

平成 23 年度研究成果発表リスト

1. 論文発表

■先端表面敏感計測技術の開発と先進材料応用

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■強磁場固体NMR計測技術の開発と先進材料応用

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5. 主要論文別刷り

■先端表層領域計測技術の開発と先進材料応用

- [19] H.Shinotsuka, S.Tanuma, C. J. Powell, D.R. Penn, “Calculations of electron stopping powers for 41 elemental solids over the 50 eV to 30 keV range with the full Penn algorithm”, NUCLEAR INSTRUMENTS & METHODS IN PHYSICS RESEARCH SECTION B-BEAM INTERACTIONS WITH MATERIALS AND ATOMS, Vol.270 (2011).

■超先端電子顕微鏡技術の開発と先進材料応用

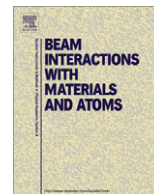
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Calculations of electron stopping powers for 41 elemental solids over the 50 eV to 30 keV range with the full Penn algorithm

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ABSTRACT

We present mass collision electron stopping powers (SPs) for 41 elemental solids (Li, Be, graphite, diamond, glassy C, Na, Mg, Al, Si, K, Sc, Ti, V, Cr, Fe, Co, Ni, Cu, Ge, Y, Nb, Mo, Ru, Rh, Pd, Ag, In, Sn, Cs, Gd, Tb, Dy, Hf, Ta, W, Re, Os, Ir, Pt, Au, and Bi) that were calculated from experimental energy-loss-function data with the full Penn algorithm for electron energies between 50 eV and 30 keV. Improved sets of energy-loss functions were used for 19 solids. Comparisons were made of these SPs with SPs calculated with the single-pole approximation, previous SP calculations, and experimental SPs. Generally satisfactory agreement was found with SPs from the single-pole approximation for energies above 100 eV, with other calculated SPs, and with measured SPs.

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1. Introduction

In previous papers [1,2], we reported calculations of collision electron stopping powers (SPs) over the 100 eV to 30 keV energy range in 41 elemental solids from their “optical” energy-loss functions (ELFs). These ELFs were obtained from experimental optical data representing the dependence of the inelastic-scattering probability on energy loss and the theoretical Lindhard dielectric function [3] to represent the dependence of the scattering probability on momentum transfer. SPs were calculated with Penn’s algorithm that was originally developed for the calculation of electron inelastic mean free paths (IMFPs) [4]. Our SPs were determined using the single-pole approximation or so-called simple Penn algorithm (SPA) that was expected to be satisfactory for electron energies greater than about 100 eV [1,2]. We have extended this earlier work in two ways. First, we have calculated SPs with the full Penn algorithm (FPA) which should be valid for electron energies down to about 50 eV [5]. Second, we have adopted better sets of optical ELF data in recent IMFP calculations with the FPA for 19 of our 41 elemental solids [6], and we make use of the improved ELF data in the present work.

We report mass collision SPs calculated with the FPA for 41 elemental solids (Li, Be, graphite, diamond, glassy C, Na, Mg, Al, Si, K, Sc, Ti, V, Cr, Fe, Co, Ni, Cu, Ge, Y, Nb, Mo, Ru, Rh, Pd, Ag, In, Sn, Cs, Gd, Tb, Dy, Hf, Ta, W, Re, Os, Ir, Pt, Au, and Bi) over the 50 eV to 30 keV energy range with the same optical ELF data sets that were used in the IMFP calculations [6]. We give a brief description of our

SP algorithm in the next section. The new SPs are presented in the following section where we compare SPs from the new ELFs (and the FPA) to SPs from the old ELFs (and the SPA) and compare SPs from the FPA to SPs from the SPA. We then make comparisons of our new SPs with values from previous SP calculations and measurements.

We note that the collision electron SP is an important parameter in radiation dosimetry [7] and in the modeling of electron transport in matter for many other applications. The SP has been used in Monte Carlo simulations of electron transport relevant to electron-probe microanalysis [8–10], Auger-electron spectroscopy [11,12], and dimensional metrology in the scanning electron microscope [13]. The Bethe SP equation [14–16] has been used extensively as a predictive tool for energies where it is expected to be valid (i.e., at energies much larger than the largest K-shell binding energy in the material of interest), but there is a scarcity of SP data at lower energies, typically less than 10 keV. SPs calculated from the Bethe equation are available from a National Institute of Standards and Technology (NIST) database [17] and ICRU Report 37 for electron energies of 10 keV and above [7]. It is thus important to have SPs available for lower energies in order to describe electron-solid interactions for a variety of applications.

2. Calculation of electron stopping powers with the full Penn algorithm

Penn developed an algorithm for the calculation of electron IMFPs from a model dielectric function $\epsilon(q, \omega)$, a function of momentum transfer q and energy loss $\hbar\omega$ [4]. The energy dependence of the energy-loss function (ELF), $\text{Im}[-1/\epsilon(q, \omega)]$, can

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be obtained from experimental optical data for the material of interest and the dependence of the ELF on q can be obtained from the Lindhard model dielectric function [3]. IMFPs were determined from a triple integration (the FPA) for energies down to about 50 eV, while a simpler procedure involving a single integration (the SPA) was judged satisfactory for electron energies larger than about 200 eV [4–6]. We utilized the SPA in our previous SP calculations, but now use the FPA to extend the energy range down to 50 eV.

We give here a summary of our implementation of the full Penn algorithm for SP calculations. We will use Hartree atomic units ($m_e = e = \hbar = 1$) where m_e is the electron rest mass, e is the elementary charge, and $\hbar$ is the reduced Planck constant.

The relativistic differential cross section (DCS) for inelastic scattering can be expressed as the sum of a longitudinal DCS and a transverse DCS [42]. For electron energies less than about 0.5 MeV, the transverse DCS can be neglected [18]. The relativistic inelastic DCS can then be written as [42]:

$$\frac{d^2\sigma}{d\omega dQ} \approx \frac{d^2\sigma_L}{d\omega dQ} = \frac{1}{v^2} \frac{1+Q/c^2}{Q(1+Q/2c^2)} \frac{1}{\pi N} \text{Im} \left(\frac{-1}{\varepsilon(Q, \omega)} \right), \quad (1)$$

where Q is the recoil energy given by $Q(Q+2c^2) = (cq)^2$ [19,42], v is the electron velocity, c is the speed of light, and N is the number of atoms or molecules per unit volume. The last factor in Eq. (1) is the ELF expressed here as a function of energy loss ω and recoil energy Q . The relativistic DCS in Eq. (1) can be conveniently written as a function of momentum transfer q :

$$\frac{d^2\sigma}{d\omega dq} \approx \frac{2}{\pi N v^2} \text{Im} \left(\frac{-1}{\varepsilon(q, \omega)} \right) \frac{1}{q}. \quad (2)$$

Stopping powers can be calculated with the FPA from the probability $p(T, \omega)$ for energy loss ω per unit distance traveled by an electron with relativistic kinetic energy T . This probability can be calculated from Eq. (2):

$$p(T, \omega) = \frac{2}{\pi v^2} \int_{q_-}^{q_+} \frac{dq}{q} \text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right] \\ = \frac{(1+T/c^2)^2}{1+T/(2c^2)} \frac{1}{\pi T} \int_{q_-}^{q_+} \frac{dq}{q} \text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right], \quad (3)$$

where $q_{\pm} = \sqrt{T(2+T/c^2)} \pm \sqrt{(T-\omega)(2+(T-\omega)/c^2)}$. The collision stopping power, S , can be calculated from the following equation [20]:

$$S = \int_0^{\omega_{\max}} \omega p(T, \omega) d\omega, \quad (4)$$

where $\omega_{\max} = T - E_f$ and E_f is the Fermi energy.

The ELF in the FPA can be expressed as:

$$\text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right] = \int_0^{\infty} d\omega_p g(\omega_p) \text{Im} \left[\frac{-1}{\varepsilon^L(q, \omega; \omega_p)} \right], \quad (5a)$$

where ε^L denotes the Lindhard model dielectric function of the free electron gas with plasmon energy $\omega_p (= \sqrt{4\pi n})$, n is the electron density, $g(\omega_p)$ is a coefficient introduced to satisfy the condition $\text{Im}[-1/\varepsilon(q=0, \omega)] = \text{Im}[-1/\varepsilon(\omega)]$, and $\text{Im}[-1/\varepsilon(\omega)]$ is the optical energy-loss function. The coefficient $g(\omega_p)$ is then given by

$$g(\omega) = \frac{2}{\pi \omega} \text{Im} \left[\frac{-1}{\varepsilon(\omega)} \right]. \quad (5b)$$

The Lindhard ELF, $\varepsilon^L = \varepsilon_1^L + i\varepsilon_2^L$, can be written as [3,21]:

$$\text{Im} \left[\frac{-1}{\varepsilon^L(q, \omega; \omega_p)} \right] = \frac{\varepsilon_2^L}{(\varepsilon_1^L)^2 + (\varepsilon_2^L)^2}, \quad (6a)$$

$$\varepsilon_1^L(q, \omega; \omega_p) = 1 + \frac{1}{\pi k_F z^2} \left[\frac{1}{2} + \frac{1}{8z} \left\{ F\left(z - \frac{x}{4z}\right) + F\left(z + \frac{x}{4z}\right) \right\} \right], \quad (6b)$$

$$\varepsilon_2^L(q, \omega; \omega_p) = \frac{1}{8k_F z^3} \times \begin{cases} x & \text{for } 0 < x < 4z(1-z) \\ 1 - (z - (x/4z))^2 & \text{for } |4z(1-z)| < x < 4z(1+z), \\ 0 & \text{otherwise} \end{cases} \quad (6c)$$

where $F(t) = (1-t^2) \ln |(t+1)/(t-1)|$, $x = \omega/E_f$, $z = q/2k_F$, and k_F is the Fermi wave vector corresponding to a given ω_p .

We used the following expressions for the real and imaginary parts of the Lindhard dielectric function in our numerical calculations to reduce numerical errors at the limiting conditions of $\omega/qk_F (= u) \ll 1$ and $u \gg z+1$. When $u \ll 1$, ε_1^L and ε_2^L can be written as:

$$\varepsilon_1^L(q, \omega; \omega_p) = 1 + \frac{1}{\pi k_F z^2} \left[\frac{1}{2} + \frac{1}{4z} \left\{ (1-z^2-u^2) \ln \left| \frac{z+1}{z-1} \right| + (z^2-u^2-1) \frac{2u^2 z}{(z^2-1)^2} \right\} \right] \quad (7a)$$

and

$$\varepsilon_2^L(q, \omega; \omega_p) = \frac{u}{2k_F z^2}. \quad (7b)$$

When $u \gg z+1$, ε_1^L and ε_2^L can be expressed as:

$$\varepsilon_1^L(q, \omega; \omega_p) = 1 - \frac{\omega_p^2}{\omega^2} \left\{ 1 + \left(z^2 + \frac{3}{5} \right) \frac{1}{u^2} \right\} \quad (8a)$$

and

$$\varepsilon_2^L(q, \omega; \omega_p) = 0. \quad (8b)$$

The energy-loss function in Eq. (5) from the FPA can be described as the sum of two contributions, one associated with the plasmon pole and the other with single-electron excitations [3,21,22]. That is,

$$\text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right] = \text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right]_{pl} + \text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right]_{se}. \quad (9)$$

The plasmon-pole contribution can be expressed as:

$$\text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right]_{pl} = g(\omega_0) \frac{\pi}{|\partial \varepsilon_1^L(q, \omega; \omega_p) / \partial \omega_p|_{\omega_p=\omega_0}} \Theta(q^-(\omega; \omega_0) - q), \quad (10a)$$

where

$$\frac{\partial \varepsilon_1^L(q, \omega; \omega_p)}{\partial \omega_p} = \frac{1}{3\pi \omega_p q z^2} \left\{ \ln \left| \frac{Y_- + 1}{Y_- - 1} \right| + \ln \left| \frac{Y_+ + 1}{Y_+ - 1} \right| \right\}, \quad (10b)$$

and $Y_{\pm} \equiv z \pm x/4z$. To reduce calculation errors for $z/x \ll 1$ and for $z/x \gg 1$, we use the following equations:

$$\ln \left| \frac{Y_- + 1}{Y_- - 1} \right| + \ln \left| \frac{Y_+ + 1}{Y_+ - 1} \right| \approx -\frac{64}{3} z a^2 \{ 3 + 48(1+z^2)a^2 + 256(3+z^2)(1+3z^2)a^4 \}, \quad (10c)$$

where $a \equiv z/x$ when $z/x \ll 1$, and

$$\ln \left| \frac{Y_- + 1}{Y_- - 1} \right| + \ln \left| \frac{Y_+ + 1}{Y_+ - 1} \right| \approx \ln \left(\frac{z+1}{z-1} \right)^2 + 4zb^2 \{ 1 + (1+z^2)b^2 + \frac{1}{3}(3+z^2)(1+3z^2)b^4 \}, \quad (10d)$$

where $b \equiv x/(z^2-1)$ when $z/x \gg 1$.

The single-electron-excitation contribution in Eq. (9) is given by:

$$\text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right]_{se} = \int_0^\infty d\omega_p g(\omega_p) \text{Im} \left[\frac{-1}{\varepsilon^L(q, \omega; \omega_p)} \right] \Theta(q^+(\omega; \omega_p) - q) \Theta(q - q^-(\omega; \omega_p)), \quad (11a)$$

where

$$q^\pm(\omega; \omega_p) = \pm k_F(\omega_p) + \sqrt{k_F^2(\omega_p) + 2\omega}. \quad (11b)$$

Finally, when $Y_- = \pm 1$ (corresponding to quadratic curves in the (q, ω) plane separating single-electron excitations), we note that $F(\pm 1)$ is theoretically zero.

Several factors limit the reliability of SPs calculated from our model [2]. First, the Lindhard dielectric function model used in the FPA is expected to provide a useful approximation for the q -dependence of valence-electron excitations in free-electron-like materials but may be less reliable for non-free-electron-like solids. Second, use of the Lindhard model for describing the q -dependence of core-electron excitations is unlikely to be correct. Third, no account has been taken of exchange effects in inelastic scattering. Nevertheless, IMFPs calculated from optical ELF with the FPA and the SPA are in good agreement with IMFPs determined experimentally by elastic-peak electron spectroscopy (EPES) in many elemental solids (including non-free-electron-like solids) for electron energies between 100 eV and 5 keV [23–25]. It is difficult to extend these comparisons to energies less than 100 eV due to limitations of the EPES technique [26].

3. Results

3.1. Stopping powers from the full Penn algorithm

We calculated values of the mass collision SP, S/ρ , where ρ is the mass density of the solid. Values of S/ρ were determined for 41 elemental solids (Li, Be, graphite, diamond, glassy C, Na, Mg, Al, Si, K, Sc, Ti, V, Cr, Fe, Co, Ni, Cu, Ge, Y, Nb, Mo, Ru, Rh, Pd, Ag, In, Sn, Cs, Gd, Tb, Dy, Hf, Ta, W, Re, Os, Ir, Pt, Au, and Bi) from the FPA. We utilized the same sets of optical ELFs as those used in our recent calculations of IMFPs for the same solids over the same energy range [6]. For 19 of our 41 solids (Ti, V, Cr, Fe, Ni, Y, Nb, Mo, Ru, Rh, Pd, Hf, Ta, W, Re, Os, Ir, Pt, and Au), we adopted improved sets of optical ELF data [6] compared to those used for our previous SP results [1,2]. We note that the new ELF data sets for Mg and Cu utilized in our recent IMFP calculations [6] had also been employed in our previous SP calculations for these solids [2]. Although we previously reported SPs for Zr [2], further analysis of its optical ELF data showed what we considered to be excessive errors in values of the f -sum and KK-sum [6] that we used to evaluate the internal consistency of a given ELF data set. Our SP values for Zr should therefore be considered only as rough estimates.

Values of S/ρ were calculated for relativistic electron kinetic energies between 10 eV and 30 keV (with respect to the Fermi level) at equal intervals on a logarithmic energy scale corresponding to increases of 10%. Table 1 shows the S/ρ values for our 41 elemental solids at electron energies between 50 eV and 30 keV. These SPs are given in units of MeVcm²/g. The mass collision SPs can be converted to collision SPs (e.g., in eV/Å or eV/nm units) by multiplying by the material densities given in Table 1 of Ref. [6].

Plots of calculated mass collision SPs from the FPA as a function of energy are shown as solid lines in Figs. 1–7. SPs are included in these plots for energies less than 50 eV to illustrate trends, but these data are not considered as reliable [2]. The solid circles in Figs. 1–7 are SPs calculated with the SPA [1,2]. The SPs calculated with the FPA and SPA show similar systematic trends with atomic number. Sometimes, a single maximum is observed in the SP-versus-energy curves, sometimes secondary structures or multiple

maxima are observed, and there are varying widths of the main maximum that generally occurs at energies between 10 eV and 300 eV. These trends have been discussed previously and are due to the varying contributions of valence-electron and different inner-shell excitations to the SP [2,27]. We also see in Figs. 1–7 that SPs from the SPA are larger than those from the FPA at energies in the vicinity of the maximum in each SP-versus-energy curve. On the other hand, SPs from the SPA are smaller than the corresponding SPs from the FPA for very low energies (typically less than 30 eV). This result is mainly due to differences in the ELF models used in each algorithm and will be discussed in more detail later.

The dashed lines in Figs. 1–7 show mass collision SPs calculated from the relativistic Bethe equation [2,14–16]:

$$S/\rho = \frac{784.58Z}{m_e v^2 A} [\ln(T/I)^2 + \ln(1 + \tau/2) + G(\tau)] \quad (\text{in MeVcm}^2\text{g}^{-1}), \quad (12a)$$

where

$$G(\tau) = (1 - \beta^2)[1 + \tau^2/8 - (2\tau + 1)\ln 2], \quad (12b)$$

Z is the atomic number of the target, β is the electron velocity divided by the velocity of light, c , $\tau = T/m_e c^2$ is the ratio of the electron relativistic kinetic energy to its rest energy, and I is the mean excitation energy (MEE). Eq. (12) omits a density-effect correction which has been assumed here to be zero since its contribution is very small (less than 0.3% for our energy range [7]). SPs were calculated from Eq. (12) using MEEs listed in Table 4.3 of Ref. [7] except for the three carbon allotropes. SPs for glassy C, graphite and diamond were calculated with MEEs from our previous analysis [2]. SPs from Eq. (12) are shown in Figs. 1–7 from the minimum energies for which S is positive to 30 keV. As expected, we see generally good agreement between SPs from the FPA and those from Eq. (12) for energies larger than 10 keV. The root-mean-square (RMS) relative deviations between these calculated SPs and those from the Bethe equation were 9.1% and 8.7% at energies of 9.897 and 29.733 keV, respectively. These RMS deviations are almost the same as the corresponding deviations of 9.8% and 8.5% found previously between our SPs from the SPA and those from the Bethe equation [2].

3.2. Comparisons of stopping powers from new and old energy-loss functions

The energy-loss function (ELF) is the critical material-dependent parameter in our SP calculations. These ELFs were obtained from experimental optical data or ELF measurements for each solid. Sources of ELF data were given and details of our ELF analyses were described in a previous paper [6]. We comment now on differences between ELFs for 19 solids (Ti, V, Cr, Fe, Ni, Y, Nb, Mo, Ru, Rh, Pd, Hf, Ta, W, Re, Os, Ir, Pt, and Au) used here and the ELFs used for our previous SP calculations [1,2].

We first point out that we utilized photoabsorption data for these 19 solids at photon energies over 50 eV from Henke et al. [28] that are more recent than the data used previously [1,2]. For eight of these solids (Cr, Fe, Mo, Hf, Ta, W, Re, and Pt), it was necessary to make interpolations between two photon-energy (or electron energy-loss) regions, and we were guided in this process by measured transmission and reflection electron energy-loss spectroscopy (EELS) data. The resulting ELFs agreed better overall with the energy-loss data and resulted in smaller sum-rule errors for most of the solids than in our earlier SP and IMFP work [1,2,5,29]. For Ti, we selected optical data from a recent analysis of reflection EELS data by Werner et al. [30] for energy losses up to 54 eV because the resulting ELF was in much better agreement

Table 1
Calculated mass collision SPs (i.e., collision SPs divided by density) for the 41 elemental solids as a function of electron relativistic kinetic energy *T*.

T (eV)	Collision stopping power/density (MeVcm ² g ⁻¹)						
	Li	Be	C (glassy)	C (graphite)	C (diamond)	Na	Mg
54.6	439.6	361.6	217.0	259.7	141.7	187.2	240.0
60.3	422.3	366.0	227.8	286.5	166.1	180.9	235.2
66.7	404.6	366.8	236.6	307.7	192.0	175.0	229.5
73.7	387.1	364.3	243.3	323.4	216.4	169.8	223.4
81.5	369.8	359.1	247.9	334.3	236.7	165.6	217.0
90.0	353.3	351.8	250.5	340.8	251.6	163.0	210.5
99.5	337.3	342.7	251.3	343.4	261.8	161.6	204.1
109.9	322.4	332.3	250.5	342.8	267.8	160.9	198.0
121.5	308.3	320.8	248.2	339.4	270.3	160.7	192.2
134.3	295.4	308.6	244.4	333.8	270.0	160.7	187.4
148.4	284.8	296.0	239.4	326.3	267.4	160.9	183.6
164.0	276.7	283.2	233.4	317.3	262.9	160.9	180.5
181.3	270.0	270.2	226.4	307.0	256.8	160.8	177.9
200.3	263.4	257.6	218.7	295.9	249.4	160.4	175.9
221.4	256.6	245.1	210.5	284.1	241.1	159.6	173.8
244.7	249.4	233.0	201.9	271.8	232.1	158.2	171.5
270.4	241.9	221.4	193.2	259.3	222.6	156.3	169.0
298.9	233.7	210.4	184.2	246.7	212.8	153.8	166.0
330.3	224.9	200.2	175.3	234.2	202.8	150.7	162.5
365.0	215.7	191.2	166.4	221.7	192.7	147.1	158.6
403.4	206.2	182.9	157.7	209.6	182.8	142.9	154.1
445.9	196.5	174.8	149.2	197.7	173.0	138.3	149.3
492.7	186.7	166.9	140.9	186.2	163.4	133.4	144.0
544.6	177.0	159.0	133.0	175.2	154.1	128.1	138.6
601.8	167.3	151.2	125.3	164.6	145.2	122.7	132.9
665.1	157.8	143.5	118.1	154.5	136.6	117.2	127.1
735.1	148.6	135.9	111.2	145.0	128.4	111.5	121.3
812.4	139.6	128.4	104.7	135.9	120.7	105.9	115.5
897.8	130.9	121.1	98.6	127.4	113.4	100.3	109.7
992.3	122.6	114.0	92.9	119.5	106.6	94.8	103.9
1096.6	114.6	107.2	87.8	112.2	100.3	89.4	98.3
1212.0	107.1	100.6	83.2	105.4	94.5	84.1	92.8
1339.4	99.9	94.2	78.8	99.0	88.9	79.1	87.4
1480.3	93.0	88.2	74.5	92.9	83.6	74.2	82.2
1636.0	86.6	82.4	70.3	87.1	78.6	69.5	77.2
1808.0	80.5	76.9	66.3	81.5	73.7	65.0	72.4
1998.2	74.8	71.7	62.4	76.3	69.1	60.7	67.8
2208.3	69.4	66.7	58.6	71.3	64.7	56.7	63.4
2440.6	64.4	62.1	55.0	66.5	60.5	52.9	59.3
2697.3	59.7	57.7	51.6	62.0	56.5	49.3	55.3
2981.0	55.3	53.6	48.3	57.8	52.7	45.9	51.5
3294.5	51.2	49.7	45.1	53.8	49.1	42.7	48.0
3640.9	47.4	46.1	42.1	50.0	45.7	39.7	44.7
4023.9	43.8	42.7	39.3	46.5	42.5	36.9	41.5
4447.1	40.5	39.6	36.6	43.2	39.5	34.3	38.6
4914.8	37.5	36.7	34.0	40.0	36.7	31.9	35.9
5431.7	34.6	33.9	31.7	37.1	34.1	29.7	33.3
6002.9	32.0	31.4	29.4	34.4	31.6	27.6	31.0
6634.2	29.5	29.0	27.3	31.9	29.3	25.7	28.8
7332.0	27.3	26.8	25.3	29.5	27.2	23.9	26.7
8103.1	25.2	24.8	23.5	27.3	25.2	22.2	24.8
8955.3	23.2	22.9	21.8	25.3	23.3	20.6	23.1
9897.1	21.4	21.2	20.2	23.4	21.6	19.1	21.4
10938.0	19.8	19.6	18.7	21.6	20.0	17.7	19.9
12088.4	18.3	18.1	17.3	20.0	18.5	16.5	18.5
13359.7	16.9	16.7	16.0	18.5	17.1	15.3	17.1
14764.8	15.6	15.4	14.9	17.1	15.8	14.2	15.9
16317.6	14.4	14.2	13.7	15.8	14.6	13.1	14.7
18033.7	13.3	13.1	12.7	14.6	13.5	12.2	13.7
19930.4	12.2	12.1	11.8	13.5	12.5	11.3	12.7
22026.5	11.3	11.2	10.9	12.5	11.6	10.5	11.8
24343.0	10.4	10.4	10.1	11.6	10.7	9.7	10.9
26903.2	9.65	9.60	9.36	10.7	9.92	9.01	10.1
29732.6	8.92	8.88	8.67	9.91	9.18	8.36	9.39
T (eV)	Al	Si	K	Sc	Ti	V	Cr
54.6	220.1	227.7	156.8	152.7	119.4	79.9	73.6
60.3	216.9	226.1	163.6	161.8	125.4	84.5	79.8
66.7	212.6	222.9	170.4	170.6	130.4	88.6	85.1
73.7	207.3	218.4	175.9	179.1	134.6	92.2	89.5
81.5	201.2	212.8	180.7	187.3	138.4	95.2	92.9
90.0	194.8	206.6	186.3	195.5	142.5	98.0	95.5
99.5	188.0	199.7	189.8	204.0	147.4	100.9	97.4

Table 1 (continued)

T (eV)	Al	Si	K	Sc	Ti	V	Cr
109.9	181.1	192.6	190.1	212.6	153.5	104.1	98.8
121.5	174.0	185.1	187.7	219.3	159.5	107.7	100.2
134.3	167.1	177.6	183.1	221.5	163.3	111.5	101.7
148.4	160.4	170.1	177.2	220.2	163.5	114.8	103.0
164.0	154.1	162.7	170.5	216.2	161.4	117.3	104.0
181.3	148.1	155.6	163.3	210.6	157.9	118.7	104.4
200.3	142.7	148.8	156.1	203.9	153.7	119.0	104.1
221.4	138.1	142.3	148.8	196.4	148.9	118.4	103.1
244.7	134.0	136.3	141.6	188.5	143.8	116.9	101.5
270.4	130.5	130.7	134.6	180.3	138.4	114.9	99.5
298.9	127.2	125.9	127.7	172.0	132.8	112.2	97.0
330.3	124.1	122.1	121.0	163.7	127.2	109.1	94.3
365.0	121.0	118.7	114.6	155.5	121.4	105.6	91.3
403.4	117.9	115.6	108.3	147.4	115.7	101.7	88.2
445.9	114.6	112.6	102.3	139.5	110.0	97.6	84.9
492.7	111.2	109.7	96.6	131.8	104.4	93.5	81.5
544.6	107.7	106.6	91.1	124.3	98.9	89.2	78.1
601.8	104.0	103.5	86.0	117.1	93.6	85.0	74.6
665.1	100.2	100.1	81.1	110.2	88.4	80.8	71.1
735.1	96.3	96.5	76.4	103.7	83.4	76.6	67.7
812.4	92.3	92.8	72.1	97.4	78.6	72.6	64.2
897.8	88.3	89.0	68.1	91.4	74.0	68.6	60.9
992.3	84.2	85.1	64.4	85.8	69.6	64.8	57.6
1096.6	80.1	81.1	61.1	80.5	65.4	61.1	54.5
1212.0	76.0	77.1	58.1	75.5	61.5	57.6	51.4
1339.4	72.0	73.1	55.4	70.8	57.7	54.2	48.5
1480.3	68.0	69.1	52.7	66.5	54.2	51.0	45.7
1636.0	64.1	65.3	50.2	62.5	51.0	48.0	43.0
1808.0	60.4	61.5	47.7	58.9	47.9	45.1	40.5
1998.2	56.7	57.8	45.2	55.5	45.2	42.5	38.1
2208.3	53.2	54.3	42.8	52.2	42.6	40.0	35.9
2440.6	49.9	50.9	40.5	49.2	40.2	37.8	33.9
2697.3	46.6	47.6	38.2	46.2	37.9	35.7	32.0
2981.0	43.6	44.5	35.9	43.4	35.7	33.7	30.3
3294.5	40.7	41.6	33.8	40.7	33.6	31.7	28.6
3640.9	37.9	38.8	31.7	38.1	31.5	29.8	27.0
4023.9	35.3	36.2	29.7	35.6	29.6	28.1	25.4
4447.1	32.9	33.7	27.8	33.3	27.7	26.3	23.9
4914.8	30.6	31.3	26.0	31.1	25.9	24.7	22.5
5431.7	28.5	29.1	24.3	29.0	24.3	23.1	21.1
6002.9	26.4	27.1	22.7	27.0	22.7	21.6	19.8
6634.2	24.6	25.2	21.1	25.2	21.1	20.2	18.5
7332.0	22.8	23.4	19.7	23.4	19.7	18.9	17.3
8103.1	21.2	21.7	18.3	21.8	18.4	17.6	16.2
8955.3	19.7	20.1	17.1	20.2	17.1	16.4	15.1
9897.1	18.3	18.7	15.9	18.8	15.9	15.3	14.1
10938.0	17.1	17.4	14.7	17.4	14.8	14.2	13.2
12088.4	15.8	16.1	13.7	16.2	13.7	13.2	12.3
13359.7	14.7	15.0	12.7	15.0	12.8	12.3	11.4
14764.8	13.7	13.9	11.8	13.9	11.9	11.4	10.6
16317.6	12.7	12.9	11.0	12.9	11.0	10.6	9.87
18033.7	11.8	12.0	10.2	12.0	10.2	9.87	9.18
19930.4	10.9	11.1	9.4	11.1	9.48	9.17	8.53
22026.5	10.2	10.3	8.8	10.3	8.80	8.52	7.93
24343.0	9.44	9.60	8.15	9.57	8.17	7.91	7.37
26903.2	8.76	8.92	7.58	8.88	7.59	7.35	6.86
29732.6	8.14	8.28	7.05	8.25	7.05	6.83	6.38
T (eV)	Fe	Co	Ni	Cu	Ge	Y	
54.6	66.9	56.8	53.3	49.6	96.0	83.4	
60.3	70.5	62.1	57.7	53.3	95.8	88.7	
66.7	73.5	67.2	62.1	57.0	95.3	94.2	
73.7	75.9	71.7	66.4	60.3	94.5	99.9	
81.5	77.7	75.6	70.4	63.2	93.4	106.1	
90.0	79.2	79.0	74.0	65.7	92.0	112.1	
99.5	80.4	81.9	77.2	68.0	90.5	117.0	
109.9	81.3	84.4	80.0	70.0	88.8	120.0	
121.5	82.0	86.6	82.4	71.7	87.1	121.0	
134.3	82.5	88.6	84.5	73.1	85.5	120.2	
148.4	83.5	90.7	86.3	74.3	83.8	118.0	
164.0	84.8	93.7	87.7	75.2	82.2	114.9	
181.3	85.8	96.2	89.0	75.9	80.7	111.0	
200.3	86.2	97.7	90.3	76.4	79.3	106.7	
221.4	86.2	98.3	91.3	76.7	77.9	102.2	
244.7	85.8	98.3	91.9	76.9	76.5	97.5	

(continued on next page)

Table 1 (continued)

T (eV)	Fe	Co	Ni	Cu	Ge	Y	
270.4	84.9	97.6	92.0	76.9	75.1	92.8	
298.9	83.8	96.4	91.6	76.5	73.8	88.2	
330.3	82.3	94.8	90.6	75.8	72.4	83.7	
365.0	80.5	92.9	89.2	74.9	71.1	79.4	
403.4	78.6	90.6	87.4	73.8	69.7	75.2	
445.9	76.4	88.0	85.4	72.5	68.2	71.3	
492.7	74.0	85.2	83.0	70.9	66.7	67.6	
544.6	71.5	82.2	80.4	69.2	65.1	64.2	
601.8	68.9	79.0	77.7	67.3	63.5	61.0	
665.1	66.2	75.8	74.8	65.2	61.7	58.1	
735.1	63.4	72.5	71.8	63.0	59.9	55.5	
812.4	60.6	69.2	68.7	60.6	57.9	53.0	
897.8	57.8	65.8	65.6	58.2	55.9	50.8	
992.3	55.0	62.5	62.5	55.7	53.8	48.6	
1096.6	52.2	59.3	59.4	53.2	51.7	46.5	
1212.0	49.5	56.1	56.3	50.6	49.5	44.6	
1339.4	46.8	52.9	53.2	48.1	47.2	42.6	
1480.3	44.2	49.9	50.3	45.6	45.0	40.8	
1636.0	41.7	47.0	47.4	43.2	42.8	38.9	
1808.0	39.3	44.2	44.7	40.8	40.6	37.1	
1998.2	37.0	41.5	42.0	38.5	38.4	35.2	
2208.3	34.8	39.0	39.5	36.2	36.3	33.5	
2440.6	32.8	36.6	37.1	34.1	34.2	31.7	
2697.3	30.8	34.3	34.8	32.1	32.2	30.0	
2981.0	29.0	32.2	32.6	30.1	30.3	28.3	
3294.5	27.3	30.2	30.6	28.3	28.5	26.7	
3640.9	25.8	28.4	28.7	26.6	26.7	25.1	
4023.9	24.3	26.7	26.9	24.9	25.0	23.6	
4447.1	22.9	25.0	25.3	23.4	23.5	22.1	
4914.8	21.6	23.5	23.8	22.0	22.0	20.7	
5431.7	20.3	22.1	22.3	20.7	20.6	19.4	
6002.9	19.0	20.7	20.9	19.5	19.3	18.2	
6634.2	17.9	19.4	19.6	18.3	18.1	17.0	
7332.0	16.7	18.2	18.4	17.1	16.9	15.8	
8103.1	15.7	17.0	17.2	16.0	15.9	14.8	
8955.3	14.6	15.9	16.1	15.0	14.9	13.8	
9897.1	13.7	14.8	15.0	14.0	13.9	12.9	
10938.0	12.8	13.8	14.0	13.1	13.0	12.0	
12088.4	11.9	12.9	13.1	12.3	12.1	11.2	
13359.7	11.1	12.0	12.2	11.4	11.3	10.5	
14764.8	10.3	11.2	11.4	10.7	10.6	9.77	
16317.6	9.62	10.4	10.6	9.94	9.87	9.12	
18033.7	8.96	9.65	9.84	9.25	9.20	8.51	
19930.4	8.33	8.98	9.16	8.62	8.58	7.94	
22026.5	7.75	8.35	8.52	8.02	7.99	7.40	
24343.0	7.21	7.76	7.92	7.47	7.44	6.90	
26903.2	6.71	7.22	7.37	6.95	6.93	6.44	
29732.6	6.24	6.72	6.86	6.47	6.45	6.00	
T (eV)	Nb	Mo	Ru	Rh	Pd	Ag	In
54.6	48.0	48.6	39.8	40.9	40.7	36.4	53.9
60.3	51.9	54.1	44.7	45.6	45.3	40.8	55.1
66.7	55.5	59.0	50.1	50.6	50.3	45.6	56.3
73.7	59.0	63.2	55.5	55.8	55.3	50.7	57.6
81.5	62.5	66.9	60.6	61.0	60.4	56.0	59.1
90.0	66.3	70.4	64.9	65.8	65.2	61.3	60.7
99.5	70.6	73.9	68.8	69.9	69.6	66.6	62.5
109.9	75.3	77.6	72.1	73.3	73.5	71.7	64.4
121.5	80.0	81.3	75.0	76.3	77.0	76.6	66.3
134.3	84.1	84.8	77.7	79.1	80.2	81.3	68.3
148.4	86.7	87.2	80.1	81.7	82.9	85.5	70.2
164.0	87.8	88.4	82.1	84.0	85.3	89.2	72.0
181.3	87.6	88.3	83.5	85.9	87.1	92.4	73.5
200.3	86.2	87.1	83.9	87.0	88.3	94.9	74.7
221.4	84.1	85.1	83.2	87.2	88.4	96.7	75.6
244.7	81.4	82.5	81.7	86.4	87.5	97.5	76.1
270.4	78.4	79.5	79.4	84.6	85.5	97.1	76.1
298.9	75.2	76.3	76.7	82.1	82.9	95.6	75.4
330.3	71.9	72.9	73.7	79.2	79.8	93.2	74.0
365.0	68.6	69.6	70.5	76.0	76.4	90.2	72.0
403.4	65.3	66.2	67.3	72.6	73.0	86.7	69.5
445.9	62.0	62.9	64.1	69.2	69.5	83.0	66.7
492.7	58.9	59.7	60.9	65.8	66.0	79.3	63.8
544.6	55.9	56.6	57.8	62.5	62.6	75.5	60.8
601.8	53.1	53.7	54.8	59.3	59.3	71.8	57.8

Table 1 (continued)

T (eV)	Nb	Mo	Ru	Rh	Pd	Ag	In
665.1	50.4	50.9	51.9	56.1	56.1	68.1	54.8
735.1	48.0	48.3	49.1	53.1	53.1	64.6	51.9
812.4	45.7	45.9	46.5	50.3	50.2	61.2	49.1
897.8	43.6	43.7	44.1	47.6	47.4	58.0	46.5
992.3	41.7	41.7	41.8	45.1	44.8	54.9	43.9
1096.6	39.9	39.8	39.6	42.7	42.4	52.1	41.5
1212.0	38.2	38.1	37.7	40.4	40.0	49.3	39.2
1339.4	36.6	36.5	36.0	38.4	37.9	46.8	37.0
1480.3	35.0	35.0	34.3	36.6	36.0	44.5	34.9
1636.0	33.5	33.5	32.8	34.8	34.2	42.5	33.0
1808.0	32.0	32.0	31.3	33.2	32.5	40.5	31.2
1998.2	30.5	30.6	29.9	31.7	31.0	38.5	29.6
2208.3	29.1	29.2	28.5	30.2	29.5	36.6	28.1
2440.6	27.7	27.8	27.2	28.8	28.1	34.8	26.6
2697.3	26.3	26.5	25.9	27.4	26.7	33.0	25.3
2981.0	24.9	25.1	24.6	26.0	25.4	31.2	24.0
3294.5	23.6	23.8	23.3	24.6	24.1	29.4	22.8
3640.9	22.3	22.5	22.1	23.3	22.8	27.7	21.6
4023.9	21.0	21.3	20.9	22.0	21.5	26.1	20.5
4447.1	19.8	20.0	19.7	20.8	20.3	24.6	19.4
4914.8	18.6	18.8	18.6	19.6	19.2	23.1	18.3
5431.7	17.5	17.7	17.5	18.4	18.1	21.6	17.2
6002.9	16.4	16.6	16.4	17.3	17.0	20.3	16.2
6634.2	15.4	15.6	15.4	16.2	15.9	19.0	15.3
7332.0	14.4	14.6	14.4	15.2	14.9	17.7	14.3
8103.1	13.4	13.6	13.5	14.3	14.0	16.5	13.4
8955.3	12.6	12.8	12.6	13.3	13.1	15.4	12.6
9897.1	11.7	11.9	11.8	12.5	12.2	14.4	11.8
10938.0	11.0	11.1	11.0	11.6	11.4	13.4	11.0
12088.4	10.2	10.4	10.3	10.9	10.7	12.5	10.3
13359.7	9.56	9.70	9.62	10.1	9.96	11.6	9.62
14764.8	8.93	9.05	8.97	9.45	9.30	10.8	8.98
16317.6	8.35	8.45	8.37	8.82	8.67	10.1	8.38
18033.7	7.80	7.90	7.81	8.23	8.09	9.38	7.81
19930.4	7.29	7.38	7.30	7.68	7.55	8.73	7.29
22026.5	6.81	6.89	6.81	7.17	7.04	8.13	6.80
24343.0	6.36	6.43	6.36	6.69	6.57	7.57	6.34
26903.2	5.94	6.01	5.94	6.25	6.14	7.06	5.92
29732.6	5.54	5.61	5.55	5.83	5.73	6.58	5.53
T (eV)	Sn	Cs	Gd	Tb	Dy	Hf	Ta
54.6	48.8	86.0	57.6	60.9	53.5	29.8	31.1
60.3	50.2	86.8	60.9	64.8	57.3	31.7	33.5
66.7	51.6	86.7	64.3	68.8	61.0	33.7	35.6
73.7	52.8	85.9	67.3	72.8	64.7	35.8	37.6
81.5	53.9	84.5	69.6	77.1	68.4	37.9	39.5
90.0	55.0	82.7	71.2	81.4	72.1	40.2	41.3
99.5	56.2	80.4	72.1	85.2	75.5	42.5	43.2
109.9	57.6	77.9	72.4	88.1	78.3	44.7	45.1
121.5	59.1	75.3	72.3	89.8	80.2	46.8	46.9
134.3	60.9	72.6	71.8	90.3	81.3	48.6	48.7
148.4	62.8	70.1	70.8	89.9	81.4	49.9	50.2
164.0	64.8	67.7	69.6	88.8	80.9	50.8	51.2
181.3	66.8	65.4	68.2	87.2	79.8	51.1	51.8
200.3	68.5	63.4	66.5	85.2	78.4	51.0	51.8
221.4	70.0	61.8	64.7	83.0	76.6	50.6	51.5
244.7	71.1	60.5	62.9	80.5	74.7	49.9	50.9
270.4	71.6	59.8	61.0	77.9	72.6	49.1	50.0
298.9	71.5	59.6	59.0	75.2	70.4	48.1	49.0
330.3	70.7	59.4	57.1	72.4	68.1	47.0	47.9
365.0	69.3	59.0	55.1	69.7	65.8	45.8	46.6
403.4	67.4	58.0	53.2	66.9	63.4	44.6	45.3
445.9	65.0	56.4	51.4	64.2	61.1	43.3	44.0
492.7	62.4	54.4	50.0	61.8	58.9	42.1	42.7
544.6	59.6	52.1	48.5	59.5	56.9	40.8	41.3
601.8	56.8	49.7	46.9	57.2	54.8	39.5	40.0
665.1	54.0	47.2	45.2	54.8	52.7	38.2	38.6
735.1	51.3	44.8	43.4	52.4	50.5	36.9	37.3
812.4	48.6	42.4	41.6	50.0	48.3	35.6	35.9
897.8	46.0	40.1	39.8	47.6	46.2	34.3	34.6
992.3	43.5	37.9	37.9	45.3	44.0	33.0	33.3
1096.6	41.1	35.8	36.1	43.0	41.9	31.7	32.0
1212.0	38.8	33.8	34.3	40.8	39.8	30.4	30.7
1339.4	36.7	31.9	32.6	38.6	37.8	29.1	29.4
1480.3	34.6	30.0	30.9	36.6	35.8	27.8	28.1

(continued on next page)

Table 1 (continued)

T (eV)	Sn	Cs	Gd	Tb	Dy	Hf	Ta
1636.0	32.7	28.3	29.3	34.5	33.9	26.6	26.9
1808.0	30.9	26.7	27.7	32.6	32.1	25.4	25.7
1998.2	29.2	25.1	26.2	30.8	30.3	24.2	24.5
2208.3	27.6	23.7	24.7	29.0	28.6	23.0	23.3
2440.6	26.2	22.3	23.3	27.3	26.9	21.8	22.1
2697.3	24.9	21.0	22.0	25.6	25.4	20.7	21.0
2981.0	23.6	19.8	20.7	24.1	23.9	19.6	19.9
3294.5	22.4	18.8	19.5	22.6	22.4	18.6	18.8
3640.9	21.3	17.8	18.3	21.2	21.1	17.5	17.8
4023.9	20.1	16.8	17.2	19.9	19.8	16.5	16.8
4447.1	19.1	15.9	16.2	18.7	18.5	15.6	15.8
4914.8	18.0	15.1	15.2	17.5	17.4	14.7	14.9
5431.7	17.0	14.3	14.3	16.4	16.3	13.8	14.0
6002.9	16.0	13.5	13.5	15.4	15.3	13.0	13.2
6634.2	15.1	12.7	12.7	14.5	14.3	12.2	12.4
7332.0	14.1	12.0	12.0	13.6	13.5	11.5	11.6
8103.1	13.3	11.3	11.3	12.8	12.7	10.8	10.9
8955.3	12.4	10.6	10.6	12.0	11.9	10.1	10.3
9897.1	11.6	10.0	10.0	11.3	11.2	9.54	9.64
10938.0	10.9	9.35	9.43	10.55	10.46	8.98	9.06
12088.4	10.2	8.77	8.87	9.89	9.81	8.45	8.52
13359.7	9.52	8.21	8.33	9.27	9.19	7.94	8.01
14764.8	8.88	7.68	7.82	8.67	8.61	7.46	7.52
16317.6	8.29	7.18	7.33	8.11	8.05	7.01	7.06
18033.7	7.74	6.71	6.87	7.58	7.53	6.58	6.62
19930.4	7.22	6.27	6.43	7.09	7.04	6.17	6.21
22026.5	6.73	5.86	6.02	6.62	6.57	5.78	5.82
24343.0	6.28	5.47	5.63	6.18	6.14	5.41	5.45
26903.2	5.86	5.11	5.27	5.77	5.73	5.07	5.11
29732.6	5.47	4.77	4.92	5.39	5.35	4.75	4.78
T (eV)	W	Re	Os	Ir	Pt	Au	Bi
54.6	27.0	24.4	21.1	21.7	23.7	24.3	40.8
60.3	30.3	28.1	23.9	24.6	26.4	26.9	42.1
66.7	33.3	32.0	26.9	27.8	29.4	29.7	43.2
73.7	36.0	35.3	29.9	31.0	32.5	32.6	44.3
81.5	38.2	38.0	32.6	34.2	35.5	35.7	45.4
90.0	39.9	40.3	35.0	36.9	38.1	38.7	46.5
99.5	41.4	42.3	36.8	39.1	40.3	41.6	47.8
109.9	42.9	44.0	38.3	41.0	42.1	44.2	49.0
121.5	44.3	45.6	39.7	42.6	43.6	46.5	50.4
134.3	45.8	47.1	41.1	44.0	44.9	48.4	51.9
148.4	47.3	48.4	42.5	45.1	45.9	49.9	53.4
164.0	48.4	49.5	43.7	46.0	46.6	51.0	54.8
181.3	49.1	50.2	44.5	46.6	47.0	51.8	56.0
200.3	49.3	50.5	44.9	47.0	47.1	52.1	56.9
221.4	49.1	50.3	44.9	46.9	46.9	52.0	57.3
244.7	48.5	49.7	44.6	46.4	46.5	51.5	57.4
270.4	47.7	48.8	43.9	45.7	45.8	50.7	57.0
298.9	46.8	47.7	43.0	44.7	44.8	49.7	56.1
330.3	45.6	46.5	42.0	43.5	43.7	48.4	54.9
365.0	44.5	45.2	40.9	42.2	42.5	47.0	53.5
403.4	43.2	43.8	39.7	40.8	41.1	45.4	51.7
445.9	41.9	42.5	38.5	39.4	39.7	43.7	49.8
492.7	40.7	41.1	37.3	38.1	38.3	42.0	47.8
544.6	39.4	39.7	36.1	36.7	37.0	40.3	45.8
601.8	38.1	38.3	34.9	35.4	35.6	38.6	43.7
665.1	36.8	37.0	33.7	34.1	34.3	37.0	41.7
735.1	35.6	35.6	32.5	32.9	33.1	35.4	39.7
812.4	34.3	34.3	31.4	31.7	31.9	34.0	37.8
897.8	33.1	33.0	30.3	30.6	30.7	32.5	36.0
992.3	31.9	31.8	29.2	29.4	29.5	31.2	34.3
1096.6	30.7	30.5	28.1	28.4	28.4	29.8	32.7
1212.0	29.5	29.3	27.0	27.3	27.3	28.6	31.2
1339.4	28.3	28.1	26.0	26.2	26.2	27.3	29.8
1480.3	27.1	26.9	24.9	25.2	25.2	26.1	28.4
1636.0	25.9	25.8	23.9	24.2	24.2	25.0	27.1
1808.0	24.8	24.6	22.9	23.2	23.1	23.9	25.9
1998.2	23.7	23.5	21.9	22.2	22.2	22.8	24.7
2208.3	22.6	22.4	20.9	21.2	21.2	21.8	23.5
2440.6	21.5	21.4	19.9	20.3	20.2	20.7	22.4
2697.3	20.4	20.3	19.0	19.3	19.3	19.7	21.3
2981.0	19.4	19.3	18.0	18.4	18.3	18.7	20.3
3294.5	18.4	18.3	17.1	17.4	17.4	17.8	19.2
3640.9	17.4	17.3	16.2	16.6	16.5	16.8	18.3

Table 1 (continued)

T (eV)	W	Re	Os	Ir	Pt	Au	Bi
4023.9	16.4	16.3	15.4	15.7	15.7	16.0	17.3
4447.1	15.5	15.4	14.5	14.8	14.8	15.1	16.4
4914.8	14.6	14.5	13.7	14.0	14.0	14.2	15.5
5431.7	13.7	13.7	12.9	13.2	13.2	13.4	14.6
6002.9	12.9	12.9	12.2	12.4	12.5	12.7	13.8
6634.2	12.2	12.1	11.5	11.7	11.7	11.9	12.9
7332.0	11.4	11.4	10.8	11.0	11.0	11.2	12.2
8103.1	10.7	10.7	10.1	10.3	10.4	10.5	11.4
8955.3	10.1	10.0	9.52	9.72	9.74	9.88	10.7
9897.1	9.46	9.42	8.94	9.13	9.14	9.27	10.0
10938.0	8.89	8.84	8.40	8.57	8.58	8.69	9.41
12088.4	8.36	8.31	7.89	8.05	8.06	8.16	8.82
13359.7	7.86	7.81	7.42	7.57	7.57	7.65	8.25
14764.8	7.38	7.34	6.98	7.11	7.11	7.19	7.73
16317.6	6.93	6.89	6.56	6.68	6.68	6.75	7.24
18033.7	6.51	6.47	6.16	6.27	6.28	6.34	6.78
19930.4	6.10	6.07	5.78	5.89	5.89	5.95	6.36
22026.5	5.72	5.69	5.42	5.53	5.53	5.58	5.95
24343.0	5.36	5.33	5.09	5.18	5.19	5.24	5.58
26903.2	5.02	5.00	4.77	4.86	4.86	4.91	5.23
29732.6	4.70	4.68	4.47	4.55	4.56	4.61	4.90

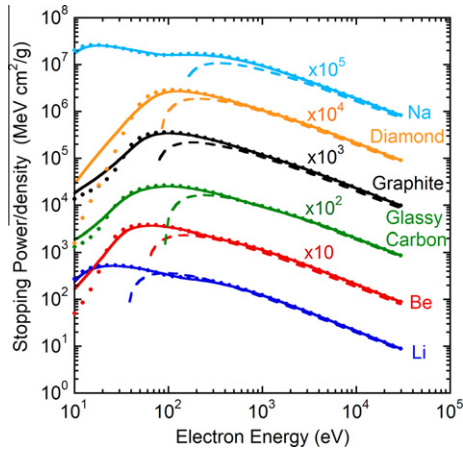


Fig. 1. Energy dependence of calculated mass collision stopping powers (or collision stopping powers divided by density) for Li, Be, glassy carbon, graphite, diamond, and Na. The solid lines show stopping powers calculated for each solid with the full Penn algorithm as a function of electron relativistic kinetic energy T . The solid circles show stopping powers calculated with the single-pole approximation. The dashed lines show stopping powers calculated from the relativistic Bethe equation (Eq. (12)).

with the reflection EELS data of Robins and Swan [31]. We also chose a set of optical data from Palik [32] for Au since this data set gave an ELF in better agreement with reflection EELS experiments [31] than the data set from Hagemann et al. [33] we used previously [1,2,5,29].

For two solids, Pd and V, there were large gaps in our previous ELFs (between 20 and 100 eV for Pd and between 40 and 100 eV for V). Our new ELF for Pd between 18 and 120 eV was obtained from an interpolation of ELF data [34] with a cubic spline function. The error in the f-sum for the new ELF was -2.3% , a value much smaller than that for the previous ELF (-12%) [29]. For V, we used a new set of optical data from Palik [32] for photon energies between 42.5 and 120 eV. We also chose new ELF data for photon energies less than 24 eV from Ref. [34] because this ELF better resembled transmission and reflection EELS data [30,35] than the previous ELF data. The resulting error in the f-sum was -0.8% (compared to the previous value of -20%) [29]. As a result, the new SPs for Pd and V are much larger than our previous SPs.

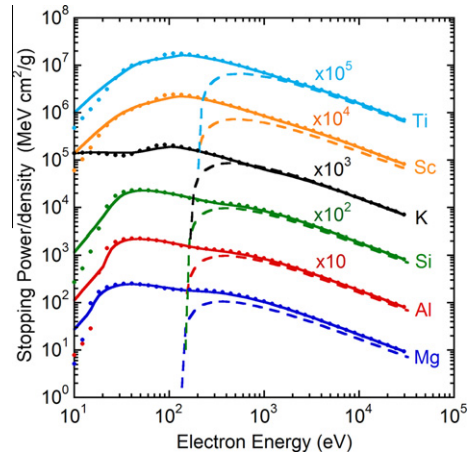


Fig. 2. Energy dependence of calculated mass collision stopping powers for Mg, Al, Si, K, Sc, and Ti. See caption to Fig. 1.

We checked the internal consistency of the new ELF data through use of the oscillator-strength sum rule (or f-sum rule) and a limiting form of the Kramers–Kronig integral (or KK-sum rule) [36–39]. The average root-mean-square (RMS) errors for the ELF data sets of our 41 solids were 4.2% and 7.7% based on the f-sum and KK-sum rules, respectively. These values are superior to the corresponding results for our previous ELFs (about 10% RMS error in both sum rules for 27 elemental solids) [5,29]. For over 80% of our 41 elemental solids, the ELFs satisfied the f-sum and KK-sum rules to better than 10%.

Fig. 8 shows ratios of SPs determined from the non-relativistic FPA, which are obtained from Eq. (3) when $T/c^2 \rightarrow 0$, and the new ELF data sets, S_{new} , to those calculated previously from the SPA and the old ELF data sets, S_{old} , [1,2] as a function of non-relativistic kinetic energy for the 19 elemental solids for which we adopted new ELF data. The largest changes occurred for Pd (where the SP increased by up to 58%), V (where the SP increased by up to 52%), and Re (where the SP decreased by up to 26%). These changes are the opposite of those found in similar comparisons of IMFPs (as shown in Fig. 11 of Ref. [6]). The changes in Fig. 8 result from two causes: use here of improved sets of ELF data [6] and differences in the SP calculation algorithms (the FPA and the SPA). The inset in

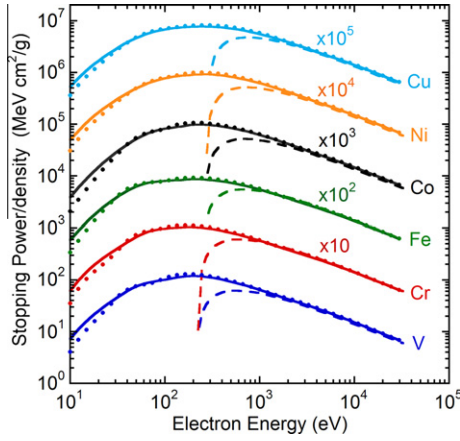


Fig. 3. Energy dependence of calculated mass collision stopping powers for V, Cr, Fe, Co, Ni, and Cu. See caption to Fig. 1.

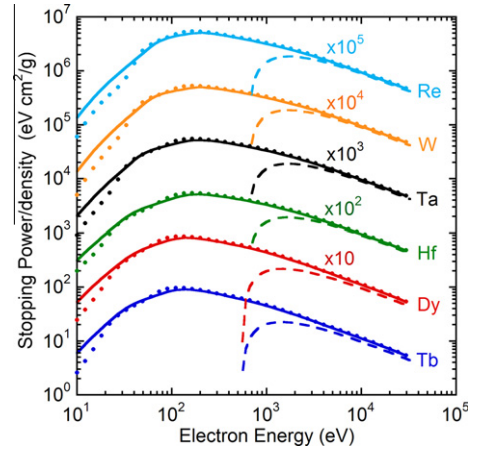


Fig. 6. Energy dependence of calculated mass collision stopping powers for Tb, Dy, Hf, Ta, W, and Re. See caption to Fig. 1.

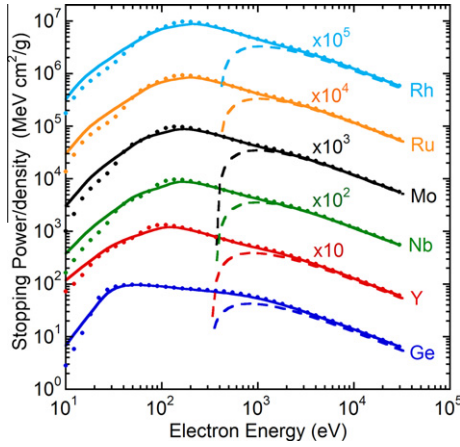


Fig. 4. Energy dependence of calculated mass collision stopping powers for Ge, Y, Nb, Mo, Ru, and Rh. See caption to Fig. 1.

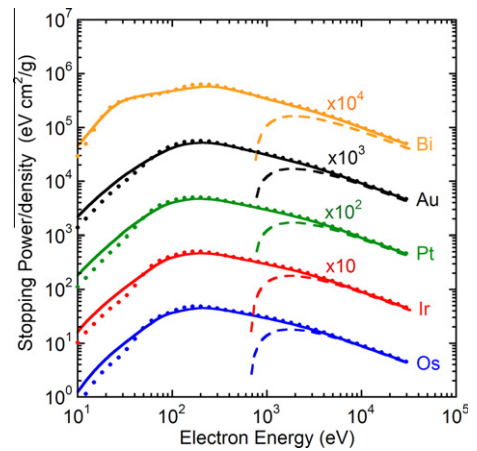


Fig. 7. Energy dependence of calculated mass collision stopping powers for Os, Ir, Pt, Au, and Bi. See caption to Fig. 1.

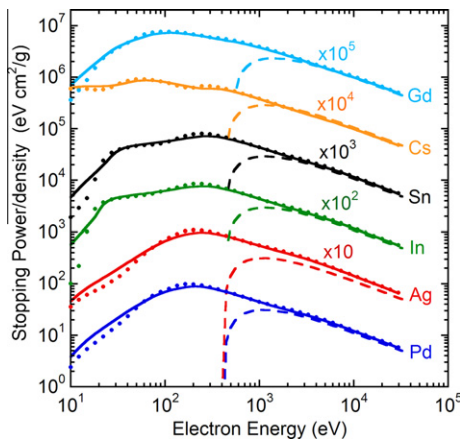


Fig. 5. Energy dependence of calculated mass collision stopping powers for Pd, Ag, In, Sn, Cs, and Gd. See caption to Fig. 1.

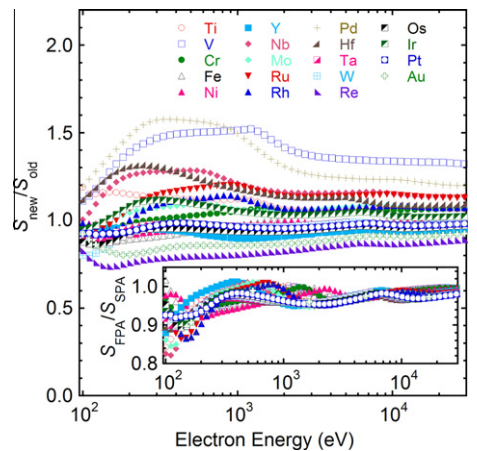


Fig. 8. Plots of ratios of SPs determined from the FPA and the new ELF data sets, S_{new} , to those calculated previously from the SPA and the old ELF data sets, S_{old} , [1,2] as a function of electron non-relativistic kinetic energy (i.e., $E = v^2/2$) for the 19 elemental solids for which we adopted new ELF data. The inset shows ratios of SPs from the FPA, S_{FPA} , to those from the SPA, S_{SPA} , for the same 19 solids. These SPs were calculated with the new ELFs for each algorithm.

Fig. 8 shows ratios of SPs from the FPA, S_{FPA} , to those from the SPA, S_{SPA} , for the 19 solids that were calculated using the new ELFs for each algorithm. We see that the SP changes are smaller than 10% and that most of the differences occur for energies less than 200 eV (as will be discussed further in the following section). Since the ratios $S_{\text{new}}/S_{\text{old}}$ in Fig. 8 generally deviate from unity by more

than the values of $S_{\text{FPA}}/S_{\text{SPA}}$ in the inset, it is clear that most of the changes in $S_{\text{new}}/S_{\text{old}}$ are due to the differences in the ELFs.

3.3. Comparison of stopping powers from the FPA and SPA

We now make comparisons of RMS differences between SPs calculated from the FPA and SPA using the same (new) ELF data set for each of our 41 solids. Relative percentage RMS differences, *RMS*, were calculated from

$$RMS = 100 \times \left[\sum_{i=1}^{41} \left(\frac{S_{SPA}(T)_i - S_{FPA}(T)_i}{S_{FPA}(T)_i} \right)^2 / 41 \right]^{0.5} (\%) \quad (13)$$

as a function of electron energy from 10 eV to 30 keV.

Fig. 9 shows plots of *RMS* as a function of relativistic kinetic energy. We see a steep decrease from about 50% for $E = 10$ eV to less than 10% for energies above 40 eV. The steep decrease must be due to the contributions of single-electron excitations to the SP that are neglected in the SPA (which only considers excitations at the plasmon pole). For energies above 40 eV, *RMS* generally decreases with increasing energy, reaching 1% at 30 keV. Three maxima are observed in Fig. 9 at energies of about 100 eV, 2 keV, and 10 keV. These maxima are due to the different energy positions of maxima and structure found in the SP plots from the FPA and SPA as a function of energy in Figs. 1–7.

3.4. Influence of electron exchange on stopping powers

It is important to know the effect of exchange between projectile and target electrons on SP calculations with the FPA. There is no consensus, however, on how to incorporate exchange effects within the dielectric formalism [6]. Nevertheless, we can estimate the influence of exchange on calculated SPs using the Born–Ochkur exchange correction [40,41].

The non-relativistic DCS with the Born–Ochkur correction can be written as [41]:

$$\frac{d^2\sigma}{dq d\omega} = \frac{C_{ex}}{\pi N E} \text{Im} \left[\frac{-1}{\varepsilon(q, \omega)} \right] \frac{1}{q}, \quad (14)$$

where E is the non-relativistic kinetic energy (i.e., $E = v^2/2$) and C_{ex} is the exchange correction factor given by

$$C_{ex} = 1 - \frac{q^2}{2E} + \left(\frac{q^2}{2E} \right)^2. \quad (15)$$

We calculated SPs of Al, Cu, Ag, and Au with the exchange correction from Eqs. (14) and (15) and compared the results with

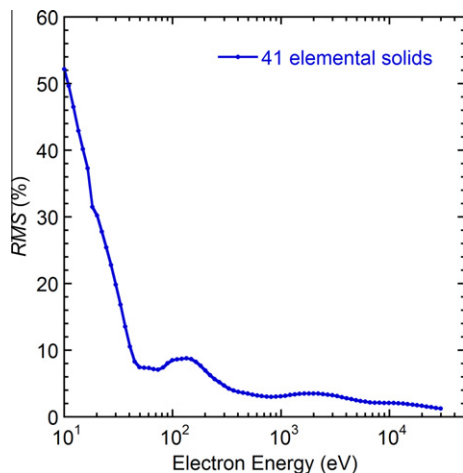


Fig. 9. The root-mean-square relative differences, *RMS*, of stopping powers calculated with the full Penn algorithm from stopping powers calculated with the single-pole approximation for the 41 elemental solids as a function of electron relativistic kinetic energy. The *RMS* differences were calculated from Eq. (13).

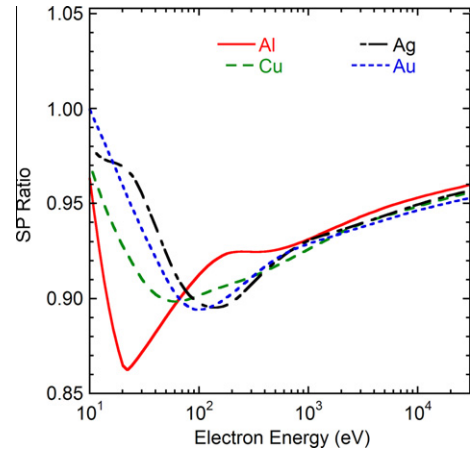


Fig. 10. Ratios of SPs calculated from the FPA with and without an exchange correction for Al, Cu, Ag, and Au as a function of non-relativistic electron kinetic energy (i.e., $E = v^2/2$).

corresponding SPs calculated without the exchange correction. Fig. 10 shows plots of ratios of SPs with the exchange correction to those without this correction as a function of non-relativistic kinetic energy. We see that SPs with the exchange correction are smaller than those without the exchange correction for these four solids and energies between 10 eV and 30 keV. Above 100 eV, SPs with the exchange correction are smaller than those without the exchange correction by less than about 10%. The SP ratios generally increase with increasing electron energy, reaching about 0.95 at 30 keV.

The influence of electron exchange on calculated SPs is almost the same as that found in IMFP calculations for electron energies between 50 and 100 eV [26]. We note, however, that the Born–Ochkur approximation is essentially a high-energy approximation. It is then not clear whether this approximation is useful for evaluating the exchange correction for energies less than 100 eV.

4. Discussion

We will compare our calculated mass collision SPs with SPs from other calculations and from experiments. Although the FPA is expected only to provide a qualitative guide to SPs for energies less than about 50 eV, we show SPs calculated from the FPA for energies as low as 3 eV in Figs. 11–15 in order to make comparisons with available SP data.

4.1. Comparisons with calculated stopping powers

Mao et al. [21] calculated SPs of Al and Cu from 1 eV to 10 keV with the FPA. Fig. 11 shows comparisons of our SPs for Al and Cu calculated from the FPA and SPA with those of Mao et al. There is excellent agreement between our FPA SPs for Al and those of Mao et al. and satisfactory agreement for Cu. The slight differences for Cu at energies less than about 30 eV are probably due to the selection of different sets of optical ELF data in each calculation. As discussed in Section 3.3, the substantial differences in the SPs for Al from the FPA and SPA at energies less than 20 eV are associated with the contributions of single-electron excitations to the SP that occur at much lower energy losses than the relatively sharp plasmon peak at about 15 eV in the ELF. In contrast, there is broad structure in the Cu ELF, and a wide range of excitation energies contribute to the SP in the SPA calculation for Cu.

Fernandez-Varea et al. [42] calculated SPs for Al, Si, Cu, and Au for electron energies between 10 eV and 100 MeV. Their

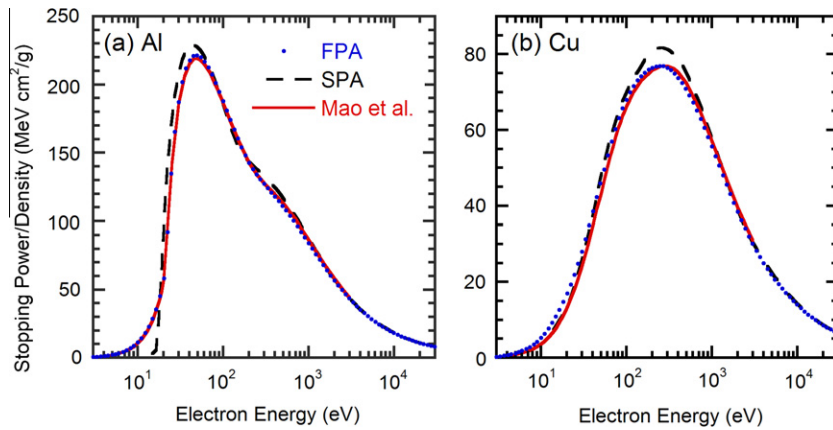


Fig. 11. Comparison of mass collision SPs calculated from optical data for (a) Al and (b) Cu with the full Penn algorithm by Mao et al. [21] (solid lines) as a function of electron relativistic kinetic energy with our SPs calculated with the single-pole approximation (long-dashed line) and the full Penn algorithm (solid circles).

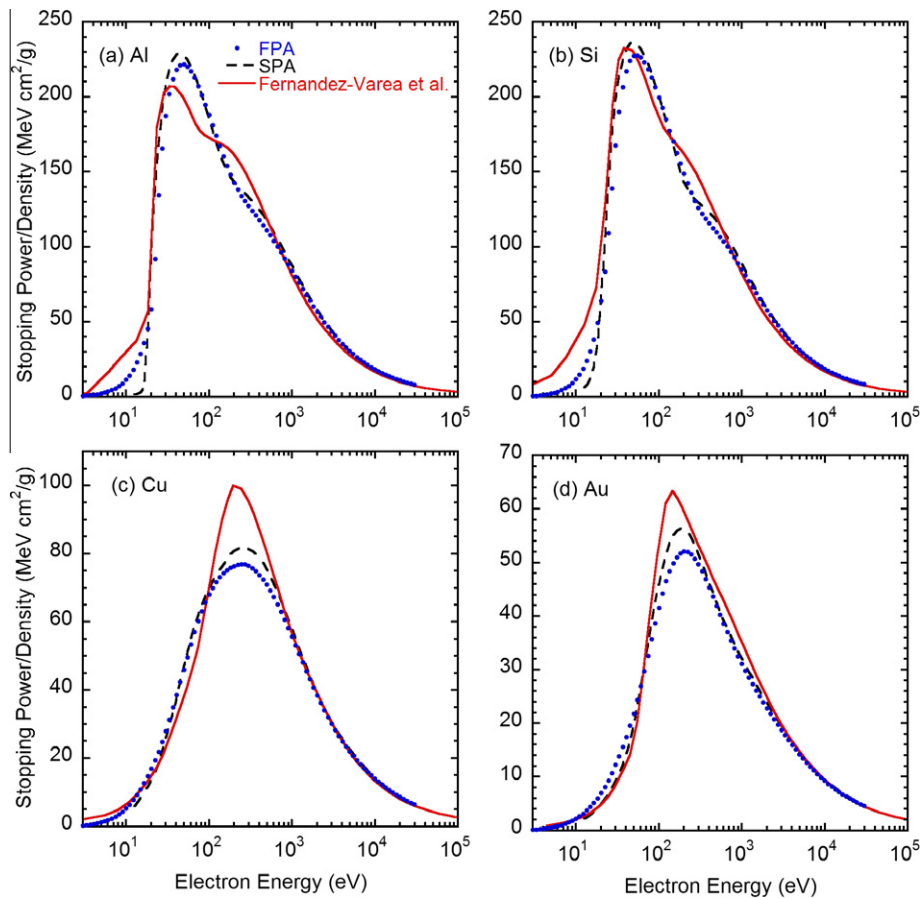


Fig. 12. Comparison of mass collision SPs calculated from optical data for (a) Al, (b) Si, (c) Cu, and (d) Au by Fernandez-Varea et al. [42] (solid lines) as a function of electron relativistic kinetic energy with our SPs that were obtained with the single-pole approximation (long-dashed line) and the full Penn algorithm (solid circles).

calculations were based on a so-called “N-oscillator” model in which different dispersion relations were applied for valence-electron excitations and inner-shell excitations. They also included a correction for electron exchange to cross sections for inner-shell excitations. We compare their SPs with our SPs from the FPA and SPA in Fig. 12. For energies over 1 keV, the SPs of Fernandez-Varea et al. for Al, Si, Cu, and Au are in excellent agreement with our values from the FPA. For Al and Si, there are differences in the shapes of the SP versus energy curves in the vicinity of 100 eV.

These differences might be associated with the “switch energies” of 73 eV and 99 eV for Al and Si, respectively, used by Fernandez-Varea et al. to represent the demarcation between their models for valence-electron and inner-shell excitations. For Cu and Au, the switch energies are 74 eV and 54 eV, respectively, but there are no obvious changes of slope in the SP-versus-energy curves for these solids in Fig. 12. This difference from the Al and Si behavior occurs because the switch energies for Cu and Au occur in a structureless region of their ELF. Tan et al. [43] reported SP

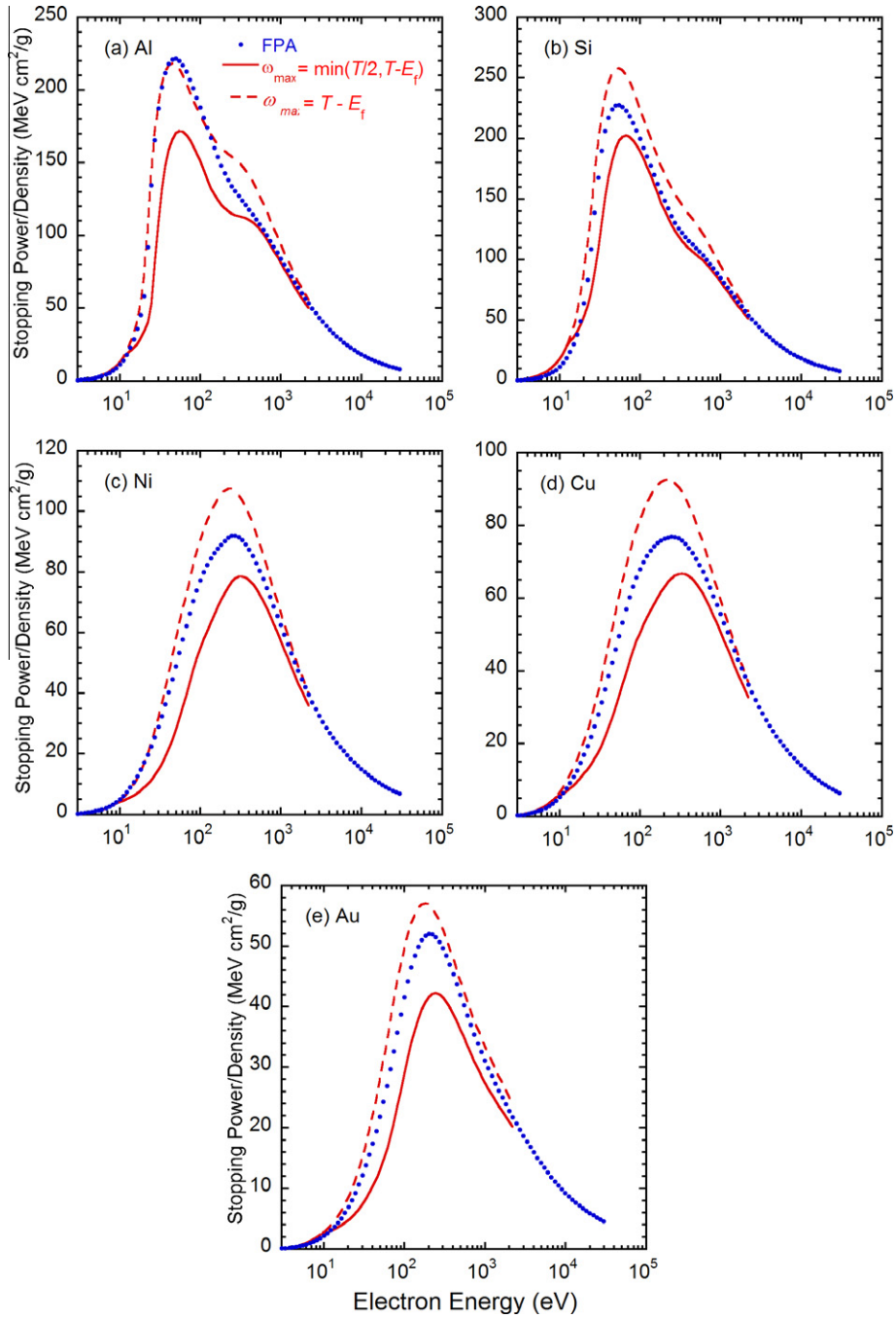


Fig. 13. Comparison of our mass collision SPs obtained from the full Penn algorithm (solid circles) as a function of relativistic kinetic energy for (a) Al, (b) Si, (c) Ni, (d) Cu, and (e) Au with SPs calculated from the Mermin-ELF model with $\omega_{\max} = \min(T/2, T - E_f)$ (solid lines) and with $\omega_{\max} = T - E_f$ (long-dashed lines).

calculations for a group of organic compounds with the SPA and two types of exchange correction [41,44]. They found that inclusion of exchange reduced their computed SPs by an average of 28%, 9%, and 9% for energies of 100 eV, 1 keV, and 10 keV, respectively. These differences are in good agreement with our estimates of electron exchange effect for SPs as shown in Fig. 10 except at 100 eV. Their calculations were done up to $\omega_{\max} = T/2$ in Eq. (4). They must then obtain larger SPs at a low energy such as 100 eV if they used the same $\omega_{\max} (= T - E_f)$ value as we used. We will refer later to this issue in comparison of SPs from the FPA and from the Mermin model. Nevertheless, the maximum SPs of Fernandez-Varea et al. (with an exchange correction) for Cu and Au are larger than our corresponding maximum SPs from the FPA (without an

exchange correction). For Si and Al, however, there is close agreement in the maximum SPs from Fernandez-Varea et al. and our calculations with the FPA. It therefore appears that the exchange correction must be smaller than differences due to other factors (e.g., differences in optical ELF and differences in the models). We have also pointed out in our related IMFP calculations that correlation and exchange should be treated in an integrated manner together with information on the band structure of the solid [6].

Abril et al. [45] proposed an algorithm for IMFP and SP calculations based largely on a Mermin-model dielectric function for the ELF (Mermin-ELF model) [46]. The Mermin function is an improvement over the Lindhard dielectric function used here in that it accounts for the finite lifetimes of the various excitations. Their

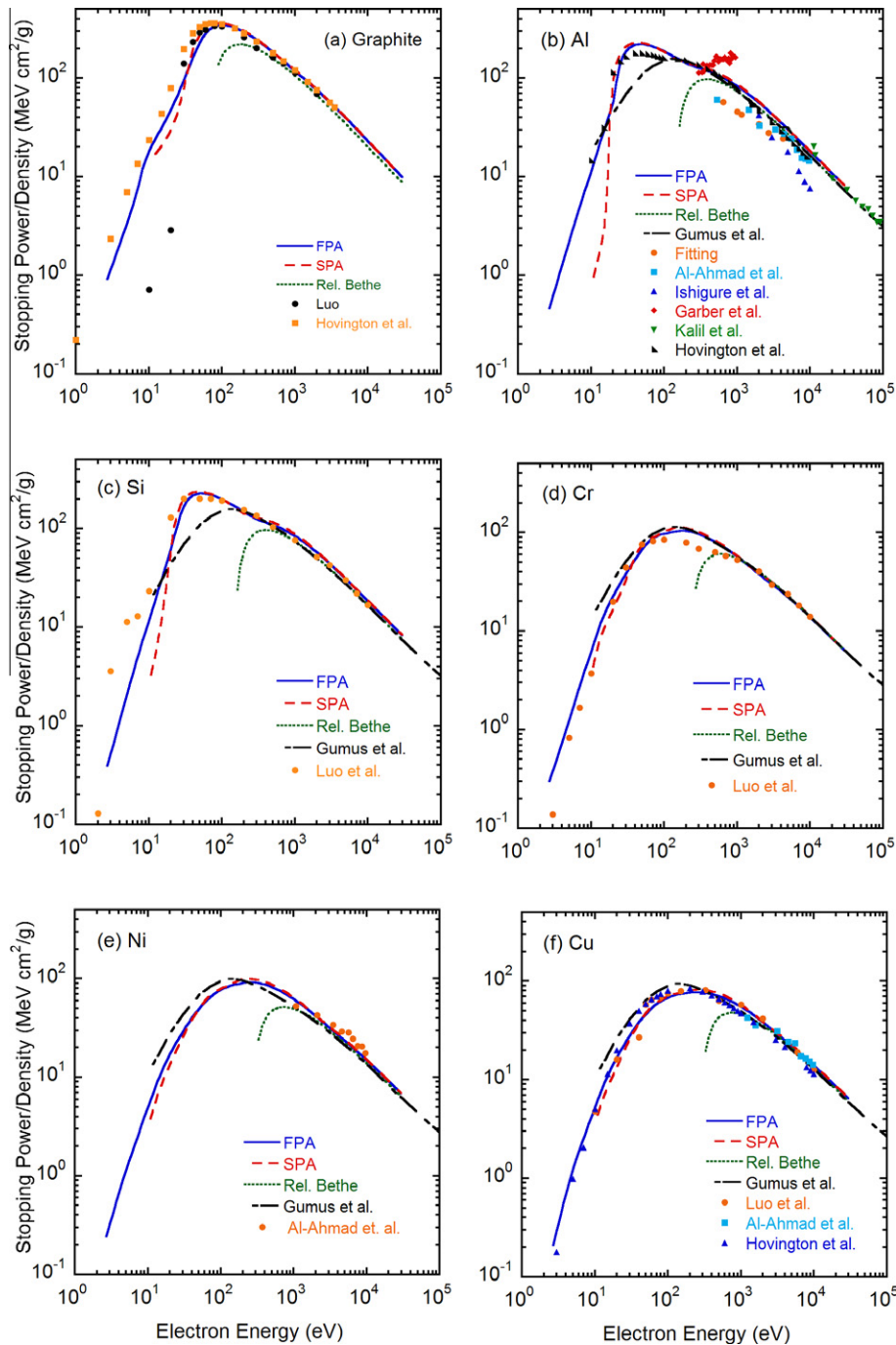


Fig. 14. Mass collision stopping powers for (a) graphite, (b) Al, (c) Si, (d) Cr, (e) Ni, and (f) Cu as a function of electron relativistic kinetic energy. The dotted lines show SPs from the relativistic Bethe formula (Eq. (12)), and the solid and long-dashed lines show our SPs from the full Penn algorithm and the single-pole approximation, respectively. The long-short dashed lines show SPs calculated from a modified Bethe-Bloch SP equation and expressions for the effective atomic electron number and the effective mean excitation energies [51]. The symbols indicated SPs derived from the experiments of Luo et al. [53] (for Si, Cr, and Cu), Luo [54] (for graphite), Hovington et al. [55] (for graphite, Al, and Cu), Kalil et al. [57] (for Al), Al-Ahmad and Watt [58] (for Al, Ni, and Cu), Garber et al. [59] (for Al), Fitting [60] (for Al), and Ishigure et al. [61] (for Al).

model of Mermin energy-loss functions was combined with generalized oscillator strengths (MELF-GOSs) to fit experimental ELF data. The part of an experimental ELF ascribed to excitations of outer-shell electrons was fitted with a linear combination of Mermin-type ELFs, and the part associated with excitations of inner-shell electrons was fitted with hydrogenic generalized oscillator strengths [47].

We have calculated SPs for Al, Si, Ni, Cu, and Au with the Mermin-ELF model using the parameters for outer-electron excitations given in Table 1 of Ref. [48] for Al, Si, Ni, and Cu and Table 1 of Ref.

[49] for Au; these parameters were determined from fits to optical ELFs for excitation energies up to about 1 keV and thus include the contributions of several inner shells. In our calculations we ignored the contributions of GOSs for inner-shell ionization of the K shells of Al and Si, of the K and L shells of Ni and Cu, and of the K, L, and M shells of Au. We estimated their contribution to be less than a few percent for electron energies less than 2 keV for Ni and Cu and less than 10 keV for Al, Si, and Au.

Fig. 13 shows comparisons between SPs from the FPA (solid symbols) and from the Mermin-ELF model (solid and dashed lines)

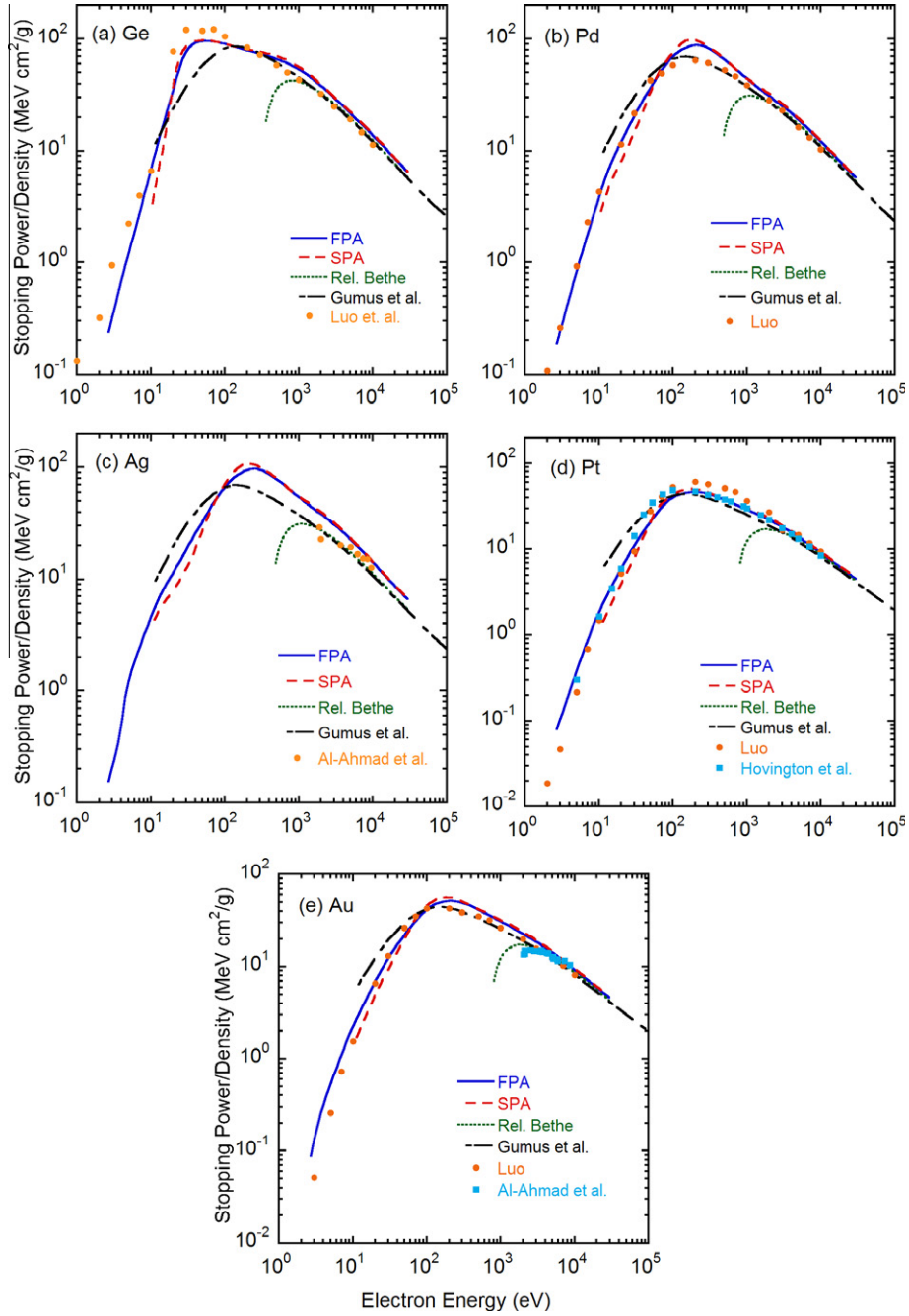


Fig. 15. Mass collision stopping powers for (a) Ge, (b) Pd, (c) Ag, (d) Pt, and (e) Au as a function of electron relativistic kinetic energy. The dotted lines show SPs from the relativistic Bethe formula (Eq. (12)), and the solid and long-dashed lines show our SPs from the full Penn algorithm and the single-pole approximation, respectively. The long-short dashed lines show SPs calculated from a modified Bethe–Bloch SP equation and expressions for the effective atomic electron number and the effective mean excitation energies [51]. The symbols indicated SPs derived from the experiments of Luo et al. [53] (for Ge), Luo [54] (for Pd, Pt, and Au), Hovington et al. [55] (for Pt), and Al-Ahmad and Watt [58] for Ag and Au).

for Al, Si, Ni, Cu, and Au using two choices for the upper limit $\omega_{\max}$ for excitation energy in the latter SP calculations. One value of $\omega_{\max}$ was $\omega_{\max} = \min(T/2, T - E_f)$, as recommended by Denton et al. [50], while the other was $\omega_{\max} = T - E_f$ as chosen here for evaluation of Eq. (2). Denton et al. chose the former limit to avoid consideration of secondary electrons having energies larger than inelastically-scattered primary electrons. Larger energy transfers are possible, however, although secondary electrons may then be indistinguishable from scattered primary electrons, either in experiments or in model calculations of electron energy spectra. We believe that the upper limit $\omega_{\max} = T - E_f$ is more appropriate

for the SP calculation, as we have chosen here for the results in Figs. 1–7.

Fig. 13 indicates that SPs from the FPA are smaller than those from the Mermin-ELF model with the same upper limit $\omega_{\max} = T - E_f$ for Si, Ni, Cu, and Au. These differences are generally less than 20% for energies over 50 eV and may be associated with different q -dependences of the ELF in the two models. For Al, SPs from the Mermin-ELF model (with $\omega_{\max} = T - E_f$) and from the FPA are in good agreement for energies less than 200 eV. This agreement, in contrast to the results for Si, Ni, Cu, and Au, may be fortuitous because of the relatively poorer fit for Al of the

Mermin-ELF to the optical ELF (particularly around the volume-plasmon energy-loss peak between 12 and 17 eV) than for the other four solids.

Fig. 13 also shows that calculated SPs from the FPA are larger by up to about 250% than those from the Mermin-ELF model with $\omega_{\max} = \min(T/2, T - E_f)$ for energies between 20 eV and 1 keV. For energies above 1 keV, there is generally good agreement between SPs from the two approaches. In contrast, we previously found much better agreement between IMFPs calculated with the FPA for Al and Au [6] and those reported by Denton et al. [50] who used the Mermin-ELF model with the same parameters as those given in Refs. [48] and [49]. This observation suggests that choice of the upper limit $\omega_{\max}$ is more significant in SP calculations than in IMFP calculations because of the greater relative contributions of possible large-energy-loss excitations to the SP [the factor ω in Eq. (4)] than to the IMFP. Fig. 13 indicates that the choice of the upper limit is most important for electron energies between about 20 eV and about 1 keV.

Gumus et al. [51] calculated SPs for Al, Si, Cr, Ni, Cu, Ge, Pd, Ag, Pt, and Au for electron energies between 10 eV and 100 MeV. Their calculations were performed using a modified Bethe–Bloch SP expression and analytical expressions for the effective atomic number and the effective mean excitation energy of each material. We show their SPs in Figs. 14 and 15 together with our SPs (and the experimental SPs discussed in the next section). For energies over 200 eV, the SPs of Gumus et al. for Al, Si, Cr, Cu, Ge, and Pt, are in good agreement with our values from the FPA and SPA. We also see that the SPs of Gumus et al. for the other solids are smaller than our SPs from the FPA for energies above 200 eV. For energies between 10 and 200 eV, the SPs of Gumus et al. for Al, Si, and Ge are smaller than our SPs while they obtain larger SPs for the other solids. Except for Ag, there is generally good agreement between the Gumus et al. SPs and our SPs from the FPA and SPA for energies over 200 eV. Since their calculations were made with a modified Bethe formula that is usually applied for much higher energies, 200 eV must be an effective low-energy limit for their approach.

For Ag, there are larger differences between our SPs from the FPA and those of Gumus et al. at energies above 200 eV, as stated above. Since the SPs of Ag for Gumus et al. are in good agreement with SPs from the relativistic Bethe formula (Eq. (12)) for energies above 5 keV, the differences with our SPs might be due to uncertainties in the experimental ELF data for Ag (despite relatively small errors in the f-sum and KK-sum rules [6]).

4.2. Comparison with experimental stopping powers

We compare calculated SPs from the FPA for graphite, Al, Si, Cr, Ni, Cu, Ge, Pd, Ag, Pt, and Au in Figs. 14 and 15 with experimental SP data that were mostly obtained from Joy's database [52]. We previously made similar comparisons of SPs from the SPA for Al, Si, Cr, Ni, Cu, Ge, Pd, Ag, Pt, and Au and for electron energies between 100 eV and 30 keV [1]. We will therefore emphasize comparisons here between SPs from the FPA and measured SPs as well as comparisons for energies less than 100 eV. Comparisons will also be made with SPs from the relativistic Bethe equation (Eq. (12)) using MEEs listed in Table 4.3 of Ref. [7] except graphite. The MEE value for graphite was obtained from our previous work [2]. We also show SPs from the SPA in Figs. 14 and 15 so that similarities and differences with the FPA results are visible.

The experimental SPs in Figs. 14 and 15 can be classified into two groups. Almost all of the experimental SPs for energies less than 1 keV were reported by Luo et al. [53], Luo [54], and Hovington et al. [55,56]. These SPs are based on measurements of transmission electron energy-loss spectra of 100 or 200 keV electrons transmitted through thin specimen films. The energy-loss spectra for energy losses up to 1 keV and for the angular acceptance of

their spectrometer were analyzed to obtain the single-scattering ELF. These ELFs were extended to larger energy losses using atomic X-ray absorption data. Checks were made to ensure that the ELFs satisfied the expected sum rules. SPs were then calculated from the experimental ELFs without consideration of any q -dependence other than that expected from the scattering kinematics. Their SP calculation, although derived from experimental ELFs, is very similar in principle to our SP calculation from optical ELFs. The other group of measured SPs in Figs. 14 and 15 was obtained from calorimetric methods for Al, Ni, Cu, Ag, and Au [57,58], from a novel thin-film method in which currents to electrodes at the top and bottom surfaces of a film were measured (and with the electrodes separated from the film by thin insulating layers) for Al [59], from analyses of energy distributions of electrons transmitted through a thin Al film with a retarding-field analyzer [60], and from analyses of energy distributions of electrons transmitted through a thin Al film at various scattering angles [61].

We see generally excellent agreement between our SPs from the FPA and the Joy SPs for energies larger than about 10 eV and in some cases (Cu, Pd, and Pt) for lower energies. However, the excellent agreement at energies near 10 eV is very likely fortuitous because our SP calculations with the FPA ignored the effects of electron exchange and correlation that must be important at such low energies. Some small but systematic differences are found at energies between 10 and 100 eV for graphite, Al, Si, and Ge, and similar differences can be seen for larger energies for Cr, Pd, Pt, and Au. Generally good agreement is found between our SPs from the FPA and the SPs measured by calorimetry for Al, Ni, Cu, Ag, and Au for energies between about 4 and 30 keV, but there are disagreements between our SPs and those of Al-Ahmad and Watt [58] for Al, Ag, and Au at lower energies.

The comparisons in Fig. 14 for Al show a wide spread in measured SPs for the same material as measured by different methods. The measured SPs at a given energy can differ by a factor of more than two, and the energy dependence of the SPs reported by Garber et al. [59] differs from those obtained by other methods (including the FPA). Given this disparity in SP results for a single material, we believe that there is satisfactory agreement between SPs from the FPA and the measured SPs. Definitive experimental tests are still required, however, to determine whether and how any exchange correction should be included in the SP calculation, as discussed in Sections 3.4 and 4.1. Further experimental tests are also required to distinguish differences in SPs corresponding to different choices of the upper limit $\omega_{\max}$ in Eq. (4), as discussed in Section 4.1.

We note that there is good agreement between SPs from the FPA and SPA in Figs. 14 and 15 for all solids except Al, Si, and Ge at energies less than 20 eV. These solids have strong and narrow plasmon peaks in their energy-loss spectra. As discussed in Sections 3.3 and 4.1, substantial differences can occur for such solids between SPs determined from the FPA and SPA because SPs from the SPA do not have contributions from single-electron excitations.

Finally, we see satisfactory agreement in Figs. 14 and 15 between SPs from the relativistic Bethe equation (Eq. (12)), SPs calculated from the FPA, and most measured SPs for energies larger than about 5 keV for low- Z and most medium- Z elements (graphite, Al, Si, Cr, Ni, Cu, Ge, and Pd) and for energies larger than about 10 keV for high- Z elements (Pt and Au). For Ag, however, there are larger differences between SPs from the FPA and those from Eq. (12) at energies above 5 keV than for the other solids. These differences might be due to uncertainties in the experimental ELF data for Ag [6].

5. Summary

We have reported mass collision electron SPs for Li, Be, graphite, diamond, glassy C, Na, Mg, Al, Si, K, Sc, Ti, V, Cr, Fe, Co, Ni, Cu, Ge, Y,

Nb, Mo, Ru, Rh, Pd, Ag, In, Sn, Cs, Gd, Tb, Dy, Hf, Ta, W, Re, Os, Ir, Pt, Au, and Bi over the 50 eV to 30 keV energy range. These SPs were calculated from ELF data determined from experimental optical data or ELF measurements [6] with the full Penn algorithm [4]. For 19 of our 41 solids, we adopted improved sets of optical ELF data [6] over those used for our previous SP calculations with the single-pole approximation or simple Penn algorithm [1,2]. The largest changes occurred for Pd (where the SP increased by up to 58%), V (where the SP increased by up to 52%), and Re (where the SP decreased by up to 26%).

We made comparisons of RMS differences between SPs calculated from the FPA and SPA using the same ELF data sets for the calculations with each algorithm. For energies above 50 eV, the RMS relative differences were less than 10% and generally decreased with increasing energy, reaching 1% at 30 keV.

We compared our calculated SPs with results from other calculations. Mao et al. [21] calculated SPs of Al and Cu from 1 eV to 10 keV with the FPA. There was excellent agreement between our SPs from the FPA for Al and those of Mao et al. and satisfactory agreement for Cu. Fernandez-Varea et al. [42] calculated SPs of Al, Si, Cu, and Au for electron energies between 10 eV and 100 MeV with their N-oscillator model. For energies over 1 keV, the SPs of Fernandez-Varea et al. for Al, Si, Cu, and Au were in excellent agreement with our values from the FPA. For Al and Si, there were differences in the shapes of the SP-versus-energy curves in the vicinity of 100 eV. These differences might be associated with the “switch energies” of 73 and 99 eV for Al and Si, respectively, used by Fernandez-Varea et al. to represent the demarcation between their models for valence-electron and inner-shell excitations. Their maximum values of the SPs for Al and Si were in reasonable agreement with our maximum values, but their maximum values for Cu and Au were larger than our maximum values.

Abril et al. [45] developed an algorithm for IMFP and SP calculations based on a Mermin-model dielectric function (Mermin-ELF model) [46]. We calculated SPs for Al, Si, Ni, Cu, and Au with the Mermin-ELF model using the parameters for outer-electron excitations adopted by the Abril group [48,49]. These calculations were performed with two choices for the upper limit $\omega_{\max}$ for excitation energy, one being $\omega_{\max} = \min(T/2, T - E_f)$, as recommended by Denton et al. [50], while the other was $\omega_{\max} = T - E_f$ as chosen here for our SP calculations. The differences between SPs with the latter choice of $\omega_{\max}$ and our SPs from the FPA were generally less than 20% for Si, Ni, Cu, and Au at energies above 50 eV and for Al at energies above 200 eV. In similar comparisons of SPs with the former choice of $\omega_{\max}$, our SPs were larger than those from the Mermin-ELF model by up to 250% for energies between 20 eV and 1 keV. There was, however, good agreement for energies less than 20 eV for Al, Si, Cu, and Au or less than 10 eV for Ni and greater than 1 keV for the five solids. We believe that $\omega_{\max} = T - E_f$ is the more appropriate choice for the upper limit.

We compared our SPs calculated from the FPA for graphite, Al, Si, Cr, Ni, Cu, Ge, Pd, Ag, Pt, and Au with values derived from available experimental data. Most of these comparisons were made with SPs derived by Joy et al. [52–56] from ELF data obtained from analyses of energy-loss spectra measured by transmission of 100 or 200 keV electrons through thin specimen films. We found generally excellent agreement between SPs derived in this way and our SPs from the FPA for energies larger than about 10 eV, although there were small but systematic differences for some solids between 10 and 100 eV. There was satisfactory agreement between our calculated SPs and values determined from calorimetry experiments for Al, Ni, Cu, Ag, and Au [57,58] at energies between 4 keV and 30 keV, but there were disagreements with the experimental SPs for Al, Ag, and Au [58] at lower energies. SPs have been measured by other methods for only one material (Al) [59–61], but

the reported SPs at a particular energy can differ by a factor of more than two.

Finally, we compared our calculated SPs with values from the relativistic Bethe equation with recommended mean excitation energies derived from a wide variety of experimental data [7] and from our previous analysis for the three carbon allotropes [2]. The RMS relative deviations between our calculated SPs and values from the Bethe equation were 9.1% and 8.7% for energies of 9.897 and 29.733 keV, respectively. Satisfactory agreement was found between SPs from the Bethe equation, our SPs, and most measured SPs for energies larger than about 5 keV for low-Z and most medium-Z elements (graphite, Al, Si, Cr, Ni, Cu, Ge, and Pd) and for energies larger than about 10 keV for high-Z elements (Pt and Au). Larger differences between SPs from the Bethe equation and from the FPA were found for Ag at energies above 5 keV, presumably due to uncertainties in the experimental ELF data set for Ag.

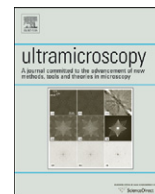
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Imaging properties of bright-field and annular-dark-field scanning confocal electron microscopy: II. Point spread function analysis

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ABSTRACT

The imaging properties of bright field and annular dark field scanning confocal electron microscopy (BF-SCEM and ADF-SCEM) are discussed based on their point spread functions (PSFs) in comparison with multislice simulations. Although the PSFs of BF-SCEM and ADF-SCEM show similar hourglass shapes, their numerical distributions are quite different: BF-SCEM PSF is always positive and shows a center of symmetry whereas the ADF-SCEM PSF is complex and has Hermitian symmetry. These PSF properties explain the large elongation effect in BF-SCEM for laterally extended object and almost no-elongation in ADF-SCEM, illustrating the importance of the numerical analysis of PSFs. The Hermitian symmetry of the ADF-SCEM PSF results in an interesting “edge enhancement effect” at the interface. Simulation using the PSF and the multislice method verified this effect at GaAs surfaces and InAs interfaces embedded in GaAs. This unique feature of ADF-SCEM can potentially be useful for depth sectioning. It is also pointed out that a PSF imaging model cannot be applicable for BF-SCEM of a phase object, when the system is symmetric and aberration free.

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1. Introduction

Scanning confocal electron microscopy (SCEM) [1,2] is the electron version of the well-established optical three-dimensional imaging technique, scanning confocal optical microscopy (SCOM) [3,4] and was firstly reported in 2002 [2]. The schematic of the system is depicted in Fig. 1. The SCEM ray path consists of scanning transmission electron microscopy (STEM) illumination system and transmission electron microscopy (TEM) imaging system with pin-hole aperture placed at the detector. The image is formed by scanning the focused probe over the sample and by de-scanning the probe image on the detector. The original report [2] demonstrated an image improvement for thick semiconductor device by SCEM in comparison with STEM or TEM.

Takeguchi et al. [5] and Hashimoto et al. [6] developed a stage-scan system that forms an image by scanning the sample instead of scanning the beam. This system avoids the difficulties associated with beam scanning such as adjusting the probe positions on the sample and detector during the scan and de-scan, which is a technically very demanding task especially at high-resolution.

Using the stage scan system they obtained for the first time high-resolution SCEM images from Au particles.

However, the depth sectioning (z-sectioning) capability of a sample has not been achieved by usual SCEM yet partially because in comparison with optical lenses used in SCOM, the available numerical aperture size is much smaller for the SCEM due to the aberrations in magnetic lenses used in SCEM. Recent progress of aberration correctors is now improving this situation rapidly [7–9].

A more intrinsic problem of z-sectioning with SCEM is, as pointed by many authors [10–12] including our previous paper [13], conventional bright field (BF)-SCEM is susceptible to the elongation effect depending on the sample lateral size, and thus the depth resolution is much worse than the vertical spread of the focused probe. Note that the point spread functions of aberration free BF-SCEM is the same with for high-angle annular dark field (HAADF)-STEM [10–12,14,15]. Therefore, the depth resolution achievable with BF-SCEM is the same with HAADF-STEM.

On the other hand, annular dark field (ADF)-SCEM, recently suggested SCEM technique that uses annular dark field aperture placed at the collector lens (Fig. 1(b)), shows more readily interpretable image contrast. Depth sectioned images of a nano-coil were obtained with ADF-SCEM using an uncorrected microscope, and no obvious elongation effects due to the sample lateral

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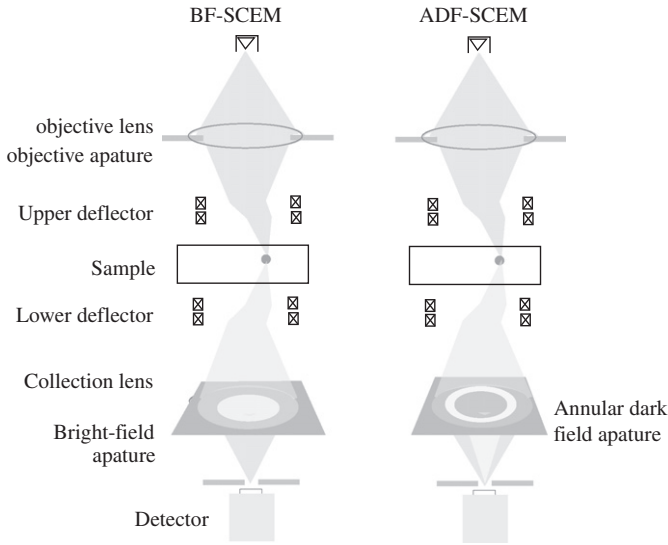


Fig. 1. Schematic drawing of BF and ADF-STEM setups. The sample is illuminated by an electron beam, which is focused by the objective lens and scanned by the upper deflection coils. The transmitted beam is de-scanned by the lower deflection coils and re-focused by the collector lens on the detector. The detector pin-hole aperture selects only electrons passing through the focus. A BF or ADF aperture is applied at the collector lens.

size were observed [16,17]. In the previous paper, we provided a geometrical explanation of the image features observed in BF and ADF-STEM [13]. It was demonstrated that the BF-STEM intensity profile consists of broad and sharp features originated from the sample shape and atomic structure close to the probe focus, respectively.

The properties of imaging system are often discussed using the contrast transfer function (CTF), and the boundaries of the CTF are used to discuss the limit of frequency transfer. However, since CTF is an expression in reciprocal space, it is difficult to predict the imaging property in real space. Furthermore, the values of the CTF within the boundary are equally important for imaging.

In this paper, we illustrate the properties of BF and ADF-STEM using the point spread function (PSF) and CTF and explain some unusual features of ADF-STEM that can be explained by taking account of the value of PSF. To accentuate the imaging nature of BF and ADF-STEM, we only discuss the case of aberration free system. This assumption may be justified by the recent developments of electron microscopes equipped with double (objective and collector lenses) aberration correctors that is capable of opening up the aperture as wide as a few 10 mrad and potentially realize a nanometer resolution in the z -direction.

2. Calculations conditions

Throughout this paper, the objective semiangles are 30 mrad for both BF and ADF-STEM and the collection semiangles are 30 mrad for BF and 30 to 40 mrad for ADF-STEM. The PSFs are calculated for the region of $2.56 \text{ nm} \times 2.56 \text{ nm} \times 25.6 \text{ nm}$ divided by $512 \times 512 \times 512$. Dynamical calculations were performed using the FFT-multislice method [18] implemented into the EMS software [19], which was modified for SCEM simulations. The incident energy was 200 keV. All aberrations but defocus were ignored and a point detector was assumed. Both BF and ADF intensities at the point detector were normalized at each illumination angle by the BF intensities without a sample. The supercell of $9.60 \times 9.58 \text{ nm}$ was used for GaAs/InAs where 1024×1024 waves were calculated; this corresponds to 17×24 unit cells of

GaAs(110). The lattice constant was assumed to be the same for InAs and GaAs. All absorption has been neglected, although the elastic potential was smeared through the use of the Debye–Waller factors.

3. Imaging model of HAADF-STEM and SCEM

In this and following sections we restrict the discussion of SCEM to coherent scattering by a weak scattering object and neglect multiple scattering. In this case the image amplitude may be expressed by a convolution between an object distribution and a PSF, which describes the system response to a point object. Then, a Fourier transform of the image amplitude is given by a simple product between a Fourier transform of the object distribution and a CTF, which is a Fourier transform of the PSF. Thus, either function expresses a system performance, if the system shows a linear response against the object distribution.

For ADF-STEM in which we can assume the incoherent imaging, the image can be described as [12,14]

$$I(x, y, z) = o_{\text{STEM}}(x, y, z) \otimes |h^{\text{obj}}(x, y, z)|^2 = o_{\text{STEM}}(\mathbf{r}) \otimes \text{PSF}(\mathbf{r}), \quad (1)$$

where o_{STEM} is the cross section of an incoherent signal (object function), such as electron intensity scattered to the ADF-detector, $h^{\text{obj}}(x, y, z)$ describes a wave field formed by the objective lens and $\otimes$ expresses the three-dimensional (3D) convolution operation. In this case, the PSF of the system is the probe intensity, namely $|h^{\text{obj}}(x, y, z)|^2$.

SCEM imaging is described as [12,14,20]

$$I(x, y, z) = |o(x, y, z) \otimes h^{\text{obj}}(-x, -y, -z) h^{\text{col}}(x, y, z)|^2 = |o(\mathbf{r}) \otimes \text{PSF}(\mathbf{r})|^2, \quad (2)$$

where $o(\mathbf{r})$ is the object function, $h^{\text{obj}}(x, y, z)$ and $h^{\text{col}}(x, y, z)$ are the PSFs of the objective and collector lenses, respectively. Thus, the PSF of the whole system is $h^{\text{obj}}(-x, -y, -z) h^{\text{col}}(x, y, z)$. As discussed below, for an aberration-free symmetric SCEM, $h^{\text{obj}}(-x, -y, -z) = h^{\text{obj}}(x, y, -z) = h^{\text{col}}(x, y, z)^*$ and the PSF becomes identical to that of ADF-STEM. However, we have to note that the object function is different between ADF-STEM and SCEM. The object function of HAADF-STEM is a HAADF absorption potential in the absorption model [18], whereas for the object function of SCEM, a transmission function derived within the phase object approximation is often used [12,14,15,21]. The phase object has the form of $t(\mathbf{r}) = \exp(i\sigma V(\mathbf{r}))$, where $V(\mathbf{r})$ is the crystal potential and σ is the interaction constant. For a single-layer or a single atom, a phase-grating works well as the transmission function and can predict an intensity distribution [12,14]. However, its value is unity in the area of zero potential, namely in vacuum, and the convolution between the phase grating and PSF depends on the volume of integration [12].

In optics, this unphysical situation is avoided by defining the object function as a sum of the unscattered and scattered components as $o(\mathbf{r}) = \delta(\mathbf{z}) + s(\mathbf{r})$ [20], where $\delta(\mathbf{z})$ is the Dirac delta function. The delta function gives the unscattered wave at the observation plane when it is convoluted with a 3D-PSF and $s(\mathbf{r})$ represents the scattering object, which is assumed to be weaker than the unscattered wave. In the case of phase object $s(\mathbf{r}) = t(\mathbf{r}) - 1 = \exp(i\sigma V(\mathbf{r})) - 1$, which becomes zero in vacuum. Using the scattering power $s(\mathbf{r})$ for an object function, as in optics, we can avoid the unphysical dependence on the integration volume.

Then, the BF-STEM intensity may be given using $s(\mathbf{r})$ as the object function:

$$I(x, y, z) = |(\delta(\mathbf{z}) + s(\mathbf{r})) \otimes \text{PSF}(\mathbf{r})|^2 = |1 + s(\mathbf{r}) \otimes \text{PSF}(\mathbf{r})|^2 \approx 1 + 2\text{Re}[s(\mathbf{r}) \otimes \text{PSF}(\mathbf{r})].$$

Therefore, the ADF-STEM and BF-SCEM intensities are both linear in the PSF, and thus the information transfer becomes identical between them as pointed out by several authors [12,14].

In the case of ADF-SCEM there is no unscattered wave at the observation plane, and thus the observed amplitude is simply given by a convolution between the scattering object and the PSF:

$$I(x,y,z) = |s(\mathbf{r}) \otimes \text{PSF}(\mathbf{r})|^2$$

The mathematical justification for this assumption is given below (Eq. (10)). Note that the intensity is not linear in PSF.

4. Shapes of PSF and CTF

A PSF of a single lens can be expressed as a stack of two-dimensional PSFs evaluated at each defocus z :

$$h(x,y,z) = \int A(\mathbf{K}_{\parallel}) \exp\{-i\chi(K_{\parallel},z)\} \exp(i\mathbf{K}_{\parallel} \cdot \mathbf{R}) d\mathbf{K}_{\parallel} \quad (3)$$

Here $\mathbf{K}_{\parallel}$ is a 2D reciprocal space vector, $\mathbf{R} = (x,y)$ is a 2D real-space vector and χ is the lens aberration function. When all aberrations are ignored as in the case of ideal (aberration free)

lens, it can be written as

$$\chi(\mathbf{K}_{\parallel},z) = \pi\lambda |\mathbf{K}_{\parallel}|^2 z$$

The corresponding CTF can be calculated numerically using Fourier transform. When all aberrations except defocusing are ignored a CTF for an individual lens becomes a section of the sphere (Ewald sphere) [20].

The PSF of SCEM is a product of the two PSFs for the objective and collector lenses:

$$\text{PSF}(x,y,z) = h^{obj}(-x,-y,-z)h^{col}(x,y,z) \quad (4)$$

Then, the CTF of SCEM is calculated by convoluting the CTFs for the objective and collector lenses:

$$\text{CTF}(\xi,\eta,\zeta) = H^{obj}(-\xi,-\eta,-\zeta) \otimes H^{col}(\xi,\eta,\zeta) \quad (5)$$

Then, the SCEM CTF is calculated by convoluting the two opposite spherical sections of the objective and collector lenses as illustrated in Fig. 2(a) for BF-SCEM case.

Fig. 3 shows the PSF and CTF of a rotationally symmetric aberration-free BF-SCEM calculated for the convergence and collection semiangles of 30 mrad. The FWHM of the PSF is 0.045 nm in the x - y plane, and 4.9 nm in the z direction. These values roughly correspond to the ranges of CTF of $4\alpha/\lambda$ for x - y and α^2/λ for the z direction [11,14,20] as shown in Fig. 2(b).

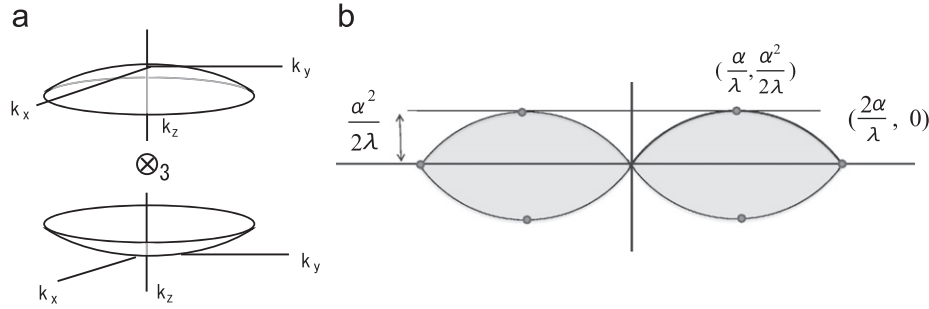


Fig. 2. (a) Schematics of the convolution of the CTFs for the objective and collector lenses that results in (a) BF-SCEM CTF. (b) The x - z section of BF-SCEM CTF. In the shaded area the CTF has non-zero value.

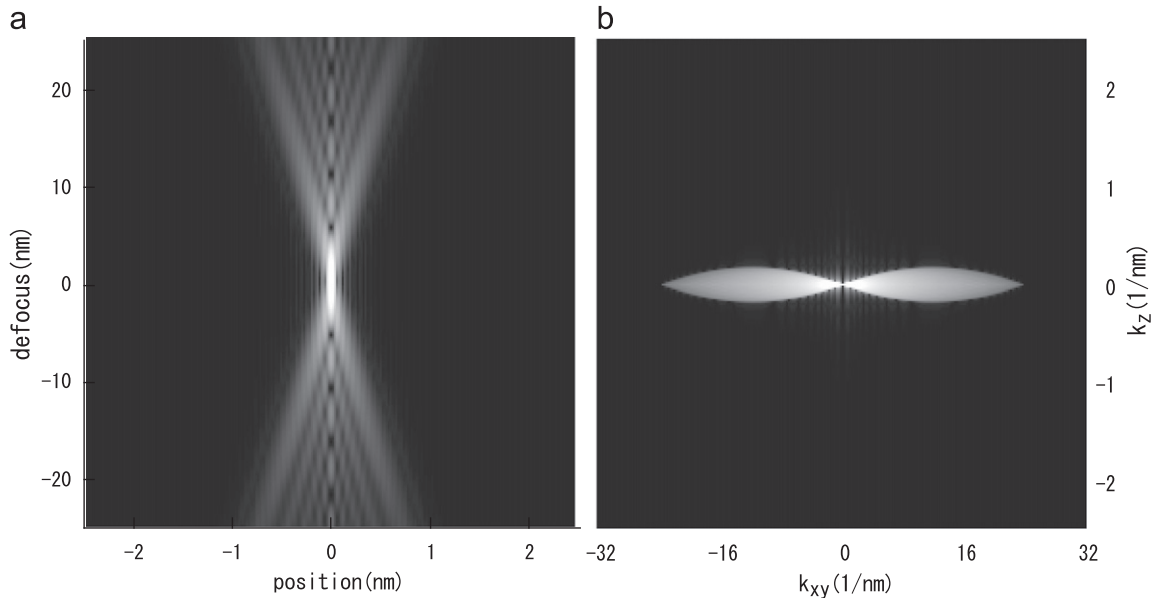


Fig. 3. (a) The x - z section of point-spread-function and (b) the k_x - k_z section of contrast transfer function of BF-SCEM calculated for 30 mrad convergence for both objective and collector apertures.

Here, the PSF shows an hourglass shape, or a head-to-head double cone, while the CTF resembles narrowed eyes' shape with so-called missing-cones near the origin [12,20]. A finite object perpendicular to the optical axis contributes to the BF-SCM intensity for a wide defocus range, and thus yields an elongated image. This elongation effect of BF-SCM has been explained by the missing cone of the CTF [10]. However, we note that contrary to SCOM the cone angle of BF-SCM is almost 90° , and that the terminology of missing cones may be confusing. On the other hand, we could explain the elongation effect of BF-SCM from simple geometrical considerations [13]. Below we will analyze the elongation of the BF-SCM imaging in terms of the PSF.

The PSF and CTF of ADF-SCM for a point detector can be calculated in the same way using different apertures for the objective and collector lenses. In this case the CTFs to be convoluted become a section and annular zone of Ewald spheres, respectively, defined by the objective and collector apertures, as shown in Fig. 4(a), where all aberrations except defocusing are ignored. Fig. 5 shows a PSF and CTF of aberration-free ADF-SCM calculated for the convergence semiangle of 30 mrad and the collection semiangles between 30 mrad and 40 mrad. An overall feature of the PSF of ADF-SCM is also characterized by a similar hourglass observed for the BF-SCM. On the other hand, the CTF

of ADF-SCM shows distinctively different shapes from that of BF-SCM [12,20]. The FWHM of the PSF is 0.035 nm in the x - y plane and 5.5 nm in the z direction. These values roughly correspond to the ranges of CTF of $2(\beta_{out} + \alpha)/\lambda$ for x - y and $\beta_{out}^2/2\lambda$ for the z direction as shown in Fig. 4(b), where $\alpha = \beta_{in}$. The limit of CTF in the x - y plane extends a little bit more due to the larger outer (40 mrad) collection semiangle, which results in the smaller FWHM of PSF in the x - y plane. On the other hand, the FWHM in the z direction becomes slightly worse since the range of the CTF along the z direction is narrower than that of BF-SCM. It may be noted that the CTF has a missing-region close to the origin as in the case of BF-SCM, although ADF-SCM does not show an elongation effect as observed in BF-SCM [13,16,17]. Thus, it is clear that a simple explanation of the elongation effect in BF-SCM using only the shape of the CTF shape, namely a missing-cone is insufficient.

When all aberrations except defocusing are ignored, $h(x, y, -z) = h(x, y, z)^*$. Thus, the PSF of each rotationally symmetric lens has Hermitian symmetry, $h(-x, -y, -z) = h(x, y, -z) = h(x, y, z)^*$, and the value of BF-SCM PSF becomes

$$PSF_{BF}(x, y, z) = h_{BF}^{obj}(-x, -y, -z)h_{BF}^{col}(x, y, z) = h_{BF}^{obj}(x, y, z)^*h_{BF}^{col}(x, y, z) = |h_{BF}^{obj}(x, y, z)|^2 \quad (6)$$

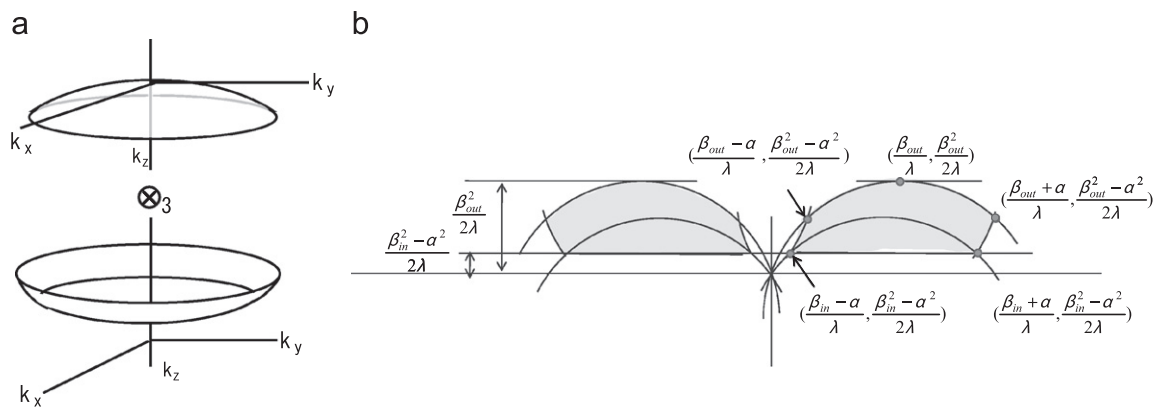


Fig. 4. (a) Schematics of the convolution of two CTFs of the objective and collector lenses that corresponds to ADF-SCM CTF. (b) The x - z section of ADF-SCM CTF. The shaded region is where the CTF has a non-zero value.

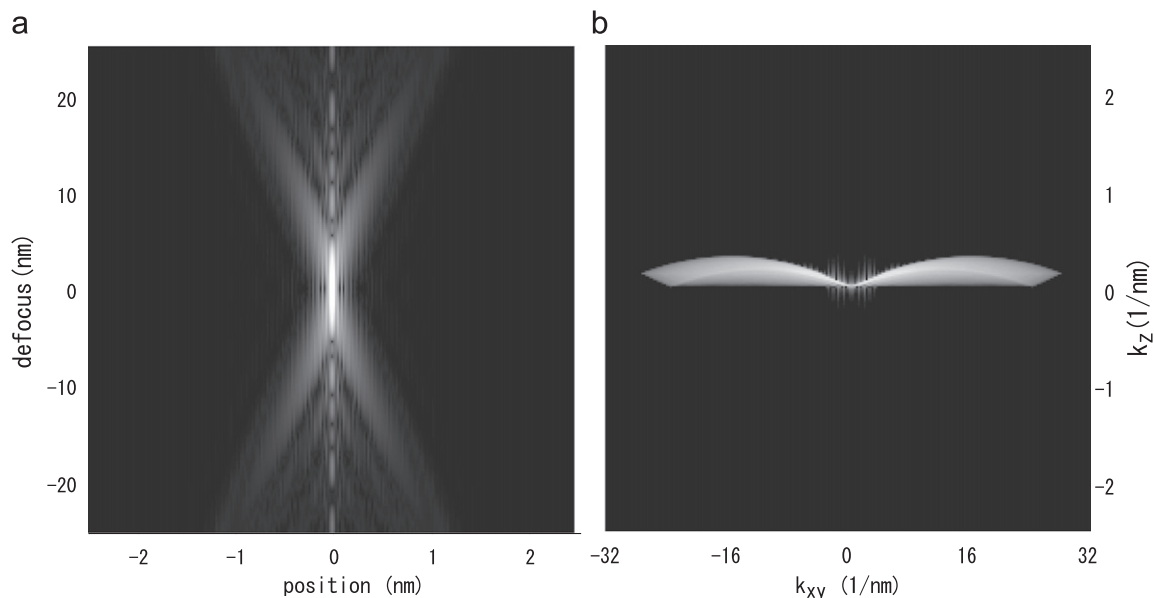


Fig. 5. (a) The x - z section of the PSF and (b) the k_x - k_z section of the CTF of ADF-SCM. The objective aperture is 30 mrad and the collector aperture is 30–40 mrad.

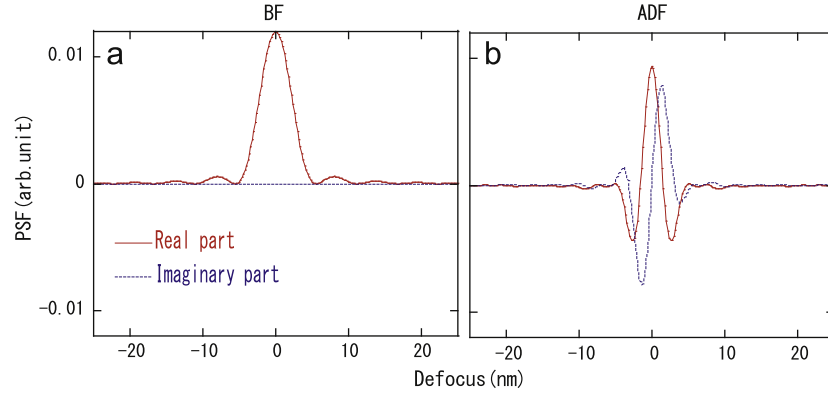


Fig. 6. The profiles of (a) BF-SCEM PSF and (b) ADF-SCEM PSF. The former is real while the latter has an imaginary part, which is asymmetric about $z=0$.

This function is real and positive, and its distribution is symmetric to reflection over the mid plane ($z=0$):

$$h_{BF}(x, y, z) = h_{BF}(x, y, -z)$$

It may be noted that the integral of the BF-SCEM PSF perpendicular to the optic axis (over the x - y plane) is unity for any defocus:

$$\int PSF_{BF}(x, y, z) dx dy = \int |h_{BF}^{obj}(x, y, z)|^2 dx dy = \frac{1}{A} \int |H_{BF}^{obj}(\xi, \eta, z)|^2 d\xi d\eta = 1, \quad (7)$$

since $|H(\xi, \eta, z)| = 1$.

On the other hand, in the case of ADF-SCEM even when all aberrations except defocusing are ignored, PSF is still complex:

$$PSF_{ADF}(x, y, z) = h_{ADF}^{obj}(-x, -y, -z) h_{ADF}^{col}(x, y, z) = h_{ADF}^{obj}(x, y, z) * h_{ADF}^{col}(x, y, z), \quad (8)$$

and its distribution becomes a complex-conjugate after reflection over the plane $z=0$:

$$h_{ADF}(x, y, z) = h_{ADF}(x, y, -z)^* \quad (9)$$

Especially for the case with no overlap between objective and annular apertures, the integral of the ADF-SCEM PSF perpendicular to the optic axis is always zero:

$$\begin{aligned} \int PSF_{ADF}(x, y, z) dx dy &= \int h_{ADF}^{obj}(x, y, z) * h_{ADF}^{col}(x, y, z) dx dy \\ &= \int H_{ADF}^{obj}(-\xi, -\eta, z) * H_{ADF}^{col}(\xi, \eta, z) d\xi d\eta = 0, \end{aligned} \quad (10)$$

Here, we have used the property of Fourier transform that an integral of the product of two functions in one space is equal to an integral of the product of two functions in other space.

Although the PSF shows the similar shape of an hourglass both for BF and ADF PSF, their profiles are quite different when the real and imaginary parts of the PSFs are plotted in the z direction as shown in Fig. 6. Note that the imaginary part of the ADF-SCEM PSF is antisymmetric along the z -direction. In the following section, we use the PSF profiles to explain some features of BF- and ADF-SCEM images.

5. Imaging properties of BF-SCEM derived from PSF

Consider the intensity of BF-SCEM signal when a thin finite object is gradually shifted along the z -axis of the probe. For the case where defocus value is large so that the probe is much larger than the size of atoms, the atomic structure of the object can be neglected and the object can be treated as a homogeneous medium. For such case, the thin object larger than the section of the PSF produces a constant signal from Eq. 7. When object

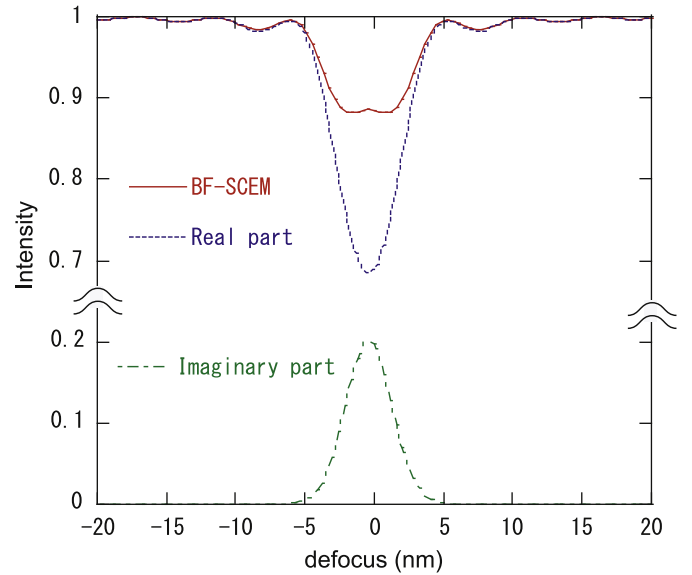


Fig. 7. BF-SCEM intensity profile of single atom calculated using the CTF, plotted together with the real and imaginary part contributions.

moves further from the confocal plane, the finite size of the object becomes smaller than the section of the PSF, and thus the signal starts to decrease from the constant value because the electron that passes through the sample reduces. This qualitatively describes the elongation effect in BF-SCEM image [13].

The fact that the BF-SCEM PSF is a real-valued function (Eq. 6) explains why a contrast of BF-SCEM for a phase object will be low. The first-order term of a phase grating expansion is imaginary, and thus its convolution with the real-valued PSF is also imaginary as discussed by Cosgriff et al. [21]. Thus, the lowest order in the BF image intensity of a phase object becomes a second order, and the contrast becomes low. This is contrary to the BF HRTEM image, where the first-order term of a phase grating expansion contributes to the image intensity at a Scherzer focus. At and near the confocal position, the probe is well focused to the atomic level. Then the atomic nature of the sample appears. Cosgriff et al. [21] showed that the contrast of a single atom as a function of z has a small rise at the position as shown in Fig. 7, and they explained this by atomic focusing.

This rise can be straightforwardly explained by a balance between contributions from the real and imaginary parts of the phase grating. Even if we include the higher-order terms in the exponential expansion of the phase object, the weak scattering approximation will hold provided all the terms included are small compared with the unscattered wave. When we include terms up

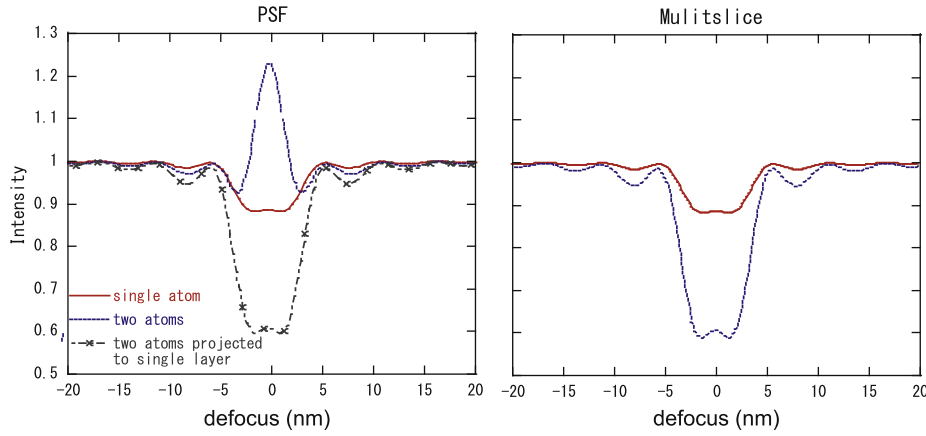


Fig. 8. BF-SCM intensity profiles of one gold atom and two atoms of gold (separated by 0.4 nm along the beam direction) calculated using the PSF (a) and multislice method (b).

to the second order, the image intensity can be given by

$$\begin{aligned}
 I(\mathbf{r}) &= |\exp\{i\sigma V(\mathbf{r})\} \otimes \text{PSF}(\mathbf{r})|^2 \\
 &\approx \left| \left(1 + i\sigma V(\mathbf{r}) - \frac{1}{2}(\sigma V(\mathbf{r}))^2 \right) \otimes \text{PSF}(\mathbf{r}) \right|^2 \\
 &\approx 1 - (\sigma V(\mathbf{r}))^2 \text{PSF}(\mathbf{r}) + (\sigma V(\mathbf{r}) \otimes \text{PSF}(\mathbf{r}))^2
 \end{aligned} \quad (11)$$

Note that the first-order term becomes imaginary after the convolution with the real-valued PSF. Thus, the lowest orders in the BF-SCM image intensity from both the real and imaginary parts of the phase object are of the second order, within the weak scattering approximation. In Fig. 7 the image intensity of a single Au atom located at $z=0$ is plotted together with the real and imaginary contributions. The contrast from the real part is negative around the atom position, whereas it is positive for the imaginary part. Thus, the resultant intensity profile shows a small rise at the focus position. When the resolution increases, both contributions become sharper and thus the width between two minima decreases as found by Cosgriff et al. [21]. However, the atomic focusing model may not be able to explain quantitatively the change of shape as a function of resolution.

Next, we consider a two-atom case of aberration-free symmetric BF-SCM imaging. In Fig. 8, we show the BF-SCM intensity profiles of one gold atom and two atoms of gold (separated by 0.4 nm along the beam direction) calculated by the multislice method and by the PSF (the convolution is actually calculated using the CTF in reciprocal space). The PSF intensity profile of two atoms is calculated by two different projected potentials; one is projecting two atoms into a single layer, and the other is projecting two atoms into two different layers. Here the PSF result for projecting two atoms into a single layer is close to the one obtained by the multislice method. On the contrary, placing two atoms in two different layers gives an unphysical intensity. This discrepancy is because multiple scattering contributions between layers are neglected in the PSF treatment.

The unphysical intensity distribution can be understood by considering the object function consisting of two layers characterized by the same potential V . The image intensity can be expressed for the real-valued PSF as

$$\begin{aligned}
 I(\mathbf{r}) &= |[\delta(\mathbf{r}) + s(\mathbf{r})] \otimes \text{PSF}(\mathbf{r})|^2 \\
 &= |1 + (\exp(i\sigma V(\mathbf{r})) - 1) + (\exp(i\sigma V(\mathbf{r})) - 1)| \otimes \text{PSF}(\mathbf{r})|^2 \\
 &\approx \left| \left[1 + \left(i\sigma V(\mathbf{r}) - \frac{1}{2}(\sigma V(\mathbf{r}))^2 \right) + \left(i\sigma V(\mathbf{r}) - \frac{1}{2}(\sigma V(\mathbf{r}))^2 \right) \right] \otimes \text{PSF}(\mathbf{r}) \right|^2 \\
 &\approx 1 - 2\sigma^2[V(\mathbf{r})^2 \otimes \text{PSF}(\mathbf{r})] + (2\sigma)^2[V(\mathbf{r}) \otimes \text{PSF}(\mathbf{r})]^2
 \end{aligned} \quad (12)$$

where we assume that those two layers are close with each other so that the difference in PSF can be ignored. The lowest terms are of second order by potentials, as in the case of a single atom.

The second term increases linearly with number of layers while the third term quadratically. Therefore, the total intensity will exceed 1 regardless of how weak is the scattering. This unphysical intensity distribution occurs because the contribution from double scattering by the second layers is ignored in the image calculated using PSF. Thus, it is evident that at least double scattering should be taken into account in the case of phase object imaging for aberration-corrected symmetrical BF-SCM. This will not happen if two layers can be projected into one as shown above. Nevertheless, a total (3D) sample may not be projected to a single layer, and the depth sectioning cannot be argued by the projected layers.

In the case where the system's aberration is not negligible, the PSF becomes complex. Then, it can be shown that the BF-SCM image intensity is of the first order in terms of the potential as in the case of BF-HRTEM by explicitly writing the real and imaginary parts of the PSF as

$$\begin{aligned}
 I(\mathbf{r}) &= |[1 + (\exp(i\sigma V_1(\mathbf{r})) - 1) + (\exp(i\sigma V_2(\mathbf{r})) - 1)] \\
 &\quad \otimes (\text{PSF}_r(\mathbf{r}) + i\text{PSF}_i(\mathbf{r}))|^2 \\
 &\approx 1 - 2\sigma(V_1(\mathbf{r}) + V_2(\mathbf{r})) \otimes \text{PSF}_i(\mathbf{r}) - \sigma^2(V_1(\mathbf{r})^2 + V_2(\mathbf{r})^2) \otimes \text{PSF}_r(\mathbf{r}) \\
 &\quad + \sigma^2[(V_1(\mathbf{r}) + V_2(\mathbf{r})) \otimes \text{PSF}_r(\mathbf{r})]^2 + \sigma^2[(V_1(\mathbf{r}) + V_2(\mathbf{r})) \otimes \text{PSF}_i(\mathbf{r})]^2
 \end{aligned} \quad (13)$$

Here, the first order term contributes mainly to the image intensity, and thus the PSF can be used to estimate the image intensity. This operation condition may be preferable in practice because the first-order term contributes to the image with a strong contrast. As we have seen in Eq. (8) the PSF of ADF-SCM is complex so that the image can always be safely approximated by the square of a convolution of the PSF and the object function.

6. Imaging properties of ADF-SCM derived from PSF

In contrast to the BF-SCM case, when a thin object is larger than the section of the ADF-SCM PSF, the signal becomes zero (Eq. 10) at any defocus value. As a consequence, ADF SCM signal does not appear from a large object at any defocused position, when the object is wider than the PSF. This quantitatively explains why ADF SCM does not show the elongation effect. When the object covers only a portion of PSF by moving further from the confocal plane, or when the object moves from the optic axis, this cancellation over the integration becomes partial, and may result in some intensity. Nevertheless, the convolution

integral of ADF-SCM may be destructive so that the resultant intensity may not be significant.

At and near the confocal position, the probe is well focused to the atomic level and again the atomic structure of the sample appears. In contrast to BF-SCM, however, the PSF of ADF-SCM is complex so that the potential of the first order dominates the image contrast so that the effect of double diffraction is unimportant. This can be seen in Fig. 9 where the image intensity profiles of one and two Au atoms are calculated with the PSF and multislice. Here, the two profiles are almost identical for both single and two atoms between PSF and multislice.

It may be noted that the integral of ADF-SCM PSF over the x - y plane is zero even on the confocal plane. It means that when the object function does not change much for the range of the probe size, the ADF-SCM signal will be small. Therefore, an ADF-SCM signal is only generated by sharp features of the sample, but not

generated by broad sample features. This result was also explained by a simple geometrical consideration that a sharp feature scatters electrons into the collector aperture [13]. It is therefore evident that the shape of the PSF does not explain the elongation effect. In the same way, the shape of the CTF, namely the missing-cone, does not simply explain the elongation effect. We need to evaluate the value of PSF or CTF.

As we noted in Section 4, the imaginary part of ADF-SCM PSF is antisymmetric in the z direction, and the positive and negative parts of PSF will cancel out, when the probe is focused within the object. On the contrary, when the probe is located at an interface or surfaces this cancellation becomes imperfect. Therefore, the ADF-SCM intensity should be sensitive for detecting interfaces as demonstrated in the following simulations.

Here, we calculated the same 30 nm thick $\langle 110 \rangle$ GaAs sample as used in Ref. [21]. The images in Fig. 10(a) and (b) are simulated

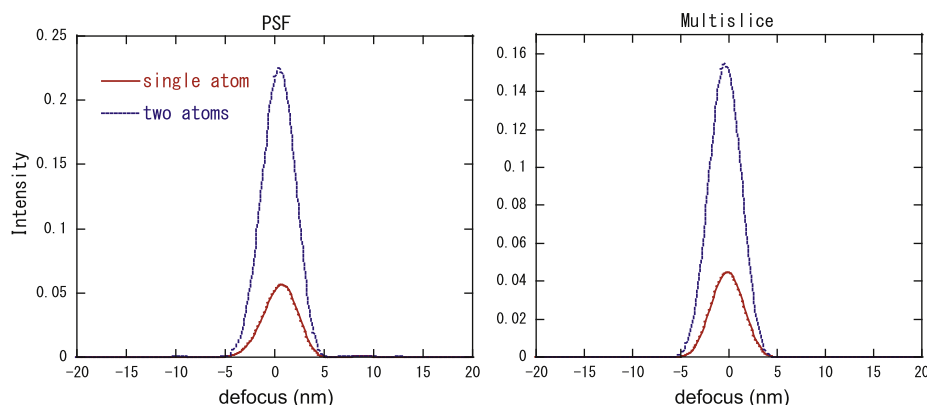


Fig. 9. The ADF-SCM image intensity profiles of one and two Au atoms calculated with the PSF and multislice method.

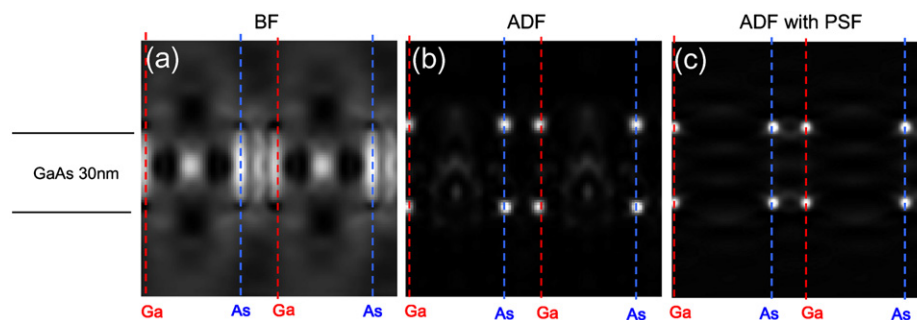


Fig. 10. Simulated x - z images of 30 nm thick $\langle 110 \rangle$ GaAs located at the center of the image calculated for 1.13 nm in the x direction and 90 nm in the z direction for (a) BF, (b) ADF and (c) ADF with the PSF. In ADF-SCM, the atoms within the slab show almost no intensity, and strong peak observed only at the surfaces.

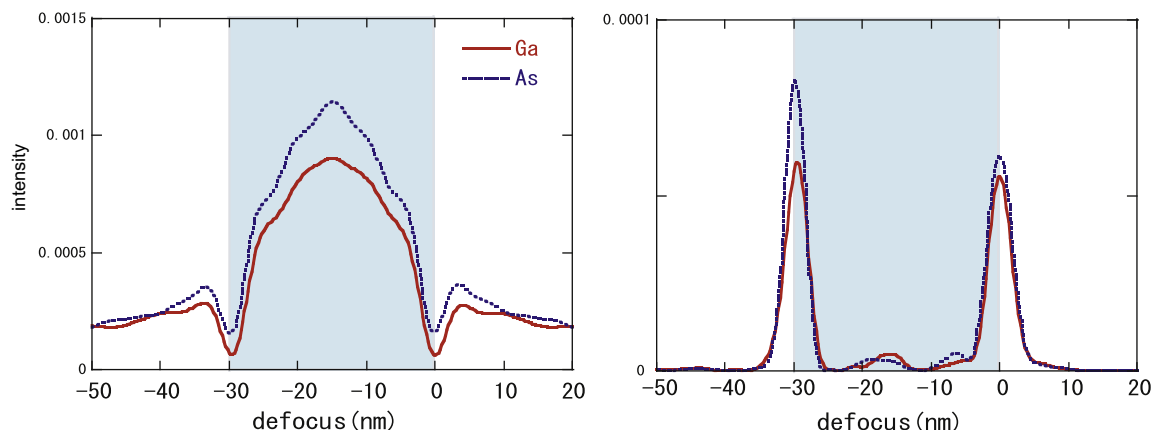


Fig. 11. The simulated (a) BF- and (b) ADF-SCM intensity profiles of a 30 nm thick $\langle 110 \rangle$ GaAs sample for Ga and As columns together with two inter-atomic sites.

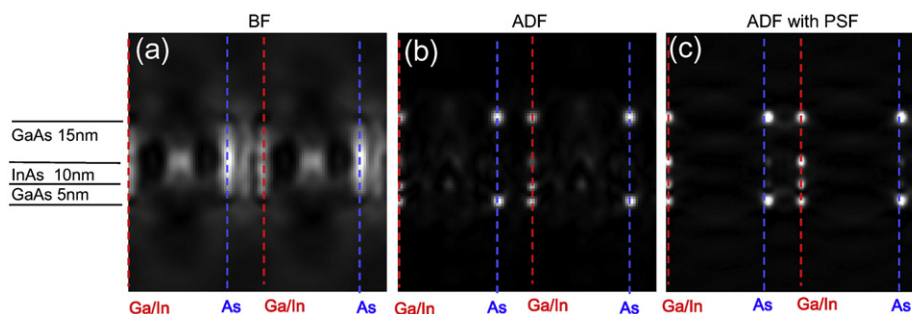


Fig. 12. The simulated x - z SCEM images of an InAs layer embedded in the 30 nm GaAs sample of Fig. 10 and located at 5 nm from the sample bottom, for (a) BF, (b) ADF and (c) ADF calculated with the PSF. The lattice constants are assumed to be the same for InAs and GaAs.

using the multislice method while the image in Fig. 10(c) is done by the PSF. The BF-SCEM image (Fig. 10a) is similar to the one given in Fig. 9 of Ref. [21], although the z -spread of the probe is three times larger here. It is striking that the atoms within the slab show almost no intensity in ADF-SCEM, while strong peaks were observed only at the surfaces (Fig. 10b). It may be noted that the PSF gives the ADF-SCEM image (Fig. 10c) similar to the one calculated with the multislice method.

Fig. 11 shows intensity profiles of (a) BF- and (b) ADF-SCEM images along the z -direction at the Ga and As columns. As already pointed by Cosgriff et al., BF-SCEM intensity at the atomic column drops at the top and bottom surfaces of the sample. On the contrary ADF-SCEM intensities have their peaks at the top and bottom surfaces for atomic column site.

Fig. 12 shows simulated images for a buried interface, where 10 nm of GaAs is replaced by InAs at 5 nm from the sample bottom as shown in the figure. Thus, the Ga/In columns have discontinuity, while the As atom columns are continuous. In the BF-SCEM image (Fig. 12a), the change from GaAs to InAs is faint, and it is difficult to identify the depth of the impurity layer as pointed by Cosgriff et al. [21]. On the other hand, a clear peak appears at the Ga/In interface in the ADF-SCEM image while the As column intensity does not show any contrast between the top and bottom interfaces (Fig. 12b). This unique edge enhancement effect of ADF-SCEM should be useful for depth sectioning. It may be noted again that the PSF gives the ADF-SCEM image (Fig. 12c) similar to the one calculated with the multislice method.

7. Conclusions

We have discussed that the imaging properties of BF- and ADF-SCEM are discussed based on their PSFs. The PSFs of both BF-SCEM and ADF-SCEM show the similar hourglass shapes, and the corresponding CTFs have comparable missing cones. Therefore, the difference in the imaging properties of BF and ADF-SCEM cannot be explained by the shape of the PSF or CTF. It is shown that the ADF-SCEM PSF is complex and has a Hermitian symmetry, while the symmetric BF-SCEM PSF is always positive and possesses a center of symmetry. The positive valued PSF explains the large elongation effect of BF-SCEM for a laterally extended object, and the complex-valued PSF accounts for the absence of the elongation effect in ADF-SCEM. This means that it is essential to consider the numerical distributions of PSF as well as its overall shape to understand the imaging characteristics.

The Hermitian symmetry of the ADF-SCEM PSF results in the interesting “edge enhancement effect” at the interface. Simulation using the PSF and multislice method verified this effect at the surfaces of a GaAs and at the InAs impurity interfaces embedded in GaAs. This unique feature of ADF-SCEM can potentially be useful for depth sectioning. It is also pointed out that the approach of PSF is not applicable for BF-SCEM of a phase object, if the lens system is symmetric and aberration free.

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Optical switching of nuclear spin–spin couplings in semiconductors

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Two-qubit operation is an essential part of quantum computation. However, solid-state nuclear magnetic resonance quantum computing has not been able to fully implement this functionality, because it requires a switchable inter-qubit coupling that controls the time evolutions of entanglements. Nuclear dipolar coupling is beneficial in that it is present whenever nuclear–spin qubits are close to each other, while it complicates two-qubit operation because the qubits must remain decoupled to prevent unwanted couplings. Here we introduce optically controllable internuclear coupling in semiconductors. The coupling strength can be adjusted externally through light power and even allows on/off switching. This feature provides a simple way of switching inter-qubit couplings in semiconductor-based quantum computers. In addition, its long reach compared with nuclear dipolar couplings allows a variety of options for arranging qubits, as they need not be next to each other to secure couplings.

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Nuclear magnetic resonance (NMR) quantum computing has attracted broad interest because it is one of the most advanced testbeds for quantum computation. Although the interest began with solution NMR^{1–3}, it is now believed that scalable NMR quantum computers in the future will be built on semiconductors based on highly developed semiconductor technology^{4–6}. The main challenges include the initialization and the creation of spin entanglement, which are essential features of quantum computation⁷. Semiconductor-based NMR quantum computers are advantageous as they can be achieved optically; that is, the initialization (nuclear-spin polarization) is provided by optical pumping^{8,9}, and the entanglement is created via internuclear (nuclear spin-spin) couplings between polarized nuclei^{10,11}. In optically pumped semiconductors, the latter manifests itself as dipolar order^{12,13} and double-quantum coherence¹⁴.

Switchability is another essential functionality required for inter-nuclear couplings, which should be ‘on’ during operations and ‘off’ otherwise. In this respect, nuclear dipolar coupling (*D*-coupling, hereafter) is not the best choice for the above-mentioned reasons. In addition, the time required for operations increases rapidly with increasing qubit number because of decoupling operations¹⁵. Other possible candidates include indirect couplings mediated by donor electrons⁴ and magnons¹⁵. Their implementations are fairly challenging, however, given the complicated switching mechanisms. By contrast, the scheme presented in this paper is rather simple; the coupling strength can be controlled externally through light power, and on/off switching can be easily implemented.

In this study, we have performed cross-polarization (CP) experiments with GaAs under light illumination, and demonstrated that a nuclear spin-spin coupling grows in strength, and extends its reach to farther nuclei as light power is increased. These futures bring about a unique transition of the CP process from oscillatory behaviour in the ‘dark’ towards exponential relaxation with increasing light power. The experiments provide us with information on the essential features of the optically induced nuclear spin-spin couplings; in particular, we find that the coupling strength is roughly proportional to light power, which is essential for the switching of the couplings.

Results

CP process in GaAs in the dark. The present mechanism is manifested in a CP process from ⁷⁵As (*I*-spin) to ⁷¹Ga (*S*-spin) in GaAs under infrared light irradiation. Before detailing this process, we first describe it in the dark (without light irradiation) as a reference. This is an ordinary CP process, for which we expect a contact time (τ_{cp}) dependence of *S*-magnetization (M_S^{eq}) of the form

$$M_S^{eq}(\tau_{cp}) = M_0^{eq}[1 - \exp(-\tau_{cp}/T_{IS})], \quad (1)$$

where T_{IS} is the cross-relaxation time. A relaxation process in the rotation frame (T_{ip}) need not be considered here, as it is sufficiently long because of high crystal symmetry^{16,17}. The reality, however, is more complicated than equation (1). Figure 1 shows $M_S^{eq}(\tau_{cp})$ obtained in the dark, which exhibits a clear transient oscillation.

Transient oscillations have been reported in some molecular crystals, and attributed to discrete *S*–*I* coupling spectra of isolated *S*–*I* pairs^{18–20}. The magnetization is transferred back and forth inside the pair with a frequency corresponding to half the flip-flop term of the *D*-coupling. The present sample, however, is not a molecular crystal, so isolated pairs are expected to be rare. Here an essential factor is the existence of two Ga isotopes, that is, ⁶⁹Ga and ⁷¹Ga with natural abundances of ⁶⁹N_A = 0.604 and ⁷¹N_A = 0.396, respectively. A local ⁷⁵As–⁷¹Ga pair appears when only one of the four nearest-neighbour sites of ⁷⁵As is occupied by ⁷¹Ga and the others are occupied by ⁶⁹Ga. The probability of finding such pairs is ${}_4C_1 \cdot {}^{71}N_A \cdot ({}^{69}N_A)^3 = 0.35$; that is, about 35% of ⁷⁵As have a single ⁷¹Ga in the vicinity and contribute to the oscillation. The pairs are coupled through indirect

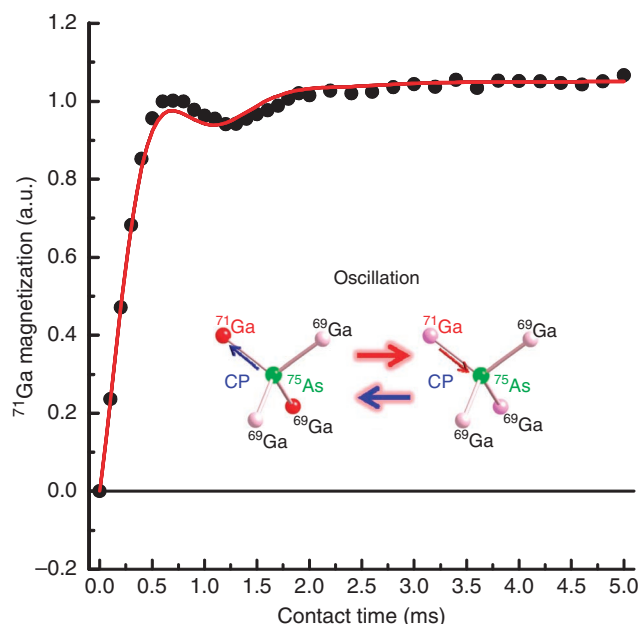


Figure 1 | CP experiment in GaAs in the dark. Contact time (τ_{cp}) dependence of ⁷¹Ga magnetization (M_S^{eq}) in a cross polarization experiment in the dark at 10 K, normalized at $\tau_{cp} = 0.7$ ms. The pulse sequence is shown in Figure 5, where the light is turned off, that is, $P_{IR} = 0$ mW. The magnetization is transferred back and forth between ⁷⁵As and ⁷¹Ga at a nearest neighbour site. This process gives rise to a transient oscillation. The solid red line represents the best-fit curve using equation (2).

scalar coupling J_{IS} , where *D*-couplings are absent because the Ga sites are situated at magic angle positions in the (100) crystal orientation^{16,17,21}. The process is described by a damping oscillation¹⁸,

$$M_S^{eq}(\tau_{cp}) = M_0^{eq} \left[1 - \frac{1}{2} \exp(-R\tau_{cp}) - \frac{1}{2} \exp(-3R\tau_{cp}/2) \cos(2\pi\Omega\tau_{cp}) \right], \quad (2)$$

where Ω is the oscillation frequency ($\Omega = J_{IS}/2$) and R is the damping factor. We fit the data using equation (2) with Ω and R as free parameters. The best result is obtained with $\Omega = 0.74$ kHz, which yields $J_{IS} = 2\Omega = 1.48$ kHz. This value is comparable with that in the InP case ($J_{IS} = 2.3$ kHz)¹⁷. The fitting curve is shown by a solid line in Figure 1. The fit is not very good at the beginning of the oscillation; this may be due to the presence of ⁷⁵As sites with more than one ⁷¹Ga nucleus in the four nearest-neighbour sites.

CP process under light irradiation. Figure 2a shows the τ_{cp} dependence of *S*-polarization, $M_S(\tau_{cp})$ under various levels of light power, P_{IR} , which exhibits new local maxima (peaks) that are not observed in the dark. The maxima form a number of series ($\alpha, \beta, \gamma, \dots$), and the maximal position in each series shifts towards smaller values of τ_{cp} as the light power is increased. For example, the series ‘ β ’ shown by the red arrows starts with a broad maximum around $\tau_{cp} = 3.1$ ms at $P_{IR} = 100$ mW, which shifts towards smaller values of τ_{cp} as P_{IR} is increased and eventually merges into a peak at $\tau_{cp} = 0.8$ ms at 166 mW. These maxima represent new polarization transfer processes that appear under light irradiation. Figure 3 shows a three-dimensional representation of Figure 2a obtained by interpolating the data in between. The continuous shift of the maximum in each series can be readily confirmed.

This phenomenon can be explained by discrete increments in the number of *S*-nuclei (⁷¹Ga) involved: in the dark ($P_{IR} = 0$ mW), the number of nuclei participating in the process is small and oscillatory behaviour is observed, as shown in Figure 1. Light illumina-

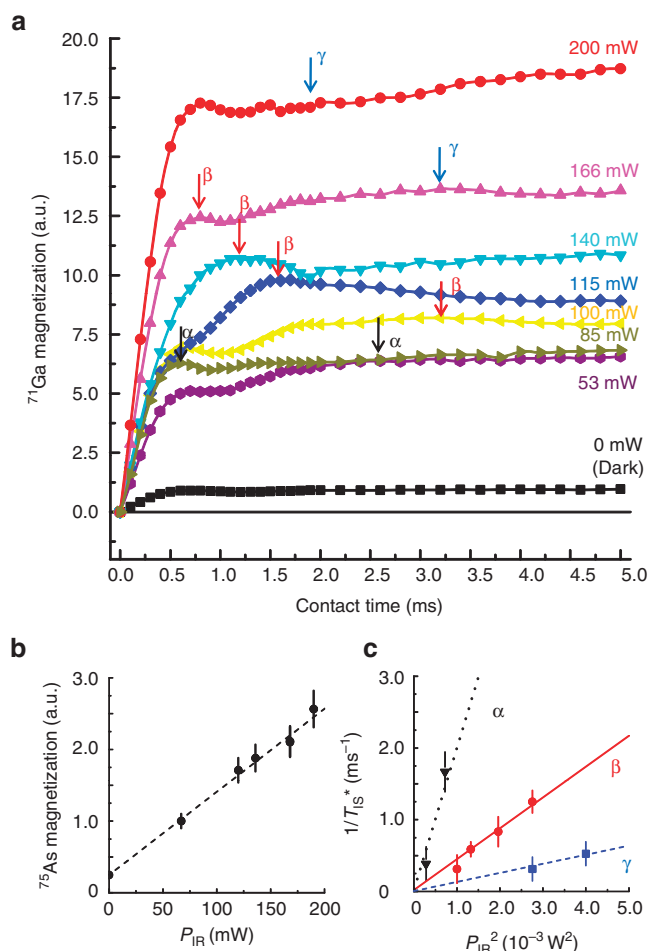


Figure 2 | CP experiments in GaAs under light irradiation. (a) Contact time (τ_{cp}) dependence of ^{71}Ga magnetization (M_s) in CP experiments at 10 K measured under light irradiation with power P_{IR} ranging from 0 to 200 mW. The pulse sequence is shown in Figure 5. The data at $P_{IR} = 0$ mW are the same as those in Figure 1. Black, red and blue arrows represent the three series of maxima. (b) The light power (P_{IR}) dependence of ^{75}As magnetization (M_I), which represents the enhancement of the nuclear spin polarization due to the optical pumping effect. The magnetization exhibits a linear dependence with P_{IR} . The error bars represent the data scatterings in a few experiments performed under the same conditions. (c) The values of $1/T_{IS}^*$ for the three series (α , β and γ) marked by the arrows in (a) plotted against P_{IR}^2 . The error bars represent the broadness around each peak. The lines are visual guides.

tion causes a series of S-nuclei batches (α , β , γ ...) to participate in sequence. Their contributions are successively added to the signal intensity one after another, whereas the speed of transfer from I- to S-nuclei in each series increases with the light power (that is, the maximal position shifts towards smaller values of τ_{cp}). As the number of nuclei involved is further increased, the cross relaxation is expected to approach the exponential relaxation behaviour given by equation (1)²⁰. That is, we see here the transition from oscillatory behaviour in the 'dark' towards exponential relaxation with increasing light power. This is a unique case in that such a transition can hardly be observed in ordinary NMR measurements.

Another factor responsible for the phenomenon is the optical pumping effect^{8,9}. It contributes to the S-magnetization (^{71}Ga) through the enhancement of the I-magnetization (^{75}As) caused by the polarization transfer from the optically oriented electrons for the duration of light irradiation $\tau_L = 60$ s. Figure 2b shows the light-power dependence of the I-magnetization (^{75}As), which increases linearly with light power. This enhancement is partly responsible for

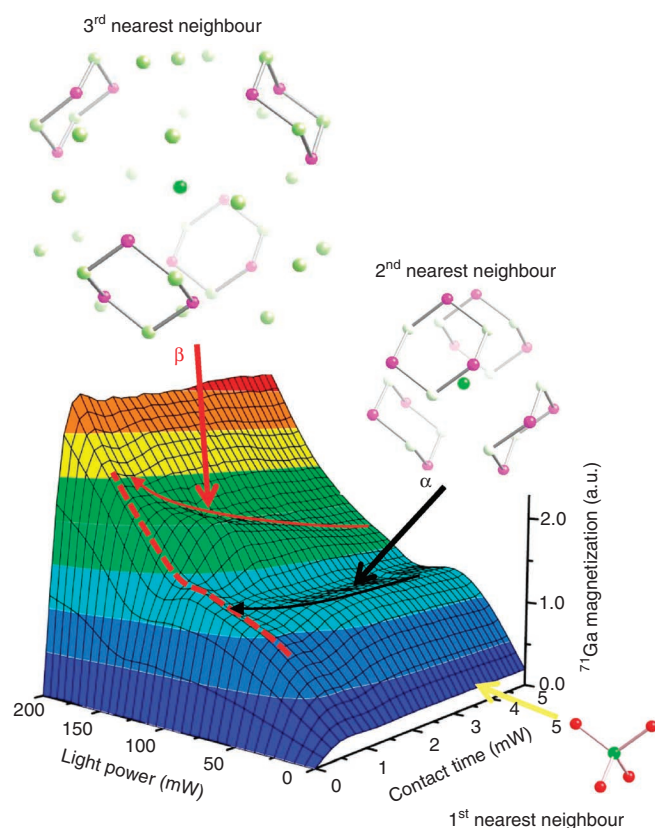


Figure 3 | Light power dependence of the CP process in GaAs. Three-dimensional representation of Figure 2a obtained by interpolating the data in between. The data shown by the yellow arrow represents those in the dark ($P_{IR} = 0$ mW), which corresponds to the polarization transfer to the nearest-neighbour ^{71}Ga through the conventional J_{IS} . The black (α) and red (β) arrows indicate the series of maxima corresponding, presumably, to the polarization transfers to the second and third nearest neighbour sites, respectively. The dotted red line represents the ^{71}Ga magnetization at $\tau_{cp} = 0.7$ ms, which exhibits a plateau-like feature (that is, less steeper slope) around 80 mW.

the increase of the S-magnetization with increasing light power, as seen in Figures 2a and 3.

The τ_{cp} dependence of M_s is expressed as,

$$M_S(\tau_{cp}) = M_I^{\text{dark}}(\tau_L) m_{I \rightarrow S}^{\text{eq}}(\tau_{cp}) + M_I^{\text{OP}}(P_{IR}, \tau_L) \times \left[m_{I \rightarrow S}^{\text{eq}}(\tau_{cp}) + \sum_{i=\alpha} m_{I \rightarrow S}^{\text{OP}, i}(P_{IR}, \tau_{cp}) \right], \quad (3)$$

where $M_I^{\text{dark}}(\tau_L)$ in the first term on the right-hand side represents the I-magnetization in the dark portion of the sample recovered during $\tau_L = 60$ s and $m_{I \rightarrow S}^{\text{eq}}$ represents the polarization transfer process to S-spins in the first nearest neighbour sites through J_{IS} . This process is essentially the same as that in Figure 1. On the other hand, $M_I^{\text{OP}}(P_{IR}, \tau_L)$ in the second term is the I-magnetization in the illuminated area generated by the optical pumping effect during $\tau_L = 60$ s, whose P_{IR} dependence is shown in Figure 2b. This magnetization is transferred to S-spins through the two processes shown in the brackets: the same process as that in the dark, $m_{I \rightarrow S}^{\text{eq}}$, and additional ones induced by light irradiation, $\sum_i m_{I \rightarrow S}^{\text{OP}, i}(P_{IR}, \tau_{cp})$ ($i = \alpha, \beta, \gamma$...). The latter are caused by the optically induced heteronuclear indirect coupling, J_{IS}^{opt} . We will discuss characteristics of this coupling in the next section.

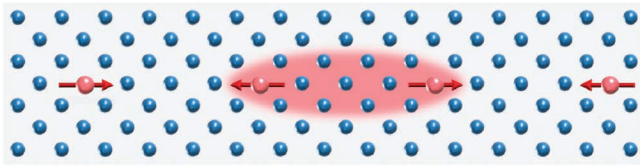


Figure 4 | Optical switching of a nuclear spin-spin coupling. A conceptual (ideal) model of optical switching of a nuclear spin-spin coupling between nuclear-spin qubits (red arrows) spaced a few lattice points apart from each other. Blue balls represent ‘inert’ (for example, spin-zero) nuclei. The nuclear dipolar couplings between the qubits are out of reach. Once the light is illuminated, a coupling emerges between the qubits. The strength of the coupling can be controlled externally through light power.

Discussion

The new coupling J_{IS}^{opt} is presumably mediated by photo-excited electrons. It is known that electrons in metals can mediate indirect nuclear spin-spin couplings, a process referred to as the RKKY interaction²². In the present case, however, J_{IS}^{opt} is observed in semiconductors where no intrinsic Fermi surfaces exist. Moreover, the lifetimes and spin relaxation times of the photo-excited electrons usually fall in the range between pico- and nanoseconds, which is more than six orders of magnitude smaller than that of the CP process, that is, milliseconds. It is intriguing that two phenomena with such different time scales are coupled with each other.

The strength of J_{IS}^{opt} in each batch ($\alpha, \beta, \gamma, \dots$) can be evaluated by the cross-relaxation time T_{IS}^{opt} in $m_{I \rightarrow S}^{\text{opt}, i}(P_{\text{IR}}, \tau_{\text{CP}})$ ($i = \alpha, \beta, \gamma, \dots$). According to Demco *et al.*^{17,20,23}, the cross-relaxation time is expressed as

$$1/T_{IS}^{\text{opt}} \approx \frac{\sqrt{\pi}}{4} M_2^{IS} \tau_c, \quad (4)$$

where τ_c is the correlation time of the CP and,

$$M_2^{IS} = \frac{1}{3} I(I+1) \sum (2\pi J_{IS}^{\text{opt}})^2 \quad (5)$$

is the second moment of the I - S heteronuclear spectrum due to J_{IS}^{opt} . Equation (5) indicates that M_2^{IS} is proportional to $(J_{IS}^{\text{opt}})^2$. Provided that J_{IS}^{opt} is proportional to P_{IR} and that τ_c is independent of J_{IS}^{opt} , equations (4) and (5) lead to the conclusion that $1/T_{IS}^{\text{opt}}$ is proportional to P_{IR}^2 . The actual value of T_{IS}^{opt} in each batch is determined from the analysis of the functional form of $m_{I \rightarrow S}^{\text{opt}, i}(P_{\text{IR}}, \tau_{\text{CP}})$ ($i = \alpha, \beta, \gamma, \dots$). Here we evaluate it from the maximal position; T_{IS}^{opt} is defined as the contact time at which the maximum is formed. Figure 2c shows $1/T_{IS}^{\text{opt}}$ for the three series of maxima (α, β and γ) plotted against P_{IR}^2 . The graph suggests that $1/T_{IS}^{\text{opt}}$ is roughly proportional to P_{IR}^2 , implying that J_{IS}^{opt} is proportional to P_{IR} .

This result provides a clue to the nature of the series of S -nuclei batches ($\alpha, \beta, \gamma, \dots$). Equation (5) indicates that the condition $M_2^{IS} \propto (J_{IS}^{\text{opt}})^2$ is fulfilled only when all I - S pairs participating in the summation of the right-hand side share the same J_{IS}^{opt} , such that J_{IS}^{opt} can be taken out of the summation Σ . Therefore, each series may be assigned to the polarization transfers to a batch of S -spins at the same distance from the I -spin at the origin. Figure 3 illustrates this situation. For example, the α -maxima may be assigned to the second nearest-neighbour S -spins, the β -maxima to the third, and so forth. It is reasonable that equivalent S -spins in the same order situated at the same distance from the I -spin share the same scalar coupling with it. Moreover, the contributions from nuclei in the same order to the correlation time are partially cancelled out^{17,20,23}. Therefore, the assumption that τ_c is independent of J_{IS}^{opt} may be a reasonable approximation.

The result also provides us with information about the reach of J_{IS}^{opt} . The dotted red line in Figure 3 shows the P_{IR} dependence of

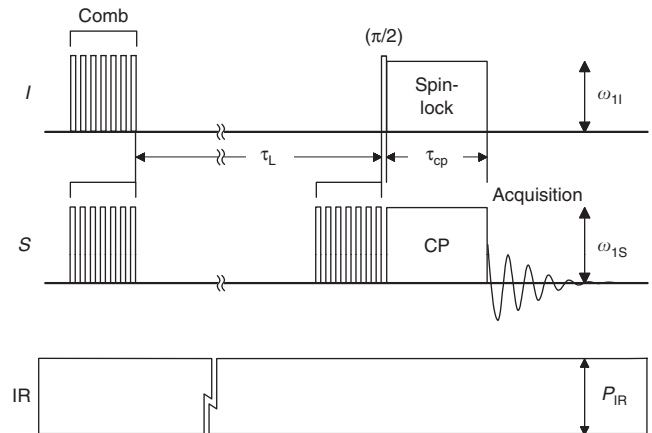


Figure 5 | Pulse sequence. Pulse sequence for optical pumping CP experiments. $I = {}^{75}\text{As}$, $S = {}^{71}\text{Ga}$ and IR = infrared light (1.50 eV, σ^-). ‘Comb’: comb pulse consisting of a number of $\pi/2$ pulses. τ_L : duration of light irradiation (≈ 60 s). τ_{CP} : contact time for CP. $\omega_{11}/2\pi$, $\omega_{IS}/2\pi$: rf-pulse strengths in terms of Rabi frequency (≈ 35 kHz). P_{IR} : infrared light power. The horizontal axis is not to scale.

$M_s(\tau_{\text{CP}})$ at a constant τ_{CP} . One finds a plateau-like feature around 80 mW, which indicates that the S -spins at the third nearest-neighbour sites participate in the process only when $P_{\text{IR}} > 80$ mW; that is, J_{IS}^{opt} can reach farther than D - and J_{IS} -couplings, as the latter two may not substantially reach the third-nearest neighbours.

Features such as the external switchability and the long reach may add variety to the qubit arrangement in semiconductor-based NMR quantum computers. Figure 4 shows an example. In an array of nuclear-spin qubits, a few lattice points apart from each other, the D -couplings are out of reach, so that the qubits are decoupled. Then, the present mechanism provides the external control of inter-nuclear couplings. As the light power is increased, the reach of couplings is extended. This would further enhance the flexibility of qubit arrangements.

This mechanism is compatible with most schemes proposed for semiconductor-based NMR quantum computing. The implantation of nuclear spin arrays in semiconductors has been studied using various techniques such as scanning probe microscopes, ion beams, and isotope engineering^{24–26}. These techniques will provide promising technologies to implement the schemes shown in Figure 4.

In conclusion, we have discovered an optically switchable indirect nuclear spin-spin coupling, which is manifested in CP experiments with GaAs under light illumination. As the strength of this coupling is externally controllable through light power, we expect it to have an essential role in the quantum gate operations of solid-state NMR quantum computers in the future.

Methods

Optical pumping double resonance NMR system and the sample. The CP experiments were performed at 10 K using an optical-pumping double-resonance NMR system developed for this study. The system consists of an NMR spectrometer, a laser system and a cryostat loaded in a 6.34-T superconducting magnet. A custom-built double-resonance probe is installed inside the cryostat. The sample used in this study is an undoped semi-insulating GaAs single-crystal wafer with a thickness of 600 μm and a crystal orientation of (100) (Mitsubishi Chemical). It is mounted on a sample stage made of sapphire located at the probe head and set with the surface normal to the magnetic field. The sample stage is cooled through thermal contact with the cold head of the cryostat. Infrared light emitted from a Ti:Sapphire laser is delivered to the cryostat through a polarization-maintaining fibre²⁷. It is converted to circularly polarized light by a quarter-wave plate at the probe head, and applied to the sample in parallel to the magnetic field. The spot size at the sample surface is about $\phi 5$ mm.

Pulse sequence. The pulse sequence used is shown in Figure 5. The magnetization of I -spins, saturated by the first comb pulse, is regenerated during the time interval

τ_1 and transferred to S-spins through the CP immediately thereafter. The infrared light is irradiated at a constant strength P_{IR} throughout the sequence. The photon energy of 1.50 eV (near the band gap) and the helicity σ^- were selected so that the optical pumping NMR signal enhancements for both ^{71}Ga and ^{75}As were nearly maximal. The experiment in the dark (Fig. 1) was obtained with the same pulse sequence as that in Figure 5, with the exception that $P_{\text{IR}} = 0$ mW.

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Author contributions

A.G. conceived and designed the experiments. All the authors jointly designed the system for the experiments, and A.G. and S.O. constructed it. A.G. carried out the main experiments. All the authors were involved in the analyses. A.G. wrote the paper, with the help of the co-authors.

Additional information

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Hyperthermic effects of dissipative structures of magnetic nanoparticles in large alternating magnetic fields

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Targeted hyperthermia treatment using magnetic nanoparticles is a promising cancer therapy. However, the mechanisms of heat dissipation in the large alternating magnetic field used during such treatment have not been clarified. In this study, we numerically compared the magnetic loss in rotatable nanoparticles in aqueous media with that of non-rotatable nanoparticles anchored to localised structures. In the former, the relaxation loss in superparamagnetic nanoparticles has a secondary maximum because of slow rotation of the magnetic easy axis of each nanoparticle in the large field in addition to the known primary maximum caused by rapid Néel relaxation. Irradiation of rotatable ferromagnetic nanoparticles with a high-frequency axial field generates structures oriented in a longitudinal or planar direction irrespective of the free energy. Consequently, these dissipative structures significantly affect the conditions for maximum hysteresis loss. These findings shed new light on the design of targeted magnetic hyperthermia treatments.

Tumour-targeted magnetic hyperthermia has recently attracted much attention¹. Preferential accumulation of magnetic nanoparticles in tumour tissue is achieved by conjugating nanoparticles with tumour-homing peptides² or antibodies³. When the accumulated nanoparticles are exposed to a large alternating magnetic field, $H = H_{ac} \sin(2\pi f \cdot t)$, where H_{ac} is the amplitude of the field, f is the frequency, and t is time, they begin to rotate because of magnetic torque. Simultaneously, the direction of the magnetic moment, μ , in each nanoparticle reverses with a certain probability. Consequently, heat equivalent to the magnetic loss dissipates locally in the tumour tissue. If the properties of the irradiated field are limited (*i.e.*, $H_{ac} \cdot f < \text{constant}$)¹ to ensure biomedical safety, then nanoparticles that maximise the *in vivo* efficiency of heat dissipation, $P_H/(H_{ac} \cdot f)$, are required, where P_H is the specific energy dissipation rate (specific loss power) per unit mass of nanoparticles. The actual rotations of the nanoparticles are disordered because the microviscosity of the local environment in cancer cells is not constant^{4,5}, and effective elasticity depends on the binding conditions between nanoparticles and membranes. To minimise the effect of irregular rotations in magnetic hyperthermia, two guiding principles have been proposed on the basis of simple models that consider a linear response of thermodynamic equilibrium states or magnetic field-driven reversals.

The first guiding principle⁶ is to use the relaxation loss in superparamagnetic iron oxide nanoparticles (SPIONs) with a sufficiently low energy barrier ΔU for reversal. If a linear response of their thermodynamic equilibrium state is considered at low H_{ac} , the out-of-phase component of AC susceptibility χ'' can be expressed as follows:

$$\chi'' = \chi_0 (2\pi f \cdot \tau) / [1 + (2\pi f \cdot \tau)^2], \quad (1)$$

where χ_0 is the initial susceptibility per unit mass of SPIONs. When reversal and rotation occur in a nanoparticle in parallel, the characteristic time τ is given by the following equation:

$$\tau^{-1} = \tau_N^{-1} + \tau_B^{-1}, \quad (2)$$

where τ_N is the Néel relaxation time for reversal, and τ_B is the Brownian relaxation time for rotation. Consequently, the heating efficiency $P_H/(H_{ac} \cdot f) = \pi \mu_0 \chi'' \cdot H_{ac}$ for individual monodisperse nanoparticles has a single maximum at the peak frequency $2\pi f_p = \tau^{-1}$. For a sufficiently small SPION, τ is determined only by τ_N because τ_N is much shorter than τ_B . In this case, it has been assumed that the conditions for maximising the efficiency are unaffected by uncontrolled rotation of the nanoparticles.

However, in some experiments, dual peaks have been observed for the frequency dependence of $\chi'' \propto P_H/(H_{ac} \cdot f)^{7,8}$ despite the prediction of a single peak at a $2\pi f_p$ value of $\tau^{-1} (= \tau_N^{-1} + \tau_B^{-1})$. For this reason, size



distribution⁷ or aggregation⁸ of the nanoparticles was considered based on the linear response theory. In an earlier study⁷, the low-frequency peak observed for the susceptibility was attributed to Brownian relaxation of larger nanoparticles, while the high-frequency peak was attributed to Néel relaxation of smaller nanoparticles. In another study⁸, the low- and high-frequency peaks were attributed to individual and agglomerated nanoparticles, respectively. Thus the observed dual peaks have been theoretically explained by the coexistence of two kinds of nanoparticles. In other words, these explanations are based on the assumption that a single kind of nanoparticle will produce only a single peak at $\tau^{-1} (= \tau_N^{-1} + \tau_B^{-1})$. However, this assumption has never been theoretically verified under a large AC magnetic field, where the linear response theory does not hold.

The second guiding principle is to use hysteresis loss in ferromagnetic nanoparticles⁹. In mechanical models such as the Stoner–Wohlfarth model for single domain particles, μ is reversed in the time scale of Larmor precession (picoseconds) when ΔU disappears at the switching field H_{sw} , because thermal fluctuations are not considered. Such fast reversals are considered to dominate the response to high frequency AC magnetic field because the Brownian relaxations of large ferromagnetic nanoparticles are generally slow compared with the oscillation of the field. In this case, the work done in one cycle is given by the area inside the hysteresis loop, $\zeta \cdot M_s \cdot H_{sw}$, where M_s is the spontaneous magnetisation and ζ is a coefficient related to the rectangularity of the loop. In the simple case of rectangular hysteresis loops, ζ is 0 for $H_{ac} < H_{sw}$ and 4 for $H_{ac} \geq H_{sw}$. Consequently, the maximum efficiency, $P_H/(H_{ac} \cdot f) = \zeta \cdot M_s \cdot (H_{sw}/H_{ac})/\rho$, where ρ is the density of magnetite, is achieved when H_{ac} is adjusted to H_{sw} . Because H_{sw} depends on the magnetic anisotropy field H_K specific to each nanoparticle, it has been assumed that, in cases where reversal is much faster than rotation, the amount of hysteresis loss is unaffected by the inhomogeneous rotations of nanoparticles in cancer cells.

In recent experimental studies^{10,11}, the observed P_H of immobilised ferromagnetic nanoparticles was lower than that of the same nanoparticles dispersed in a fluid. Kim *et al.*¹⁰ attributed the difference to variation in the rates of convective heat transfer. Müller *et al.*¹¹ suggested that the orientation or agglomeration of the nanoparticles, or interaction effects, may be responsible for the observed difference. The orientation of nanoparticles is important because it is related to the magnetic torque intrinsic in rotatable ferromagnetic nanoparticles. However, there has been no theoretical study on magnetic field-driven reversals of μ in ferromagnetic nanoparticles with easy axes that simultaneously rotate under the magnetic torque.

There are many reported inconsistencies between experimental results and predictions based on the above two guiding principles for optimising hyperthermia treatment. These guiding principles are based on the simple models established at the two limits: in zero magnetic field or at zero temperature. Under the conditions for hyperthermia ($H_{ac} \neq 0$, $T \neq 0$), where T is temperature, the validity of the guiding principles has not been theoretically verified even for an ideal system of non-interacting monodisperse nanoparticles. Consequently, we attempted to simulate the thermally assisted magnetic response of individual superparamagnetic/ferromagnetic iron oxide nanoparticles exposed to a large AC magnetic field like that

used in hyperthermia treatment. The simulation was performed in the following two extreme cases: non-rotatable nanoparticles strongly anchored to structures resembling organelles, and rotatable nanoparticles in an aqueous phase mimicking cytoplasm. In the simulations, the thermally activated reversals of μ were calculated between the meta-stable directions. Simultaneously, the rotations of the spheroidal nanoparticles were computed in the inertialess limit (Brownian dynamics simulation), where the frictional torque always balances with magnetic torque and with Brownian torque (details are reported in the Methods section). The results allow examination of whether the relaxation loss for $\tau_N \ll \tau_B$ and the hysteresis loss at $H_{ac} \approx H_{sw}$ are independent of the ability of the nanoparticles to rotate under the conditions for hyperthermia treatment.

Results

The magnetic response to an AC magnetic field $H_{ac} \sin(2\pi f \cdot t)$ at $T = 310$ K was simulated for individual monodisperse spheroidal magnetite nanoparticles with non-magnetic surfactant layers in non-rotatable and rotatable situations (see the Methods section for details). Results are presented for the following representative nanoparticles: nearly spherical nanoparticles with an aspect ratio, κ , of 1.1 and an equatorial diameter, $2R_M$, of 18 nm, and elongated spheroidal particles with $\kappa = 1.4$ and $2R_M = 24$ nm. The parameters of these nanoparticles are summarised in Table 1. The former nanoparticles with τ_N ($H_{ac}=0$) of 20 ns can be considered as typical SPIONs, while the latter with τ_N ($H_{ac} = 0$) of 2×10^7 s (1 year) can be regarded as typical ferromagnetic nanoparticles in the frequency range of hyperthermia treatment ($(2\pi f)^{-1}$ of approximately 1 μ s). Results for nanoparticles with other sizes and shapes are shown in the Supplementary Information.

The magnetisation curves of the non-rotatable nearly spherical nanoparticles at low H_{ac} (1 kA/m) are shown in Fig. 1. A linear response without hysteresis was observed at $f = 100$ kHz. Such superparamagnetic behaviour is reasonable because the estimated τ_N ($H_{ac} = 0$) for the nanoparticles is 20 ns. Hysteresis appeared in the curves at $f = 1,000$ kHz. As f increased further, the area inside the hysteresis loop grew. This area corresponds to the work done in one cycle. Therefore, $P_H/(H_{ac} \cdot f)$ also increased with f , and a single maximum was observed at a peak frequency, f_p , of 10,000 kHz (Fig. 1 (b)). Figure 2(a) shows the H_{ac} dependence using a contour plot of $P_H/(H_{ac} \cdot f)$. As H_{ac} increased, f_p shifted towards higher frequencies. As indicated by the dashed line in Fig. 2(a), this shift can be explained by the values of $\tau_N(H_{ac})$ calculated using the conventional Brown's equation as follows¹²:

$$[\tau_N(H_{ac})]^{-1} = f_0 \{ (1 - h^2) \{ (1 + h) \exp[(-K_d V / k_B T)(1 + h)^2] + (1 - h) \exp[(-K_d V / k_B T)(1 - h)^2] \} \} \quad (3)$$

where h is $\mu H / (2 K_d V)$, K_d is the shape anisotropy constant, V is the volume of the magnetic core, and k_B is the Boltzmann constant. Therefore, the emergence of a single peak in $P_H/(H_{ac} \cdot f)$ can be attributed to Néel relaxation loss, as expected for SPIONs.

For the nearly spherical nanoparticles with a low H_{ac} (1 kA/m), the magnetisation curves in the rotatable case were the same as those in the non-rotatable case (see Fig. 1). An equivalent maximum appeared in the f -dependence of $P_H/(H_{ac} \cdot f)$ in the linear response

Table 1 | Parameters of the simulated nanoparticles: including the aspect ratio κ , equatorial diameter $2R_M$, thickness of the surfactant layer δR , density of magnetite ρ , spontaneous magnetisation M_s , anisotropy field H_K , viscosity η , Néel relaxation time τ_N ($H = 0$), and Brownian relaxation time τ_B . The value of τ_N ($H = 0$) was estimated using equation (3), while the value of τ_B was estimated using equation (4).

	κ	$2R_M$ (nm)	δR (nm)	ρ (kg/m ³)	M_s (kA/m)	H_K (kA/m)	η (mPa s)	τ_N ($H = 0$) (sec)	τ_B (sec)
Nearly spherical nanoparticle	1.1	18	4.5	5200	450	17	1.0 / ∞	2×10^{-8}	8×10^{-6}
Elongated spheroidal nanoparticle	1.4	24	6.0	5200	450	57	1.0 / ∞	2×10^7	2×10^{-5}

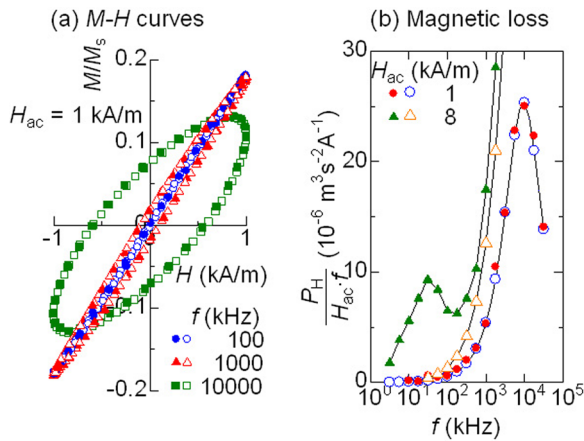


Figure 1 | Magnetic response of nearly spherical nanoparticles (typical SPIONs). The open and solid symbols show the values of non-rotatable and rotatable nanoparticles, respectively. (a) Steady magnetisation curves with low H_{ac} (1 kA/m) at various frequencies, (b) frequency dependence of the efficiency of heat dissipation with low and intermediate values of H_{ac} .

range ($H_{ac} = 1$ kA/m) (Fig. 1(b)). This behaviour is consistent with the above assumption because the estimated τ_N ($H_{ac} = 0$) of 20 ns is much shorter than $\tau_B = 8$ μ s (Table 1). The shift of this peak with increasing H_{ac} is analogous to that in the non-rotatable case (Fig. 2(b)). However, another maximum of $P_H/(H_{ac} \cdot f)$ was observed at $H_{ac} = 8$ –16 kA/m and $f = 30$ kHz in the contour plot shown in Fig. 2(b). A secondary maximum like this has not previously been theoretically predicted for individual monodisperse nanoparticles. Figure 3(a) shows the magnetisation curves calculated under these conditions. Unlike the non-rotatable case, an S-shaped hysteresis loop without remanence existed. At the same time, the mean orientation of the long (easy) axes of the nanoparticles showed butterfly-shaped hysteresis, as shown Fig. 3(b). Because such behaviour cannot be explained using the present linear response theory, its origin is discussed in the next section from the viewpoint of the rotation of the long axis of SPION.

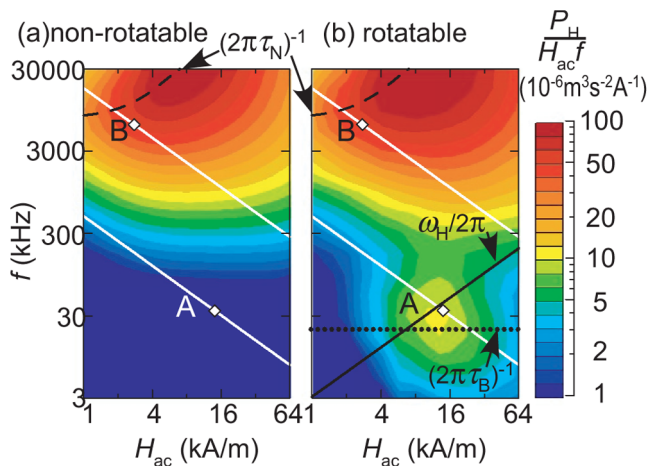


Figure 2 | Efficiency of heat dissipation in the nearly spherical nanoparticles (typical SPIONs) that are (a) non-rotatable and (b) rotatable, where P_H is the specific energy dissipation rate. Dashed lines represent the Néel relaxation time $(2\pi\tau_N)^{-1}$, dotted lines show the Brownian relaxation time $(2\pi\tau_B)^{-1}$, and solid lines indicate typical angular velocity, $\omega_H(H = H_{ac}, \psi = \pi/4)/2\pi$, of the rotation caused by magnetic torque. White lines show the thresholds for biomedical safety. Diamonds A and B on the white lines denote the conditions for maximum $P_H/(H_{ac} \cdot f)$ in the rotatable nanoparticles.

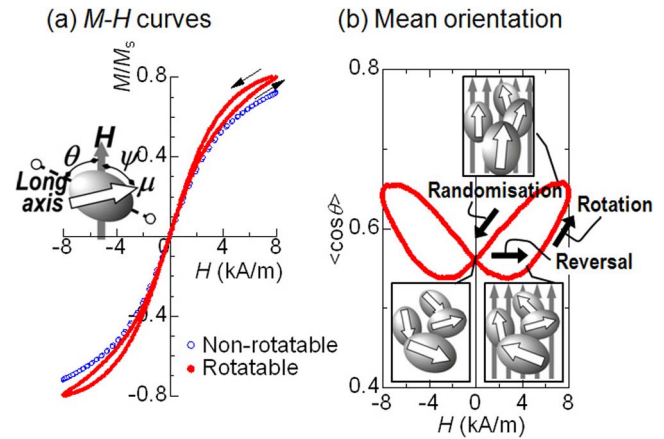


Figure 3 | Magnetic response of nearly spherical nanoparticles (typical SPIONs) with an applied AC field with $H_{ac} = 8$ kA/m and $f = 30$ kHz. (a) Steady magnetisation curves, (b) mean orientation of the long (easy) axis of the nanoparticles, $\langle \cos\theta \rangle$. Orientations are indicated in the inset images.

The magnetisation curves at $f = 10,000$ kHz for the elongated spheroidal nanoparticles, which are typical ferromagnetic nanoparticles, in the non-rotatable case are shown in Fig. 4(a). The curve was reversible at $H_{ac} = 20$ kA/m, and hysteresis appeared in the curve at $H_{ac} = 26$ kA/m. As H_{ac} increased further, the area inside the hysteresis loop grew. When H_{ac} became larger than 32 kA/m, the expansion of the area was saturated and the shape of the magnetisation curve approached that predicted by the Stoner–Wohlfarth model. $P_H/(H_{ac} \cdot f)$ was almost zero at low H_{ac} , then at approximately 30 kA/m it began increasing rapidly with H_{ac} , followed by a gradual decrease with increases in H_{ac} (Fig. 4(b)). This behaviour depends weakly on the frequency (Fig. 5(a)). In mechanical models that do not consider thermal fluctuation, a hysteresis loop appeared when H_{ac} was higher than H_{sw} . Because H_{sw} of ferromagnetic nanoparticles with randomly oriented easy axes ranges from $H_K/2 = 29$ kA/m to $H_K = 57$ kA/m, and is often close to $0.5 H_K$, that $P_H/(H_{ac} \cdot f)$ is almost independent of frequency at $H_{ac} > 30$ kA/m in the non-rotatable

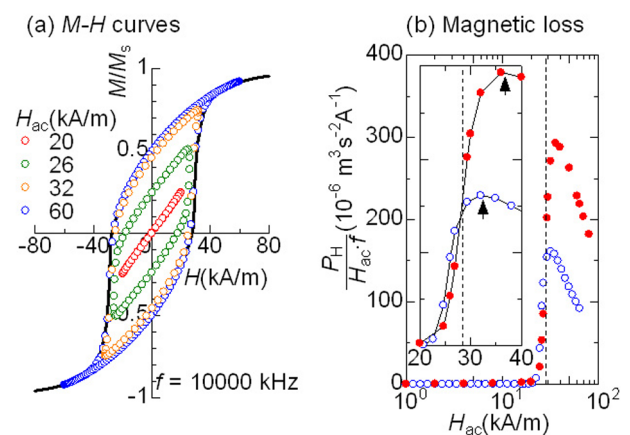


Figure 4 | Magnetic response of elongated spheroidal nanoparticles (typical ferromagnetic nanoparticles) in high-frequency AC fields with $f = 10,000$ kHz and various values of H_{ac} . (a) Steady magnetisation curves in the non-rotatable case; the corresponding curves for rotatable nanoparticles are presented in Fig. 6(a) and (b). Solid lines indicate Stoner–Wohlfarth model curves. (b) H_{ac} –dependence of the efficiency of heat dissipation. The open and solid symbols show the values of the non-rotatable and rotatable nanoparticles, respectively. The arrows indicate the peak maxima of $P_H/(H_{ac} \cdot f)$ in both cases, and the broken line shows half of the anisotropic field, $H_K/2$. The inset shows an enlarged view.

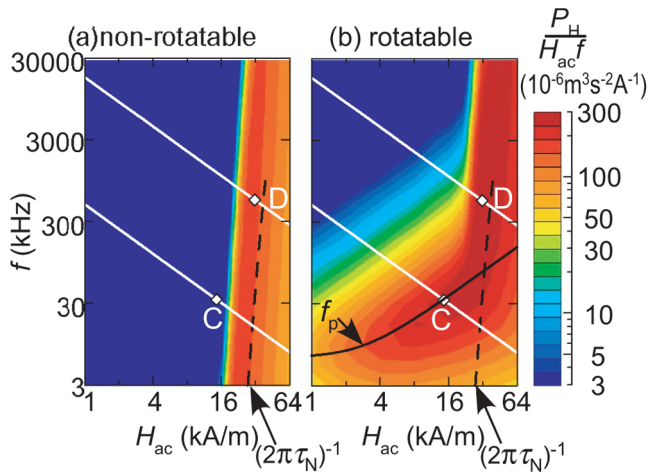


Figure 5 | Efficiency of heat dissipation of elongated spheroidal nanoparticles (typical ferromagnetic nanoparticles) that are (a) non-rotatable and (b) rotatable. Dashed lines represent the Néel relaxation time $(2\pi\tau_N)^{-1}$ and the solid line indicates f_p , which was calculated using equation (5). White lines show the thresholds for biomedical safety. Diamonds C and D on the white lines denote the conditions for maximum $P_H/(H_{ac} \cdot f)$ in the rotatable nanoparticles.

ferromagnetic nanoparticles is consistent with the properties expected for the hysteresis loss.

Figure 6 shows the magnetisation curves for rotatable elongated spheroidal nanoparticles at $f = 10,000$ kHz. Because the magnetic response slowly changed after the AC magnetic field was applied at $t = 0$, transient variations of the hysteresis loops are observed. The shape of the major hysteresis loop at $H_{ac} = 60$ kA/m was initially consistent with that predicted by the Stoner–Wohlfarth model with randomly oriented easy axes. However, the remanence of the major loop gradually increased from $0.5 M_s$ to M_s . In other words, the major loop became squarer, and the area inside the loop increased with time. In comparison, the remanence of the minor loop at $H_{ac} = 26$ kA/m gradually decreased and the area became smaller over time. As shown in Fig. 6(c) and (d), the long (easy) axes of the nanoparticles gradually turned when the variations of the loops proceeded (see the next section for details). Consequently, the increased area of the major hysteresis loops and decreased area of the minor loops caused the maximum of $P_H/(H_{ac} \cdot f)$ to shift towards higher H_{ac} compared with the non-rotatable case (see arrows in Fig. 4(b)). Note that reversals occurred every hundred nanoseconds ($\sim 1/f$), while rotations took several microseconds (Fig. 6(c)). Thus, the assumption that the amount of hysteresis loss is unaffected by the rotation of nanoparticles when reversal is significantly faster is invalid for ferromagnetic nanoparticles in large AC magnetic fields at high frequencies.

Before discussing this novel phenomenon observed at high frequencies, the other important variation in the contour plot of $P_H/(H_{ac} \cdot f)$ (Fig. 5) that occurred because of the ability of the ferromagnetic nanoparticles to rotate at lower frequencies shall be examined. The maximum of $P_H/(H_{ac} \cdot f)$ shifted toward lower H_{ac} below 100 kHz for the rotatable elongated spheroidal nanoparticles, while it stayed between $H_K/2$ and H_K in the non-rotatable case. Figure 7(a) shows the magnetisation curve obtained when $H_{ac} = 16$ kA/m and $f = 30$ kHz. The curve in the rotatable case had an obvious hysteresis loop with a large remanence in the steady state, but there was no hysteresis observed for the non-rotatable situation. Because $H_{ac} = 16$ kA/m is much smaller than $H_K/2$, no reversals of μ occur at any orientation of the easy axis so hysteresis is not observed for the latter case. Figure 7(b) shows the variation of $\langle \cos\theta \rangle$ in the rotatable case, where $\langle \cos\theta \rangle$ ($0 < \theta < \pi/2$) is the mean angle between

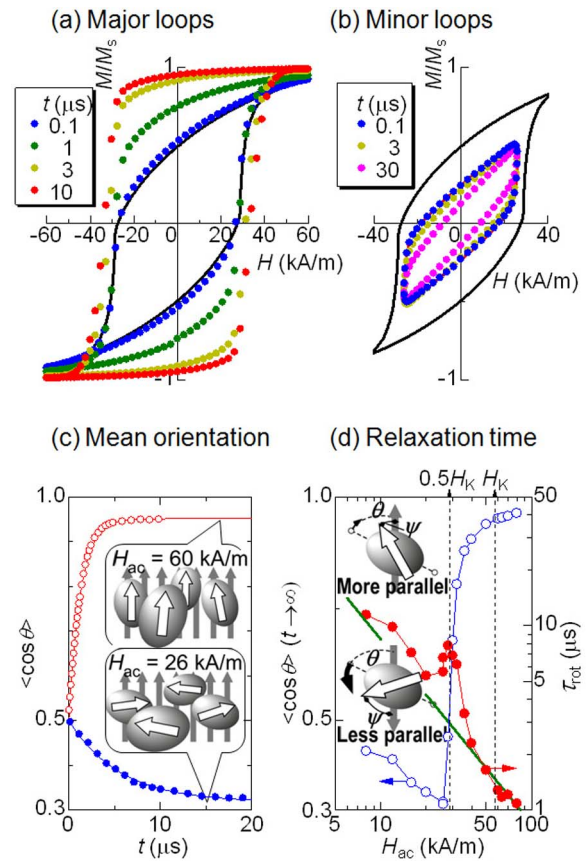


Figure 6 | Magnetic response of elongated spheroidal nanoparticles (typical ferromagnetic nanoparticles) in high-frequency AC fields with $f = 10,000$ kHz and various values of H_{ac} . (a) Transient major hysteresis loops of the rotatable nanoparticles after application of a field with $H_{ac} = 60$ kA/m at $t = 0$. (b) Transient minor loops of the rotatable nanoparticles after application of a field with $H_{ac} = 26$ kA/m at $t = 0$. Solid lines indicate Stoner–Wohlfarth model curves. (c) Relaxation of the mean orientation of the long axis, $\langle \cos\theta \rangle$, (d) steady values of $\langle \cos\theta \rangle$ and relaxation time τ_{rot} . Dashed lines in (d) show $H_K/2$ and H_K ; and the solid line in (d) indicates the reciprocal, $[\omega_H(H = H_{ac} \sin\psi = 0.43)]^{-1}$, of the typical angular velocity of rotation caused by magnetic torque. Orientations are indicated in the inset images.

the magnetic field and the long axes of the spheroidal nanoparticles. Note that $\langle \cos\theta \rangle$ is synchronised with $|M/M_s| = |\cos\psi|$, where ψ is the angle between μ and H . This fact indicates that the hysteresis in the rotatable case (Fig 7(a)) is mainly caused by the rotation of the easy axis where the direction of μ is fixed. Consequently, heat equivalent to the hysteresis loss dissipates even at $H_{ac} < H_K/2$.

For Brownian relaxation, τ_B can be expressed as follows:

$$\tau_B = (3\eta V_H (0.8 + 0.2\kappa) / (k_B T)), \quad (4)$$

where η is the viscosity of the surrounding medium, and V_H is the hydrodynamic volume. In equation (4), the frictional torque for spheroids, described in the Methods section, is considered. For the elongated spheroidal nanoparticle, $(2\pi\tau_B)^{-1}$ is calculated to be 8 kHz. This value is too low to cause the nanoparticle to rotate at 30 kHz. Therefore, Yoshida *et al.*¹³ also took into account the rotation caused by magnetic torque, $\mu(t) \times H(t)$. They concluded that the area of the hysteresis loop was maximised as follows:

$$2\pi f_p = \tau_B^{-1} [1 + 0.07(\mu H_{ac} / k_B T)^2]^{0.5}. \quad (5)$$

The location of the peak in $P_H/(H_{ac} \cdot f)$ below 100 kHz can be explained by this equation, as shown in Fig. 5(b). An expression that describes all of the variation in the position of the primary peak of

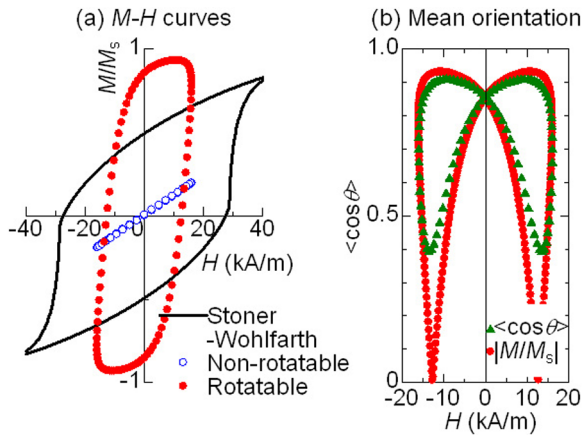


Figure 7 | Magnetic response of elongated spheroidal nanoparticles (typical ferromagnetic nanoparticles) in a low-frequency AC field with $f = 30$ kHz and $H_{ac} = 16$ kA/m. (a) Steady magnetisation curves, where the open and solid symbols show the values of the non-rotatable and rotatable nanoparticles, respectively. Solid lines are Stoner–Wohlfarth model curves. (b) Mean orientation of the long (easy) axis of nanoparticles, $\langle \cos\theta \rangle$.

$P_H/(H_{ac} \cdot f)$ is desirable, and equation (5) can be rewritten as follows:

$$2\pi f_p \sim [\tau_N(H_{ac})]^{-1} + \tau_B^{-1} [1 + 0.07(\mu H_{ac}/k_B T)^2]^{0.5}. \quad (6)$$

This equation is an extended relationship of $\tau^{-1} = \tau_N^{-1} + \tau_B^{-1}$ for a large AC magnetic field. This expression is for the primary maximum; the secondary maximum is discussed later. The second term of equation (6) can be approximated to $0.1\mu H_{ac}/(\eta V_H)$ for $\mu H_{ac}/k_B T \gg (0.07)^{-0.5}$ and $\kappa \sim 1$. On the other hand, $\tau_N(H_{ac})$ of ferromagnetic nanoparticles becomes extremely short only when the energy barrier disappears between $H_K/2$ and H_K . Therefore, the changeover between the two terms in the equation (6) generally occurs at $H_{ac} \approx H_K/2$ and $2\pi f \approx 0.1\mu H_K/(2\eta V_H) = 0.1 K_d V/(\eta V_H)$. For the elongated spheroidal particles ($\kappa = 1.4$, $K_d = 16$ kJ/m³, and $V/V_H = 0.3$) in a liquid phase with $\eta = 1$ mPa·s, the values of H_{ac} and f are 29 kA/m and 76 kHz. Such a changeover around 100 kHz occurs for ferromagnetic nanoparticles of any size, as long as the conditions, K_d , V/V_H , and η , are constant. We must keep in mind that, even when ferromagnetic nanoparticles are large enough for their Brownian relaxation to be negligible, the magnetic torque caused by the large AC magnetic field can easily rotate such nanoparticles in the liquid phase at a time scale of microseconds. This knowledge is helpful when considering the frequency for hyperthermia treatment, even if it is obtained for a simplified system.

In summary, most of the simulated results, including significant variations for ferromagnetic nanoparticles exposed to a low-frequency AC magnetic field, can be explained using the existing models. The two essential exceptions are as follows:

- a secondary maximum in the relaxation loss for SPIONs exposed to a low-frequency AC magnetic field, and
- a shift of the maximum hysteresis loss caused by the ability of typical ferromagnetic nanoparticles exposed to a high-frequency AC magnetic field to rotate.

These novel phenomena are discussed in detail in the following section.

Discussion

The two novel phenomena of rotatable nanoparticles in a large AC magnetic field described above cannot be explained by simple models that consider a linear response of thermodynamic equilibrium states or magnetic field-driven reversals. In this section, we shall further

discuss these atypical responses from the viewpoint of the orientation of the long (easy) axis. First, we begin with the appearance of a secondary maximum of $P_H/(H_{ac} \cdot f)$ near $H_{ac} = 8$ kA/m and $f = 30$ kHz for the nearly spherical nanoparticles (typical SPIONs) where an S-shaped hysteresis loop without remanence was obtained (Fig. 3). We must recall that the variation in $\langle \cos\theta \rangle$ showed butterfly-shaped hysteresis under these conditions. This behaviour explains the atypical magnetic response in the period f^{-1} (33 μ s) (Fig. 3(b)). Initially (at $t = 0$), no magnetisation exists because the occupation probabilities of μ in the two stable directions parallel to the long (easy) axis are equalised in a zero magnetic field. As H increases, the occupation probability in the more stabilised direction immediately increases because of reversals on a time scale of τ_N (≤ 20 ns). The reversed μ in the stabilised direction is not completely parallel to H , $\psi \neq 0$, and the magnetic torque $\mu H \sin\psi$ turns the long (easy) axis towards the direction of the field. If we neglect Brownian torque $\lambda(t)$ (see equation (11) in the Methods section), the angular velocity of the rotation due to magnetic torque can be expressed as

$$\omega_H(H(t), \psi(t)) = [\mu H(t) \sin\psi(t)] / [6\eta V_H (0.8 + 0.2\kappa)]. \quad (7)$$

Hence, $\omega_H(H(t), \psi(t))$ increases in proportion to the field amplitude $H = H_{ac} \sin(2\pi f t)$. For example, $\omega_H(H, \psi = \pi/4)$ is 0.15×10^6 rad/s when $H(t = 1/4f \approx 8 \mu\text{s})$ is 8 kA/m. Therefore, rotation is not negligible in the peak period of the oscillations of H . Subsequently, H decreases to zero at $t = 1/2f \approx 17 \mu\text{s}$, and the occupation probabilities are again equalised because reversal is rapid, so the magnetic torque disappears. Alternatively, the Brownian torque randomises the orientation of the long axis on a time scale of τ_B ($= 8 \mu\text{s}$). Therefore, competition between the magnetic and Brownian torques can cause the butterfly-shaped hysteresis of $\langle \cos\theta \rangle$. Because the equilibrium magnetisation of SPIONs with easy axes parallel to H is higher than that of randomly oriented SPIONs¹⁴, the magnetisation curve shows hysteresis without remanence. Consequently, a secondary maximum appears for the rotatable SPIONs even though $\tau_N \ll \tau_B$.

Next, we investigate the influence of the ability of elongated spheroidal nanoparticles to rotate under a high-frequency AC magnetic field on the shift of the maximum $P_H/(H_{ac} \cdot f)$. As shown in Fig. 6(a) and (b), the magnetisation curves varied after an AC magnetic field was applied at $t = 0$. At the same time, transient variations also occurred in $\langle \cos\theta \rangle$, as shown in Fig. 6(c). In the case of the major loop at $H_{ac} = 60$ kA/m, $\langle \cos\theta \rangle$ gradually increased from 0.5 to 0.95. In other words, the long (easy) axis became oriented towards the direction parallel to H . The characteristic time, τ_{rot} , was estimated to be 1.3 μs using the approximation of exponential decay. Note that the direction of μ is not completely parallel to H even for $H \geq H_K$, even though μ is already reversed for all of the nanoparticles. Because $\sin\psi$ is 0.43 when $\cos\psi$ is 0.9, a large magnetic torque can turn the long axis even if the magnetisation is almost saturated after reversals at $H \sim H_K$. Indeed, τ_{rot} at $H_{ac} \geq H_K$ is comparable to the reciprocal of typical values of $\omega_H(H = H_{ac}, \sin\psi = 0.43)$ (Fig. 6(d)). Therefore, these transient variations can be attributed to the longitudinal orientation being adopted preferentially because of the magnetic torque.

In the minor loops at $H_{ac} = 26$ kA/m, the remanence of the rotatable nanoparticles decreased gradually with time (Fig. 6(b)) and $\langle \cos\theta \rangle$ simultaneously decreased from 0.5 (Fig. 6(c)). The long axis was oriented perpendicular to H during this period, although the longitudinal orientation is preferred when the Zeeman energy is considered. It should be noted that the angle ψ for μ in a stable direction more parallel to H is smaller than that in a metastable direction less parallel to H (Fig. 6(d)). In other words, the magnitude of the magnetic torque toward the longitudinal orientation in the former is weaker than that toward the perpendicular orientation in the latter. This difference makes the orientation of the long axis planar on average, because the stable and metastable states alternate every half period when the reversal of μ is blocked in the minor loop. These arguments suggest that the slowing of the rotation for



$0.5 H_K \leq H_{ac} \leq H_K$ (Fig. 6(d)) can be attributed to compensation between two magnetic torques in the intermediate range. Briefly, in ferromagnetic nanoparticles in the aqueous phase, longitudinal or planar orientations are adopted, irrespective of the free energy, as dissipative structures under a high-frequency AC magnetic field. Consequently, $P_H/(H_{ac} \cdot f)$ increases gradually in major hysteresis loops and decreases in minor loops. These variations cause the maximum of $P_H/(H_{ac} \cdot f)$ to shift towards higher H_{ac} .

Finally, we return to the contour plots of $P_H/(H_{ac} \cdot f)$ in Figs. 2 and 5, and discuss the effect of rotation on the design for maximising $P_H/(H_{ac} \cdot f)$. If a safety limit of $H_{ac} \cdot f < 4.85 \times 10^8 \text{ Am}^{-1} \text{ s}^{-1}$ is applied⁹, then maximum values of $P_H/(H_{ac} \cdot f)$ for rotatable SPIONs and ferromagnetic nanoparticles are obtained at the conditions shown by diamonds A and C in Figs. 2(b) and 5(b), respectively. However, no heat dissipation occurs under the same conditions (A and C) if the rotation of these nanoparticles is blocked (Figs. 2(a) and 5(a)). If a highly amplified AC magnetic field $H_{ac} \cdot f$ of $1.74 \times 10^{10} \text{ Am}^{-1} \text{ s}^{-1}$ is allowed¹⁵, a maximum $P_H/(H_{ac} \cdot f)$ of $3.0 \times 10^{-4} \text{ m}^{-3} \text{ s}^{-2} \text{ A}^{-1}$ (5.2 MW/kg) for the rotatable ferromagnetic nanoparticles can be obtained (diamond D in Fig. 5(b)). However, $P_H/(H_{ac} \cdot f)$ halves when the rotation of these nanoparticles is blocked (Fig. 5(a)) because oriented structures are not formed. In contrast, condition B for the primary maximum of $P_H/(H_{ac} \cdot f)$ in the rotatable SPIONs remains the optimum condition when these nanoparticles cannot rotate (Fig. 2(a) and (b)). This is because the long (easy) axes of SPIONs are randomly oriented in rotatable SPIONs as Brownian torque has more effect than magnetic torque in a weak magnetic field. As demonstrated here, rotation generated by the magnetic torque caused by a large alternating magnetic field greatly affects the conditions for maximising heat dissipation in magnetic nanoparticles.

In this study, we simulated the magnetic responses of superparamagnetic and ferromagnetic magnetite nanoparticles in a large alternating magnetic field. The results show that both the relaxation loss for $\tau_N \ll \tau_B$ and the hysteresis loss at $H_{ac} \approx H_{sw}$ are affected by the formation of dissipative structures because of the ability of nanoparticles to rotate. Consequently, the conditions for maximising heat dissipation depend strongly on the inhomogeneous microviscosity of the surrounding medium.

Compared with the simplified model used for our simulation, actual magnetic nanoparticles used for targeted magnetic hyperthermia treatment are not ideal. For this reason, the factors affecting more realistic situations need to be evaluated. First, the effects of crystalline and surface anisotropy energy are considered. In this case, the potential energy with respect to the direction of μ is complicated. Even if multiple valleys appear in the energy surface, the easy axes are not parallel to H , because the orientations of the nanoparticles are randomised by Brownian torque in the liquid phase. For this reason, slow rotations inevitably occur after fast reversals because of the magnetic torque in an AC magnetic field. These rotations lead to secondary relaxation loss in SPIONs in a low frequency AC magnetic field and shift the hysteresis loss in ferromagnetic nanoparticles in a high frequency AC magnetic field. Next, the variation in the size and shape of actual nanomagnets must be considered. In this case, Néel relaxation times, τ_N , differ significantly because they depend exponentially on the volume of each nanoparticle and the shape anisotropy constant. In contrast, the dependence of frictional torque on the size and shape of nanoparticles is weak. Because the S-shaped hysteresis loop of SPIONs appears in the frequency range of rotation, the secondary loss peak becomes less diffuse compared with the primary relaxation loss peak. For ferromagnetic nanoparticles, the shift of the hysteresis loss at high frequencies should still be significant even if the size of nanoparticles is not uniform, because the anisotropy field is independent of nanoparticle size. Finally, the effect of dipole–dipole interactions is considered, because the density of nanoparticles accumulated in cancer cells might be inhomogeneous if they are trapped at specific sites. In such a case, chain structures of

longitudinally aligned nanoparticles have been conventionally discussed in a magnetic field, although their details are still controversial¹⁶. Our findings illuminate this conventional view, because, in some cases, formation of structures with a planar orientation is predicted even for individual ferromagnetic nanoparticles. In future studies, we will clarify a variety of dissipative structures, which are different from ordinary chains, for interacting nanoparticles. As discussed here, knowledge of the heat dissipation in the non-equilibrium steady states of rotatable nanoparticles is essential for the design of targeted magnetic hyperthermia treatments using large AC magnetic fields.

Methods

Model for the simulation. The magnetic response to an AC magnetic field $H_{ac} \sin(2\pi f \cdot t)$ was simulated for individual superparamagnetic/ferromagnetic magnetite nanoparticles in two extreme cases: non-rotatable nanoparticles strongly anchored to structures resembling organelles and rotatable nanoparticles in an aqueous phase resembling the cytoplasm. We considered the nanoparticles to be monodisperse prolate spheroids with equatorial diameters, $2R_M$, from 12 to 24 nm, and aspect ratios, κ , between 1.1 and 1.4. Because these dimensions are smaller than the typical exchange length of magnetite, 27 nm¹⁷, all of the spins are parallel to one another in each nanoparticle. In other words, we can assume that each nanoparticle has a single magnetic moment $\mu = M_s V$ (coherent rotation/ macro-spin approximations¹⁴), where V is the volume of each nanoparticle, $[(4/3)\pi \kappa R_M^3]$. The magnitude of spontaneous magnetisation $M_s = |M_s|$ was set to the value of bulk magnetite⁶, 450 kA/m, because the dependence of M_s on particle size has not been well established¹⁸.

The magnitude of μ is important; for example, μ of a nanoparticle with $2R_M = 24 \text{ nm}$ and $\kappa = 1.4$ is $1.5 \times 10^5 \mu_B$. Hence, such nanoparticles aggregate because of the large dipole–dipole interactions between μ unless a sufficient non-magnetic surfactant layer exists. The required thickness of this layer is approximately 6 nm for nanoparticles with $2R_M = 24 \text{ nm}$ ¹⁹. Therefore, in our model, we used non-magnetic layers with a thickness δR of 0.5 R_M . Consequently, we can assume that the nanoparticles are uniformly dispersed and do not aggregate. In this case, the typical distance between nanoparticles is $n^{-1/3}$, where n is the number density of nanoparticles. Because the actual mass fraction of nanoparticles accumulated in cancer cells does not exceed 1% (approximately 10 mg/mL), the magnitude of the interaction between nanoparticles with $2R_M = 24 \text{ nm}$, $n \cdot \mu^2/k_B$, is estimated to be less than $1 \times 10^2 \text{ K}$. Consequently, we did not consider minor effects caused by dipole–dipole interactions in our simulations.

In such an individual nanoparticle, the potential energy with respect to the direction of μ , $U(\Omega)$, is given by the following equation:

$$U(\Omega) = U_d(\Omega) + U_c(\Omega) + U_s(\Omega) - \mu \cdot H \quad (8)$$

where $U_d(\Omega)$, $U_c(\Omega)$, $U_s(\Omega)$ and $-\mu \cdot H$ are the shape, crystalline, and surface anisotropy energies, and Zeeman energy, respectively, and Ω is the solid angle of μ . For spheroidal particles, the first term, $U_d(\Omega)$, can be described as $K_d V \sin^2 \phi$ and has two minima separated by the energy barrier $\Delta U = K_d V$, where K_d is the shape anisotropy constant and ϕ is the angle between the long (easy) axis and μ . The magnitudes of K_d , given by $(N_{\text{easy}} - N_{\text{hard}}) \mu_0 / M_s^2$, are 5 and 16 kJ/m³ for spheroidal particles with κ of 1.1 and 1.4, respectively. N_{easy} and N_{hard} are the demagnetising factors for the long and short axes, respectively. In comparison, magnetite has cubic crystalline anisotropy²⁰ and an anisotropy constant $K_1 = -11 \text{ kJ/m}^3$. The energy landscape of cubic anisotropy ($K_1 < 0$) is gentle and ΔU is $(1/12) K_1 V \approx (1 \text{ kJ/m}^3) \cdot V^{1/3}$. Additionally, the effects of surface anisotropy are generally insignificant if the particles are larger than 10 nm¹⁸, although the rationale for this is still unknown. For these reasons, we assumed that the uniaxial shape anisotropy dominated. Consequently, equation (8) can be simplified as follows:

$$U(\phi, \psi) = K_d V \sin^2 \phi - \mu H_{ac} \sin(2\pi f \cdot t) \cos \psi, \quad (9)$$

where ψ is the angle between μ and H (see Fig. 3).

Little is known about the local environments of magnetic nanoparticles accumulated in tumour tissue. For example, nanoparticles coated with dextran are completely immobilised in tissues²¹, whereas dextran nanoparticles appear to be mobile in cells⁵. For this reason, we simulated the magnetic response of the monodisperse spheroidal nanoparticles in the following two extreme cases: non-rotatable nanoparticles with randomly oriented easy axes, and rotatable nanoparticles in a Newtonian fluid with a viscosity η of 1 mPa·s.⁴

Simulation of the reversal. The detailed trajectories of μ in a magnetic field applied at an oblique angle θ to the long easy axis of a spheroidal particle can be precisely simulated by solving the stochastic Landau–Lifshitz–Gilbert equations. However, we are only interested in the reversal of μ once every microsecond because the frequency used for hyperthermia is limited. Therefore, we can use a well-known coarse-grained approach, or “two-level approximation”¹⁴, that considers thermally activated reversals between the meta-stable directions, (ϕ_1, ψ_1) and (ϕ_2, ψ_2) , via the midway saddle point at (ϕ_3, ψ_3) in $U_0(\phi, \psi)$. The reversal probability from (ϕ_1, ψ_1) to (ϕ_2, ψ_2) , v_{12} , is given by $f_0^{-1} \cdot \exp[(U_0(\phi_3, \psi_3) - U_0(\phi_1, \psi_1))/k_B T]$, while the backward reversal



probability v_{21} is $f_0^{-1} \cdot \exp[(U_0(\phi_3, \psi_3) - U_0(\phi_2, \psi_2))/k_B T]$, where f_0 is the attempt frequency of 10^9 s^{-1} .

In the simulation, the time evolution of the occupation probabilities, $p_1(\theta)$, $p_2(\theta) = 1 - p_1(\theta)$, at the two stable directions of a nanoparticle tilted at θ was computed using the following relationship:

$$\partial p_1 / \partial t = v_{21} p_2(\theta) - v_{12} p_1(\theta). \quad (10)$$

$p_1(\theta)$ was simply set to either zero or one when there was only one minimum in $U_0(\phi, \psi)$. This calculation was continued until the transient factors depending on the initial conditions disappeared. The time step Δt was typically $10^{-4}/f_s$ but was shorter when $v_{12}\Delta t$ (or $v_{21}\Delta t$) became large compared with one. At each step, magnetisation was obtained as $[p_1(\theta)\mu\cos\psi_1 + p_2(\theta)\mu\cos\psi_2]\sin\theta d\theta$. Test simulations were performed to check the validity of this method using the same parameters as in an earlier study¹⁴ to calculate the reversals of magnetic nanoparticles in large AC magnetic fields. As detailed in the Supplementary Information, our results agree with the reported behaviour¹⁴.

Simulation of reversal and rotation. The rotation of spheroidal nanoparticles was simulated in synchronisation with the reversal of μ . In Newtonian fluids, the frictional torque for rotation can be expressed as $6\eta V_H \cdot (0.8 + 0.2\kappa) \cdot \omega(t)$,²² where $V_H = [(4/3)\pi \cdot \kappa(R_M + \delta R)^3]$ is the hydrodynamic volume and $\omega(t)$ is the angular velocity of rotation; $d\epsilon/dt = \omega(t) \times \epsilon(t)$, where $\epsilon(t)$ is the unit vector along the long axis of the spheroid; and $\mu(t) \cdot \epsilon(t) = \mu\cos\phi$. Under typical conditions, where $\eta = 1 \text{ mPa}\cdot\text{s}$, $V_H \approx 10^3 \text{ nm}^3$, and $\omega(t) \approx 1 \times 10^5 \text{ rad/s}$, the inertia of the nanoparticle can be neglected (Brownian dynamics simulation). In this inertia-less limit²³, the frictional torque balances with magnetic torque $\mu(t) \times H(t)$ and Brownian torque $\lambda(t)$ as follows:

$$6\eta V_H \cdot (0.8 + 0.2\kappa) \cdot \omega(t) = \mu(t) \times H(t) + \lambda(t) \quad (11)$$

$$\langle \lambda_i(t) \rangle = 0, \quad (12)$$

$$\langle \lambda_i(t_1) \lambda_j(t_2) \rangle = 2k_B T \cdot (6\eta V_H \cdot (0.8 + 0.2\kappa)) \cdot \delta(t_1 - t_2), \quad (13)$$

where $\delta(t_1 - t_2)$ is the Dirac delta function.

At the beginning of the simulation for the rotatable nanoparticles, an assembly of randomly oriented nanoparticles was generated, where their number ensures an optimal compromise between calculation time and precision. Then, the time evolution of the direction of μ and the orientation of the long easy axis were computed by the following steps. (i) Using equation (9), reversible variations of the meta-stable directions ($\phi_i(t)$, $\psi_i(t)$) caused by the latest changes in the field strength and direction of the easy axis were calculated, (ii) $\mu(t)$ at ($\phi_1(t)$, $\psi_1(t)$) was reversed if $x < v_{12}\Delta t$, but otherwise not. In this case, $x \in [0, 1]$ is a pseudorandom number generated by the Xorshift algorithm²⁴. The backward reversal was computed in a similar manner. (iii) Substituting the reversed (or held) $\mu(t)$ into equation (11), $\omega(t)$ was calculated. (iv) $\epsilon(t)$ was finally computed using the relationship $d\epsilon/dt = \omega(t) \times \epsilon(t)$. This calculation was continued until transient factors depending on the initial conditions disappeared. In this simulation, Δt was typically $10^{-4}/f_s$ but was shorter unless $v_{12}\Delta t$, or the changes in $\epsilon(t)$ were sufficiently small compared to one. Magnetisation was obtained as $\Sigma(n_i \mu \cos\psi_i)$ at each step. Test simulations were performed to check the validity of this method. There have been no prior theoretical studies on systems where both reversal and rotation occur simultaneously in a large AC magnetic field. Consequently, comparisons with prior studies were performed under two extreme conditions. In the first case, high viscosities were assumed. Because reversal dominates rotation under these conditions, the results were compared with those reported by Carrey *et al.*¹⁴. In the second case, a high anisotropic field was assumed. Because rotation dominates reversal in this situation, the results were compared with the numerical simulations of Yoshida *et al.*, where nonlinear Brownian rotational relaxation of magnetic fluids with a large excitation field was studied using the Fokker–Planck equation¹³. The results obtained from our simulation of reversal and rotation were consistent with those of earlier studies (Supplementary Figs. S1–S4). Therefore, we can now take a first step toward understanding the roles that rotation of a nanoparticle and reversal of its magnetic moment play together in large AC magnetic fields.

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Author contributions

H. M. performed the simulation. H. M. and B. J. wrote the manuscript.

Additional information

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