

## Auxiliary Density Matrix Methods for Hartree–Fock Exchange Calculations

Manuel Guidon, Jürg Hutter, and Joost VandeVondele\*

*Physical Chemistry Institute, University of Zurich, Winterthurerstrasse 190,  
CH-8057 Zurich, Switzerland*

Received April 27, 2010

**Abstract:** The calculation of Hartree–Fock exchange (HFX) is computationally demanding for large systems described with high-quality basis sets. In this work, we show that excellent performance and good accuracy can nevertheless be obtained if an auxiliary density matrix is employed for the HFX calculation. Several schemes to derive an auxiliary density matrix from a high-quality density matrix are discussed. Key to the accuracy of the auxiliary density matrix methods (ADMM) is the use of a correction based on standard generalized gradient approximations for HFX. ADMM integrates seamlessly in existing HFX codes and, in particular, can be employed in linear scaling implementations. Demonstrating the performance of the method, the effect of HFX on the structure of liquid water is investigated in detail using Born–Oppenheimer molecular dynamics simulations (300 ps) of a system of 64 molecules. Representative for large systems are calculations on a solvated protein (Rubredoxin), for which ADMM outperforms the corresponding standard HFX implementation by approximately a factor 20.

### 1. Introduction

The success of density functional theory (DFT) can be attributed to the fact that it can provide an accurate description of the electronic structure at a moderate computational cost. DFT has become a unique tool to describe systems containing hundreds to thousands of atoms. Not only is it possible to describe molecules in the gas phase, properties of condensed phase systems such as liquids and solids can also be computed. For these systems, using contemporary computer resources, it has become possible to go beyond a static description of matter, and finite temperature effects can be included directly through ab initio molecular dynamics (MD) simulations. Large-scale, condensed phase and dynamical simulations have mostly adopted a relatively simple form for the exchange and correlation functional, namely, the semilocal generalized gradient approximation (GGA). However, it becomes increasingly clear that an improved description of the electronic structure, and thus more accurate results, can only be obtained by functionals that go beyond the GGA form and incorporate a nonlocal term such as Hartree–Fock exchange (HFX). The

computational cost of these nonlocal terms is typically much larger than that of the local terms. Consequently, there is significant interest in finding efficient approaches to deal with these nonlocal forms.

The efficiency of a HFX calculation depends strongly on the algorithm employed. A straightforward implementation based on localized basis sets scales with the fourth power of the system size. However, integral screening<sup>1</sup> reduces the scaling with system size to quadratic and, for short-range operators, such as screened<sup>2</sup> or truncated exchange,<sup>3–6</sup> to linear scaling. Nonmetallic systems furthermore allow for a screening on the density matrix,<sup>7</sup> which leads to linear scaling also for long-range operators. Using these techniques, HFX can be evaluated also for condensed phase systems containing a few thousand atoms<sup>3</sup> and can be used to perform ab initio molecular dynamics simulations.<sup>8</sup> Despite the favorable scaling with system size, HFX calculations scale very poorly with basis set quality. This is an important issue, since high-quality results not only require accurate functionals but also good basis sets. There are several reasons why the cost of HFX depends very strongly on the basis employed. Indeed, even in a linear scaling code, the cost increases with the fourth power of the number of (primitive) basis functions

\* Corresponding author e-mail: Joost.VandeVondele@pci.uzh.ch.

**Table 1.** Impact of the Basis Set Quality for the Wavefunction Optimization of a 20 Water Cluster<sup>a</sup>

basis	$\kappa(S)$	threshold	cost [ERIs]	cost [s]
3-21G*	$4.9 \times 10^1$	$1.0 \times 10^{-4}$	$2.3 \times 10^7$	0.06
6-31G**	$2.1 \times 10^2$	$1.0 \times 10^{-5}$	$5.2 \times 10^8$	0.35
6-311G++G**	$1.2 \times 10^5$	$1.0 \times 10^{-7}$	$1.1 \times 10^{10}$	11.71
pc-0	$5.2 \times 10^1$	$1.0 \times 10^{-4}$	$1.7 \times 10^7$	0.07
pc-1	$4.5 \times 10^3$	$1.0 \times 10^{-5}$	$4.4 \times 10^8$	0.50
pc-2	$5.7 \times 10^5$	$1.0 \times 10^{-7}$	$2.0 \times 10^{10}$	11.21
aug-pc-1	$1.4 \times 10^6$	$1.0 \times 10^{-8}$	$5.0 \times 10^{10}$	53.23
aug-pc-2	$3.9 \times 10^8$	$1.0 \times 10^{-9}$	$1.5 \times 10^{12}$	766.92
def2-QZVP	$7.1 \times 10^4$	$1.0 \times 10^{-8}$	$3.2 \times 10^{11}$	127.16
aug-def2-QZVP	$8.5 \times 10^5$	$1.0 \times 10^{-8}$	$6.2 \times 10^{11}$	331.61

<sup>a</sup> The condition number  $\kappa(S)$  of the overlap matrix determines the maximal possible screening threshold. The latter needs to be chosen more tightly, if  $\kappa(S)$  gets large. This is reflected in the cost of a calculation, which is given once by the number of Cartesian four-center electron repulsion integrals (ERIs) that need to be evaluated and the time in seconds that is spent in building the Fock matrix in the first self-consistent field (SCF) step. 3-21G\*, 6-31G\*\*, and 6-311G++G\*\* refer to basis sets by Pople and co-workers.<sup>10–13</sup> The polarization consistent (pc) basis sets have been developed by Jensen,<sup>14–16</sup> the def2-QZVP basis by Ahlrichs and Weigend.<sup>17</sup> Timings are obtained on 128 cores of a CRAY-XT5.

per atom. Basis sets with a high  $l$ -quantum number (polarization functions) are therefore costly, as the number of basis functions per atom grows quadratically with  $l$ . Heavily contracted basis functions, such as the molecularly optimized basis sets proposed in ref 9, are expensive since for each quartet of basis functions a very larger number of primitive integrals needs to be considered. Very flexible basis sets, or basis sets with diffuse primitives, are costly for several reasons. First, diffuse primitives are nonzero in a larger part of space, and thus screening becomes less efficient. This is particularly important in condensed phase systems, where periodic boundary conditions provide a potentially unlimited number of interacting atomic sites. Second, uncontracted diffuse primitives influence the condition number of the overlap matrix ( $S$ ) strongly, and a poor condition number in turn implies that a tighter screening threshold has to be employed<sup>3</sup> to obtain a stable self-consistent calculation. Third, the sparsity of the matrix representation of the density matrix ( $P$ ) also depends strongly on the condition number of  $S$ , making density matrix screening less efficient for poorly conditioned basis sets. Table 1 illustrates this problem by providing costs and maximal thresholds needed in order to get converged results for a water cluster containing 20 water molecules.

Clearly, a technique which reduces the impact of the basis set on the computational cost is a significant progress.

During the past decade, much effort has been invested in solving this problem, and many different techniques have been proposed. Among them are methods that apply an approximate resolution of identity, for example, RI<sup>18</sup> or Cholesky decomposition.<sup>19</sup> These schemes rely on the introduction of auxiliary basis functions in terms of which the four center integrals can be approximated by corresponding two- and three-center terms. In order to improve efficiency, Sodt and Head-Gordon<sup>20</sup> developed a local variant of RI, atomic resolution of identity (ARI). A slightly different post-Hartree–Fock approach in a dual basis was

introduced in ref 21, where a reference calculation in a small basis set is perturbatively corrected to a large basis set. A different approximation for the two-electron integrals has been proposed by Friesner<sup>22</sup> and has been termed the “pseudo-spectral” method. Recently, Neese et al.<sup>23</sup> presented an algorithm called COSX that is a combination of semi-numerical methods and RI. Furthermore, there exist several schemes to achieve linear scaling in the context of plane wave basis sets such as the multiwavelet-based ansatz of Harrison et al.<sup>24</sup> or FFT-based algorithms as presented in refs 25–27.

In this work, we propose to employ an auxiliary density matrix to evaluate the expensive nonlocal part of the functional, while all other energy components are computed with the primary (original) density matrix. The auxiliary density matrix will be constructed in a way that allows for a rapid evaluation of the HFX energy, using any algorithm, including traditional or linear scaling approaches. In order to ensure that the quality of the calculation is influenced as little as possible by the quality of the auxiliary density matrix, a correction term is added to the exchange and correlation functional. On the basis of a GGA for exchange, this correction takes the difference between auxiliary and primary density matrices into account. All terms of the resulting density functional are straightforward to compute, but there is considerable freedom in how to obtain from a given primary density matrix a suitable auxiliary density matrix. In this paper, various procedures are discussed and tested. Tests are presented in section 3 and include gas phase thermochemistry; basis set superposition error; and electronic structure including band gaps, large systems, and liquid water. The theory is introduced in the following section, but for mathematical derivations and technical details, we refer to the Appendix.

## 2. Theory

**2.1. Basic Concepts.** In Kohn–Sham DFT, the total energy of a system consisting of  $N_e$  electrons can be written in terms of the electron density

$$\rho(\mathbf{r}) = \sum_{i=1}^{N_e} |\psi_i(\mathbf{r})|^2 \quad (1)$$

where  $\psi_i$  denotes the single particle wave functions, which are assumed to be real-valued. The total energy is then expressed in terms of a functional of the electron density as

$$E[\rho] = T_s[\rho] + J[\rho] + E_{xc}[\rho] + \int v(\mathbf{r}) \rho(\mathbf{r}) d\mathbf{r} \quad (2)$$

with the standard abbreviations for kinetic, Hartree, and exchange-correlation energy and the part due to the external potential. In hybrid DFT, the exchange-correlation functional is augmented by a certain fraction of Hartree–Fock exchange based on the wave functions  $\{\psi_i\}$

$$E_{xc}[\rho] = \alpha E_x^{\text{HFX}}[\{\psi_i\}] + (1 - \alpha) E_x^{\text{DFT}}[\rho] + E_c^{\text{DFT}}[\rho] \quad (3)$$

where  $\alpha$  denotes the fraction of HFX and  $E_x$  and  $E_c$  are the density functionals for exchange and correlation, respectively. In the presence of an atomic centered basis set  $\{\phi_\mu(\mathbf{r})\}$

$$\psi_i(\mathbf{r}) = \sum_{\mu} C^{\mu i} \phi_{\mu}(\mathbf{r}) \quad (4)$$

the Hartree–Fock exchange energy can be expressed in terms of a density matrix and two-electron integrals (ERIs)

$$E_x^{\text{HFX}}[P] = -\frac{1}{2} \sum_{\lambda\sigma\mu\nu} P^{\mu\sigma} P^{\nu\lambda} (\mu\nu|\lambda\sigma) \quad (5)$$

where the density matrix elements  $P^{\mu\nu}$  are obtained from the molecular (MO) coefficients as

$$P^{\mu\nu} = \sum_i C^{\mu i} C^{\nu i} \Leftrightarrow P = CC^T \quad (6)$$

and the ERIs are defined as

$$(\mu\nu|\lambda\sigma) = \int \int \phi_{\mu}(\mathbf{r}_1) \phi_{\nu}(\mathbf{r}_1) g(|\mathbf{r}_2 - \mathbf{r}_1|) \phi_{\lambda}(\mathbf{r}_2) \phi_{\sigma}(\mathbf{r}_2) d\mathbf{r}_1 d\mathbf{r}_2 \quad (7)$$

with the interaction potential  $g(r)$  that is Coulombic ( $1/r$ ) in standard Hartree–Fock theory. The fourth order scaling of HFX with basis set size can be directly inferred from eq 5.

By introducing an auxiliary density matrix  $\hat{P} \approx P$  that is either smaller in size or more rapidly decaying than the original one, the evaluation of HFX can be sped up significantly. The HFX energy can be written as

$$\begin{aligned} E_x^{\text{HFX}}[P] &= E_x^{\text{HFX}}[\hat{P}] + (E_x^{\text{HFX}}[P] - E_x^{\text{HFX}}[\hat{P}]) \\ &\approx E_x^{\text{HFX}}[\hat{P}] + (E_x^{\text{DFT}}[P] - E_x^{\text{DFT}}[\hat{P}]) \end{aligned} \quad (8)$$

The assumption behind this approximation is that *the difference* in the exchange energy between primary and auxiliary density matrices is well captured by a GGA, even in those cases where GGA exchange and HFX might be qualitatively different. Equation 8 amounts to computing the HFX energy with an auxiliary density matrix, while a GGA correction is introduced which takes the difference between auxiliary and primary density matrices into account. As shown in section 3, applying this correction indeed improves upon uncorrected results. Clearly, our approach yields the original HFX energy as the quality of either the auxiliary density matrix or the correcting functional improves. In this work, we have based the GGA correction on PBE exchange<sup>28,29</sup> and have not explored other parametrizations or other functionals. The introduction of eq 8 in hybrid density functionals is natural and straightforward and, usually, because only a fraction of exchange is needed, will introduce a smaller error. If hybrid functionals employ a non-Coulombic operator, the exchange functional needs to be chosen consistently with the shape of the interaction potential ( $g(r)$ ) in the ERI calculation. Currently, the GGA correction for the standard Coulomb potential, the short-range (erfc) and the truncated Coulomb potential have been implemented and tested.

**2.2. Auxiliary Density Matrices.** The performance and accuracy of the ADMM scheme depends on how the auxiliary density matrix is constructed, and various approaches seem possible. In this section, we present methods that either rely on the use of an auxiliary basis set or directly manipulate the sparsity of the density matrix.

The size of  $P$  obtained from a high-quality primary basis set (PBS)  $\{\phi_{\mu}(\mathbf{r})\}$  can be reduced by introducing an auxiliary basis set (ABS)  $\{\hat{\phi}_{\mu}(\mathbf{r})\}$  for the description of the underlying wave function

$$\hat{\psi}_i(\mathbf{r}) = \sum_{\mu} \hat{C}^{\mu i} \hat{\phi}_{\mu}(\mathbf{r}) \quad (9)$$

i.e.

$$\hat{P}^{\mu\nu} = \sum_i \hat{C}^{\mu i} \hat{C}^{\nu i} \Leftrightarrow \hat{P} = \hat{C}\hat{C}^T \quad (10)$$

An optimal value for the MO coefficients can be obtained by requiring that the square difference for the occupied wave functions in ABS and PBS representation is minimized

$$\min_{\hat{C}} = \sum_i \int (\psi_j(\mathbf{r}) - \hat{\psi}_j(\mathbf{r}))^2 d\mathbf{r} \quad (11)$$

This yields the following expression for the auxiliary MO coefficients

$$\hat{C} = AC \quad (12)$$

where  $A$  is defined as the projector between the two basis sets

$$A = \hat{S}^{-1}Q \quad (13)$$

with the overlap matrices

$$\hat{S}_{nn'} = \int \hat{\phi}_n(\mathbf{r}) \hat{\phi}_{n'}(\mathbf{r}) d\mathbf{r} \text{ and } Q_{nm} = \int \hat{\phi}_n(\mathbf{r}) \phi_m(\mathbf{r}) d\mathbf{r} \quad (14)$$

A slightly more complicated formula is obtained when the auxiliary wave functions are required to minimize eq 11 subject to the constraint that they remain orthonormal. This constraint can be enforced introducing Lagrangian multipliers ( $\Lambda_{kl}$ ) in eq 11 as

$$\min_{\tilde{C}} \left[ \sum_j \int (\psi_j(\mathbf{r}) - \tilde{\psi}_j(\mathbf{r}))^2 d\mathbf{r} + \sum_{k,l} \Lambda_{kl} \left( \int \tilde{\psi}_k(\mathbf{r}) \tilde{\psi}_l(\mathbf{r}) d\mathbf{r} - \delta_{kl} \right) \right] \quad (15)$$

The coefficients  $\tilde{C}$  that minimize this expression can be obtained as

$$\tilde{C} = \hat{C}\Lambda^{-1/2} \text{ with } \Lambda = \hat{C}^T \hat{S} \hat{C} \quad (16)$$

where  $\hat{C}$  is defined by eq 12.

Of course, there is significant freedom in selecting the auxiliary basis set, and the choice need not to be homogeneous in space. For example, for large systems with a chemically active region, such as enzymes, it is natural to retain the high-quality primary basis where exchange matters most, while a lower-quality auxiliary basis can be used for the bulk. Furthermore, note that the explicit shape of the basis functions (Gaussian functions, Slater functions, etc.) is not important and indeed need not to be the same in the auxiliary and primary basis sets. The method thus provides an interesting approach for computing exchange contributions

in programs that do not employ Gaussian basis functions and for which the calculation of exchange is relatively difficult.

The two different sets of MO coefficients,  $\hat{C}$  and  $\tilde{C}$ , correspond to two different density matrices:

$$\tilde{P} = \tilde{C}\tilde{C}^T = \hat{C}\Lambda^{-1}\hat{C}^T \quad (17)$$

and

$$\hat{P} = \hat{C}\hat{C}^T = APA^T \quad (18)$$

that can be used as an auxiliary density matrix. We will refer to the first as purified wave function fitting or ADMM1 and to the second as nonpurified wave function fitting or ADMM2 (see section 2.3 for an explanation of the nomenclature).

A strategy directly aimed at obtaining a sparse auxiliary density matrix relies on a blocking of the primary density matrix. This strategy is applicable if the system of interest can be divided into subsystems that have no important exchange interactions beyond what is captured with a GGA. In this case, the nonrelevant blocks in the auxiliary density matrix can just be zeroed, and to some extent this method can be considered a subsystem-based neglect of diatomic differential overlap (NDDO). If all intersubsystem blocks are zeroed, the approximate density matrix will be positive definite; however, we employ the slightly generalized form of the auxiliary density matrix as

$$\hat{P} = P \otimes B \quad (19)$$

where  $B$  is a blocking matrix with  $B_{ij} \in \{1, 0\}$  and  $\otimes$  denotes the Hadamard product of two matrices. In this case,  $B$  can reflect the molecular topology and allow for connections between subsystems. In the general case, the resulting  $\hat{P}$  need not to be positive definite. This method of obtaining an auxiliary density matrix will be referred to as blocking or ADMM3 in the following.

**2.3. Density Matrix Purification.** As already mentioned, an approximate density matrix might not fulfill the properties of a pure density matrix:

$$P = P^T \quad (20)$$

$$PSPS = PS \quad (21)$$

$$\text{tr}(PS) = N_e \quad (22)$$

that is, symmetry, idempotency, and particle conservation. For the three approximations mentioned in the previous section, all three conditions are only fulfilled by ADMM1, i.e., the purified wave function fitting scheme. ADMM2 and the block diagonal version of ADMM3 fulfill a property of ensemble averaged (finite temperature) density matrices, i.e., that the eigenvalues of  $\hat{P}$  are bounded by 0 and 1, which is a relaxed version of the idempotency condition. In order to compute a GGA correction for exchange, it is essential that the approximate density matrix is at least positive semidefinite. Fortunately, there exist purification algorithms that can restore the idempotency of an approximate density matrix.

Well known is the McWeeny purification algorithm,<sup>30</sup> which, in the presence of an overlap matrix, is defined as follows

$$\bar{P}_{n+1} = f(\bar{P}_n) = 3\bar{P}_n S \bar{P}_n - 2\bar{P}_n S \bar{P}_n S \bar{P}_n \quad (23)$$

for an initial guess  $\bar{P}_0 = \hat{P}$ . The pure density matrix is then given as

$$\tilde{P} = \lim_{n \rightarrow \infty} \bar{P}_n \quad (24)$$

An interesting property of this algorithm is that it can be implemented in a linear scaling fashion.<sup>31,32</sup> In the current context, we prefer an extension of the McWeeny procedure based on a Cauchy integral representation<sup>33</sup>

$$\tilde{P} = S^{-1} \left[ \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{S^{-1}z - \hat{P}} dz \right] S^{-1} \quad (25)$$

where  $\Theta(z)$  denotes the Heaviside function. This scheme yields a pure density matrix for all input matrices, is noniterative, but is not easily incorporated in a linear scaling procedure. Through eq 25, a purified  $\tilde{P}$  can be interpreted as a matrix functional of a nonpure  $\hat{P}$ . This is an important property, which will be used to derive an expression for the Kohn–Sham matrix in the following section. Equation 25 can be easily computed using basic linear algebra techniques as

$$\tilde{P} = S^{-1} R L R^T S^{-1} \quad (26)$$

where  $R$  is the matrix of eigenvectors of the generalized eigenvalue problem

$$\hat{P}R = S^{-1} R \lambda \quad (27)$$

and  $L$  is the diagonal matrix  $L_{ii} = \Theta(\lambda_i - 0.5)$  with the corresponding eigenvalues  $\lambda_i$ . At this point, and as shown in the Appendix, we remark that purification by eq 25 of the density matrix obtained from nonpurified wave function fitting (ADMM2) exactly yields the density matrix derived from the purified wave function fitting (ADMM1).

**2.4. Kohn–Sham Matrix and the SCF Procedure.** In a standard SCF procedure, an improved density matrix is obtained from a diagonalization of the Kohn–Sham matrix. The Kohn–Sham matrix itself is obtained as the derivative of the total energy with respect to the density matrix. In ADMM, the total energy can be considered to consist of two parts, one part depending explicitly on the primary density matrix ( $E[P]$ ) and one part depending explicitly on the auxiliary matrix ( $\tilde{E}[\tilde{P}]$ ):

$$E_{\text{total}} = E[P] + \tilde{E}[\tilde{P}] \quad (28)$$

The Kohn–Sham matrix associated with this expression

$$K_{\text{total}} = \frac{dE[P]}{dP} + \frac{d\tilde{E}[\tilde{P}]}{dP} = K + \frac{d\tilde{E}[\tilde{P}]}{dP} \quad (29)$$

contains one nontrivial term

$$\frac{d\tilde{E}[\tilde{P}]}{dP} = \frac{d\tilde{E}}{dP} = \frac{d\tilde{E}}{d\tilde{P}} \frac{d\tilde{P}}{dP} = \tilde{K} \frac{d\tilde{P}}{dP} \quad (30)$$



where  $\tilde{K}$  is the Kohn–Sham matrix constructed from the purified density matrix.  $(d\tilde{P})/(dP)$  is readily evaluated for wave function fitting and blocking, while  $(d\tilde{P})/(d\tilde{P})$  can be obtained through the Cauchy integral eq 25. We obtain in the case of purified wave function fitting (for details, see Appendix part B)

$$\frac{d\tilde{E}}{dP} = A^T R [(R^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R) \otimes M] R^T A \quad (31)$$

with  $R$  as defined above, and

$$M_{kj} = \begin{cases} \frac{\Theta(\lambda_k - 0.5) - \Theta(\lambda_j - 0.5)}{\lambda_k - \lambda_j} & k \neq j \\ \delta(\lambda_k - 0.5) & k = j \end{cases} \quad (32)$$

In the Appendix, computationally more efficient expressions are presented for optimization schemes that only require the derivative of the energy with respect to the MO coefficients ( $dE/dC$ ) or that exploit the special structure of  $\tilde{P}$ .

At this point, it is important to point out that the eigenvalues of the Kohn–Sham matrix in ADMM might be very different from the eigenvalues of the Kohn–Sham matrix in the primary basis. This is not an indication of the inaccuracy of the scheme, nor is it a problem for the SCF procedure, but it is related to the fact that purification as part of the energy functional partially accounts for the orthonormality constraint of the wave function (see also Appendix part G). In order to use the eigenvalues of the ADMM Kohn–Sham matrix directly as orbital energies, e.g., to calculate the band gaps of a system, an ADMM scheme without purification needs to be employed. For the nonpurified wave function fitting (ADMM2), the corresponding Kohn–Sham matrix is given by

$$K_{\text{total}} = K[P] + A^T \hat{K} A \quad (33)$$

where  $\hat{K}$  is built from  $\hat{P}$ . This simple expression suggests an expression for use with purified wave function fitting (ADMM1); i.e., orbital energies can be obtained from eigenvalues of

$$K_{\text{total}} = K[P] + A^T \tilde{K} A \quad (34)$$

where  $\tilde{K}$  is constructed from  $\tilde{P}$ . We will show in section 3.6 that this expression can be accurate.

### 3. Assessment and Validation of the Method

**3.1. Computational Details.** All algorithms have been implemented in CP2K,<sup>34</sup> a freely available molecular simulation package. CP2K is well suited for these calculations as the density functional module Quickstep,<sup>35</sup> implements a linear scaling and fast scheme for calculations on the basis of local functionals. Indeed, the Gaussian and plane waves (GPW) scheme<sup>36</sup> and its augmented (GAPW) variant<sup>37</sup> provide an efficient method to evaluate the Coulomb energy for pseudopotential and all-electron calculations, respectively. These approaches use Fourier transform based techniques, i.e., a plane wave auxiliary basis, and scale favorably with basis set size. Recently, an efficient, massively parallel and

linear scaling implementation of Hartree–Fock exchange has been incorporated into the CP2K code.<sup>3,8</sup> Despite this efficiency, calculations including HFX and employing high-quality basis sets are at least one order of magnitude more expensive than calculations based on GGAs. ADMM aims at resolving this issue. Currently, the GGA correction term required for ADMM has only been implemented for use with the GPW method, and consequently all calculations are based on Goedecker, Teter, Hutter (GTH) pseudopotentials.<sup>38</sup> Pseudopotentials<sup>39</sup> constructed for the PBE functional have been used throughout. This is an approximation that poorly describes core–valence exchange and that is known to introduce errors in excess of 0.1 eV in the computation of band gaps<sup>40,41</sup> but appears to give reasonable results for ground state properties (see, e.g., section 3.2). In this work, both ADMM and the standard HFX implementation employ the same pseudopotential approximation, so that a meaningful comparison can be made. The all-electron implementation of ADMM and the development of pseudopotentials for hybrid functionals are beyond the scope of the current work.

Calculations based on pseudopotentials use split valence Gaussian basis sets as discussed in ref 35, the fully contracted molecularly optimized (MOLOPT) basis sets discussed in ref 9, or a reference basis (GTH-def2-QZVP), which combines the pseudoatomic orbitals of the MOLOPT basis, with uncontracted valence, and polarization exponents of the Ahlrichs quadruple- $\zeta$  (aug-)def2-QZVP<sup>17</sup> basis set. The latter basis can be considered close to the basis set limit. The choice of auxiliary basis for the ADMM method will in general be dictated by accuracy and performance requirements of a particular calculation. Indeed, the gain in performance for the hybrid part of the calculation might allow for better primary basis sets, or large systems can be simulated by more aggressively using a smaller auxiliary basis. Here, we are interested in exploring the accuracy of relatively small auxiliary basis sets, of which a library of eight different basis sets per atom have been constructed. This basis employs three Gaussian exponents for the valence orbitals, optimized in atomic calculations. We will refer to this uncontracted basis, without polarization functions, as FIT3, while a contraction of this basis (to double- $\zeta$  quality) is referred to as cFIT3. In order to improve accuracy, polarization functions from the standard 6-31G\*\* basis sets have been added, yielding pFIT3 and cpFIT3 basis sets. Finally, an augmented version has been constructed by adding a “diffuse” function (typical exponents are 0.03 for hydrogen and 0.09 for oxygen), yielding aug-FIT3, aug-cFIT3, aug-pFIT3, and aug-cpFIT3 (see Table 2).

**3.2. GMTKN24 Database.** The GMTKN24 database is a compilation of 24 different chemically relevant benchmarks collected and established by Goerigk and Grimme.<sup>42,43</sup> It is based on 1049 atomic and molecular single point energies that are combined to yield 731 relative energies. These energies can be compared to available benchmark data, derived from either theory or experiment. In order to judge the quality of a computational method using a single number, the authors defined a weighted total mean absolute deviation (WTMAD) that combines all mean absolute deviations (MADs). This convenient measure is adopted here to judge

**Table 2.** Cost for Using the FIT3 Basis Sets on a Cluster of 20 Water Molecules<sup>a</sup>

basis	$\kappa(S)$	threshold	cost [ERIs]	cost [s]
cFIT3	$1.3 \times 10^2$	$1.0 \times 10^{-4}$	$1.8 \times 10^7$	0.08
FIT3	$1.5 \times 10^2$	$1.0 \times 10^{-4}$	$1.6 \times 10^7$	0.10
cpFIT3	$1.7 \times 10^2$	$1.0 \times 10^{-4}$	$1.0 \times 10^8$	0.13
pFIT3	$2.3 \times 10^2$	$1.0 \times 10^{-4}$	$9.0 \times 10^7$	0.15
aug-cFIT3	$5.5 \times 10^4$	$1.0 \times 10^{-7}$	$1.1 \times 10^9$	3.38
aug-FIT3	$6.1 \times 10^4$	$1.0 \times 10^{-7}$	$1.1 \times 10^9$	4.14
aug-cpFIT3	$5.9 \times 10^4$	$1.0 \times 10^{-7}$	$3.2 \times 10^9$	6.09
aug-pFIT3	$6.4 \times 10^4$	$1.0 \times 10^{-7}$	$3.0 \times 10^9$	6.78

<sup>a</sup> For comparison and details, see Table 1.**Table 3.** WTMAD and WTMAD<sub>ref</sub> in kcal/mol for the GMTKN24 Database and the PBE0 Functional<sup>a</sup>

method	PBS	ABS	WTMAD	WTMAD <sub>ref</sub>
STD	GTH-def2-QZVP		5.0	0.0
	FIT3		15.3	10.8
	pFIT3		7.1	4.2
ADMM1	GTH-def2-QZVP	cFIT3	5.3	1.0
	GTH-def2-QZVP	FIT3	5.3 (6.1)	0.7 (1.8)
	GTH-def2-QZVP	cpFIT3	5.0	0.7
	GTH-def2-QZVP	pFIT3	4.9 (5.5)	0.5 (1.2)
ADMM2	GTH-def2-QZVP	cFIT3	5.3	1.1
	GTH-def2-QZVP	FIT3	5.3	0.8
	GTH-def2-QZVP	cpFIT3	4.9	0.7
	GTH-def2-QZVP	pFIT3	4.9	0.5

<sup>a</sup> Whereas WTMAD refers to the weighted mean absolute deviation with respect to experimental and theoretical benchmark results, WTMAD<sub>ref</sub> refers to deviations with respect to PBE0 reference results obtained using a standard HFX implementation and the high-quality GTH-def2-QZVP basis. Standard (STD) HFX calculations with the GTH-def2-QZVP, FIT3, and pFIT3 basis sets are employed to establish the quality of these basis sets as a primary basis set (PBS). Wavefunction fitting results with purification (ADMM1) and without purification (ADMM2) are provided using four different auxiliary basis sets (ABS), while the GTH-def2-QZVP has been employed as a primary basis in all of these cases. The results in parentheses have been obtained using ADMM, but ignoring the GGA correction.

the quality of the wave function fitting methods ADMM1 and ADMM2 for various basis sets. Results, summarized in Table 3, are based on the hybrid PBE0 functional<sup>44–46</sup> without empirical dispersion correction.<sup>47</sup> In a first step, reference results using a standard HFX implementation have been generated for the GTH-def2-QZVP basis. As in ref 42, an augmented basis set has been used for two of the subsets. ADMM results can be directly compared to these reference results, and deviations with respect to this data is referred to as WTMAD<sub>ref</sub>. WTMAD without a subscript is used to refer to the deviations with respect to the experimental and theoretical benchmark results. Second, to quantify the expected poor quality of the FIT3 family as a primary basis, these basis sets have been used with a standard HFX implementation. These calculations yield a WTMAD<sub>ref</sub> in the range 4–11 kcal/mol and WTMADs in the range 7–15

kcal/mol, far worse than the typical performance of local functionals, with a good basis set, on this database.<sup>42</sup> Third, ADMM calculations have been performed using the FIT3 family as auxiliary basis sets. Whenever the primary basis is augmented, an augmented auxiliary basis has been used as well. The results obtained with ADMM are in very close agreement with the reference calculations. In particular, both ADMM1 and ADMM2 using the better auxiliary basis set (pFIT3 or cpFIT3) are basically indistinguishable in terms of error with respect to the benchmark data (WTMAD) and have an error of less than 1 kcal/mol compared to the reference run (WTMAD<sub>ref</sub>). In the case of FIT3, ADMM results improve by 10 kcal/mol as compared to standard HFX calculations with the same basis. In Table 3, it is also shown that including the GGA correction term in ADMM more than halves the WTMAD<sub>ref</sub>, thus emphasizing the benefit of the correction term. Finally, we observe that ADMM1 and ADMM2 perform equally well, suggesting that in this case the purification is not essential. In the cases we have verified,  $\hat{P}$  had eigenvalues close to 0 and 1, even for the small cFIT3 basis. These data show that results of def2-QZVP quality can be obtained at a cost similar to 6-31G\*\*. The relatively modest cost of computing the full database with ADMM has been exploited to benchmark the quality of the PBE0-TC-LRC functional proposed in ref 3. This functional uses a truncated operator for the calculation of exchange but, like HSE,<sup>48,49</sup> corrects for the long-range part using a density functional. As shown in ref 3, the PBE0-TC-LRC is useful in the condensed phase but can also reduce the computational cost for (large) molecules. In Table 4, the effect of varying the range of exchange has been studied systematically, using ADMM1 with the pFIT3 basis, for PBE0-TC-LRC functionals including 20% and 25% of nonlocal exchange.

These WTMADs clearly show that the range of the truncated operator can be reduced to 2 Å without affecting the quality of the results. The lowest WTMAD, slightly smaller than the WTMAD for PBE0, is found for 20% nonlocal exchange and a range of 2.5 Å.

**3.3. Basis Set Superposition Error.** In this section, the impact of ADMM on the basis set superposition error (BSSE) for the water dimer is investigated. Indeed, the BSSE is a concern as soon as small, lower-quality basis sets are employed. Here, it is shown that small auxiliary basis sets introduce only a moderate BSSE, especially if compared to the BSSE in standard HFX calculations with the same basis. In order to quantify the BSSE, the counterpoise correction<sup>50</sup> has been computed for a water dimer at fixed equilibrium geometry, for various methods. These results are summarized in Table 5. As expected, using the nonaugmented FIT3 basis sets as primary bases leads to errors of approximately 3 kcal/mol. This error is large when compared to a basis using

**Table 4.** WTMADs in kcal/mol for the GMTKN24 Database and the PBE0\_TC\_LRC Functional for Several Different Cutoff Radii in the Range of 0.5–6.0 Å<sup>a</sup>

$\alpha$	0.50	0.75	1.00	1.25	1.50	1.75	2.00	2.25	2.50	3.00	4.50	6.00	$\infty$
0.20	5.5	5.3	5.6	5.6	5.3	5.0	4.9	4.8	4.7	4.7	4.8	4.8	N/A
0.25	5.5	5.5	6.1	6.3	5.9	5.4	5.2	5.0	4.9	4.9	4.9	4.9	5.0

<sup>a</sup> The column labeled with  $\infty$  refers to the standard PBE0 hybrid functional. All calculations have been performed twice for different fractions of Hartree–Fock exchange,  $\alpha = 0.2$  and  $\alpha = 0.25$ .

**Table 5.** Counterpoise Corrections in kcal/mol to the PBE0 Binding Energy of a Water Dimer<sup>a</sup>

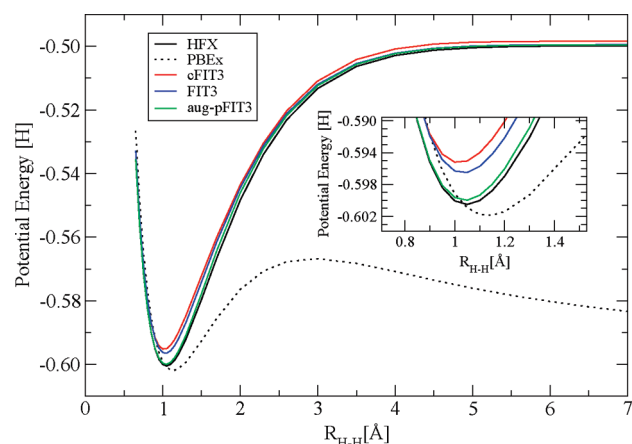
basis set	STD	ADMM1	ADMM2
cFIT3	-3.112	0.771	0.223
FIT3	-3.128	0.520	-0.006
cpFIT3	-3.468	0.882	0.248
pFIT3	-3.448	0.604	0.004
aug-cFIT3	-1.889	-0.193	-0.346
aug-FIT3	-1.744	-0.095	-0.253
aug-cpFIT3	-1.023	-0.246	-0.325
aug-pFIT3	-1.005	-0.162	-0.247
TZV2P-MOLOPT	-0.123	-0.123	-0.123

<sup>a</sup> STD refers to traditional hybrid calculations, using the shown basis set as the primary basis set. ADMM1 and ADMM2 refer to the wavefunction fitting methods, using the TZV2P-MOLOPT basis set as the primary basis and the shown basis as the auxiliary basis set.

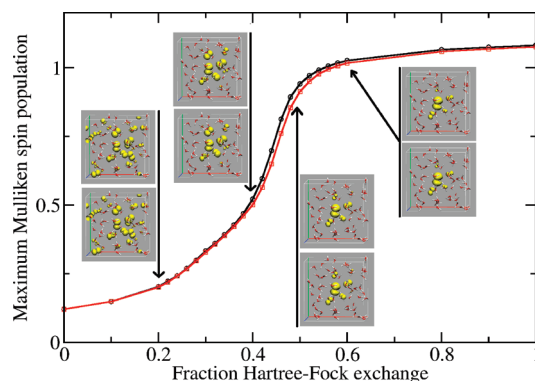
diffuse primitives, such as the TZV2P-MOLOPT basis, which has a BSSE of only 0.1 kcal/mol. However, within the ADMM scheme, the error reduces to 0.8 kcal/mol, approximately a 4-fold reduction. Using the augmented auxiliary basis sets reduces the error to approximately 0.3 kcal/mol, similar to, but not quite as good as, the quality of the primary basis set. Note that, since an auxiliary basis set method is not necessarily variational in the auxiliary basis, the counterpoise corrections can be of both signs. This can lead to an error cancellation, which is presumably the reason why ADMM2 performs surprisingly well with the lower-quality auxiliary basis sets. It can thus be concluded that both wave function fitting methods do not suffer from the large BSSE associated with the inferior quality of the auxiliary basis even though the BSSE does not reduce to the extent of the primary basis in all cases.

**3.4.  $H_2^+$  Dissociation Curve.** ADMM calculations that are GGA corrected might be biased from deficiencies of the underlying GGA functional. In order to investigate this effect, dissociation curves for  $H_2^+$  at different levels of theory have been calculated. As is well-known, GGA functionals, such as PBE exchange, describe the dissociation of this system incorrectly.<sup>51</sup> Figure 1 compares results obtained from a Hartree–Fock reference calculation, which is exact for this system, with results obtained from ADMM1. The primary basis was chosen to be the same as in the reference calculation (TZV2P-MOLOPT) while several different ABSs have been applied. The results clearly show that the wave function fitting is not biased by the GGA correction. Furthermore, as shown in the inset, better quality ABSs consistently improve the description of the potential around the minimum. It can thus be concluded that the qualitatively important effects of HFX are properly retained and that the GGA correction does not introduce artifacts of the underlying functionals.

**3.5. The Cationic Hole in Liquid Water.** In order to probe the effect of the dual basis set approach on the electronic structure directly, the spin density distribution of the cationic hole in bulk liquid water has been computed. The poor performance of local functionals for the radical cation water dimer was discussed in detail by Sodupe et al. in ref 52 and attributed to the self-interaction error, which favors configurations with a delocalized spin density distribution. Hybrid functionals with a relatively large fraction



**Figure 1.** Dissociation curves for  $H_2^+$  obtained from different calculations. The black solid line depicts the reference Hartree–Fock run with the TZV2P-MOLOPT basis set. Red, blue, and green lines represent results for ADMM1 for auxiliary basis sets of increasing quality, cFIT3, FIT3, and aug-pFIT3, respectively. The dotted black line shows the dissociation curve obtained from a pure GGA exchange calculation (PBEx). In the inset, a magnification of the potential energy around the minimum is presented.



**Figure 2.** Localization of the spin density distribution after ionization of bulk liquid water as a function of the fraction of Hartree–Fock exchange employed in the density functional. The Mulliken spin populations of the oxygen atom on which the hole localizes is shown with a solid line, while the insets show a contour plot at 0.001 a.u. of the spin density for selected fractions (0.2, 0.4, 0.5, and 0.6) of exchange. Results obtained with the auxiliary FIT3 basis (black line and upper panels of the inserts) are almost indistinguishable from the results obtained with the primary basis only, despite the pronounced sensitivity of this system toward the use of Hartree–Fock exchange.

of exchange, for example, BH&HLYP,<sup>53,54</sup> perform significantly better. In ref 55, ionization of bulk liquid water was probed, and the difficulty of DFT to properly describe the electronic structure was discussed. In particular, it has been found that the electron hole, or similarly the spin density, is delocalized over the full simulation cell with local functionals, whereas it localizes on a single water molecule with Hartree–Fock exchange. Hybrid functionals with varying amounts of exchange yield intermediate degrees of localization. This is illustrated in Figure 2 for a bulk sample of liquid water (64 molecules), where the localization of the spin



**Table 6.** Band Gaps of Diamond As Obtained from Different Methods<sup>a</sup>

method	number of integrals	gap [eV]
PBE (PBS)		4.17
PBE (ABS)		4.37
PBE0 (PBS)	40 787 850 778 591	6.07
PBE0 (ABS)	23 561 509 497	6.25
PBE0 ADMM1	24 816 897 009	6.03
PBE0 ADMM2	24 795 460 638	6.02

<sup>a</sup> All calculations have been performed using the  $3 \times 3 \times 3$  repetition of the basic unit cell in the  $\Gamma$ -point approximation. For the hybrid PBE0 calculations, also the number of Cartesian integrals is shown. See the text for details on the primary and auxiliary basis set ((PBS) and (ABS)). ADMM1 is purified wavefunction fitting and ADMM2 is non-purified wavefunction fitting.

density is shown as a function of the amount of HFX in the PBE0 functional.

As a quantitative measure, the maximum value of the Mulliken spin population is reported, ranging from approximately 0.1 in the local functional to more than 1.0 in a functional containing 100% HFX. Contour plots of the spin density distribution emphasize this radical change in the electronic structure. Given this very strong dependence on the amount of Hartree–Fock exchange, this is a very stringent test for the auxiliary basis method presented in this work. Furthermore, this calculation has been performed with the relatively small FIT3 basis, i.e., without polarization functions. The results shown in Figure 2 are therefore very reassuring, since the spin distribution obtained with the auxiliary basis set approach essentially reproduces the reference density in all details for all fractions of exchange.

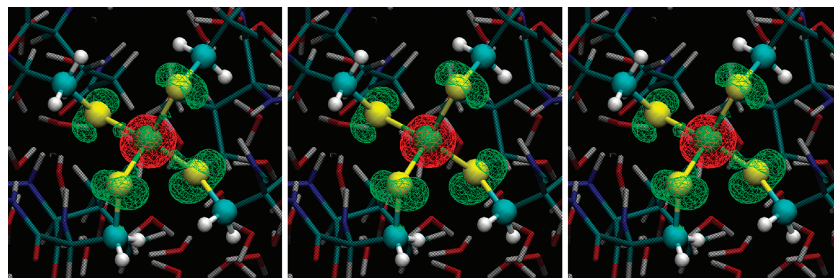
**3.6. Diamond Band Gap.** Both wave function fitting methods (ADMM1 and ADMM2) have been benchmarked with respect to their capability of predicting the band gap in diamond. The basic cubic unit cell with lattice parameter  $a = 3.576$  Å containing eight carbon atoms has been extended to a large supercell in order to apply the  $\Gamma$ -point approximation. In a first step, PBE band gaps for supercell sizes ranging from  $1 \times 1 \times 1$  to  $6 \times 6 \times 6$  repetitions of the basic unit cell have been determined with a high-quality basis set. The band gap calculation was found to be converged for the  $3 \times 3 \times 3$  repetition, yielding a band gap of 4.17 eV in agreement with the literature.<sup>41</sup> This supercell has been used to calculate the PBE0 reference band gap of this system, applying the same high-quality basis set. Since the condition number of the overlap matrix with the FIT3 basis is unfavorable in the case of bulk C ( $1.7 \times 10^5$ ), an optimized FIT3 (optFIT3) basis has been constructed that served as the ABS for the two wave function fitting methods. optFIT3 was obtained by minimization of the total energy of the PBE  $2 \times 2 \times 2$  supercell with respect to the constraint of a well behaved overlap matrix (the final condition number is on the order of  $10^2$ ). This allows for rather loose screening thresholds ( $10^{-6}$ ) and thus significantly reduces the amount of work in the Fock matrix construction. Results are summarized in Table 6. Both wave function fitting methods are in good agreement with the reference band gaps of the PBE0 run in the high-quality basis. In order to illustrate the cost savings, the total number of Cartesian integrals that needs to be

calculated has been added to the table. The ADMM calculations are by 3 orders of magnitude more efficient than the reference PBE0 run. Both methods give very similar results, suggesting that the approximate Kohn–Sham matrix (eq 34) is a valid approximation.

### 3.7. Performance and Embedding for Large Systems.

In order to illustrate the impact of ADMM for large systems, the electronic structure of Rubredoxin has been computed. Rubredoxin is a relatively small iron–sulfur protein that is an excellent benchmark system for electronic structure calculations, since it features an interesting active site. A realistic model including solvent and using periodic boundary conditions is comprised of only 2825 atoms and fits in a unit cell with edges  $31.1 \times 28.1 \times 30.5$  Å<sup>3</sup>. This system has been used extensively in our earlier work. In ref 56, ab initio simulations of the full system have been combined with statistical sampling to quantify the effect of mutations on the redox potential of the active site. In ref 9, the feasibility of computing the electronic structure with accurate, molecularly optimized, basis sets has been demonstrated. In ref 3, hybrid density functional calculations using an all-electron description and a polarized triple- $\zeta$  valence basis set<sup>57</sup> have been performed. Molecularly optimized basis sets<sup>9</sup> have been employed as a primary basis for hybrid calculations (B3LYP<sup>54,58,59</sup>) of the same system, and the performance and accuracy of the ADMM scheme have been evaluated. Using a traditional HFX implementation with the MOLOPT basis set requires significant computational effort, despite the fact that the DZVP-MOLOPT-SR-GTH basis has been employed (22 910 basis functions), which has fewer and less diffuse primitives than the basis sets originally presented in ref 9. Indeed, the reference calculation has been run using 48 000 cores on a Cray XT5. The first SCF step required 45 min to compute  $3.7 \times 10^{14}$  primitive Cartesian integrals after screening with a threshold of  $10^{-6}$ . Successive SCF steps spent only 25 s in the Hartree–Fock routines, since these calculations could be run in-core using integral compression<sup>8</sup> and 6.8 Tb of RAM. Due to the contracted and diffuse nature of the basis sets, this calculation is significantly more expensive than the calculations performed in ref 3. The difference in spin density between the B3LYP and a BLYP calculations is shown in the left panel of Figure 3. ADMM1 calculations using the cFIT3 basis (12 311 basis functions) require far fewer resources and have been run on 1152 cores only. The Hartree–Fock routines used 75 and 25 s in the first and successive SCF steps, respectively, and in-core operation only required 5.2 Gb of RAM. The time spent in dense linear algebra for the wave function fitting (15s, eq 16) and corresponding derivative calculation (15s, eq 101) is similar to the time spent in the HFX, suggesting that this system might benefit from linear scaling techniques for this part of the calculation. For this system, ADMM thus improves the efficiency of the calculation by a factor 20 to 1000, depending on the measure. As shown in Figure 3, the obtained spin density reproduces the reference calculation very well, even though some small differences near the Fe–S bond can be observed. To improve the accuracy, we have employed the simple embedding strategy in which the auxiliary basis for the five central atoms (Fe and S) was set





**Figure 3.** Isosurfaces at  $\pm 0.001$  a.u. of the difference between the spin density as computed with BLYP and B3LYP for the iron–sulfur protein Rubredoxin. Left panel: Traditional calculation using only a primary basis. Middle panel: ADMM1 calculation using the cFIT3 auxiliary basis. Right panel: ADMM1 calculations using an embedding-like strategy, where the bulk of the system is described using the cFIT3 basis, but Fe and S use the primary basis as the auxiliary basis. Both ADMM calculations clearly capture the effect of Hartree–Fock exchange, a reduced delocalization of the spin density, at a small fraction of the cost of the traditional approach. The embedding strategy faithfully reproduces all details, including the change in spin density along the Fe–S bonds.

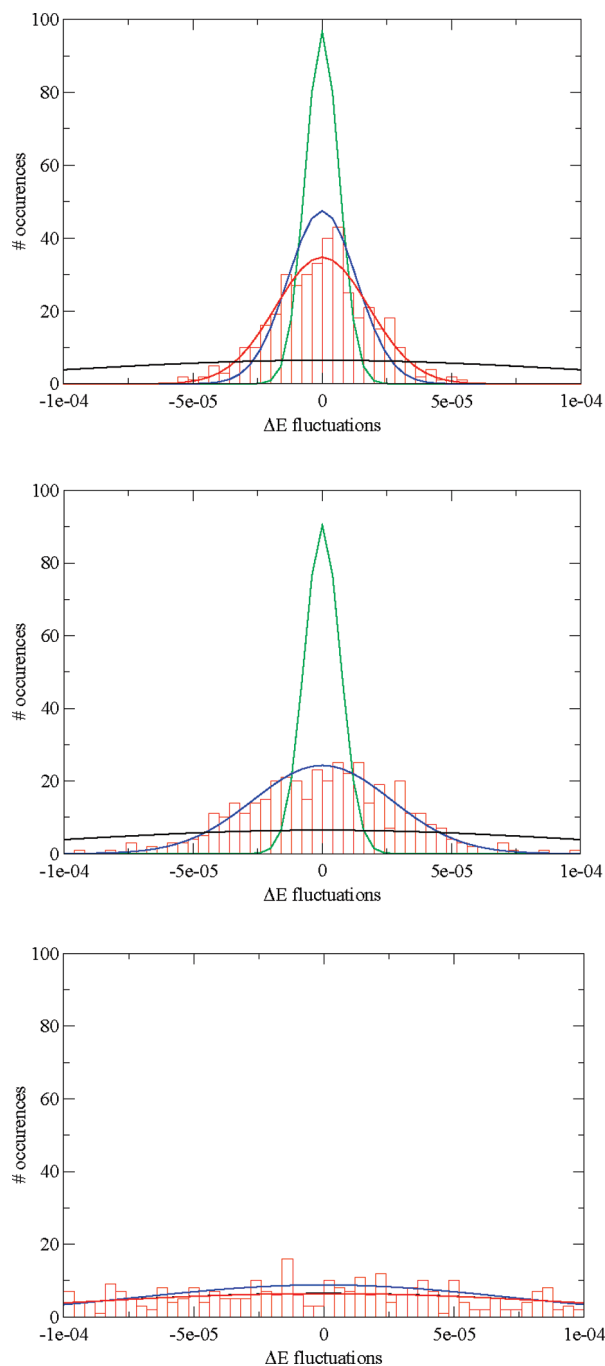
equal to the primary basis. These calculations can be performed without any significant increase in computational cost, and the right panel of Figure 3 shows that full quantitative agreement can be obtained in this way.

**3.8. The Effect of Hartree–Fock Exchange on the Structure of Liquid Water.** In this section, ADMM is employed to study the effect of changing the fraction of Hartree–Fock exchange in the PBE0 functional on the structure of liquid water. First, the accuracy of ADMM for describing bulk water is investigated. Second, we perform ab initio molecular dynamics simulations based on ADMM for various values of the fraction of exchange. The model system is a sample of 64 water molecules in a cubic box with edges of 12.42 Å that has previously been equilibrated using PBE0.<sup>8</sup> The primary basis is in all cases a TZV2P basis (2560 basis functions in total).

In order to investigate the accuracy of ADMM, the following procedure has been adopted. In a first step, a reference molecular dynamics trajectory of 2 ps starting from an equilibrated configuration has been produced. In a second step, 400 equipartitioned configurations have been chosen, and for all of these, single point ADMM calculations have been performed. The error has been quantified by computing the distribution of the difference between the reference energy and the ADMM energy. The important quantity is the variance of this difference, i.e., the energy fluctuations between the two potential energy surfaces. ADMM1 and ADMM2 have been benchmarked for various basis sets. ADMM3, which starts from a blocked density matrix, has been employed with blocked purification (eq 70) or full purification (eq 69). Nonpurified ADMM3 was found to be unstable. The subsystems have been defined as containing exactly one water molecule per block; i.e., the whole system consists of 64 diagonal sub-blocks. As shown in Figure 4, the fluctuations have approximately a Gaussian distribution. For ADMM1 and ADMM2, the associated variance gets consistently smaller when improving the quality of the auxiliary basis set. The variance for the purified wave function fitting (ADMM1) is slightly lower than the variance from nonpurified wave function fitting (ADMM2). The variance of the energy fluctuations per water molecule is below 30 micro-Hartree for all auxiliary basis sets. This variance is significantly below the variance obtained applying

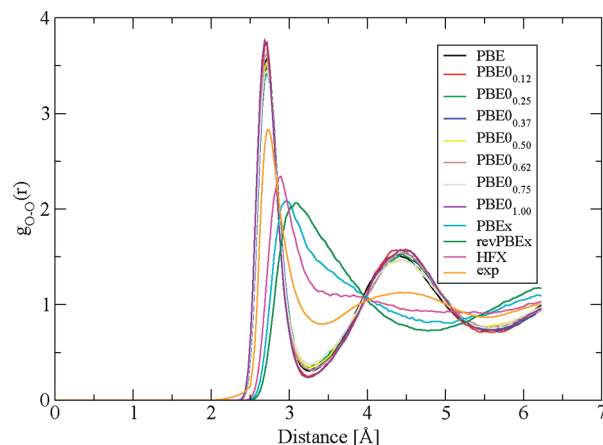
the same procedure with the pure density functional PBE; i.e., the difference between PBE0 and PBE is captured correctly with the ADMM1 and ADMM2 procedures. ADMM3 shows a relatively large variance, similar to direct use of the PBE functional, and its accuracy is not competitive.

With the aim of studying the effect of the fraction of exchange on the structure of the liquid, simulations employing the following functionals have been performed: PBE0 with various amounts of Hartree–Fock exchange ( $\alpha \in \{0.12, 0.25, 0.37, 0.5, 0.62, 0.75, 1.00\}$ ), PBE, pure Hartree–Fock, PBE exchange (PBE<sub>ex</sub>), and a revised parametrization<sup>61</sup> of PBE<sub>ex</sub> (revPBE<sub>ex</sub>). With these settings, trajectories longer than 30 ps have been obtained for all cases at a rate of 7 and 20 s per MD step (0.5 fs) for the pure and the hybrid functionals, respectively, on 64 cores of a Nehalem-based cluster. Compared to a standard hybrid functional calculation in the PBS without ADMM and multiple time-step MD,<sup>8</sup> this is a speed-up of a factor 16 per MD step. All MD simulations have been done within the isokinetic ensemble<sup>62</sup> at a temperature of 330 K, using ADMM1 and the FIT3 auxiliary basis. The structure has been analyzed using the oxygen–oxygen pair correlation function using the last 28 ps for each run, binning with a width of 0.03 Å. As shown in Figure 5, PBE and all variants of PBE0 yield very similar pair correlation functions. Compared to experimental results,<sup>60</sup> the location of the peak is correct, but the liquid is overstructured. In order to quantify the structure, the maximum value of the pair correlation function is shown in Figure 6. For PBE and all variants of PBE0, the height of the first peak falls in the range 3.45–3.75. There is no systematic trend with respect to the fraction of exchange, and the differences between the peak heights must be attributed to the limited statistics that can be collected within 30 ps for a structured liquid. Within these statistical uncertainties, these ADMM results agree with the PBE0 results obtained using traditional HFX and the same basis in ref 8, where a maximum height of 3.4 was found for PBE and PBE0 ( $\alpha = 0.25$ ) using 7.5 ps of data. On the other hand, the liquid is significantly understructured for the pure Hartree–Fock, PBE<sub>ex</sub>, and revPBE<sub>ex</sub> runs. The maximum pair correlation height obtained from the Hartree–Fock simulation is in agreement with the results in ref 26, 2.34 and 2.35, respectively, where a plane wave basis set has been

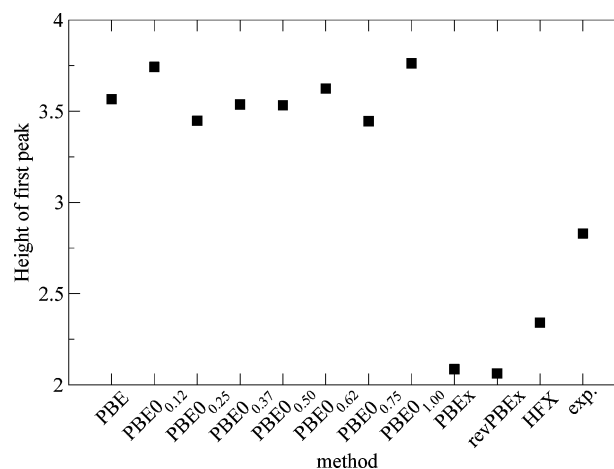


**Figure 4.** Centered distributions of energy differences (a.u.) between a standard PBE0 reference run and an ADMM method for bulk water. The top, middle, and bottom panels were obtained with ADMM1, ADMM2, and ADMM3, respectively. ADMM1 and ADMM2 results have been computed using cFIT3 (red), FIT3 (blue), and aug-pFIT3 (green) auxiliary basis sets. ADMM3 employs blocking on a molecular level, with blocked (red) and full purification (blue). For clarity, Gaussian distributions are shown instead of binned data, except for one data set per panel.

employed. The large difference between a Hartree–Fock simulation and a PBE0 ( $\alpha = 1.00$ ) simulation can only be attributed to the correlation functional, since all other terms in the Kohn–Sham equations are the same. Consistent with this and the observations made above, we find that the PBEx and revPBEx simulations, which do not include a correlation



**Figure 5.** Oxygen–oxygen pair correlation functions as obtained for bulk liquid water, based on a sample of 64 water molecules. Seven variants of PBE0, using various amounts of Hartree–Fock exchange, and PBE almost superimpose but are overstructured as compared to the experimental result from ref 60. PBE exchange (PBEx) only, revised PBE exchange (revPBEx), and pure Hartree–Fock yield pair correlations that are similar and understructured as compared to experimental results. A detailed comparison of the maximum values of the pair correlation functions is shown in Figure 6.



**Figure 6.** Height of the first peak for the oxygen–oxygen pair correlations shown in Figure 5. For the variants of PBE0, no trend in peak height with respect to the fraction of exchange can be observed.

functional and employ a density functional to model exchange, qualitatively reproduce the Hartree–Fock simulations. The deviation between the Hartree–Fock and the revPBEx pair correlation function is somewhat larger than the deviation between the Hartree–Fock and the PBEx results.

Finally, it is important to emphasize three limitations of our simulations. First, due to the fact that the stress tensor is currently not implemented for hybrid functionals, these simulations have been performed at constant volume and not at constant pressure. In recent work, see, e.g., refs 63 and 64, it has been shown that constant volume simulations might differ significantly from constant pressure simulations for this system. Indeed, the density of the liquid, and several other macroscopic quantities,<sup>65</sup> most of them challenging to compute ab initio, might be more revealing about the quality

of the underlying density functional than the pair correlation function. Second, whereas the structure of the liquid for the PBE0 functional does not depend strongly on the fraction of exchange, this dependence might be different for other hybrid functionals. Third, the fact that the structure of the liquid is effectively unchanged as the fraction of exchange is varied does not imply that the properties as a solvent, i.e., the interaction of the liquid with solutes, is unchanged. In the future, the efficiency of ADMM might contribute to addressing some of these important issues.

#### 4. Summary

We presented auxiliary density matrix methods that aim at reducing the cost of simulations based on hybrid density functionals. By constructing an approximate density matrix, which allows for a fast calculation of exchange, and by correcting the error introduced using a density functional, significant speedups have been achieved while accuracy has been retained. Wave function fitting methods that employ a small auxiliary basis to reduce the size of the density matrix appear to be a simple yet successful way to obtain an approximate density matrix. The accuracy of this approach has been investigated using a variety of tests. Calculations on the GMTKN24 database suggest that the predictivity of calculations based on wave function fitting essentially equals that of the more expensive traditional approach. Test calculations specifically aimed at difficult systems, such as BSSE calculations for the water dimer and the dissociation profile of  $\text{H}_2^+$ , have demonstrated that neither the deficiencies of the small auxiliary basis nor the correcting functional impact the quality of the results significantly. Two variants of wave function fitting, either with purification (ADMM1) or without purification (ADMM2), have been tested, and no significant differences in accuracy have been found so far. Whereas ADMM1 has the advantage of yielding a pure auxiliary density matrix, ADMM2 is particularly simple to implement and is directly suitable for a linear scaling code. ADMM3, which relies on a blocking of the density matrix, has not been tested thoroughly yet but might find its application in cases where clear subsystems, such as a solute in solution, can be easily defined. Exploiting the efficiency of the ADMM scheme, the effect of the range of exchange has been investigated for the PBE0-TC-LRC functional. The performance of this functional on the GMTKN24 database is optimal for 20% of exchange, and a range of 2.5 Å. Furthermore, ADMM has been used to perform extensive simulations of bulk water, showing that for PBE0-like functionals the amount of Hartree–Fock exchange does not directly influence the structure of the liquid. In this case, the role of correlation is more significant. Finally, a calculation on a solvated protein has been used to demonstrate that speedups in excess of a factor of 20 can be observed in actual applications.

**Acknowledgment.** The authors acknowledge Goerigk and Grimme for making the GMTKN24 database publically available in a convenient format. Calculations were enabled by a 2008–2010 INCITE award on the CRAY XT5 using resources of the National Center for Computational Sciences

at Oak Ridge National Laboratory (ORNL), which is supported by the Office of Science of the U.S. DOE under Contract No. DE-AC05-00OR22725, and by the Swiss National Supercomputer Centre (CSCS). This work has been funded by the Swiss University Conference through the High Performance and High Productivity Computing (HP2C) Programme.

#### Appendix

**A. Wave Function Fitting.** The one-particle wave functions represented with the high-quality primary basis set (PBS)  $\{\phi_\mu\}$  can be written in terms of molecular coefficients  $C^{\mu i}$

$$\psi_i(\mathbf{r}) = \sum_{\mu} C^{\mu i} \phi_{\mu}(\mathbf{r}) \quad (35)$$

These wave functions are assumed to be orthonormal, i.e.

$$\int \psi_i(\mathbf{r}) \psi_j(\mathbf{r}) \, \mathrm{d}\mathbf{r} = \delta_{ij} \quad (36)$$

For the wave function fitting, a lower-quality auxiliary basis set (ABS)  $\{\hat{\phi}_{\mu}\}$  is introduced which yields a second set of molecular coefficients  $\hat{C}^{\mu i}$  and auxiliary one-particle wave functions in the following form

$$\hat{\psi}_i(\mathbf{r}) = \sum_{\mu} \hat{C}^{\mu i} \hat{\phi}_{\mu}(\mathbf{r}) \quad (37)$$

The molecular coefficients  $\hat{C}^{\mu i}$  are a priori unknown but can be determined by requiring that the corresponding occupied wave functions resemble as well as possible the original ones by minimizing their square difference over all space

$$\sum_j \int (\psi_j(\mathbf{r}) - \hat{\psi}_j(\mathbf{r}))^2 \, \mathrm{d}\mathbf{r} \quad (38)$$

Optionally, the auxiliary wave functions can be restricted to obey the orthonormality constraint

$$\int \hat{\psi}_i(\mathbf{r}) \hat{\psi}_j(\mathbf{r}) \, \mathrm{d}\mathbf{r} = \delta_{ij} \quad (39)$$

These two possibilities give rise to two slightly different minimization problems:

$$\min_{\tilde{C}} \left[ \sum_j \int (\psi_j(\mathbf{r}) - \tilde{\psi}_j(\mathbf{r}))^2 \, \mathrm{d}\mathbf{r} \right] \quad (40)$$

and

$$\min_{\tilde{C}} \left[ \sum_j \int (\psi_j(\mathbf{r}) - \tilde{\psi}_j(\mathbf{r}))^2 \, \mathrm{d}\mathbf{r} + \sum_{k,l} \Lambda_{kl} \left( \int \tilde{\psi}_k \tilde{\psi}_l - \delta_{kl} \right) \right] \quad (41)$$

where in the latter case, the Lagrangian multipliers  $\Lambda_{kl}$  enforce condition eq 39 and the notation  $\tilde{C}$  has been introduced in order to distinguish the two different sets of molecular coefficients. The overlap matrices associated with the two basis set representations are given as



$$S_{mm'} = \int \phi_m(\mathbf{r}) \phi_{m'}(\mathbf{r}) d\mathbf{r} \text{ and } \hat{S}_{nn'} = \tilde{S}_{nn'} = \int \hat{\phi}_n(\mathbf{r}) \hat{\phi}_{n'}(\mathbf{r}) d\mathbf{r} \quad (42)$$

In order to retain a consistent notation,  $\hat{S}$  and  $\tilde{S}$  have been introduced, even though both matrices are identical. Furthermore, a mixed overlap matrix  $Q$  needs to be defined that takes the overlap of both sets of basis functions into account:

$$Q_{nm} = \int \hat{\phi}_n(\mathbf{r}) \phi_m(\mathbf{r}) d\mathbf{r} \quad (43)$$

Within this notation, the Lagrange functions associated with the two minimization problems eqs 40 and 41 can conveniently be expressed as

$$\hat{L} = \sum_j \left( \sum_{m,m'} C_{mj} C_{m'} S_{mm'} + \sum_{n,n'} \hat{C}_{nj} \hat{C}_{n'} \hat{S}_{nn'} - 2 \sum_{m,n} C_{mj} \hat{C}_{nj} Q_{nm} \right) \quad (44)$$

and

$$\begin{aligned} \tilde{L} = & \sum_j \left( \sum_{m,m'} C_{mj} C_{m'} S_{mm'} + \sum_{n,n'} \tilde{C}_{nj} \tilde{C}_{n'} \tilde{S}_{nn'} - 2 \sum_{m,n} C_{mj} \tilde{C}_{nj} Q_{nm} \right) \\ & + \sum_{m,n} \sum_{k,l} \Lambda_{kl} (\tilde{C}_{nj} \tilde{C}_{mk} \tilde{S}_{nm} - \delta_{kl}) \end{aligned} \quad (45)$$

Because of eq 36, or, equivalently,  $C^T S C = 1$ , and due to eq 39 or  $\tilde{C}^T \tilde{S} \tilde{C} = 1$  in the second case, this simplifies to

$$\hat{L} = \sum_j \left( \sum_{n,n'} \hat{C}_{nj} \hat{C}_{n'} \hat{S}_{nn'} - 2 \sum_{m,n} C_{mj} \hat{C}_{nj} Q_{nm} \right) \quad (46)$$

and

$$\tilde{L} = -2 \sum_j \sum_{m,n} C_{mj} \tilde{C}_{nj} Q_{nm} + \sum_{k,l} \Lambda_{kl} (\tilde{C}_{nk} \tilde{C}_{ml} \tilde{S}_{nm} - \delta_{kl}) \quad (47)$$

respectively. From that, the unknown auxiliary molecular coefficients can be determined by taking the partial derivatives and equating them to zero. This yields

$$\frac{\partial \hat{L}}{\partial \hat{C}_{pq}} = -2(QC)_{pq} + 2(\hat{S}\hat{C})_{pq} \doteq 0 \quad (48)$$

and

$$\frac{\partial \tilde{L}}{\partial \tilde{C}_{pq}} = -2(QC)_{pq} + 2(\tilde{S}\tilde{C})_{pq} \doteq 0 \quad (49)$$

Thus, the final results for the MO coefficients are given by

$$\hat{C} = \hat{S}^{-1} Q C \text{ and } \tilde{C} = \tilde{S}^{-1} Q C \Lambda^{-1} \quad (50)$$

with the matrix of the Lagrangian multipliers

$$\Lambda = [(QC)^T \tilde{S}^{-1} QC]^{1/2} \quad (51)$$

Defining  $A := \hat{S}^{-1} Q = \tilde{S}^{-1} Q$  to be the projector between the PBS and ABS directly yields eqs 12 and 16 presented in section 2.2.

**B. Purification.** Most of the calculations that follow take advantage of the Cauchy integral theorem for matrix functions.<sup>33</sup> For an arbitrary matrix  $F$ , it states

$$f(F) = \frac{1}{2\pi i} \oint f(z) \frac{1}{zI - F} dz \quad (52)$$

which, since

$$\frac{d}{dx} F^{-1} = -F^{-1} \frac{dF}{dx} F^{-1} \quad (53)$$

transforms into an explicit formula for the derivative of a matrix function

$$\frac{df(F)}{dx} = \frac{1}{2\pi i} \oint f(z) \frac{1}{F - zI} \frac{dF}{dx} \frac{1}{F - zI} dz \quad (54)$$

Applying this formula, matrix function derivatives can be calculated through residues of its eigenvalues. Applying this to  $f(x) = \Theta(z)$ , where  $\Theta(z)$  denotes the Heaviside function, the expression for the purified density matrix becomes

$$\tilde{P} = \hat{S}^{-1/2} \left[ \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{zI - \hat{S}^{1/2} \hat{P} \hat{S}^{1/2}} dz \right] \hat{S}^{-1/2} \quad (55)$$

or, after some rearrangements

$$\tilde{P} = \hat{S}^{-1} \left[ \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{\hat{S}^{-1} z - \tilde{P}} dz \right] \hat{S}^{-1} \quad (56)$$

The evaluation of the contour integral can easily be performed via diagonalization. For that purpose, the following generalized eigenvalue problem needs to be solved:

$$\hat{P} R = \hat{S}^{-1} R \lambda \quad (57)$$

where  $R$  defines the matrix containing the generalized eigenvectors of  $\hat{P}$ . Indeed, inserting  $RR^{-1} = 1$  and  $(\hat{S}^{-1} R)(\hat{S}^{-1} R)^{-1}$  from left and right into eq 56 gives

$$\begin{aligned} \tilde{P} &= \hat{S}^{-1} \left[ \frac{1}{2\pi i} \oint \frac{RR^{-1} \Theta(z - 0.5)}{\hat{S}^{-1} z - \tilde{P}} (\hat{S}^{-1} R)(\hat{S}^{-1} R)^{-1} dz \right] \hat{S}^{-1} \\ &= \hat{S}^{-1} R \left[ \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{zI - D} dz \right] R^{-1} \\ &= \hat{S}^{-1} R \left[ \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{zI - D} dz \right] R^T \hat{S}^{-1} \end{aligned} \quad (58)$$

where in the last step, the relation  $R^T \hat{S}^{-1} R = 1$ , which is valid under the assumption that  $\hat{P}$  is a symmetric matrix. The integral in brackets is evaluated using the Cauchy residue theorem and can be written in terms of a diagonal matrix  $L$ . Component-wise, this yields

$$L_{ii} = \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{z - \lambda_i} dz = \text{Res} \left( \frac{\Theta(z - 0.5)}{z - \lambda_i}, z = \lambda_i \right) = \Theta(\lambda_i - 0.5) \quad (59)$$

The final expression for  $\tilde{P}$  is therefore

$$\tilde{P} = \hat{S}^{-1} R L R^T \hat{S}^{-1} \quad (60)$$

In a similar fashion, the derivative of  $\tilde{P}$  with respect to  $\hat{P}$ , which is needed in the expression for the Kohn–Sham matrix, can be evaluated. After diagonalization of  $\hat{P}$ , this derivative reads

$$\frac{d\tilde{P}}{d\hat{P}} = \hat{S}^{-1} R \left[ \frac{1}{2\pi i} \oint \Theta(z - 0.5) \left( \frac{1}{D - zI} \right) R^{-1} \hat{S} \frac{d\hat{P}}{d\hat{P}} \times \right. \\ \left. R \left( \frac{1}{D - zI} \right) dz \right] R^{-1} \quad (61)$$

Again, the contour integral in brackets is computed via the Cauchy residue theorem. Since, in this case, the diagonal matrix  $D$  with the eigenvalues appears twice, the result is now a matrix  $M$  that also contains off-diagonal elements

$$M_{kj} = \frac{1}{2\pi i} \oint \frac{\Theta(z - 0.5)}{(\lambda_k - z)(\lambda_j - z)} dz = \frac{1}{2\pi i} \oint g(z) dz \\ = \text{Res}(g, \lambda_k) + \text{Res}(g, \lambda_j) \\ = \begin{cases} \frac{\Theta(\lambda_k - 0.5) - \Theta(\lambda_j - 0.5)}{\lambda_k - \lambda_j} & k \neq j \\ \delta(\lambda_k - 0.5) & k = j \end{cases} \quad (62)$$

The derivative eq 61 thus becomes

$$\frac{d\tilde{P}}{d\hat{P}}_{cd} = [\hat{S}^{-1} R (M \otimes G_{ef}) R^{-1}]_{cd} \quad (63)$$

with

$$G_{ef} = R^{-1} \hat{S} \frac{d\hat{P}}{d\hat{P}}_{ef} R \quad (64)$$

This result can now be applied to the purified wave function fitting in order to obtain an expression for the Kohn–Sham matrix. In this case,  $\hat{P}$  is a function of  $P$ ; i.e., it holds that  $\hat{P} = A P A^T$ , and the Kohn–Sham matrix written in terms of a derivative of the energy with respect to  $P$  is given as

$$\frac{d\tilde{E}[\tilde{P}]}{dP_{ab}} = \frac{d\tilde{E}}{d\tilde{P}_{cd}} \frac{d\tilde{P}_{cd}}{d\hat{P}_{ef}} \frac{d\hat{P}_{ef}}{dP_{ab}} = \tilde{K}_{cd} \frac{d\tilde{P}_{cd}}{d\hat{P}_{ef}} \frac{d\hat{P}_{ef}}{dP_{ab}} \quad (65)$$

where summation over same indices is assumed. The last derivative trivially amounts to

$$\frac{d\hat{P}_{ef}}{dP_{ab}} = \frac{d}{dP_{ab}} [A P A^T]_{ef} = A_{ea} A_{fb} \quad (66)$$

Under utilization of eq 63, this term simplifies to

$$\frac{d\tilde{E}[\tilde{P}]}{dP_{ab}} = [(A^T \tilde{S} R^{-1}) [(R^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R^{-T}) \otimes M] R^T A]_{ab} \quad (67)$$

Since  $R^T \tilde{S}^{-1} R = 1$ , this can be rewritten as

$$\frac{d\tilde{E}[\tilde{P}]}{dP} = A^T R [(R^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R) \otimes M] R^T A \quad (68)$$

If  $\hat{P}$  is obtained from a blocking procedure, the above expression needs to be filtered through  $B$

$$\frac{d\tilde{E}[\tilde{P}]}{dP} = [A^T R [(R^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R) \otimes M] R^T A] \otimes B \quad (69)$$

For the purification of the blocked density matrix, the McWeeny procedure based on the overlap matrix  $S$  can be replaced by a blocked McWeeny procedure where the overlap matrix is replaced by its blocked counterpart  $S^\dagger = S \otimes B$ . Equation 23 thus becomes

$$\bar{P}_{n+1} = f(\bar{P}_n) = 3\bar{P}_n S^\dagger \bar{P}_n - 2\bar{P}_n S^\dagger \bar{P}_n S^\dagger \bar{P}_n \quad (70)$$

If the matrix  $B$  is chosen to be block diagonal, the eigenvalue problem eq 27 can thus be solved within the smaller diagonal subspaces, which significantly reduces the computational workload.

**C. Wave Function Fitting with and without Purification.** As mentioned in section 2.3, applying the purification scheme eq 23 to the density matrix  $\hat{P}$  obtained from wave function fitting without the orthogonality constraint yields exactly the density matrix  $\bar{P}$  obtained through the fitting procedure including the constraint. This can easily be seen by plugging  $\hat{P}$  into the McWeeny purification algorithm. The first two iterations amount to

$$\begin{aligned} \bar{P}_1 &= 3\hat{P}\hat{S}\hat{P} - 2\hat{P}\hat{S}\hat{P}\hat{S}\hat{P} \\ &= 3\hat{C}\hat{C}^T\hat{S}\hat{C}\hat{C}^T - 2\hat{C}\hat{C}^T\hat{S}\hat{C}\hat{C}^T\hat{S}\hat{C}\hat{C}^T \\ &= 3\hat{C}\Lambda\hat{C}^T - 2\hat{C}\Lambda^2\hat{C}^T \\ &= \hat{C}(3\Lambda - 2\Lambda^2)\hat{C}^T =: \hat{C}g_1\hat{C}^T \end{aligned} \quad (71)$$

and

$$\begin{aligned} \bar{P}_2 &= 3\hat{P}_1\hat{S}\hat{P}_1 - 2\hat{P}_1\hat{S}\hat{P}_1\hat{S}\hat{P}_1 \\ &= 3\hat{C}(3\Lambda - 2\Lambda^2)\hat{C}^T\hat{S}\hat{C}(3\Lambda - 2\Lambda^2)\hat{C}^T \\ &\quad - 2\hat{C}(3\Lambda - 2\Lambda^2)\hat{C}^T\hat{S}\hat{C}(3\Lambda - 2\Lambda^2)\hat{C}^T\hat{S}\hat{C}(3\Lambda - 2\Lambda^2)\hat{C}^T \\ &= 3\hat{C}g_1^2\Lambda\hat{C}^T - 2\hat{C}g_1^3\Lambda^2\hat{C}^T \end{aligned} \quad (72)$$

Recursively, that yields

$$\bar{P}_{n+1} = \hat{C}g_{n+1}\hat{C}^T \quad (73)$$

with

$$g_{n+1} = 3g_n^2\Lambda - 2g_n^3\Lambda^2 \quad (74)$$

If the McWeeny procedure converges in the limit for  $n \rightarrow \infty$ ,  $g_n(\Lambda)$  is required to become a fixed point. Obviously this condition is fulfilled in the case of  $g_n \rightarrow \Lambda^{-1}$ , i.e.

$$\bar{P} = \hat{C}\Lambda^{-1}\hat{C}^T = \bar{P} \quad (75)$$

which matches exactly the corresponding equation for the purified wave function fitting.

**D. Diagonalization in the Occupied Subspace for Wave Function Fitting.** As shown in Appendix part B,

in order to obtain an expression for the Kohn–Sham matrix, a general eigenvalue problem of size  $N_{\text{ABS}}$

$$\hat{P}R = \tilde{S}^{-1}R\lambda \quad (76)$$

needs to be solved. This might become the bottleneck in an ADMM calculation because  $N_{\text{ABS}}$  is not necessarily a small quantity. However, if  $\hat{P}$  can be expressed in terms of molecular coefficients, as is the case in wave function fitting, it is sufficient to diagonalize the occupied subspace only, which is typically much smaller than  $N_{\text{ABS}}$ . This can be achieved by introducing the following substitution  $R \rightarrow R_{\Lambda}$ :

$$R = \tilde{S}\hat{\Lambda}^{-1/2}R_{\Lambda} = \tilde{S}\tilde{C}R_{\Lambda} \quad (77)$$

Equation 76 thus transforms into

$$\begin{aligned} \hat{P}R &= \tilde{S}^{-1}R\lambda \\ \hat{C}\hat{C}^TR &= \tilde{S}^{-1}R\lambda \\ \hat{C}\hat{C}^T\tilde{S}\tilde{C}R_{\Lambda} &= \tilde{S}^{-1}\tilde{S}\tilde{C}R_{\Lambda}\lambda \\ \tilde{C}\Lambda^{1/2}\Lambda^{1/2}\tilde{C}^T\tilde{S}\tilde{C}R_{\Lambda} &= \tilde{C}R_{\Lambda}\lambda \\ (\tilde{S}\tilde{C})^T\tilde{C}\Lambda\tilde{C}^T\tilde{S}\tilde{C}R_{\Lambda} &= (\tilde{S}\tilde{C})^T\tilde{C}R_{\Lambda}\lambda \\ \Lambda R_{\Lambda} &= R_{\Lambda}\lambda \end{aligned} \quad (78)$$

where, in the last step, the fact that  $\tilde{C}\tilde{S}\tilde{C}^T = 1$  has been used. This eigenvalue problem is of the size  $N_{\text{mo}} \times N_{\text{mo}}$ , with  $N_{\text{mo}}$  being the number of occupied orbitals in the system, and therefore significantly smaller in size than the general one. Unfortunately, its solution will only provide the eigenvectors of the occupied subspace

$$R_o = \tilde{S}\hat{\Lambda}^{-1/2}R_{\Lambda} \quad (79)$$

and the eigenvectors  $R_n$  for the null space are unknown. However, since  $RR^T = \tilde{S}$ , it follows for the decomposition into occupied and unoccupied subspaces that

$$R_n R_n^T + R_o R_o^T = \tilde{S} \quad (80)$$

which motivates the notation  $R = (R_n \ R_o)$ . Furthermore, the matrix  $M$  has a very characteristic structure

$$M = \begin{pmatrix} M_n & M_{no} \\ M_{on} & M_o \end{pmatrix} \quad (81)$$

with

$$M_o = M_n = \begin{pmatrix} 0 & 0 & \dots \\ 0 & 0 & \dots \\ \vdots & \vdots & \ddots \end{pmatrix}, M_{no} = M_{on}^T = (M_{no} \ M_{no} \ M_{no} \dots) \quad (82)$$

which directly follows from its definition in eq 62. Using this decomposition, eq 68 can be rewritten in terms of occupied and unoccupied parts:

$$\begin{aligned} \frac{d\tilde{E}[\tilde{P}]}{dP} &= A^T(R_n R_o) \left[ \begin{pmatrix} R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R_n & R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R_o \\ R_o^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R_n & R_o^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R_o \end{pmatrix} \otimes M \right] (R_n \ R_o)^T A \\ &= A^T(R_n R_o) \begin{pmatrix} 0 & R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o \\ (R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o)^T & 0 \end{pmatrix} (R_n \ R_o)^T A \end{aligned} \quad (83)$$

with  $\tilde{R}_o$  denoting the eigenvectors of the occupied subspace with the columns scaled by the vectors  $M_{no}$ . This can further be simplified, yielding an expression that only depends on the eigenvectors  $R_o$  of the occupied subsystem:

$$\begin{aligned} \frac{d\tilde{E}[\tilde{P}]}{dP} &= A^T(R_o(R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o)^T \ R_n R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o) (R_n \ R_o)^T A \\ &= A^T(R_o \tilde{R}_o^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} R_n R_n^T + R_n R_n^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o R_o^T) A \\ &= A^T(R_o \tilde{R}_o^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} (\tilde{S} - R_o R_o^T) \\ &\quad + (\tilde{S} - R_o R_o^T) \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o R_o^T) A \end{aligned} \quad (84)$$

An explicit expression for the products  $R_o \tilde{R}_o^T$  is still required. Again, due to the special structure of  $M$ , the matrix  $\tilde{R}_o$  resulting from a column scaling of  $R_o$  with  $M_{no}$  can conveniently be obtained from the eigenvalues of  $\Lambda$

$$\tilde{R}_o = R_o D_{\Lambda}^{-1} \quad (85)$$

where  $D_{\Lambda}$  contains the eigenvalues of the matrix  $\Lambda$  and fulfills  $R_{\Lambda} D_{\Lambda} R_{\Lambda}^T = \Lambda$  and similarly  $R_{\Lambda} D_{\Lambda}^{-1} R_{\Lambda}^T = \Lambda^{-1}$ , yielding

$$\begin{aligned} R_o \tilde{R}_o^T &= \tilde{S}\hat{\Lambda}^{-1/2}R_{\Lambda}(\tilde{S}\hat{\Lambda}^{-1/2}R_{\Lambda}D_{\Lambda}^{-1})^T \\ &= \tilde{S}\hat{\Lambda}^{-1/2}R_{\Lambda}D_{\Lambda}^{-1}R_{\Lambda}^T\Lambda^{-1/2}\hat{C}^T\tilde{S} \\ &= \tilde{S}\hat{\Lambda}^{-1/2}\Lambda^{-1}\Lambda^{-1/2}\hat{C}^T\tilde{S} \\ &= \tilde{S}\hat{\Lambda}^{-2}\hat{C}^T\tilde{S} \end{aligned} \quad (86)$$

Inserting this result into eq 84 yields

$$\begin{aligned} \frac{d\tilde{E}[\tilde{P}]}{dP} &= A^T(R_o \tilde{R}_o^T \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} (\tilde{S} - R_o R_o^T) \\ &\quad + (\tilde{S} - R_o R_o^T) \tilde{S}^{-1} \tilde{K} \tilde{S}^{-1} \tilde{R}_o R_o^T) A \\ &= A^T(\tilde{S}\hat{\Lambda}^{-2}\hat{C}^T\tilde{S}\tilde{S}^{-1}\tilde{K}\tilde{S}^{-1}(\tilde{S} - R_o R_o^T) \\ &\quad + (\tilde{S} - R_o R_o^T)\tilde{S}^{-1}\tilde{K}\tilde{S}^{-1}\tilde{S}\hat{\Lambda}^{-2}\hat{C}^T\tilde{S})A \end{aligned} \quad (87)$$

For the last step, this equation needs to be back-transformed applying the substitution from eq 77. Since  $\Lambda$  is symmetric, its eigenvectors are orthonormal, i.e.,  $R_{\Lambda} R_{\Lambda}^T = 1$ , and one finds

$$\begin{aligned} R_o R_o^T &= \tilde{S}\hat{\Lambda}^{-1/2}R_{\Lambda}R_{\Lambda}^T\Lambda^{-1/2}\hat{C}^T\tilde{S} \\ &= \tilde{S}\hat{\Lambda}^{-1}\hat{C}^T\tilde{S} = \tilde{S}\tilde{P}\tilde{S} \end{aligned} \quad (88)$$

Thus, the final expression for the Kohn–Sham matrix is given as

$$\begin{aligned} \frac{d\tilde{E}[\tilde{P}]}{dP} &= A^T(\tilde{S}\hat{\Lambda}^{-2}\hat{C}^T\tilde{K}(1 - \tilde{P}\tilde{S}) \\ &\quad + (1 - \tilde{S}\tilde{P})\tilde{K}\hat{\Lambda}^{-2}\hat{C}^T\tilde{S})A \end{aligned} \quad (89)$$

which indeed depends only on the inverse square of  $\Lambda$ , which has the size  $N_{\text{mo}} \times N_{\text{mo}}$ , which can be evaluated



through efficient Cholesky decomposition and does not require a diagonalization.

**E. MO Derivatives.** For wave function optimization algorithms that do not rely on the existence of a Kohn–Sham matrix but rather utilize the MO derivatives

$$U = \frac{dE}{dC} \quad (90)$$

such as the orbital transformation (OT) method<sup>66</sup> in Quickstep,<sup>35</sup> the explicit construction of a Kohn–Sham matrix can be omitted. This is certainly the case for the purified wave function fitting because in that case the corresponding auxiliary density matrix can be obtained from molecular coefficients. Instead of calculating the derivative of the energy with respect to the density matrix, it is thus sufficient to compute the MO derivatives

$$U_{\text{total}} = \frac{dE[P]}{dC} + \frac{d\tilde{E}[\tilde{P}]}{dC} \quad (91)$$

where only the second term is of interest here. Notice that the auxiliary density matrix  $\tilde{P}$  can be expressed either in terms of purified molecular coefficients

$$\tilde{P} = \tilde{C}\tilde{C}^T \text{ with } \tilde{C} = \hat{C}\Lambda^{-1} = A\hat{C}\Lambda^{-1} \quad (92)$$

or in terms of nonpurified molecular coefficients

$$\tilde{P} = \hat{C}\Lambda^{-1}\hat{C}^T \quad (93)$$

As a consequence, there exist two different approaches for calculating the desired MO derivative. The first method involves the auxiliary Kohn–Sham matrix

$$\frac{d\tilde{E}[\tilde{P}]}{dC} = \frac{d\tilde{E}}{dC} = \frac{d\tilde{E}}{d\tilde{P}} \frac{d\tilde{P}}{d\hat{C}} \frac{d\hat{C}}{dC} = \tilde{K} \frac{d\tilde{P}}{d\hat{C}} \frac{d\hat{C}}{dC} \quad (94)$$

and the second method directly takes the derivative of the energy with respect to the purified MO coefficients into account

$$\frac{d\tilde{E}[\tilde{P}]}{dC} = \frac{d\tilde{E}}{dC} = \frac{d\tilde{E}}{d\tilde{C}} \frac{d\tilde{C}}{dC} = \tilde{U} \frac{d\tilde{C}}{dC} \quad (95)$$

with

$$\tilde{U} = \frac{d\tilde{E}[\tilde{P}]}{d\tilde{C}} \quad (96)$$

The first case is algebraically straightforward, leading to

$$\frac{dE[\tilde{P}]}{dC} = 2(A^T \tilde{H} \hat{C} \Lambda^{-1}) - 2(A^T \tilde{S} \hat{C} \Lambda^{-1} \hat{C}^T \tilde{H} \hat{C} \Lambda^{-1}) \quad (97)$$

The second case is slightly more involved and requires the usage of the Cauchy integral formalism. The required derivative  $(d\tilde{C})/(dC)$  involves terms such as

$$\frac{d\Lambda^{-1/2}}{dC} \quad (98)$$

which can conveniently be expressed by

$$\frac{d\Lambda^{-1/2}}{dC} = \frac{1}{2\pi i} \oint f(z) \frac{1}{\Lambda - zI} \frac{d\Lambda}{dC} \frac{1}{\Lambda - zI} dz \quad (99)$$

with  $f(z) = z^{-1/2}$ . For the evaluation of the contour integral, the matrix  $\Lambda$  needs to be diagonalized. Using the same notation as in Appendix part D, i.e.,  $R_\Lambda D_\Lambda R_\Lambda^T = \Lambda$  and  $D_\Lambda$  defined through the eigenvalues  $\mu_i$  of  $\Lambda$ , the resulting matrix reads

$$\begin{aligned} N_{kj} &= \frac{1}{2\pi i} \oint \frac{z^{-1/2}}{(\mu_k - z)(\mu_j - z)} dz = \frac{1}{2\pi i} \oint g(z) dz \\ &= \text{Res}(g, \mu_k) + \text{Res}(g, \mu_j) \\ &= \begin{cases} \frac{\mu_k^{-1/2} - \mu_j^{-1/2}}{\mu_k - \mu_j} & k \neq j \\ -\frac{1}{2\mu_k^{-3/2}} & k = j \end{cases} \end{aligned} \quad (100)$$

The final result for the MO derivatives is thus given by

$$\frac{d\tilde{E}[\tilde{P}]}{dC} = A^T \tilde{U} \Lambda^{-1/2} + Q^T A C (Y + Y^T) \quad (101)$$

with

$$Y = R_\Lambda ([R_\Lambda^T C^T A^T \tilde{U} R_\Lambda] \otimes N) R_\Lambda^T \quad (102)$$

The first method, eq 97, has the advantage that only the inverse of  $\Lambda$  is needed, whereas the second method, eq 101, requires the diagonalization of  $\Lambda$ .

**F. Analytical Ionic Forces.** The derivative of the energy with respect to the atomic positions  $\mathbf{R}$

$$\frac{dE}{d\mathbf{R}} = \frac{dE[P]}{d\mathbf{R}} + \frac{d\tilde{E}[\tilde{P}]}{d\mathbf{R}} \quad (103)$$

can be calculated via the MO derivatives given in eq 101. That is, for the interesting term,

$$\frac{d\tilde{E}[\tilde{P}]}{d\mathbf{R}} = \frac{d\tilde{E}[\tilde{P}]}{d\tilde{C}} \frac{d\tilde{C}}{d\mathbf{R}} = \tilde{U} \frac{d\tilde{C}}{d\mathbf{R}} \quad (104)$$

Component-wise, this yields

$$\frac{d\tilde{E}[\tilde{P}]}{d\mathbf{R}} = \tilde{U}_{ab} \left[ \frac{dA}{d\mathbf{R}} C \Lambda^{-1/2} \right]_{ab} + \tilde{U}_{ab} \left[ A C \frac{d\Lambda^{-1/2}}{d\mathbf{R}} \right]_{ab} \quad (105)$$

with implicit summation over repeated indices. In order to evaluate the second term, it is possible to apply the same mathematical formalism as in eq 99, and the final result reads

$$\begin{aligned} \frac{d\tilde{E}[\tilde{P}]}{d\mathbf{R}} = & -\tilde{U}_{ab} \left[ \tilde{S}^{-1} \frac{d\tilde{S}}{d\mathbf{R}} \tilde{S}^{-1} Q C \Lambda^{-1/2} \right]_{ab} \\ & + \tilde{U}_{ab} \left[ \tilde{S}^{-1} \frac{dQ}{d\mathbf{R}} C \Lambda^{-1/2} \right]_{ab} \\ & + Y_{ab} \left[ C^T \frac{dQ^T}{d\mathbf{R}} A C \right]_{ab} \\ & - Y_{ab} \left[ C^T Q^T \tilde{S}^{-1} \frac{d\tilde{S}}{d\mathbf{R}} \tilde{S}^{-1} Q C \right]_{ab} \\ & + Y_{ab} \left[ C^T Q^T \tilde{S}^{-1} \frac{dQ}{d\mathbf{R}} C \right]_{ab} \end{aligned} \quad (106)$$

with  $Y_{ab}$  as in eq 102. This can further be simplified in terms of traces and becomes

$$\frac{d\tilde{E}}{d\mathbf{R}} = -\text{tr} \left( W_S^T \frac{d\tilde{S}}{d\mathbf{R}} \right) + \text{tr} \left( W_Q^T \frac{dQ}{d\mathbf{R}} \right) \quad (107)$$

with the weighted density matrices

$$W_{\tilde{S}} = \tilde{S}^{-1} \tilde{U} \Lambda^{-T/2} C^T A^T + A C Y C^T A^T \quad (108)$$

and

$$W_Q = \tilde{S}^{-1} \tilde{U} \Lambda^{-T/2} C^T + A C Y^T C^T + A C Y C^T \quad (109)$$

**G. Eigenvalues.** For purified wave function fitting, the Kohn–Sham matrix obtained through the McWeeny procedure or the Cauchy integral is not suitable for the calculation of orbital energies. This problem can be illustrated by evaluating eq 89 for identical primary and auxiliary basis sets, i.e.,  $A = 1$ ,  $\Lambda = 1$ ,  $\hat{C} = C$ ,  $\tilde{S} = S$ , and  $\tilde{P} = C C^T$ . In that case, the eigenvalues are given as

$$\begin{aligned} C^T \frac{d\tilde{E}[\tilde{P}]}{d\mathbf{P}} C = & C^T [A^T (\tilde{S} \hat{C} \Lambda^{-2} \hat{C}^T \tilde{K} (1 - \tilde{P} \tilde{S}) + \\ & (1 - \tilde{S} \tilde{P}) \tilde{K} \hat{C} \Lambda^{-2} \hat{C}^T \tilde{S}) A] C \\ = & C^T [S C C^T \tilde{K} - S C C^T \tilde{K} P S + \tilde{K} C C^T S - S P \tilde{K} C C^T S] C \\ = & C^T \tilde{K} C - C^T \tilde{K} P S + C^T \tilde{K} C - S P \tilde{K} C \\ = & C^T \tilde{K} C - C^T \tilde{K} C + C^T \tilde{K} C - C^T \tilde{K} C \end{aligned} \quad (110)$$

which is identically zero. Thus, eq 89 is not a suitable candidate for the calculation of the orbital energies, and a different approach needs to be taken into account. The obvious choice is to derive a similar expression as in eq 33 for the nonpurified wave function fitting, i.e., omitting the purification procedure. This assumption leads to the approximated Kohn–Sham matrix given in eq 34.

## References

- (1) Strout, D. L.; Scuseria, G. E. *J. Chem. Phys.* **1995**, *102*, 8448–8452.
- (2) Izmaylov, A. F.; Scuseria, G. E.; Frisch, M. J. *J. Chem. Phys.* **2006**, *125*, 104103.
- (3) Guidon, M.; Hutter, J.; VandeVondele, J. *J. Chem. Theory Comput.* **2009**, *5*, 3010–3021.
- (4) Spencer, J.; Alavi, A. *Phys. Rev. B* **2008**, *77*, 193110.
- (5) Williamson, A. J.; Rajagopal, G.; Needs, R. J.; Fraser, L. M.; Foulkes, W. M. C.; Wang, Y.; Chou, M.-Y. *Phys. Rev. B* **1997**, *55*, R4851–R4854.
- (6) Kent, P. R. C.; Hood, R. Q.; Williamson, A. J.; Needs, R. J.; Foulkes, W. M. C.; Rajagopal, G. *Phys. Rev. B* **1999**, *59*, 1917–1929.
- (7) Ochsenfeld, C.; White, C. A.; Head-Gordon, M. *J. Chem. Phys.* **1998**, *109*, 1663–1669.
- (8) Guidon, M.; Schiffmann, F.; Hutter, J.; VandeVondele, J. *J. Chem. Phys.* **2008**, *128*, 214104.
- (9) VandeVondele, J.; Hutter, J. *J. Chem. Phys.* **2007**, *127*, 114105.
- (10) Binkley, J. S.; Pople, J. A.; Hehre, W. J. *J. Am. Chem. Soc.* **1980**, *102*, 939–947.
- (11) Pietro, W. J.; Francl, M. M.; Hehre, W. J.; Defrees, D. J.; Pople, J. A.; Binkley, J. S. *J. Am. Chem. Soc.* **1982**, *104*, 5039–5048.
- (12) Harihara, P. C.; Pople, J. A. *Theor. Chim. Acta* **1973**, *28*, 213–222.
- (13) Krishnan, R.; Binkley, J. S.; Seeger, R.; Pople, J. A. *J. Chem. Phys.* **1980**, *72*, 650–654.
- (14) Jensen, F. *J. Chem. Phys.* **2001**, *115*, 9113.
- (15) Jensen, F. *J. Chem. Phys.* **2002**, *116*, 7372.
- (16) Jensen, F. *J. Phys. Chem. A* **2007**, *111*, 11198–11204.
- (17) Weigend, F.; Ahlrichs, R. *Phys. Chem. Chem. Phys.* **2005**, *7*, 3297–3305.
- (18) Weigend, F. *Phys. Chem. Chem. Phys.* **2002**, *4*, 4285–4291.
- (19) Boman, L.; Koch, H.; de Merás, A. S. *J. Chem. Phys.* **2008**, *129*, 134107.
- (20) Sodt, A.; Head-Gordon, M. *J. Chem. Phys.* **2008**, *128*, 104106.
- (21) Liang, W.; Head-Gordon, M. *J. Phys. Chem. A* **2004**, *108*, 3206–3210.
- (22) Friesner, R. A. *Chem. Phys. Lett.* **1985**, *116*, 39–43.
- (23) Neese, F.; Wennmohs, F.; Hansen, A.; Becker, U. *Chem. Phys.* **2009**, *356*, 98–109.
- (24) Yanai, T.; Fann, G. I.; Gan, Z.; Harrison, R. J.; Beylkin, G. *J. Chem. Phys.* **2004**, *121*, 6680–6688.
- (25) Gygi, F.; Baldereschi, A. *Phys. Rev. B* **1986**, *34*, 4405–4408.
- (26) Todorova, T.; Seitsonen, A. P.; Hutter, J.; Kuo, I.-F. W.; Mundy, C. J. *J. Phys. Chem. B* **2006**, *110*, 3685–3691.
- (27) Sorouri, A.; Foulkes, W. M. C.; Hine, N. D. M. *J. Chem. Phys.* **2006**, *124*, 064105.
- (28) Perdew, J. P.; Burke, K.; Ernzerhof, M. *Phys. Rev. Lett.* **1996**, *77*, 3865–3868.
- (29) Ernzerhof, M.; Perdew, J. P. *J. Chem. Phys.* **1998**, *109*, 3313–3320.
- (30) McWeeny, R. *Rev. Mod. Phys.* **1960**, *32*, 335–369.
- (31) Palser, A. H. R.; Manolopoulos, D. E. *Phys. Rev. B* **1998**, *58*, 12704–12711.
- (32) Niklasson, A. M. N.; Tymczak, C. J.; Challacombe, M. *J. Chem. Phys.* **2003**, *118*, 8611–8620.
- (33) Rinehart, R. F. *Proc. Amer. Math. Soc.* **1956**, *7*, 2–5.
- (34) The CP2K developers group. <http://cp2k.berlios.de/> (accessed June 3, 2010).
- (35) VandeVondele, J.; Krack, M.; Mohamed, F.; Parrinello, M.; Chassaing, T.; Hutter, J. *Comput. Phys. Commun.* **2005**, *167*, 103.

- (36) Lippert, G.; Hutter, J.; Parrinello, M. *Theor. Chem. Acc.* **1999**, *103*, 124.
- (37) Krack, M.; Parrinello, M. *Phys. Chem. Chem. Phys.* **2000**, *2*, 2105–2112.
- (38) Goedecker, S.; Teter, M.; Hutter, J. *Phys. Rev. B* **1996**, *54*, 1703–1710.
- (39) Krack, M. *Theor. Chem. Acc.* **2005**, *114*, 145–152.
- (40) Gómez-Abal, R.; Li, X.; Scheffler, M.; Ambrosch-Draxl, C. *Phys. Rev. Lett.* **2008**, *101*, 106404.
- (41) Duchemin, I.; Gygi, F. *Comput. Phys. Commun.* **2010**, *181*, 855–860.
- (42) Goerigk, L.; Grimme, S. *J. Chem. Theory Comput.* **2010**, *6*, 107–126.
- (43) Goerigk, L.; Grimme, S. GMTKN24. <http://toc.uni-muenster.de/GMTKN/GMTKNmain.html> (accessed June 3, 2010).
- (44) Perdew, J. P.; Ernzerhof, M.; Burke, K. *J. Chem. Phys.* **1996**, *105*, 9982–9985.
- (45) Perdew, J. P.; Ernzerhof, M.; Burke, K. *Int. J. Quantum Chem.* **1997**, *64*, 285–295.
- (46) Ernzerhof, M.; Scuseria, G. E. *J. Chem. Phys.* **1999**, *110*, 5029–5036.
- (47) Grimme, S. *J. Comput. Chem.* **2006**, *27*, 1787–1799.
- (48) Heyd, J.; Scuseria, G. E.; Ernzerhof, M. *J. Chem. Phys.* **2003**, *118*, 8207–8215.
- (49) Heyd, J.; Scuseria, G. E.; Ernzerhof, M. *J. Chem. Phys.* **2006**, *124*, 219906; Erratum.
- (50) Boys, S. F.; Bernardi, F. *Mol. Phys.* **1970**, *19*, 553–&.
- (51) Ruzsinszky, A.; Perdew, J. P.; Csonka, G. I. *J. Phys. Chem. A* **2005**, *109*, 11006–11014.
- (52) Sodupe, M.; Bertran, J.; Rodriguez-Santiago, L.; Baerends, E. *J. Phys. Chem. A* **1999**, *103*, 166–170.
- (53) Becke, A. D. *J. Chem. Phys.* **1993**, *98*, 1372–1377.
- (54) Lee, C.; Yang, W.; Parr, R. G. *Phys. Rev. B* **1988**, *37*, 785–789.
- (55) Maršálek, O.; Elles, C. G.; Pieniazek, P. A.; VandeVondele, J.; Bradforth, S. E.; Jungwirth, P. Manuscript in preparation.
- (56) Sulpizi, M.; Raugei, S.; VandeVondele, J.; Carloni, P.; Sprik, M. *J. Phys. Chem. B* **2007**, *111*, 3669.
- (57) Schäfer, A.; Huber, C.; Ahlrichs, R. *J. Chem. Phys.* **1994**, *100*, 5829.
- (58) Becke, A. D. *J. Chem. Phys.* **1993**, *98*, 5648–5852.
- (59) Vosko, S. H.; Wilk, L.; Nusair, M. *Can. J. Phys.* **1980**, *58*, 1200–1211.
- (60) Hura, G.; Sorenson, J. M.; Glaeser, R. M.; Head-Gordon, T. *J. Chem. Phys.* **2000**, *113*, 9140–9148.
- (61) Zhang, Y.; Yang, W. *Phys. Rev. Lett.* **1998**, *80*, 890.
- (62) McGrath, M. J.; Siepmann, J. I.; Kuo, I. F. W.; Mundy, C. J.; VandeVondele, J.; Hutter, J.; Mohamed, F.; Krack, M. *ChemPhysChem* **2005**, *6*, 1894–1901.
- (63) Schmidt, J.; VandeVondele, J.; Kuo, I. F. W.; Sebastiani, D.; Siepmann, J. I.; Hutter, J.; Mundy, C. J. *J. Phys. Chem. B* **2009**, *113*, 11959–11964.
- (64) McGrath, M. J.; Siepmann, J. I.; Kuo, I. F. W.; Mundy, C. J.; VandeVondele, J.; Hutter, J.; Mohamed, F.; Krack, M. *J. Phys. Chem. A* **2006**, *110*, 640–646.
- (65) Vega, C.; Abascal, J. L. F.; Conde, M. M.; Aragonés, J. L. *Faraday Discuss.* **2009**, *141*, 251–276.
- (66) VandeVondele, J.; Hutter, J. *J. Chem. Phys.* **2002**, *118*, 4365–4369.

CT1002225