, -1)$ can be immediately obtained from Eq. (48). Therefore, the formal integration of $I(1, 1, -1)$ over the parameter γ gives us an expression for $I_{\epsilon}(1, 1, -2)$, which is correct up to a γ -independent constant. The simplest way to fix this constant is to examine the limit of Eq. (A5) for $\gamma \rightarrow \infty$. For very large γ , only the region of small r contributes, and therefore

$$\begin{aligned} & \frac{1}{16\pi^2} \int d^3r_1 \int d^3r_2 \frac{e^{-\alpha r_1 - \beta r_2 - \gamma r}}{r^3} \Theta(r - \epsilon) \\ &= 2 \int_{\epsilon}^{\infty} dr r \int_0^{\infty} dr_1 r_1 \int_{|r_1 - r|}^{r_1 + r} dr_2 r_2 \frac{e^{-\alpha r_1 - \beta r_2 - \gamma r}}{r^3} \\ &\approx \frac{2}{(\alpha + \beta)^3} \int_{\epsilon}^{\infty} dr \frac{e^{-\gamma r}}{r}. \end{aligned} \quad (\text{A7})$$

Hence,

$$I_\epsilon(1, 1, -2) \underset{\gamma \rightarrow \infty}{=} -\frac{2}{(\alpha + \beta)^3} \ln \gamma. \quad (\text{A8})$$

We conclude that the γ -independent constant in the limit $\gamma \rightarrow \infty$ vanishes. It is interesting that this simple prescription holds also for other singular integrals. Fixing the γ -independent constant, we arrive at the final result

$$I_\epsilon(1, 1, -2) = \frac{1}{(\alpha + \beta)^3} \left[-\ln [(\alpha + \gamma)(\beta + \gamma)] - \frac{8\alpha\beta}{(\alpha - \beta)^2} + \frac{(\alpha + \beta)^3 + 8\alpha\beta\gamma}{(\alpha - \beta)^3} \ln \frac{\alpha + \gamma}{\beta + \gamma} \right]. \quad (\text{A9})$$

$$I_\epsilon(1, 1, -3) = \frac{1}{(\alpha^2 - \beta^2)^3} \left\{ 2(\alpha - \beta) [\alpha\beta(\alpha + \beta) - (\alpha^2 - 4\alpha\beta + \beta^2)\gamma] - [2\alpha\beta(\alpha^2 + \beta^2) + (\alpha + \beta)^3\gamma + 4\alpha\beta\gamma^2] \ln \left(\frac{\alpha + \gamma}{\beta + \gamma} \right) \right\} + \frac{\gamma}{(\alpha + \beta)^3} \ln [(\alpha + \gamma)(\beta + \gamma)]. \quad (\text{A11})$$

Repeating the same procedure once more, we obtain also a result for the $1/r^5$ integral,

$$I_\epsilon(1, 1, -4) = \frac{1}{6(\alpha^2 - \beta^2)^3} [\alpha\beta(\alpha + \beta)^3 + 12\alpha\beta(\alpha^2 + \beta^2)\gamma + 3(\alpha + \beta)^3\gamma^2 + 8\alpha\beta\gamma^3] \ln \left(\frac{\alpha + \gamma}{\beta + \gamma} \right) + \frac{1}{18(\alpha - \beta)^2(\alpha + \beta)^3} [-\alpha\beta(17\alpha^2 - 10\alpha\beta + 17\beta^2) - 24\alpha\beta(\alpha + \beta)\gamma + 3(9\alpha^2 - 26\alpha\beta + 9\beta^2)\gamma^2] + \frac{1}{6(\alpha + \beta)^3} (\alpha\beta - 3\gamma^2) \ln [(\alpha + \gamma)(\beta + \gamma)]. \quad (\text{A12})$$

Appendix B: Tensor decomposition of a product of irreducible tensors

In calculations of the Bethe logarithm and the second-order perturbation corrections, we encounter a problem of decomposition of products of irreducible Cartesian tensors into the irreducible parts. In this section we collect formulas required for such decompositions. The product of two vectors is represented as a sum of a symmetric and traceless second-rank tensor, a vector, and a scalar,

$$P^i Q^j = (P^i Q^j)^{(2)} + \frac{1}{2} \epsilon^{ijk} (\vec{P} \times \vec{Q})^k + \frac{\delta^{ij}}{3} \vec{P} \cdot \vec{Q}. \quad (\text{B1})$$

We note that this expression has a spurious singularity at $\alpha = \beta$. It can be easily removed if we separate the first two Taylor expansion terms of the logarithm function by introducing $\overline{\ln}_2(x)$.

We now turn to the integral with $1/r^4$. Analogously to $1/r^3$, we write

$$-\frac{\partial}{\partial \gamma} I_\epsilon(1, 1, -3) = I_\epsilon(1, 1, -2). \quad (\text{A10})$$

So, integrating Eq. (A9) over γ and setting the γ -independent constant in the limit $\gamma \rightarrow \infty$ to zero (this time we omit the justification), we obtain

The product of a vector and a symmetric and traceless second-rank tensor is decomposed as

$$P^i Q^{jk} = (P^i Q^{jk})^{(3)} + \epsilon^{ijl} T^{kl} + \epsilon^{ikl} T^{jl} + \delta^{ij} T^k + \delta^{ik} T^j - \frac{2}{3} \delta^{jk} T^i, \quad (\text{B2})$$

where

$$T^k = \frac{3}{10} P^i Q^{ik}, \quad (\text{B3})$$

$$T^{kl} = \frac{1}{6} P^i (\epsilon^{ijl} Q^{jk} + \epsilon^{ijk} Q^{jl}). \quad (\text{B4})$$

This identity can be verified by contracting Eq. (B2) with δ^{ij} and ϵ^{ijk} . It can be easily extended to the higher-rank tensors Q .

Finally, we present the decomposition of the product of two symmetric and traceless tensors P^{ij} and Q^{kl} , required for calculations of second-order corrections involving D -states,

$$P^{ij} Q^{kl} = (P^{ij} Q^{kl})^{(4)} + \epsilon^{ika} T^{jal} + \epsilon^{jka} T^{ial} + \epsilon^{ila} T^{jak} + \epsilon^{jla} T^{iak} + \delta^{ik} T^{jl} + \delta^{il} T^{jk} + \delta^{jk} T^{il} + \delta^{jl} T^{ik}$$

$$-\frac{4}{3}\delta^{ij}T^{kl}-\frac{4}{3}\delta^{kl}T^{ij}+T^a(\epsilon^{ika}\delta^{jl}+\epsilon^{ila}\delta^{jk}+\epsilon^{jka}\delta^{il}+\epsilon^{jla}\delta^{ik})+T(\delta^{ik}\delta^{jl}+\delta^{il}\delta^{jk}-\frac{2}{3}\delta^{ij}\delta^{kl}), \quad (\text{B5})$$

where

$$T^{jbl}=\frac{1}{4}(\epsilon^{ikb}P^{ij}Q^{kl})^{(3)}, \quad (\text{B6})$$

$$T^{jl}=\frac{3}{7}(P^{ij}Q^{il})^{(2)}, \quad (\text{B7})$$

$$T^b=\frac{1}{10}\epsilon^{jlb}P^{ij}Q^{il}, \quad (\text{B8})$$

$$T=\frac{1}{10}P^{ij}Q^{ij}. \quad (\text{B9})$$

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