Redox Entropy of Plastocyanin: Developing a Microscopic View of Mesoscopic Polar Solvation

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We report applications of analytical formalisms and Molecular Dynamics (MD) simulations to the calculation of redox entropy of plastocyanin metalloprotein in aqueous solution. The goal of our analysis is to establish critical components of the theory required to describe polar solvation at the mesoscopic scale. The analytical techniques include a microscopic formalism based on structure factors of the solvent dipolar orientations and density and continuum dielectric theories. The microscopic theory employs the atomistic structure of the protein with force-field atomic charges and solvent structure factors obtained from separate MD simulations of the homogeneous solvent. The MD simulations provide linear response solvation free energies and reorganization energies of electron transfer in the temperature range 280–310 K. We found that continuum models universally underestimate solvation entropies, and a more favorable agreement is reported between the microscopic calculations and MD simulations. The analysis of simulations also suggests that difficulties of extending standard formalisms to protein solvation are related to the inhomogeneous structure of the solvation shell at the protein-water interface combining islands of highly structured water around ionized residues along with partial dewetting of hydrophobic patches. Quantitative theories of electrostatic protein hydration need to incorporate realistic density profile of water at the protein-water interface.

I. INTRODUCTION

Calculations of the thermodynamics of hydrated biopolymers present a significant challenge to theoretical algorithms. In many cases the problem can be treated by numerical simulations with force fields assigned to the biomolecule (solute) and water (solvent). The obvious difficulty is the large computational load and still existing uncertainties in the treatment of the long-range electrostatics. The problem, however, becomes more nontrivial when derivatives of thermodynamic potentials, e. g. redox entropy, need to be computed or when the solvation thermodynamics changes on the time- and length-scale unattainable to standard Molecular Dynamics (MD) protocols, e. g. in problems related to protein folding. In all such cases, coarse graining of the system is required and that can be done on various length-scales. Dielectric continuum algorithms, solving the boundary Poisson problem, are computationally very efficient. In this approximation, all length-scales below the largest distance of microscopic correlations (density and/or polarization) are averaged out into a continuum surrounding the solute. These approaches are normally represented by either direct solution of the Poisson-Boltzmann equation on the real-space grid or even more approximate formalisms under the umbrella of the generalized Born approximation.

When the cavity cut out by the solute from the continuum dielectric is properly parametrized, equations of continuum electrostatics provide a reasonable estimate of the solvation Gibbs energy. The fundamental problem of this approach is that the local structure of the solvent around the solute, averaged out in the continuum limit, is what effectively forms the dielectric cavity. While this structure can be parametrized by choosing proper van der Waals (vdW) radii this parametrization needs to be re-done every time the thermodynamic state of the solvent changes. This difficulty makes continuum formalisms unreliable for the calculation of derivatives of the Gibbs energy, for instance the entropy of solvation. In addition, the surface of a protein is highly heterogeneous combining hydrophobic patches exposing non-polar residues and hydrophilic patches made of ionized/polar residues. While the water structure is rigid around ionized residues, probably resembling the well-studied case of solvation around simple ions, water is much less structured at hydrophobic patches with the potential for dewetting or/and oscillations of the water occupation. It is clear that simplistic continuum does not represent this complex reality and one needs to incorporate the ability of the solvent to fluctuate into the solvation model.

The goal of this paper is to extend the microscopic view of solvation in polar solvents, which we have been developing in the past in application to small and medium-size solutes, to solvation of solutes of mesoscopic dimension, biopolymers in the first place. The length-scale of this problem presents an obvious obstacle to numerical simulation techniques. On the other hand, the same length-scale allows one to hope that some of the short-range features of the solvent structure around the solute, making solvation of small molecules so specific, might average out on a larger scale. If this averaging is realized for solvation of biopolymers, it would allow coarse-grained models to efficiently operate in this field complementing direct numerical simulations. Our approach to the problem is to coarse-grain the solvent response into a number of solvent correlation functions (structure factors) representing the nuclear modes affecting electrostatic solvation. The microscopic nature of the solvent response
is then incorporated into the wave-vector dependence of these structure factors efficiently filtering out the length-scales insignificant for solvation.

This study poses the central question for the future development of such techniques: What are the solvent modes which play the central role in the thermodynamics of mesoscopic polar solvation and what are the theory ingredients critical for capturing the basic physics of large-scale solvation? We study this problem here by carrying out extensive MD simulations of solvation of plastocyanin (PC) in TIP3P water in the temperature range 280–310 K. This fully atomistic approach is compared to continuum electrostatics and to our microscopic algorithm, operating with \( k \)-space correlation functions, which was designed to scale efficiently on the mesoscopic length-scale.

Plastocyanin from spinach is a single polypeptide chain of 99 residues forming a \( \beta \)-sandwich, with a single copper ion coordinated by 2 sulfurs from cysteine and methionine and 2 nitrogens from histidine residues (Fig. 1). The presence of the copper ion, which can change redox state, allows PC to function as a mobile electron carrier in the photosynthetic apparatus of plants and bacteria. It accepts an electron from ferrocytochrome \( f \) and diffusionaly carries it to another docking location where the electron is donated to the oxidized form of Photosystem I.

FIG. 1: Structure of plastocyanin: the active site includes copper ion (green), 2 histidines (blue), methionine (red), and cysteine (orange) residues.

The redox thermodynamics of PC has been characterized experimentally\(^21,22,23\) and combined quantum/simulation calculations have been done as well\(^24,25\). The early focus of the theoretical studies had been on unusually high redox potentials of copper proteins, which was assigned to the non-traditional distorted tetrahedral coordination on the copper ion\(^26,27,28\). In particular, the Cu-S bond to methionine is unusually long and is actually broken in the reduced state of PC at pH \(< 3.8\)\(^29\). The protein is also highly charged at pH \(\approx 7\) \((-9.0\) in reduced state and \(-8.0\) in oxidized states). The charge is made by 15 negatively charged deprotonated residues (9 glutamic and 6 aspartic acids) and six positively charged lysine residues with amino groups protonated (Fig. 2). The asymmetric charge distribution located on the protein surface creates the dipole moment of 2200 D in the oxidized state (Ox) and of 2470 D in the reduced state (Red), both numbers are calculated relative to the center of partial charges.

The redox potential of the protein includes a component from the local ligand field of the active site and the Gibbs energy of solvation. The computation of the former requires quantum mechanics, making the problem of calculating the overall redox potential a very non-trivial exercise\(^25,30,31\). Calculations of solvation thermodynamics can be reasonably accomplished using partial atomic charges parametrized from quantum calculations in the vacuum. The experimental input comes from measurements of redox entropy\(^21,22,32\) since the temperature-independent ligand-field component is expected to vanish in the temperature derivative.

II. MICROSCOPIC SOLVATION MODEL

The principal idea of the microscopic solvation model is to reduce the problem of solvation of an arbitrary solute in a polar solvent to a formalism combining two major blocks: electrostatics of an isolated solute and non-local correlation functions of the pure solvent. The idea of assembling separate solute and solvent properties in a solvation model is obviously not new going back to Born\(^33\) and Onsager\(^34\) and all the subsequent development of continuum electrostatics in application to solvation\(^6,7,8,9,35\). The advantage of our approach is in avoiding the necessity to know the microscopic solute-solvent structure, which is the main complexity of microscopic solvation models and is also their main advantage when the problem is successfully resolved by either solving integral equation\(^36\) or by applying time-dependent\(^37,38\) or equilibrium\(^39,40,41\) density functional methods. Inserting a solute into a dense liquid creates a
significant distortion of its structure, and the incomplete account of the coupling between the short-range density profile around the solute with the long-range polarization field is perhaps the weakest part of our formulation when applied to small solutes.\textsuperscript{44} This deficiency is almost completely off-set by averaging of the density profile around a nano-scale solute (see below). On the other hand, a strong side of our formalism, its ability to treat solvation of large solutes of irregular shape and arbitrary charge distribution\textsuperscript{44,45} becomes particularly useful in application to protein solvation.

The reduction of the many-body solvation problem to an irreducible representation in terms of a few basic correlation functions depends on the symmetry of the solute-solvent interaction potential. The number of correlation functions is known to grow with increasing the rank of the many-body solvation problem to an irreducible representation in terms of a few basic correlation functions.\textsuperscript{45} The main challenge of this formalism, as well as of other Gaussian solvation theories,\textsuperscript{46} is how to connect the inhomogeneous response function \( \chi(k_1, k_2) \) to the shape of the solute repulsive core and the self-correlation functions of the solvent modes affecting solvation. Two modes naturally appear in most theories: dipolar (orientational) polarization and density fluctuations.\textsuperscript{19,40,59,60}

This observation opens up a significant simplification of the calculation algorithms. Instead of solving the inhomogeneous problem of restructuring the solvent in an external field of the solute, it appears to be sufficient to look at the statistics of solvent fluctuations around the repulsive core of the solute. This strategy is used here and we will base our calculations on the relation

\[
- \mu_{0s} = (\beta/2)\langle (\delta V_{0s})^2 \rangle_0,
\]

where \( \delta V_{0s} = V_{0s} - \langle V_{0s} \rangle_0 \) and \( \beta = 1/(k_B T) \).

By using the interaction potential according to Eq. (11), one can re-write Eq. (2) in the form typical for Gaussian (LRA) models of solvation\textsuperscript{53,56}

\[
- \mu_{0s} = \frac{1}{2} \tilde{E}_0(k_1) \ast \chi(k_1, k_2) \ast \tilde{E}_0(k_2).
\]

Here, the 2-rank tensor \( \chi(k_1, k_2) \) is the response function\textsuperscript{57} of the system composed of a dipolar solvent and a solute to a weak field of the solute. The inhomogeneous character of the problem is reflected by the fact that \( \chi(k_1, k_2) \) depends on two wave-vectors, \( k_1 \) and \( k_2 \), separately and not on \( k_1 - k_2 \), as is the case with response functions of homogeneous solvents. The asterisk in Eq. (3) refers to both the tensor contraction and integration in inverted \( k \)-space. In addition, \( \tilde{E}_0(k) \) is the Fourier transform of the electric field of the solute defined by the integral limited to the solvent volume \( \Omega \):

\[
\tilde{E}_0(k) = \int_{\Omega} E_0(r) e^{i k \cdot r} dr.
\]

The shape of the solute thus enters both the response function \( \chi(k_1, k_2) \) and the field Fourier transform \( E_0(k) \). The charge distribution of the solute, which determines the electric field \( \tilde{E}_0(k) \), is given by its electronic density and is commonly represented by partial atomic charges.

The following Eq. (5) allows one to split the 2-rank tensor \( \chi_s(k) = (\beta/\Omega) \langle |\delta P(k)|^2 \rangle \) into the longitudinal and transverse dyads\textsuperscript{61}

\[
\chi_s(k) = \frac{3y}{4\pi} \left[ J^L S^L(k) + J^T S^T(k) \right],
\]

where \( J^L = \hat{k} \cdot \hat{\nabla} \), \( J^T = 1 - \hat{k} \cdot \hat{\nabla} \). In Eq. (5), \( y \) is the effective density of both permanent and induced dipoles in the liquid which commonly appears in theories of dielectrics\textsuperscript{62}

\[
y = (4\pi/9)3m^2 \rho + (4\pi/3)\alpha.
\]

In Eq. (5), \( \alpha \) is the dipolar polarizability of the solvent particle.
The scalar functions $S^L(k)$ and $S^T(k)$ in Eq. (1) are, correspondingly, the longitudinal and transverse structure factors of dipolar fluctuations of the homogeneous solvent (see below). The $k = 0$ values of these structure factors are related to the dielectric constant, $\varepsilon_s$, by the following equations:

\[
S^L(0) = (\varepsilon_s - 1)/(3y\varepsilon_s),
\]

\[
S^T(0) = (\varepsilon_s - 1)/(3y).
\]  

(7)

Also, the trace of $\chi_s(0)$ over the Cartesian projections is proportional to the Kirkwood $g$-factor,\textsuperscript{62}

\[
g_k = \frac{1}{3} \left[ S^L(0) + 2S^T(0) \right].
\]  

(8)

The expansion of the solvation chemical potential in the Mayer functions corresponding to the solute-solvent interaction potential leads to the following form for the response function,\textsuperscript{13,19}

\[
\chi(k_1, k_2) = \chi_p(k_1, k_2) + \chi_d(k_1, k_2),
\]  

(9)

where

\[
\chi_p(k_1, k_2) = \chi_s(k_1)\delta_{k_1, k_2},
\]

(10)

In Eq. (10), $\Omega_k$ is the Kronecker symbol and $\chi_d(k_1, k_2)$ in Eq. (9) is the component of the response originating from the local fluctuations of the solvent density around the solute.\textsuperscript{14}

\[
\chi_d(k_1, k_2) = (3y/8\pi)(1 - S(k_1))\delta_{k_1, k_2}.
\]  

(11)

Here, $S(k) = N^{-1} \langle |\delta\rho(k)|^2 \rangle$ is the density-density structure factor of the homogeneous solvent and $N$ is the number of solvent molecules. In addition, $\delta_{0}(k)$ is the Fourier transform of the step function $\delta_{0}(r)$ defining the solute shape. It is equal to unity for $r$ inside the solute and is zero otherwise.

The problem with the direct perturbation result in Eq. (9) is that it contains the transverse polarization response function $S^T(k)$ diverging in its continuum, $k \to 0$, limit as the solvent dielectric constant goes to infinity [Eq. (7)]. The problem is really caused by the non-spherical shape of the solute. The electric field of the solute charges is longitudinal. However, when the symmetry of the solute is different from the symmetry of the charge distribution in a sense that the cavity boundary does not coincide with the equipotential surface of the Mayer function, the Fourier integral in Eq. (4) generates a transverse component of $\tilde{E}_0(k)$. Notice that this is always the case when electron transfer reactions are considered.\textsuperscript{63} A transverse component in $\tilde{E}_0(k)$ then results in a “transverse catastrophe” for solvents of high polarity. The problem was well recognized in early studies\textsuperscript{19,59,64} which suggested to use only the longitudinal component of the field $\tilde{E}_0(k)$. As a matter of fact, the problem lies in the response function of the dipolar polarization field which needs to be re-normalized with the account of the solute repulsive core, a procedure similar to applying boundary conditions to the Poisson equation of continuum electrostatics.

The Li-Kardar-Chandler\textsuperscript{56,64} Gaussian model allows one to achieve a correct renormalization of the inhomogeneous polarization response function $\chi_p(k_1, k_2)$ eliminating the “transverse catastrophe”\textsuperscript{65} This approach introduces another simplification by replacing all the short-range solute-solvent interactions by hard-core repulsions. This simplification, however, leads to an exact solution for the $k$-space response functions with the result\textsuperscript{65}

\[
\chi_p(k_1, k_2) = \chi_s(k_1)\delta_{k_1, k_2} - \chi''(k_1)\theta_0(k_1 - k_2)\chi_s(k_2).
\]  

(12)

The second summand in Eq. (12) is the correction of the response function of the homogeneous solvent, appearing in Eq. (10), by the repulsive core of the solute. The response function $\chi''(k_1)$ then incorporates both $\chi_s$ and the information about the solute shape.\textsuperscript{13,65}

A direct substitution of Eq. (12) into Eq. (3) results in a 6D integral convolution in $k$-space which is not numerically tractable. In order to arrive at a computationally efficient procedure, a mean-field approximation was introduced\textsuperscript{65} which replaces the inhomogeneous electric field of the solvent inside the solute by a mean cavity field:

\[
F_0 = \frac{3}{8\pi} \int_{\Omega} E_0 \cdot \mathbf{D}_r \frac{dr}{r^3},
\]  

(13)

where

\[
f = \frac{2(\varepsilon_s - 1)}{2\varepsilon_s + 1}.
\]  

(14)

Here, $\mathbf{D}_r = 3\hat{r} - 1$ is the 2-rank dipolar tensor with $\hat{r} = \mathbf{r}/r$. $F_0$ becomes the Onsager reaction field\textsuperscript{13,65} for a spherical solute with point dipole located at the center.

The mean-field approximation reduces the problem of calculating the solvation thermodynamics to a numerically tractable 3D integral in $k$-space. The chemical potential of solvation then becomes a sum of two components arising from the longitudinal (L) and transverse (T) polarization fluctuations, $\mu_{0s}^L$, and a third component arising from the density fluctuations, $\mu_{0s}^d$:

\[
\mu_{0s} = \mu_{0s}^L + \mu_{0s}^T + \mu_{0s}^d.
\]  

(15)

The transverse component $\mu_{0s}^T$ is defined by $k$-integral of the transverse projection of the solute filed, $\tilde{E}_0^T(k)$, with the transverse polarization structure factor

\[
-\mu_{0s}^T = g_k^{-1} S^L(0) \frac{3y}{8\pi} \int \frac{dk}{(2\pi)^3} |\tilde{E}_0^T(k)|^2 S^T(k).
\]  

(16)

The transverse field component is defined by subtracting the longitudinal projection

\[
\tilde{E}_0^T(k) = \tilde{k} (\tilde{k} \cdot \tilde{E}_0)
\]  

(17)

from the total inverted-space field $\tilde{E}_0$:

\[
\tilde{E}_0^T(k) = \tilde{E}_0(k) - \tilde{E}_0^L(k).
\]  

(18)
Equation (16) is the main result of the application of the Gaussian model to polar solvation. It replaces $S^T(k)$ of the direct perturbation expansion in Eqs. (9) and (12) with the renormalized function:

$$3yS^T(k) \rightarrow (3yS^L(0)/g_k)S^T(k).$$  \hspace{2cm} (19)

The $k = 0$ limit of the transverse response changes from $3yS^T(0) = \epsilon_s - 1$ to $3(\epsilon_s - 1)/(2\epsilon_s + 1)$ thus eliminating the “transverse catastrophe” of direct perturbation expansions.

The component $\mu^L_{0s}$ of the solvation chemical potential in Eq. (15) is obtained by inverted-space integration with the longitudinal polarization structure factor:

$$-\mu^L_{0s} = 3y \frac{S^L}{8\pi} \int \frac{dk}{(2\pi)^3} S^L(k) \left[ |\tilde{E}_0^L|^2 - |\tilde{E}_0^T|^2 \frac{f}{F_0} \cdot \frac{\tilde{E}_0^L}{F_0} \right].$$  \hspace{2cm} (20)

There is a significant physics behind the appearance of the transverse field in the brackets of Eq. (20). Longitudinal dipolar polarization is short-ranged and thus does not propagate over macroscopic distances. On the contrary, transverse polarization is long-ranged. Therefore, inducing transverse polarization modifies the electric field acting on the solvent dipoles resulting in the second term in the brackets in Eq. (20).

Solute

\begin{footnotesize}
| Structure: | Charge Distribution: |
|---|---|
| X-ray, NMR, Molecular mechanics | Empirical Force Field, QM calculations |
\end{footnotesize}

\begin{itemize}
\item $E_s(0)$
\item $FFT$
\item $\tilde{E}_s(k)$
\item $\int |\tilde{E}_s(k)|^2 S(k)dk \rightarrow$ Solvation Free Energy
\end{itemize}

Solvent

Polarization structure factors:

Liquid-state theories, Computer Simulations

FIG. 3: Diagram of the computational algorithm.

The density component in Eq. (15) can formally be obtained by multiplying the response function $\chi_d(k_1,k_2)$ [Eq. (11)] with the Fourier transforms of the electric field and integrating over $k_1$ and $k_2$. This, however, results in a 6D convolution integral to be avoided in numerical applications. An alternative approach is to use direct-space integration when the density component becomes

$$-\mu^d_{0s} = 3yF(r) * F^{-1} [(1 - S(k))\theta_0(k)].$$  \hspace{2cm} (21)

Here, $F(r) = (8\pi)^{-1}E_0^2(r)$ is the density of the electrostatic field energy and the asterisk indicates integration in real space over the volume $\Omega$ occupied by the solvent. In addition, $F^{-1}$ is the inverse Fourier transform of the function indicated in the brackets.

Solvation by the overall dipolar polarization, including nuclear and electronic components, was considered in the formalism outlined above. For solvation problems relevant to spectroscopy and charge-transfer reactions nuclear component of polarization needs to be extracted. This is achieved by replacing the density $y$ [Eq. (6)] of all, permanent and induced, dipoles in the equations above with the density of permanent dipoles only, $y \rightarrow y_p = (4\pi/9)\beta m^2\rho$. In addition, the $k = 0$ values of the structure factors need to be modified to account for screening of the dipolar interactions by the high-frequency dielectric constant $\epsilon_\infty$. The $k = 0$ values for these nuclear structure factors, $S_{n}^{L,T}(k)$, now become

$$S_{n}^{L}(0) = (\epsilon_\infty^{-1} - \epsilon_s^{-1})/(3y_p),$$  \hspace{2cm} (22)

Once the $k = 0$ values for the structure factors are fixed by Eqs. (7) and (22), the scalar functions $S^L(k)$ and $S^T(k)$ can be calculated from our parametrization scheme, parametrized polarization structure factors (PPSF). This analytical route to the polarization structure factors is tested here by comparing the results of solvation calculations employing the PPSF to the direct use of $S^{L,T}(k)$ from MD simulations (see below).

III. COMPUTATIONAL ALGORITHM

The computational algorithm is outlined in Fig. 3. The solute is parametrized by coordinates $r_j$, vdW radii $a_j$, and partial charges $q_j$ of the atoms. The electric field of the solute is calculated at points $r_n$ of the $N \times N \times N$ grid built on the $L \times L \times L$ cube:

$$E_0(r_n) = \sum_{j=1}^{N_q} \frac{q_j(r_n - r_j)}{|r_n - r_j|^3},$$  \hspace{2cm} (23)

where $N_q$ is the number of solute charges. The array $E_0(r_n)$ is converted to inverted space by using fast Fourier transform technique. The field $\tilde{E}_0(k)$ is split into longitudinal and transverse components and used in $k$-integration in Eqs. (10) and (20) with the corresponding structure factors of the dipolar polarization. As is illustrated in Fig. 3 the calculation input is subdivided into two separate components related to the solute and solvent properties. Details of calculation for each of these are given below.

A. Solute

The Fourier transform of the Coulomb field is conditionally convergent. Therefore, in order to avoid numer-
tical divergence in the Fourier integral, real space is divided into three regions: hard core of the solute (region 1), region outside a sphere of radius \( R \) (region 3), and the region between the solute surface and the sphere (region 2, Fig. 3). The Fourier integral then becomes

\[
\tilde{E}_0(k) = \int_{r<R} E_0(r)e^{-i\mathbf{k}\cdot\mathbf{r}}\,d\mathbf{r} + \tilde{E}_R,
\]

(24)

where the first integral is taken over region 2 and

\[
\tilde{E}_R(k) = \int_{r>R} E_0(r)e^{-i\mathbf{k}\cdot\mathbf{r}}\,d\mathbf{r}.
\]

(25)

The center of the sphere is taken at the center of the charge distribution defined by the relation

\[
\mathbf{r}_q = \frac{\sum_{j=1}^{N_q} q_j \mathbf{r}_j}{\sum_{j=1}^{N_q} q_j}.
\]

(26)

The radius of the cut-off sphere \( R \) is chosen to minimize the part of the grid which is used in numerical calculation of the Fourier transform. In our calculations, the radius \( R \) is chosen by adding the solvent diameter \( \sigma \) to the largest distance from \( \mathbf{r}_q \) to the solvent-accessible surface (SAS) of the solute (vdW radii of the surface atoms plus the radius of the solvent molecule).

![FIG. 4: Separation of real space into regions for the calculation of the Fourier transform of the solute electric field [Eq. (24)]. The Fourier transform is calculated numerically in region 2 and analytically [Eq. (27)] in region 3. The field is set equal to zero within the hard repulsive core of the solute (region 1).](image)

The Fourier transform outside the sphere can be evaluated analytically. For the location of charges relative to the center of charge given as \( \mathbf{s}_j = \mathbf{r}_j - \mathbf{r}_q \), the solution for \( \tilde{E}_R(k) \) can be obtained by expanding \( E_0(r) \) in \( s_j/R < 1 \):

\[
\tilde{E}_R(k) = -4\pi e \sum_{j} q_j \sum_{n=1}^{\infty} \left( \frac{s_j}{R} \right)^{n-1} j_{n-1}(kR) \cdot \frac{k}{k}
\]

\[
\left[ \hat{s}_j P_{n-1}^\prime(\cos \theta_j) - \hat{k} P_n^\prime(\cos \theta_j) \right].
\]

Here, \( \cos \theta_j = \hat{s}_j \cdot \hat{k} \), \( j_n(x) \) is the spherical Bessel function, and \( P_n(\cos \theta_j) \) is the Legendre polynomial.

### B. Charging Scheme

The formal charge of the copper ion is +2 and of the cysteine sulfur is −1 in the oxidized state of PC. The charge is, however, delocalized among the ligands and the metal center. The main factor in this delocalization is a strong covalency of a copper-sulfur (Cys) bond. Calculations by Solomon and co-workers\(^{27}\) assign 40% of spin density of an unpaired electron to copper and 36% to cysteine’s sulfur in the ground state of oxidized protein. The extent of delocalization varies significantly depending on the level of quantum mechanical calculations used\(^{26,69,70}\). The electron-nuclear double resonance (ENDOR) experiments\(^{26}\) which require additional calibration on quantum calculations, result in the following net charges on the residues coordinating copper\(^{27}-0.25\) (His), −0.51 (Cys), and −0.04 (Met). The more recent mapping of the electron spin density to NMR relaxation\(^{26}\) gives a much lower extent of delocalization: −0.11 (Cys), −0.025 (His), 0 (Met).

The uncertainties in the extent of electron delocalization pose the question of their impact on the calculation of the redox thermodynamics. In order to study this question, we have performed calculations of the solvation part of the redox potential and the corresponding entropy using different charge sets. Set I is chemically fake assuming charge +2 on copper in Ox state and the net charge of −1 on cysteine. The negative charge is placed on cysteines sulfur in addition to −0.23 from CHARMM22 protein parametrization. The rest of the protein charges are from the standard CHARMM parametrization. The reduced state for Set I is obtained by changing the metal charge to +1. The charges for copper and its four ligands are summarized in Table I. Set II is based on the charging scheme used by Ullmann et al\(^{25}\) for the oxidized state of PC. The reduced form is obtained by placing an extra negative charge on copper and its three ligands, \( N_\delta \) (His87), \( N_\delta \) (His37), \( S_1 \) (Cys84), and \( S_2 \) (Met92) in proportion extracted from NMR experiments (Table II). Finally, a third charge distribution is completely parametrized at the DFT level for the charges and force constants of the copper and ligand atoms and consistent with the Amber force field\(^{26}\). In addition, Amber FF03 parametrization\(^{25}\) was applied to all non-ligand residues (Set II). There were various numbers of TIP3P water molecules for each of the charge distributions: 5,874 (Set I), 5,886 (Set II), and 4,628 (Set III).

We ran separate simulations (ca. 5 ns) for each charging scheme to find that the results are not strongly affected by the choice of atomic charges (Table III). This was also noticed in some other recent simulations\(^{22,26}\). We have therefore implemented charge scheme II in all simulations reported here since it presents a reasonable balance between being simple and realistic.
TABLE I: Atomic partial charges for copper and its four ligands in the reduced (Red) and oxidized (Ox) states of PC.

| Set | Cu | N$_a$ | N$_b$ | S$_c$ | S$_d$ | Cu | N$_a$ | N$_b$ | S$_c$ | S$_d$ |
|-----|----|------|------|------|------|----|------|------|------|------|
| I   | 1.0 | -0.7 | -0.7 | -1.23 | -0.09 | 2.0 | -0.7 | -0.7 | -1.23 | -0.09 |
| II  | -0.49 | -0.445 | -0.495 | -0.369 | -0.24 | 0.35 | -0.42 | -0.47 | -0.26 | -0.24 |

$^a$His87
$^b$His37
$^c$Cys84
$^d$Met92

TABLE II: Temperature dependent $\langle V_{DD}/Ox \rangle$ (eV) for PC(Ox). The results are obtained from MD simulations, DelPhi calculations (with vdW and SS cavities), and from NRFT calculations. The calculations were done with three charge distributions of the active site (I-III, see text for description).

| T/K  | MD    | DelPhi(vdW) | DelPhi(SS) | NRFT$^a$ |
|------|-------|-------------|------------|----------|
|      | I     | II          | III         | I        | II       | III       |          |
| 280  | -68.29 | -69.46      | -68.37     | -104.86  | -102.07  | -99.76    | -46.37   | -46.37   | -104.6(-37.2) |
| 285  | -67.69 | -66.01      | -65.76     | -104.7   | -101.93  | -99.62    | -46.84   | -46.41   | -100.7(-35.1) |
| 290  | -67.73 | -66.91      | -65.84     | -104.65  | -101.88  | -99.57    | -46.81   | -47.39   | -99.3(-34.5)  |
| 295  | -67.67 | -66.97      | -65.84     | -104.59  | -101.82  | -99.52    | -46.80   | -47.37   | -98.3(-33.9)  |
| 300  | -67.06 | -66.09      | -65.84     | -104.53  | -101.77  | -99.46    | -46.79   | -47.35   | -97.1(-33.3)  |
| 305  | -68.58 | -66.89      | -65.84     | -104.53  | -101.77  | -99.46    | -46.79   | -47.35   | -97.1(-33.3)  |
| 310  | -67.51 | -66.68      | -65.84     | -104.53  | -101.77  | -99.46    | -46.79   | -47.35   | -97.1(-33.3)  |

$^a$The numbers in the parentheses indicate the density component of the equilibrium interaction energy, $\langle V_{DD}/Ox \rangle$.

C. Solvent

The polarization structure factors entering the equations for the solvation chemical potential are characteristics of the homogeneous solvent. They can be obtained numerically by averaging the projections of dipole moments $\hat{\mathbf{e}}_j$ on an arbitrary chosen direction of the $\mathbf{k}$-vector, $\hat{\mathbf{k}} = \mathbf{k}/k$:

$$S^L(k) = \frac{3}{N} \left\langle \sum_{i,j} (\hat{\mathbf{e}}_j \cdot \hat{\mathbf{k}})(\hat{\mathbf{k}} \cdot \hat{\mathbf{e}}_i) e^{i\mathbf{k} \cdot \mathbf{r}_{ij}} \right\rangle,$$

$$S^T(k) = \frac{3}{2N} \left\langle \sum_{i,j} [ (\hat{\mathbf{e}}_j \cdot \hat{\mathbf{e}}_i) - (\hat{\mathbf{e}}_j \cdot \hat{\mathbf{k}})(\hat{\mathbf{k}} \cdot \hat{\mathbf{e}}_i) ] e^{i\mathbf{k} \cdot \mathbf{r}_{ij}} \right\rangle,$$

(28)

where $\mathbf{r}_{ij} = \mathbf{r}_i - \mathbf{r}_j$ and $N$ is the number of liquid dipoles.

Unfortunately, experiment does not provide spatially resolved correlators of dipoles in polar liquids and one has to resort to using either computer simulations or liquid-state theories. Parametrizing homogeneous structure factors by computer simulations is a very attractive avenue for studying hydration because of continuously improving empirical potentials for water on one hand and the reliance of biological applications on aqueous solvation on the other. For the problems related to derivatives of thermodynamic potentials, e. g. entropy and volume of solvation, the structure factors need to be tabulated at different temperatures and/or pressures. In this paper, polarization structure factors $S^L,T(k)$ were obtained from $\approx 25$ ns MD trajectories of TIP3P water$^{28}$ at different temperatures (Fig. 5).

![PPSF](image)

FIG. 5: Longitudinal (L) and transverse (T) polarization structure factors of TIP3P water [Eq. (28)] calculated at different temperatures from $\approx 25$ ns MD trajectories. Also shown is the PPSF calculation at $T = 300$ K.

In parallel to simulations, we have used dipolar structure factors from our PPSF parametrization scheme$^{13}$ This approach is based on the analytical solution of the mean-spherical approximation for the fluid of dipolar hard spheres$^{29}$ which is parametrized to give $k = 0$ values from Eqs. 79 and 1314. The PPSF parametrization
gives solvation free energies essentially identical to those obtained with the structure factors from simulations, as was also found for a smaller polypeptide solute in our previous publication.\textsuperscript{14} It appears therefore that the local tetrahedral order of water, which is of course not captured by dipolar hard spheres, is not significant for the energetics of polar solvation dominated by orientational correlations ruled by dipole-dipole forces. The density fluctuations are, on the other hand, dominated by repulsions. We used, therefore, the density structure factor from the Percus-Yevick solution\textsuperscript{57} for a hard-sphere fluid of the same density as water in calculations of the density response in Eq. \textsuperscript{21}.

We need to emphasize here that using dielectric constants of model dipolar fluids would give us wrong results. The dielectric constant of a molecular liquid is affected by short range molecular correlations through the Kirkwood factor, hydrogen bonds and molecular quadrupoles are among significant factors.\textsuperscript{36} We account for all these effects in the PPSF scheme by using the experimental, either from computer or laboratory data, dielectric constants in Eqs. \textsuperscript{7} and \textsuperscript{22}. Once the $k = 0$ limit is set up by the experimental input, the behavior of $S_{L,T}^2(k)$ in the range of $k < 2\pi/\sigma$ ($\sigma$ is the solvent diameter) is well reproduced by solutions obtained for model dipolar fluids. These solutions will fail at $k \geq 2\pi/a$, where $a$ is the characteristic distance between partial charges within the solvent molecule. This part of the spectrum of polarization fluctuations is correctly captured by models based on interaction-site integral equations\textsuperscript{36} but that range of wave-vectors normally does not contribute to the solvation energy. In fact, the range of $k$-values relevant for the solvation problem is limited by $k < 2\pi/R$, where $R$ is the characteristic dimension of the solute. For large solutes, only the long-wavelength part of the polarization structure factors is really needed for the solvation energy calculations. As is shown in Fig. \textsuperscript{5} there is a mismatch between the PPSF longitudinal structure factor and MD simulations. However, this difference makes no effect on the calculated solvation energies.

We can summarize our results on parametrizing the solvent properties by stating that the model fluid of dipolar hard spheres can serve as a reliable reference system for calculations of polar solvation given the macroscopic properties, the density of dipoles $y$ and the dielectric constant $\epsilon_s$, have been taken from experiment (either laboratory or computer). The theory thus adds an additional parameter $y$ to the dielectric constant used in electrostatic solvation theories to produce a fully microscopic solvent response. In practical applications of the theory (e.g. in case of solvation in ambient water presented below), the parameter $y$ needs to be calculated from the molecular properties of the solvent. We use the 1-$R$ Wertheim theory\textsuperscript{82} to calculate the effective dipole moment of the solvent (see Ref. \textsuperscript{81} for comparison to simulations). The solvent input is thus made by five parameters: $\{\sigma, \rho, m, \alpha, \epsilon_s\}$. One needs in addition the high-frequency dielectric constant $\epsilon_\infty$ for the reorganization energy calculations and the temperature slopes of two dielectric constants, as well as the isobaric expansivity, for the solvation entropy calculations. The big advantage of the PPSF scheme is that all these parameters have been tabulated for many solvents commonly used in solution chemistry making our method broadly applicable to solvation calculations in polar molecular solvents.

Despite the fact that the dielectric constant is sensitive to local correlations, the polarization structure factors in the long-wavelength limit are fully determined by dipolar correlations general for all polar liquids and not much sensitive to details of the local structure which is of course very different in water than in a hard-sphere dipolar fluid. There are several advantages to using dipolar hard spheres as the reference system. First, all thermodynamic and structural properties are controlled by only two parameters, the reduced density $\rho\sigma^3$ and the dipolar density $y$. Second, this system is well characterized both analytically and numerically. It has served many times as a starting point for developing theories of polar liquids\textsuperscript{82} similarly to the role played by the fluid of hard spheres in theories of non-polar liquids.\textsuperscript{36} Once that stated, we however want to stress that the theory itself is based on the structure factors of an arbitrary polar medium with the Gaussian fluctuation spectrum and is not limited to a choice of any particular reference system.

**IV. SIMULATIONS PROTOCOL**

Amber 8.0\textsuperscript{82} was used for all MD simulations. The initial configuration of PC was created using a protonated version of the X-ray crystal structure at 1.7 Å resolution (PDB: 1ag6\textsuperscript{82}). This initial configuration of the protein was first minimized in vacuum by the conjugate gradient method for 10,000 steps to allow the protein to remove any bad initial contacts. Then the system was solvated in a rectilinear box with several thousand TIP3P molecules\textsuperscript{78} providing at least two-three solvation shells around the protein. To neutralize the charge, a number of sodium ions equal to the total charge of the protein were added. The protein was then relaxed for a few thousand steps while water and sodium were positionally constrained. Finally, the entire system containing solvent, counterions, and protein was energy minimized in 100,000 steps.

Next, the system was heated in a NVT ensemble for 30 ps from 0 K to the desired temperature followed by volume expansion in a 1 ns NPT run. NVT production runs, following density equilibration, lasted from 6 to 18 ns. The last 5–10 ns at the end of each trajectory were used to calculate the averages. The timestep for all MD simulations was 2 fs, and SHAKE was employed to constrain bonds to hydrogen atoms. Constant pressure and temperature simulations employed Berendsen barostat and thermostat, respectively\textsuperscript{24}. The long-range electrostatic interactions were handled using a smooth particle mesh Ewald summation with a 9 Å limit in the direct space.
sum. The total charge for the protein was $-9.0$ for the reduced state and $-8.0$ for the oxidized state.

V. RESULTS

The calculations presented here are focused on two properties: the solvent portion of the redox chemical potential, $\Delta \mu_s$, and the solvent reorganization energy $\lambda_s$, both corresponding to the half reaction

$$PC(Ox)^8 + e^- \rightarrow PC(\text{Red})^9.$$  

(29)

The former can in principle be calculated as the difference of solvation chemical potentials in the Red and Ox states. However, this approach involves calculating the difference in two large numbers, which is computationally unreliable. Instead, we use the linear response approximation to calculate $\Delta \mu_s$ according to the equation:

$$\Delta \mu_s = \mu_{0,\text{Red}} - \mu_{0,\text{Ox}} = -\Delta \tilde{E}_0 * \chi * \Delta \tilde{E}_0. \quad \text{(30)}$$

Here, $\tilde{E}_0 = (\tilde{E}_0^\text{Ox} + \tilde{E}_0^\text{Red})/2$ and $\Delta \tilde{E}_0 = \tilde{E}_0^\text{Red} - \tilde{E}_0^\text{Ox}$ are the mean and the difference of the electric fields in the Red and Ox states. Similarly, the solvent reorganization energy is calculated from

$$\lambda_s = \frac{1}{2} \Delta \tilde{E}_0 * \chi * \Delta \tilde{E}_0. \quad \text{(31)}$$

Equation (31) applies to the reorganization energy of non-polarizable solvents employed in computer simulations. For laboratory data, nuclear polarization should be separated from the overall solvent polarization and the response function $\chi$ is replaced by the nuclear response function $\chi_n$ as explained above and in more detail in Ref. 14.

A. Redox Thermodynamics

The solvation thermodynamics calculated here can be related to experimental redox entropies reported by measuring the temperature dependence of the standard or midpoint electrode potentials. An electrochemical experiment corresponds to bringing a solution containing given numbers of oxidized and reduced reagents, which are not necessarily in equilibrium (the ratio of their numbers is not a Boltzmann factor), in contact with a metal electrode. The equilibrium is established between the electronic subsystem of the redox pair and the electrode in such a way that the electrode is charged and its electrochemical potential $\mu$ is shifted from the vacuum Fermi energy $\epsilon_F$ by the electrostatic potential $\phi$: $\mu = \epsilon_F - \phi$.

The numbers of the oxidized and reduced forms of the redox pair, $N_{\text{Ox}}$ and $N_{\text{Red}}$, are assumed to be large enough so that they are not affected by charging the electrode. The electrochemical potential of the electrode than becomes equal to the absolute electrochemical potential of the redox couple in the solution. The latter can be found from simple statistical arguments. The grand-canonical free energy of two fermionic subsystems of $N_{\text{Ox}}$ and $N_{\text{Red}}$ electronic levels is

$$\beta \Omega = -N_{\text{Ox}} \ln \left( 1 + e^{\beta (\mu - \epsilon_{\text{Ox}})} \right) - N_{\text{Red}} \ln \left( 1 + e^{\beta (\mu - \epsilon_{\text{Red}})} \right), \quad \text{(32)}$$

where $\epsilon_{\text{Ox}}$ and $\epsilon_{\text{Red}}$ are the average energies of the electronic levels in the corresponding redox states. The chemical potential is then found by requiring that the derivative $-(\partial \Omega/\partial \mu)_T$ is equal to the total number of electrons $N_{\text{Red}}$. For the energy gap between Ox and Red states greater than $k_B T$, this requirement results in the Nernst equation:

$$\mu = \frac{\epsilon_{\text{Ox}} + \epsilon_{\text{Red}}}{2} - k_B T \ln \left( \frac{N_{\text{Ox}}}{N_{\text{Red}}} \right), \quad \text{(33)}$$

in which the standard potential is given by the mean of the average electronic energies

$$\phi^0 = -\frac{\epsilon_{\text{Ox}} + \epsilon_{\text{Red}}}{2e}. \quad \text{(34)}$$

The same result follows from the use of the stationary condition (zero electrode current) for the rates of reduction and oxidation. The electronic energies are given by the sums of their vacuum components, $\epsilon_{\text{Ox/Red}}$, and the interaction of the electric field of the electron $E_e$ with the polarization of the solvent in equilibrium with the total electric field of the molecule in the solution

$$\epsilon_{\text{Ox/Red}} = \epsilon_{\text{Ox/Red}}^0 - E_e * \chi * E_{\text{Ox/Red}}^0. \quad \text{(37)}$$

Taking into account that

$$E_e = E_0^\text{Red} - E_0^\text{Ox} = \Delta E_0, \quad \text{(38)}$$

one gets from Eqs. (34), (37), and (38) the commonly used connection between the standard electrode potential and the solvation part of the redox free energy

$$\phi^0 = -\frac{\epsilon_{\text{Red}}^0 + \epsilon_{\text{Ox}}^0}{2e} - \frac{\Delta \mu_s}{e}, \quad \text{(39)}$$

with $\Delta \mu_s$ the difference in two large numbers, which is computationally unreliable. Instead, we use the linear response approximation to calculate $\Delta \mu_s$ according to the equation:

$$\Delta \mu_s = \mu_{0,\text{Red}} - \mu_{0,\text{Ox}} = -\Delta \tilde{E}_0 * \chi * \Delta \tilde{E}_0. \quad \text{(30)}$$

Here, $\tilde{E}_0 = (\tilde{E}_0^\text{Ox} + \tilde{E}_0^\text{Red})/2$ and $\Delta \tilde{E}_0 = \tilde{E}_0^\text{Red} - \tilde{E}_0^\text{Ox}$ are the mean and the difference of the electric fields in the Red and Ox states. Similarly, the solvent reorganization energy is calculated from

$$\lambda_s = \frac{1}{2} \Delta \tilde{E}_0 * \chi * \Delta \tilde{E}_0. \quad \text{(31)}$$

Equation (31) applies to the reorganization energy of non-polarizable solvents employed in computer simulations. For laboratory data, nuclear polarization should be separated from the overall solvent polarization and the response function $\chi$ is replaced by the nuclear response function $\chi_n$ as explained above and in more detail in Ref. 14.

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The numbers of the oxidized and reduced forms of the redox pair, $N_{\text{Ox}}$ and $N_{\text{Red}}$, are assumed to be large enough so that they are not affected by charging the electrode. The electrochemical potential of the electrode than becomes equal to the absolute electrochemical potential of the redox couple in the solution. The latter can be found from simple statistical arguments. The grand-canonical free energy of two fermionic subsystems of $N_{\text{Ox}}$ and $N_{\text{Red}}$ electronic levels is

$$\beta \Omega = -N_{\text{Ox}} \ln \left( 1 + e^{\beta (\mu - \epsilon_{\text{Ox}})} \right) - N_{\text{Red}} \ln \left( 1 + e^{\beta (\mu - \epsilon_{\text{Red}})} \right), \quad \text{(32)}$$

where $\epsilon_{\text{Ox}}$ and $\epsilon_{\text{Red}}$ are the average energies of the electronic levels in the corresponding redox states. The chemical potential is then found by requiring that the derivative $-(\partial \Omega/\partial \mu)_T$ is equal to the total number of electrons $N_{\text{Red}}$. For the energy gap between Ox and Red states greater than $k_B T$, this requirement results in the Nernst equation:

$$\mu = \frac{\epsilon_{\text{Ox}} + \epsilon_{\text{Red}}}{2} - k_B T \ln \left( \frac{N_{\text{Ox}}}{N_{\text{Red}}} \right), \quad \text{(33)}$$

in which the standard potential is given by the mean of the average electronic energies

$$\phi^0 = -\frac{\epsilon_{\text{Ox}} + \epsilon_{\text{Red}}}{2e}. \quad \text{(34)}$$

The same result follows from the use of the stationary condition (zero electrode current) for the rates of reduction and oxidation. The electronic energies are given by the sums of their vacuum components, $\epsilon_{\text{Ox/Red}}$, and the interaction of the electric field of the electron $E_e$ with the polarization of the solvent in equilibrium with the total electric field of the molecule in the solution

$$\epsilon_{\text{Ox/Red}} = \epsilon_{\text{Ox/Red}}^0 - E_e * \chi * E_{\text{Ox/Red}}^0. \quad \text{(37)}$$

Taking into account that

$$E_e = E_0^\text{Red} - E_0^\text{Ox} = \Delta E_0, \quad \text{(38)}$$

one gets from Eqs. (34), (37), and (38) the commonly used connection between the standard electrode potential and the solvation part of the redox free energy

$$\phi^0 = -\frac{\epsilon_{\text{Red}}^0 + \epsilon_{\text{Ox}}^0}{2e} - \frac{\Delta \mu_s}{e}, \quad \text{(39)}$$
lists the results of NRFT and DelPhi program suite in the solvation calculations. Dielectric constant of ambient water was used for the solvent continuum and $\epsilon_s$ = 1 for the protein. This latter choice was driven by our desire to compare continuum and microscopic calculations of solvation thermodynamics since the latter does not assume any polarization of the protein. Table III lists the results of NRFT and DelPhi calculations of the average energy $\langle V_{0s}\rangle_{\text{ox}}$ of PC in

| T/K | MD | NRFT | vdW | SAS | MD | NRFT | vdW | SAS | NRFT | vdW | SAS |
|-----|----|------|-----|-----|----|------|-----|-----|------|-----|-----|
| 280 | −3.05 | −6.21 | −9.63 | −7.15 | 0.845 | 1.37(0.59) | 3.645 | 0.690 | 0.82 | 2.728 | 0.417 |
| 285 | −2.73 | −6.15 | −9.62 | −7.14 | 0.814 | 1.35(0.58) | 3.641 | 0.685 | 0.80 | 2.725 | 0.417 |
| 300 | −2.71 | −6.10 | −9.62 | −7.14 | 0.646 | 1.33(0.57) | 3.636 | 0.685 | 0.79 | 2.722 | 0.417 |
| 305 | −3.14 | −6.04 | −9.61 | −7.14 | 0.565 | 1.32(0.56) | 3.630 | 0.685 | 0.78 | 2.718 | 0.417 |
| 310 | −2.82 | −6.01 | −9.60 | −7.14 | 0.435 | 1.30(0.55) | 3.623 | 0.684 | 0.77 | 2.714 | 0.417 |
| 315 | −2.94 | −5.94 | −9.59 | −7.13 | 0.426 | 1.29(0.54) | 3.620 | 0.684 | 0.76 | 2.710 | 0.417 |

$^a$Poisson-Boltzmann calculations with vdW radii assigned to the protein atoms (standard vdW cavity).

$^b$Poisson-Boltzmann calculation with solvent radius added to the radii of the protein atoms exposed to the solvent (solvent-accessible cavity, SAS).

The continuum calculations are done with the dielectric constant of TIP3P water $\epsilon_s$ = 95 and $\epsilon_{\infty}$ = 1.0. The dielectric constant of the protein interior was put equal to unity in order to be consistent with the microscopic calculations. The temperature variation of the dielectric constant of $d\epsilon_s/dT = −0.654$ K$^{-1}$ (Ref. 12) was adopted for the entropy calculations listed in Tab. IV.

$^c$Calculations in ambient water using $\epsilon_{\infty}$ = 1.78, $\epsilon_s$ = 78, $d\epsilon_s/dT = −0.398$ K$^{-1}$, and $d\epsilon_{\infty} /dT = −2.75 \times 10^{-4}$ K$^{-1}$. In addition, temperature expansion was included with the constant-pressure expansivity coefficient $\alpha_p = 2.6 \times 10^{-4}$ K$^{-1}$.

We need to stress here that redox entropies in polar solutions are sensitive to the presence of electrolyte and our calculations at zero ionic strength.

From Eqs. (34), (37), and (38) one can directly derive the equation for the solvation redox free energy

$$\Delta \mu_s = \left(\langle \Delta V_{0s}\rangle_{\text{ox}} + \langle \Delta V_{0s}\rangle_{\text{red}}\right) / 2,$$  

(41)

where $\Delta V_{0s}$ is the difference in the solute-solvent interaction energies in the Red and Ox states and the averages are taken over the corresponding ensembles. The same average vertical gaps can be used to calculate the reorganization energy as

$$\lambda_s = \langle \Delta V_{0s}\rangle_{\text{ox}} - \langle \Delta V_{0s}\rangle_{\text{red}} \rangle / 2.$$  

(42)

We need to caution here is that while Eq. (31) is a statistical-mechanical result, Eqs. (37)–(42) are based on the LRA for the solute-solvent interaction energy and might be affected by deviations from this approximation.

B. Solute-solvent average energy

In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations. In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations. In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations. In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations. In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations. In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations. In addition to our NRFT formalism, we have used the dielectric continuum approximation implemented in the DelPhi program suite in the solvation calculations.
of the NRFT result to the standard (vdW) continuum calculation and (ii) a good agreement between the polarization portion of the NRFT calculations and MD results. The continuum electrostatics does not reproduce the slope of the average energy as we also discuss below in relation to the redox entropy.

In order to understand the origin of the close agreement between MD and the polarization component of the solvent response, one needs to recall what comes to the calculation of the polarization and density components of the solvation free energy. The polarization response is calculated by assuming that the only influence of the solute on the polarization field is to exclude it from the solute volume represented by the step function $\delta_0(\mathbf{r})$ equal to one inside the solute and zero otherwise [Eq. (12)]. The density component corrects this result by taking into account the inhomogeneous density profile formed at the surface of a hard-wall solute. In dense liquids, such a profile is characterized by a sharp peak of the radial distribution function in the first solvation shell of the solute. Correspondingly, reflecting the belief that the short-range structure of liquids is primarily determined by repulsions\cite{94} the density structure factor $S(k)$ in Eq. (21) was taken in our calculations from the Percus-Yevick solution for hard spheres\cite{97}. The close proximity of the full NRFT calculation to the standard DelPhi/vdW algorithm (Fig. 7) illustrates the fact that the common parametrization of the atomic radii is based on the experience learned for hydration of small ions with tightly bound first solvation shell. In the present algorithm, this physics is accommodated by the density component of the solvation free energy.

The structure of water at the protein surface is quite different from what is normally obtained by inserting a small solute in a molecular solvent. The structure is heterogeneous including islands of highly structured water around polar and ionized residues combined with much softer density profile at the hydrophobic patches. This reality is illustrated in Fig. 8 which shows pair distribution functions between ionized and non-polar residues of the water’s oxygens. While the distribution functions of ionized residues are reminiscent of the structures typically observed around small solutes in dense solvents, the water structure around non-polar residues is quite different: there is no first-shell peak and water interface is shifted by $\simeq 1$ Å, in accord with simulations of nanoscale hydrophobic solutes\cite{95,96}.

The stronger attraction of the surface water molecules to the bulk than to a non-polar hydrophobic patch of the protein (cavity expulsion potential\cite{95,96}) results in a weak dewetting of the surface\cite{95} with the density at the interface lower than in the bulk (Fig. 8). Since there are only a few charged residues on the protein surface, the average surface structure is closer to a step-wise cut-off introduced in the polarization component of the response function than to a structured liquid at the surface of a small polar/ionic solute\cite{97}. This observation explains a good agreement between MD and polarization calcula-
C. Solvent Gibbs and reorganization free energies

The results of calculations of the redox solvation energy and solvent reorganization energy [Eqs. (41) and (42)] using different levels of the theory are listed in Table III. Redox and reorganization entropies are given in Table IV. In addition, the temperature dependence of $\Delta \mu_s$ and $\lambda_s$ are visualized in Figs. IX and X. The differences in the theoretical results arise from the different level of structural solvent information incorporated in each formalism. For completeness, we have also listed in Table III the NRFT calculations with the solvent parameters of ambient water.

The continuum electrostatics gain access to the solvation entropy through the temperature variation of the solvent dielectric constant. Therefore, since the dielectric constant commonly decreases with heating, the solvent becomes effectively less polar and the solvation free energy of a charge distribution increases, i.e. becomes less negative. If one considers redox species positively charged in both redox states, the oxidized state carries a larger charge and hence the difference of Red and Ox solvation energies has a positive value decreasing with increasing temperature. The redox entropy in Eq. (40) is then positive as is typically observed for simple inorganic ions. By the same arguments, the species carrying negative charge in both redox states should have a negative redox entropy, which is the case for the negatively charged PC in our calculations.

Despite the right sign of the redox entropy, the magnitudes of both the Gibbs solvation energy and the entropy are markedly different in continuum and microscopic/simulation approaches: $\Delta \mu_s$ is higher in the standard implementation of DelPhi (vdW cavity) than the NRFT value by a factor of 1.5 while the redox entropy is lower by a factor of ten. The use of the solvent-accessible cavity brings the value of $\Delta \mu_s$ in a closer proximity to the NRFT, but the redox entropy is lowered even more (Table IV). The magnitude of $\Delta \mu_s$ from NRFT is significantly higher than from MD even if the density component is subtracted from the total response. This initially comes a bit of surprise given a good agreement between the polarization-NRFT and MD values of $(V_{ox})_Ox$ reported in Table III and Fig. 9. We do not currently have a good explanation of this disagreement (see Discussion below).

Concerning the reorganization energy calculations, the standard DelPhi/vdW algorithm gives $\lambda_s$ three times larger than NRFT and almost an order of magnitude larger than the MD simulations (Table IV). The origin of large $\lambda_s$ in the standard (vdW) implementation of the continuum model is in placing highly polar dielectric into the small pocket near copper which water molecules do not visit in MD simulations. Figure 11 shows the pair distribution function between the Cu ion and oxygen of water testifying to the fact that water never comes to Cu closer than 6 Å and the maximum of the first solvation shell appears at 6.7 Å. The addition of the solvent radius to the cavity corrects this error reducing the reorganization entropy to the level of 0.64 eV consistent with the polarization component of the NRFT (0.75 eV, Table IV).
TABLE IV: Redox solvation free energy $\Delta \mu_s$ and redox entropy $\Delta s_s$ in for the Red/Ox states of PC. Also listed are the reorganization energy and reorganization entropy, $s_\lambda = -\partial \lambda / \partial T$. All energies are in eV and entropies are in meV/K, $T = 300$ K.

| Method                                      | $\Delta \mu_s$ | $\Delta s_s$ | $\lambda_s$ | $s_\lambda$ |
|---------------------------------------------|-----------------|--------------|-------------|-------------|
| DelPhi with vdW cavity                      | $-9.92$         | $-1.25$      | 3.62        | 0.97        |
| DelPhi with solvent-accessible cavity       | $-7.12$         | $-0.45$      | 0.64        | 0.005       |
| Non-local polarization response functions$^a$| $-6.01(-1.33)$  | $-10.5(4.8)$ | 1.30(0.55)  | 3.2(2.0)    |
| Molecular Dynamics$^b$                      | $-2.81$         | $-7.40$      | 0.54        | 10.7        |
| Experiment                                  | $-0.4^c$        |              |             |             |
|                                             | $-1.4^d$        |              |             |             |

$^a$The numbers in the parentheses indicate contributions to the solvation free energy and entropy from density fluctuations.

$^b$MD results refer to linear fits through the simulation points.

$^c$Ref. 22.

$^d$Ref. 21.

FIG. 12: Reorganization energy of charge-transfer transition in p-nitroaniline dissolved in SPC/E water. The results are obtained by MD simulations$^{15}$ (closed circles) and NRFT calculations (diamonds). Closed and open diamonds refer to the full and polarization response, respectively. Triangles denote half of the Stokes shift from MD simulations; in the LRA, $\Delta u_{st}/2 = \lambda_s$. The dielectric constants of SPC/E water at different temperatures, required for the NRFT input, were taken from MD simulations$^{15}$.

The combination of the absolute values of the reorganization energy with the reorganization entropies clearly indicates that re-scaling of the dielectric cavity, often employed in various continuum formulations, does not solve the solvation problem. Entropies calculated by including the effect of density fluctuations on the solvent response are generally in better agreement with MD simulations than the results obtained from polarization response only. All solvation free energies obtained from such calculations, however, significantly exceed the simulation results. The problem lies in the assumed dense structure of the liquid around the solute, which is obviously not realized for hydrated protein. Capturing density fluctuations is essential for the correct calculations of the entropies, but the effective density should be adjusted to that of the structurally loose hydration layer at the protein surface.

One might argue that the failure to reproduce the results of the simulations by the NRFT calculations can be traced back to the deficiencies of the model and not to the specific structure created by the protein in water. We believe that this objection cannot explain the differences between our calculations and MD simulations. In order to illustrate this point, we show in Fig. 12 the comparison of the NRFT calculations with our recent MD results for solvation of a small charge-transfer molecule p-nitroaniline in SPC/E water$^{15}$. The partial charges for this molecule, used in the calculations were tabulated in Ref. 99. The comparison with the NRFT method is somewhat complicated in this case by nonlinear solvation effects seen in the deviation of the half of the Stokes shift from the reorganization energy. However, the NRFT calculation with the density component included is in reasonable agreement with the simulation data for both the reorganization energy and entropy, while the polarization response alone clearly underestimates the reorganization energy.

VI. DISCUSSION

The formalism developed here is based on the LRA suggesting that knowing the variance of electrostatic fluctuations around a neutral repulsive solute is sufficient to calculate the solvation chemical potential [Eq. (2)]. One can arrive at Eq. (2) from the following simple considerations. The chemical potential of solvation can be written as the integral over the magnitude of the solute-solvent electrostatic interaction $\epsilon = V_{0s}$ as follows

$$e^{-\beta \mu_{0s}} = \int d\epsilon P(\epsilon, \beta) e^{-\beta \epsilon}. \quad (43)$$

Here the probability density of reaching the value $\epsilon$ is obtained by taking the statistical average over the reference Hamiltonian $H_0$ which includes all the interaction potentials in the system except the solute-solvent electrostatic potential $V_{0s}$:

$$P(\epsilon, \beta) = Q_0^{-1} \int \delta(\epsilon - V_{0s}) e^{-\beta H_0} d\Gamma, \quad (44)$$

where $Q_0 = \int \exp[-\beta H_0] d\Gamma$ and $\Gamma$ denotes the phase space volume. Since the solute-solvent component of $H_0$
includes mostly isotropic short-range interactions, one can put \((V_{0s})_0 = 0\). In addition, the Gaussian approximation can be applied to \(P(\epsilon, \beta)\)

\[
P(\epsilon, \beta) \propto \exp \left[ -\frac{\epsilon^2}{2\sigma^2} \right] \tag{45}
\]

with

\[
\sigma^2 = \langle (\delta V_{0s})^2 \rangle_0. \tag{46}
\]

Combining Eqs. (43) and (45), one immediately arrives at Eq. (2).

One of the important lessons of Eqs. (13) and (44) is that all the thermodynamic information required for the solvation problem is contained in the distribution of electrostatic potential fluctuations around a “non-polar” solute in which all the electrostatic interactions with the solvent have been switched off. For large solutes, the spectrum of electrostatic fluctuations around the repulsive solute core will ultimately determine the thermodynamics of solvation. It is important to stress that Eqs. (43) and (44) are exact and this statement is not limited to linear solvation only.

The reasoning outlined above works well for small and medium-size solutes, as is exemplified in Fig. 12. However, the application of the same procedure to protein solvation studied in this paper has encountered some difficulties suggesting that methods developed over several decades and successfully applied to solvation of small solutes in dense polar solvents are probably not directly transferable without significant modifications to mesoscopic hydration of proteins. The application of our algorithm to the calculation of the average solute-solvent interaction energy turned out to be quite successful when the density profile of water around the protein is approximated by a step function (Fig. 7). The calculations agree with MD within simulation uncertainties (\(\leq 5\%\)). As mentioned above, this comes as a result of averaging between tight and loose water structures at the protein surface. The application of the same procedure to solvation of the difference charges of the active site (reorganization energy, Fig. 10) was less successful, but the agreement is probably still acceptable, in particular at lowest temperatures.

Where the calculations and simulations come in significant disagreement is for the free energy of solvation obtained in simulations as the mean of two transition energies [Eq. (11)]. Some recent simulations of cavities in force-field water models\(^{100,101}\) and of uncharged protein\(^{102}\) have suggested a possible origin of this effect. It was found that water structured at the protein surface creates a positive potential within an uncharged cavity/protein. In terms of Eq. (45) this implies a constant shift of the solute-solvent energy \(\epsilon \rightarrow \epsilon - \epsilon_0\). Since we have assumed random orientations of water around an uncharged solute, as is indeed the case for small solutes\(^{23}\) our calculations do not include the effect of a positive background potential and include only changes of the potential in response to protein’s charges. In the case of a constant background potential \(\Phi\) Eq. (11) modifies to

\[
\Delta \mu_s = \langle (\Delta V_{0s})_{\text{ox}} + (\Delta V_{0s})_{\text{Red}} \rangle / 2 - \Delta q \Phi, \tag{47}
\]

where \(\Delta q = q_{\text{Red}} - q_{\text{ox}} = -1\). Ashbaugh\(^{100}\) reported an average positive potential of about \(e\Phi \approx 9\) kcal/mol for cavities in SPC water comparable in size with PC. In terms of our calculations, it amounts a positive shift of the simulation data by about 0.4 eV which will increase the current gap of about 3.2 eV (300 K) between the MD and NRFT.

The origin in the difference in solvation free energies between calculations and MD simulations might be related to the weakly dewetted water density profile near the active site. This would imply that some of the properties of water structure around large cavities expelled by the protein from its volume are quite different from the common experience gained with small solutes. This qualitative difference between small-size and large-size solvation has recently gained appreciation for hydrophobic solvation\(^{24}\), as we discuss next.

The Gaussian model of hydrophobic solvation goes back to the Pratt-Chandler theory of hydrophobicity\(^{103}\) recently extended by Pratt and co-workers\(^{104,105}\). The formulation of the theory of hydrophobic solvation follows a path similar to the one outlined in Eqs. (13)–(40), asking what is the free energy \(\mu_s\) needed to solvate a solute of volume \(\Omega_0\). It is given by the Gaussian probability\(^{97,104}\)

\[
\beta \mu_s \approx \frac{\rho^2 \Omega_0^2}{2 \chi \Omega} \tag{48}
\]

with the fluctuation of the number of solvent particles in volume \(\Omega_0\)

\[
\chi = \langle (\delta N)^2 \rangle - \rho \Omega_0 + \rho^2 \int_{\Omega_0} d\mathbf{r} d\mathbf{r}' h_{ss}(|\mathbf{r} - \mathbf{r}'|). \tag{49}
\]

In Eq. (49), \(h_{ss}(r)\) is the pair correlation function of the homogeneous solvent.

The Gaussian probability of electron transfer carries a close similarity with Eq. (48) giving the activation free energy as

\[
\beta \mu_{\text{act}} = \frac{X_0^2}{2 \sigma^2}, \tag{50}
\]

where \(X_0\) is the average vertical donor-acceptor energy gap and \(\sigma^2\) is the variance of the solute-solvent interaction potential when \(V_{0s} = \Delta V_{0s}\) is used in Eqs. (44) and (45). Not surprisingly, the structure of the equation for \(\sigma^2\) resembles Eq. (40):

\[
\sigma^2 = \rho \int V_{0s}(1) d\Gamma_1 + \rho^2 \int V_{0s}(1)V_{0s}(2) h_{ss}(1,2) d\Gamma_1 d\Gamma_2 + \rho^2 \int V_{0s}^2(1) h_{ss}(1,2) d\Gamma_1 d\Gamma_2. \tag{51}
\]
The last summand in this equation represents the density component of the response since it transforms into the k-space integral with the density structure factor of the liquid [Eq. (21)].

Equation (51) carries a close resemblance with the Pratt-Chandler theory of hydrophobicity. The sum of the first and the third terms is the average of the squared solute-solvent potential \( V_{0s}^2 (1) \) over the solute-solvent density profile.

\[
\rho (r) = \rho (1 + c_{0s} (r)) + \rho^2 \int c_{0s} (r') h_{ss} (|r - r'|) \, dr'
\]  

in which the solute-solvent direct correlation function \( c_{0s} (r) \) is replaced by its lowest density expansion \( \theta_0 (r) \) (\( \theta_0 (r) \) is one inside the solute and zero otherwise). The observation that the density profile around solutes of size \( \geq 1 \) nm can be approximated by a step function (Fig. 11 or see Fig. 3 in Ref. 97) amounts to neglecting the second term in Eq. (52) and, correspondingly, the density components \( \chi_d \) in the solute-responsive function [Eq. (10)].

The fact that the spectrum of electrostatic potential fluctuations around a non-polar solute gives complete information about polar solvation brings this latter problem in close relation to the problem of hydrophobic solvation. It was realized in recent years that hydrophobic solvation of small and large solutes are qualitatively different. Solvation character changes at the critical size of \( \approx 1 \) nm from entropy-dominated solvation of small solutes (Gaussian statistics) to enthalpy-dominated solvation of large solutes driven by the creation of the solute-solvent interface (non-Gaussian statistics). The solvent interface around large solutes involves partial dewetting, strongly sensitive to the strength of solute-solvent attractions and the appearance of large-size interfacial density fluctuations. In case of protein solution, surface water creates a nonzero potential around uncharged proteins discussed above, while density fluctuations lead to complex protein-solvent dynamics and are probably connected to “slaving” of the protein dynamics by the solvent.

These new features observed for hydrophobic solvation at the nano-scale will affect the thermodynamics of polar solvation. It is currently not clear how the crossover from the Gaussian regime of small solutes to the interface-dominated regime of large solutes will translate to the problem of electrostatic solvation. The calculations and simulations performed in this study gave some insights into the kind of problems which the theory needs to address. It is clear that the inhomogeneous nature of the protein surface requires algorithms for the local density profile or the solvent-accessible surface to be a part of a qualitative solvation theory. The current formulation can be improved by using a local density approximation as, for instance, applied by Ramirez and Borgis. In this approach, the bulk dipolar density \( y (\rho) \) is replaced by the local dipolar density \( y (\rho(r)) = (4\pi/9)3m^2 \rho(r) + (4\pi/3)\alpha \rho(r) \) with the local density profile calculated from, for instance, the Lum-Chandler-Weeks theory. Our computational algorithm will then require the Fourier transform of the field \( E_0 (r)\rho (r)/\rho \) instead of \( E_0 (r) (1 - \theta_0 (r)) \) in the current implementation. In addition, density fluctuations of the interfacial region need to be addressed since the mean position of the interface has no physical significance in the presence of large-amplitude fluctuations. What is also clear is that the density fluctuations of the interfacial region present a nuclear mode, largely diminished for solvation of small solutes, which grows in significance for solvation of biopolymers. Future theoretical development needs to address this physical reality.

Both NRFT and MD simulations predict a substantial temperature dependence of the redox potential (ca. \( \approx 5 - 7 \) mV/K). This magnitude of the temperature variation is prohibitively high since a temperature change of \( \approx 15 \) K would shift the redox potential by about 100 mV potentially terminating many enzymatic redox reactions. The theoretically predicted redox entropies refer to zero ionic strength and the solvent contribution to the redox potential. The contribution from the protein to the overall redox potential is available from our simulations, but it depends weakly on temperature contributing only \( \approx -0.9 \) mV/K to the redox entropy. The combined solvent/protein entropy is then significantly higher than redox entropies experimentally observed in buffered solutions (Table IV). This difference raises the question of the role of the ionic atmosphere in stabilizing the redox potential of metalloproteins. The existing experimental evidence for small redox molecules and simulations of metalloproteins all indicate a substantial effect of the ionic atmosphere on the redox entropy, which might compensate for the large entropy due to water and protein. Since simulations have little to say about the ionic strength effects, laboratory measurements of redox entropy at different buffer concentrations are required to shed more light on this problem.

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1. C. L. Brooks, M. Karplus, and B. M. Pettitt, Adv. Chem. Phys. **71**, 1 (1988).
2. D. Eisenberg and A. D. McLachlan, Nature **319**, 199
3 R. Zhou, G. Krilov, and B. J. Berne, J. Phys. Chem. B 108, 7528 (2004).
4 M. Mueller, K. Katsov, and M. Schick, Phys. Rep. 434, 113 (2006).
5 H. Gohlke and M. F. Thorpe, Biophys. J. 91, 2115 (2006).
6 W. Rocchia, S. Sridharan, A. Nicholls, E. Alexov, A. Chiabarrera, and B. Honig, J. Comp. Chem. 23, 128 (2002).
7 M. Schaefer and M. Karplus, J. Phys. Chem. 100, 1578 (1996).
8 M. R. Gunner and B. Honig, Proc. Natl. Acad. Sci. 88, 9151 (1991).
9 K. A. Sharp, Biophys. J. 73, 1241 (1998).
10 K. Siriwong, A. A. Voityuk, M. D. Newton, and N. Rösch, J. Phys. Chem. B 107, 2595 (2003).
11 D. Qiu, P. Shenkin, F. Hollinger, and W. Still, J. Phys. Chem. A 101, 3005 (1997).
12 P. Vath, M. B. Zimmer, D. V. Matyushov, and G. A. Voth, J. Phys. Chem. B 103, 9130 (1999).
13 D. V. Matyushov, J. Chem. Phys. 120, 7532 (2004).
14 A. A. Milischuk, D. V. Matyushov, and M. D. Newton, Chem. Phys. 324, 172 (2006).
15 P. K. Ghori and D. V. Matyushov, J. Phys. Chem. A 110, 8857 (2006).
16 R. Zhou, X. Huang, C. J. Margulis, and B. J. Berne, Science 305, 1605 (2004).
17 M. Tarek and D. J. Tobias, Phys. Rev. Lett. 88, 138101 (2002).
18 N. Choudhury and B. Pettitt, J. Am. Chem. Soc. 129, 4847 (2007).
19 D. V. Matyushov, Chem. Phys. 174, 199 (1993).
20 M. Ubbink, M. Eldëbàck, B. G. Karlsson, and D. S. Bendlall, Structure 6, 323 (1998).
21 N. Sailasuta, F. C. Anson, and H. B. Gray, J. Am. Chem. Soc. 101, 455 (1979).
22 G. Battistuzzi, M. Borsari, L. Loschi, F. Righi, and M. Sola, J. Am. Chem. Soc. 121, 501 (1999).
23 G. Battistuzzi, M. Bellei, M. Borsari, G. W. Canters, E. de Waal, L. J. C. Jeukens, A. Ranieri, and M. Sola, Biochemistry 42, 9214 (2003).
24 D. M. Lockwood, Y.-K. Cheng, and P. J. Rossky, Chem. Phys. Lett. 345, 159 (2001).
25 S. N. Datta, J. Sudhansu, and A. Pandey, J. Phys. Chem. B 108, 8007 (2004).
26 D. F. Hansen and J. J. Led, J. Am. Chem. Soc. 126, 1247 (2004).
27 E. Solomon, R. Szilagyi, S. DeBeerGeorge, and L. Basumallick, Chem. Rev. 104, 419 (2004).
28 E. I. Solomon, Inorg. Chem. 45, 8012 (2006).
29 J. M. Guss, P. R. Harrowell, M. Murata, V. A. Norris, and H. C. Freeman, J. Mol. Biol. 192, 361 (1986).
30 P. J. Stephens, D. R. Jollie, and A. Warshel, Chem. Rev. 96, 2491 (1996).
31 M. Olsson, G. Hong, and A. Warshel, J. Am. Chem. Soc. 125, 5025 (2003).
32 E. L. Yee, R. J. Cave, K. L. Guyer, P. D. Tyma, and M. J. Weaver, J. Am. Chem. Soc. 101, 1131 (1979).
33 M. Born, Z. Phys. 1, 45 (1920).
34 L. Onsager, J. Am. Chem. Soc. 58, 1486 (1936).
35 J. G. Kirkwood, J. Chem. Phys. 2, 351 (1934).
36 F. O. Raineri and H. L. Friedman, Adv. Chem. Phys. 107, 81 (1999).
37 A. Yoshimori, J. Theor. Comp. Chem. 3, 117 (2004).
38 I. Burghardt and B. Bagchi, Chem. Phys. 329, 343 (2006).
39 T. Biben, J. P. Hansen, and Y. Rosenfeld, Phys. Rev. E 57, R3727 (1998).
40 R. Ramirez, R. Gebauer, M. Mareschal, and D. Borgis, Phys. Rev. E 66, 031206 (2002).
41 R. Ramirez and D. Borgis, J. Phys. Chem. B 109, 6754 (2005).
42 A. A. Milischuk and D. V. Matyushov, J. Chem. Phys. 124, 204502 (2006).
43 D. N. LeBard, M. Lülichenko, D. V. Matyushov, Y. A. Berlin, and M. A. Ratner, J. Phys. Chem. B 107, 14509 (2003).
44 A. A. Milischuk and D. V. Matyushov, J. Chem. Phys. 123, 044501 (2005).
45 D. V. Matyushov and M. D. Newton, J. Phys. Chem. A 105, 8516 (2001).
46 G. Stell, G. N. Patey, and J. S. Høye, Adv. Chem. Phys. 18, 183 (1981).
47 G. G. Ferencezy and C. A. Reynolds, J. Phys. Chem. A 105, 11470 (2001).
48 E. A. Carter and J. T. Hynes, J. Chem. Phys. 110, 994 (1999).
49 R. A. Kuharski, J. S. Bader, D. Chandler, M. Sprik, M. L. Klein, and R. W. Impey, J. Chem. Phys. 89, 3248 (1988).
50 J. Blumberger and M. Sprik, J. Phys. Chem. B 109, 6793 (2005).
51 T. Fonseca and B. M. Ladanyi, J. Mol. Liq. 60, 1 (1994).
52 E. I. Solomon, R. A. Kornyshev, and M. A. Vorotyntsev, Faraday Trans. II 72, 361 (1976).
53 H. Li and M. Kardar, Phys. Rev. A 46, 6490 (1992).
54 D. V. Matyushov, J. Chem. Phys. 120, 1375 (2004).
55 W. H. Press, S. A. Teukolsky, W. T. Vetterling, and B. P. Flannery, Numerical recipes in Fortran 77: The art of scientific computing (Cambridge University Press, Cambridge, 1996).
56 E. I. Solomon and M. D. Lowery, Science 259, 1575 (1993).
57 L. W. Ungar, N. F. Scherer, and G. A. Voth, Biophys. J. 72, 5 (1997).
58 G. M. Ullmann, E.-W. Knapp, and N. M. Kostić, J. Am. Chem. Soc. 119, 42 (1997).
59 P. Comba and R. Rementy, J. Comput. Chem. 23, 697 (2002).
60 M. M. Wester, C. E. Davoust, and B. M. Hoffman, J. Am.
C. A. P. Libeu, M. Kikimoto, M. Nishiyama, S. Hori-nouchi, and E. T. Adman, Biochemistry 36, 13160 (1997).

J. C. P. Libeu, M. Kikimoto, M. Nishiyama, S. Hori-nouchi, and E. T. Adman, Biochemistry 36, 1533 (1991).

Y. Duan, C. Wu, S. Chowdhury, M. C. Lee, G. Xiong, W. Zhang, R. Yang, P. Cieplak, R. Luo, T. Lee, et al., J. Comp. Chem. 24, 1999 (2003).

M. Cascella, A. Magistrato, I. Tavernelli, P. Carloni, and U. Rothlisberger, Proc. Natl. Acad. Sci. 103, 19641 (2006).

J. Dzubiella, J. M. J. Swanson, and J. A. McCammon, Phys. Rev. Lett. 96, 087802 (2006).

J. T. Hupp, Y. Dong, R. L. Blackbourn, and H. Lu, J. Phys. Chem. 97, 3278 (1993).

D. M. D’Alessandro, P. C. Junk, and F. R. Keene, Supramolecular Chemistry 17, 529 (2005).

P. D. Swartz and T. Ichiye, Biochemistry 35, 13772 (1996).

G. R. Moore, G. W. Pettigrew, and N. K. Rogers, Proc. Natl. Acad. Sci. USA 83, 4998 (1986).

T. Hotta, A. Kimura, and M. Sasai, J. Phys. Chem. B 109, 18600 (2005).

G. Hummer and S. Garde, Phys. Rev. Lett. 80, 4193 (1998).

D. Chandler, Nature 437, 640 (2005).

J. Florian and A. Warshel, J. Phys. Chem. B 103, 10282 (1999).

P. K. Ghorai and D. V. Matyushov, J. Phys. Chem. B 110, 1866 (2006).

H. S. Ashbaugh, J. Phys. Chem. B 104, 7235 (2000).

S. Rajamani, T. Ghosh, and S. Garde, J. Chem. Phys. 120, 4457 (2004).

D. S. Cerutti, A. E. Garcia, M. E. Paulaitis, and L. R. Pratt, Phys. Rev. Lett. 77, 4966 (1996).

H. S. Ashbaugh and L. R. Pratt, Rev. Mod. Phys. 78, 159 (2006).

K. Lum, D. Chandler, and J. Weeks, J. Phys. Chem. B 103, 4570 (1999).

D. M. Huang and D. Chandler, Phys. Rev. E 61, 1501 (2000).

P. W. Fenimore, H. Frauenfelder, B. H. McMahon, and R. D. Young, Proc. Natl. Acad. Sci. 101, 14408 (2004).

J. Dzubiella, J. M. J. Swanson, and J. A. McCammon, Phys. Rev. Lett. 96, 087802 (2006).

D. M. D’Alessandro, P. C. Junk, and F. R. Keene, Supramolecular Chemistry 17, 529 (2005).