Dislocation Lines in Three-Dimensional Solids at Low Temperature

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Abstract. We propose a model for three-dimensional solids on a mesoscopic scale with a statistical mechanical description of dislocation lines in thermal equilibrium. The model has a linearized rotational symmetry, which is broken by boundary conditions. We show that this symmetry is spontaneously broken in the thermodynamic limit at small positive temperatures.

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1. Introduction

1.1. Motivation and Background

The perhaps most fundamental mathematical problem of solid-state physics is that of crystallization, which in a classical version could be formulated as follows. Let $v : \mathbb{R}_{>0} \to \mathbb{R}$ be a two-body potential of Lennard–Jones type and consider the $N$-particle energy

$$H_N(x_1, \ldots, x_N) = \sum_{i \neq j} v(|x_i - x_j|), \quad (x_i \in \mathbb{R}^3). \tag{1.1}$$

The Grand Canonical Gibbs measure at inverse temperature $\beta > 0$ and fugacity $z > 0$ in finite volume $\Omega \subset \mathbb{R}^3$ is the point process on $\Omega$ given by

$$P_{\Omega, \beta, z}(B) = \frac{1}{Z_{\Omega, \beta, z}} \sum_{N \in \mathbb{N}_0} \int_{\Omega \cap B} \frac{z^N}{N!} e^{-\beta H_N(x)} \, dx \tag{1.2}$$

for measurable $B \subseteq \bigcup_{N \in \mathbb{N}_0} \Omega^N$ modulo permutations of particles with the appropriate normalizing constant $Z_{\Omega, \beta, z}$. The crystallization problem is to
prove that at low temperature and high density, i.e., large but finite inverse temperature $\beta$ and fugacity $z$, there exist corresponding infinite volume Gibbs measures that are non-trivially periodic with the symmetry of a crystal lattice. This problem remains far out of reach. Unfortunately, even the zero temperature case, i.e., studying the limiting minimizers of the finite volume energy, is open and appears to be very difficult. For the zero temperature case in two dimensions, see Theil [21] and references therein; for important progress on the problem in three dimensions, see Flatley and Theil [9] and references therein. Detailed understanding of the zero temperature case is a prerequisite for understanding the low temperature regime, but in addition, any description at finite temperature must explain spontaneous breaking of the rotational symmetry and take into account the possibility of crystal dislocations. This significantly complicates the problem because proving spontaneous breaking of continuous symmetries is already notoriously difficult in models where the ground states are obvious, such as in the $O(3)$ spin model, for which the only robust method is the very difficult work of Balaban, see [3] and references therein.

With a realistic microscopic model out of reach, we start from a mesoscopic rather than microscopic perspective to understand the effect of the dislocations. We expect the latter to be a fundamental aspect also of the original problem. Our model does not account for the full rotational symmetry, but is only invariant under linearized rotations. More precisely, our model for deformed solids in three dimensions consists of a gas of closed vector-valued defect lines which describe crystal dislocations on a mesoscopic scale. For this model, we show that the breaking of the linearized rotational symmetry persists in the thermodynamic limit.

Our model is strongly motivated by the one introduced and studied by Kosterlitz and Thouless [17], and refined by Nelson and Halperin [19], and Young [23]. The Kosterlitz–Thouless model has an energy that consists of an elastic contribution and a contribution due to crystal dislocations. These two contributions are assumed independent. This KTHNY theory explains crystallization and the melting transition in two dimensions, as a transition mediated by vector-valued dislocations effectively interacting through a Coulomb interaction. For a textbook treatment of this phenomenology, see Chaikin and Lubensky [6]. The Kosterlitz–Thouless model for two-dimensional melting is closely related to the two-dimensional rotator model, studied by Kosterlitz and Thouless in the same paper as the melting problem, following previous insight by Berezinskiǐ [4]. For their model of a two-dimensional solid, the assumption that the energy consists of elastic and dislocation contributions, which can be assumed to be essentially independent, is not derived from a realistic microscopic model. On the other hand, the rotator model admits an exact description in terms of spin waves (corresponding to the elastic energy) and vortices described by a scalar Coulomb gas. In this description, the spin wave and vortex contributions are not far from independent, and, in fact, in the Villain version of the rotator model [22], they become exactly independent. Based on a formal renormalization group analysis, Kosterlitz and Thouless proposed
a novel phase transition mediated by unbinding of the topological defects, the Berezinskii–Kosterlitz–Thouless transition. In the two-dimensional rotator model, the existence of this transition was proved by Fröhlich and Spencer [12]. For recent results on the two-dimensional Coulomb gas, see Falco [8]. In higher dimensions, the description of the rotator model in terms of spin waves and vortex defects remains valid, except that the vortex defects, which are point defects in two dimensions, now become closed vortex lines [13] in our solid model. Using this description and the methods they had introduced for the two-dimensional case, Fröhlich and Spencer [12,13] proved long-range order for the rotator model at low temperature in dimensions \( d \geq 3 \), without relying on reflection positivity. The latter is a very special feature used in [11] to establish long-range order for the \( O(n) \) model exactly with nearest neighbor interaction on \( \mathbb{Z}^d, d \geq 3 \). In general, proving spontaneous symmetry breaking of continuous symmetries remains a difficult problem. However, aside from the most general approach of Balaban and reflection positivity, for abelian spin models, several other techniques exist [13,16].

As discussed above, our model, defined precisely in (1.25), is closely related to the Kosterlitz–Thouless model, see for example [6, (9.5.1)]. Our analysis is based on the Fröhlich–Spencer approach for the rotator model [12,13].

In a parallel study of Giuliani and Theil [14] following Ariza and Ortiz [1], a model very similar to ours is examined, but with a microscopic interpretation, describing locations of individual atoms. In particular, it also has a linearized rotational symmetry.

In [15,18] (see also [2]), some of us studied other simplified models for crystallization. These models have full rotational symmetry, but do not permit dislocations. In [18] defects were excluded, while in the model in [15] isolated missing single atoms were allowed.

1.2. A Linear Model for Dislocation Lines on a Mesoscopic Scale

1.2.1. Linearized Elastic Deformation Energy. An elastically deformed solid in continuum approximation can be described by a deformation map \( f : \mathbb{R}^3 \to \mathbb{R}^3 \) with the interpretation that for any point \( x \) in the undeformed solid \( f(x) \) is the location of \( x \) after deformation. The Jacobi matrix \( \nabla f : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3} \) describes the deformation map locally in linear approximation. Only orientation preserving maps, \( \det \nabla f > 0 \), make sense physically. The elastic deformation energy \( E_{el}(f) \) is modeled to be an integral over a smooth elastic energy density \( \rho_{el} : \mathbb{R}^{3 \times 3} \to \mathbb{R} \) (respectively \( \tilde{\rho}_{el} : \mathbb{R}^{3 \times 3} \to \mathbb{R} \)):

\[
E_{el}(f) = \int_{\mathbb{R}^3} \rho_{el}(\nabla f(x)) \, dx = \int_{\mathbb{R}^3} \tilde{\rho}_{el}(\nabla f(x)^t \nabla f(x)) \, dx, \quad (1.3)
\]

where the second representation holds under the assumption of rotation invariance; see Appendix A.1.1. From now on, we consider only small perturbations \( f = \text{id} + \varepsilon u : \mathbb{R}^3 \to \mathbb{R}^3 \) of the identity map as deformation maps. The parameter \( \varepsilon \) corresponds to the ratio between the microscopic and the mesoscopic scale. We Taylor-expand \( \tilde{\rho}_{el} \) around the identity matrix \( \text{Id} \) using that \( \tilde{\rho}_{el} \) is
smooth near $\text{Id}$, obtaining
\begin{equation}
\rho_{\text{el}}(\text{Id} + \varepsilon A) = \varepsilon^2 F(A) + O(\varepsilon^3), \quad (\varepsilon \to 0, \ A = A^t \in \mathbb{R}^{3 \times 3}), \tag{1.4}
\end{equation}
with a positive definite quadratic form $F$ on symmetric matrices. Under the assumption of isotropy (see Appendix A.1), writing $|\cdot|$ for the Euclidean norm, the general form for $F$ is
\begin{equation}
F(U) = \frac{\lambda}{2} (\text{Tr } U)^2 + \mu |U|^2 \quad \text{with} \quad \mu > 0 \text{ and } \mu + 3\lambda/2 > 0. \tag{1.5}
\end{equation}
In elasticity theory, the constants $\lambda$ and $\mu$ are the so-called Lamé coefficients. Even for cubic monocrystals, the isotropy assumption is restrictive for realistic models. While it is not important for our analysis, we nonetheless assume isotropy to keep the notation somewhat simpler. We refer to [6, Chapters 6.4.2 and 6.4.3] for a discussion on the number of elastic constants necessary in order to describe various crystal systems. Summarizing, we have the following model for the linearized elastic deformation energy:
\[ E_{\text{el}}(\text{id} + \varepsilon u) = \varepsilon^2 H_{\text{el}}(\nabla u) + O(\varepsilon^3) \quad \text{with} \tag{1.6}
\]
\begin{equation}
H_{\text{el}}(w) = \int_{\mathbb{R}^3} F(w(x) + w(x)^t) \, dx, \tag{1.7}
\end{equation}
for measurable $w : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3}$ and for $F$ as in (1.5).

1.2.2. Burgers Vector Densities. The following model is intended to describe dislocation lines on a mesoscopic scale as they appear in solids at positive temperature. We describe the solid by a smooth map $w : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3}$ replacing the map $\nabla u : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3}$ from Sect. 1.2.1. If dislocation lines are absent, the model described now boils down to the setup of Sect. 1.2.1 with $w = \nabla u$ being a gradient field. The field
\begin{equation}
b : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3 \times 3}, \quad b_{ijk} = (d_1 w)_{ijk} := \partial_i w_{jk} - \partial_j w_{ik} \tag{1.8}
\end{equation}
is intended to describe the Burgers vector density. It vanishes if and only if $w = \nabla u$ is a gradient field. One can interpret $b_{ijk}$ as the $k$-th component of the resulting vector per area if one goes through the image in the deformed solid of a rectangle which is parallel to the $i$-th and $j$-th coordinate axis. The antisymmetry $b_{ijk} = -b_{jik}$ can be interpreted as the change of sign if the orientation of the rectangle is changed. Any smooth field $b : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3 \times 3}$ which is antisymmetric in its first two indices is of the form $b = d_1 w$ with some $w : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3}$ if and only if
\begin{equation}
d_2 b = 0, \tag{1.9}
\end{equation}
where
\begin{equation}
d_2 b : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3 \times 3 \times 3}, \quad (d_2 b)_{ij} = \partial_i b_{jk} + \partial_j b_{ilk} + \partial_l b_{ijk} \tag{1.10}
\end{equation}
denotes the exterior derivative with respect to the first two indices. Being antisymmetric in its first two indices, it is convenient to write the Burgers
vector density $b$ in the form

$$b_{ijk} = \sum_{l=1}^{3} \varepsilon_{ijl} \tilde{b}_{lk}, \quad (1.11)$$

where $\tilde{b} : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3}$ and $\varepsilon_{ijk} = \det(e_i, e_j, e_k)$ with the standard unit vectors $e_i \in \mathbb{R}^3$, $i \in [3] := \{1, 2, 3\}$. The integrability condition (1.9) can be written in the form

$$\text{div} \tilde{b} = 0 \quad \text{with} \quad (\text{div} \tilde{b})_k := \sum_{l=1}^{3} \partial_l \tilde{b}_{lk}. \quad (1.12)$$

In view of this equation, one may visualize $\tilde{b}$ to be a sourceless vector-valued current.

1.2.3. Model Assumptions. In linear approximation, the leading order total energy of a deformed solid described by $w : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3}$ is modeled to consist of an “elastic” part and a local “dislocation” part:

$$H(w) = H_{\text{el}}(w) + \mathcal{H}_{\text{disl}}(d_1 w), \quad (1.13)$$

where $H_{\text{el}}(w)$ was introduced in (1.7). The field $w$ consists of an exact contribution (modeling purely elastic fluctuations) and a co-exact contribution representing the elastic part of the energy induced by dislocations. Both these contributions are contained in $H_{\text{el}}(w)$, while $\mathcal{H}_{\text{disl}}(d_1 w)$ is intended to model only the local energy of dislocations. The dislocation part $\mathcal{H}_{\text{disl}}(b) \in [0, \infty]$ is defined for measurable $b : \mathbb{R}^3 \to \mathbb{R}^{3 \times 3 \times 3}$ being antisymmetric in its first two indices.

We describe now a formally coarse-grained model for dislocation lines: Dislocation lines are only allowed in the set $\Lambda$ of undirected edges of a mesoscopic lattice in $\mathbb{R}^3$. Let $V_\Lambda$ denote its vertex set. As a lattice, the graph $(V_\Lambda, \Lambda)$ is of bounded degree. To model boundary conditions, we only allow dislocation lines on a finite subgraph $G = (V, E)$ of $(V_\Lambda, \Lambda)$, ultimately taking the thermodynamic limit $E \uparrow \Lambda$. We write $E \subset \Lambda$ if $E$ is a finite subset of $\Lambda$. We denote the edge between adjacent vertices $x, y \in V_\Lambda$ by $\{x, y\}$. The graph $(V_\Lambda, \Lambda)$ is not intended to describe the atomic structure of the solid, as it lives on a mesoscopic scale. Rather, it is just a tool to introduce a coarse-grained structure which eventually makes the model discrete.

To every edge $e = \{x, y\}$, we associate a counting direction, which has no physical meaning but serves only for bookkeeping purposes. The Burgers vectors on the finite subgraph $G = (V, E)$ are encoded by a family $I = (I_e)_{e \in E} \in (\mathbb{R}^3)^E$ of vector-valued currents flowing through the edges in counting direction. A vector $I_e$ means the Burgers vector associated with a closed curve surrounding the dislocation line segment $[x, y]$ in positive orientation with respect to the counting direction. The family of currents $I$ should fulfill Kirchhoff’s node law

$$\sum_{e \in E} s_{ve} I_e = 0, \quad (v \in V), \quad (1.14)$$
where $s \in \{1, -1, 0\}^{V_\Lambda \times \Lambda}$ is the signed incidence matrix of the graph $(V_\Lambda, \Lambda)$, defined by its entries

$$s_{ve} = \begin{cases} 
1 & \text{if } e \text{ is an incoming edge of } v, \\
-1 & \text{if } e \text{ is an outgoing edge from } v, \\
0 & \text{otherwise.}
\end{cases} \quad (1.15)$$

The distribution of the current in space encoded by $I$ is supported on the union of the line segments $[x, y]$, with $\{x, y\} \in E$. Thus, it is a rather singular object having no density with respect to the Lebesgue measure on $\mathbb{R}^3$. We describe it as follows by a matrix-valued measure $J(I) : \text{Borel(} \mathbb{R}^3) \rightarrow \mathbb{R}^{3 \times 3}$ on $\mathbb{R}^3$, supported on the union of all edges: For $e = \{x, y\} \in \Lambda$, let $\lambda_e : \text{Borel(} \mathbb{R}^3) \rightarrow \mathbb{R}_{>0}$ denote the 1-dimensional Lebesgue measure on the line segment $[x, y]$; it is normalized by $\lambda_e(\mathbb{R}^3) = |x - y|$. Furthermore, let $n_e \in \mathbb{R}^3$ denote the unit vector pointing in the counting direction of the edge $e$. The matrix-valued measure $J(I)$ is then defined by

$$\text{Borel(} \mathbb{R}^3) \ni B \mapsto J_{jk}(I)(B) = \sum_{e \in E} (n_e)_j(I_e)_k \lambda_e(B), \quad j, k \in [3]. \quad (1.16)$$

Thus, the index $k$ encodes a component of the Burgers vector and the index $j$ a component of the direction of the dislocation line.

Heuristically, the current distribution $J(I)$ is intended to describe a coarse-grained picture of a much more complex microscopic dislocation configuration: On an elementary cell of the mesoscopic lattice this microscopic configuration is replaced by a vector-valued current on a single dislocation line segment, encoding the effective Burgers vector. Because the outcome $J$ of this heuristic coarse-graining procedure is such a singular object supported on line segments, its elastic energy close to the dislocation lines would be ill-defined. Hence, the coarse-graining must be accompanied by a smoothing operation.

More precisely, the Burgers vector density $\tilde{b}(I)$ associated with $I$ is modeled by the convolution of $J(I)$ with a form function $\varphi$:

$$\tilde{b}(I) = \varphi * J(I). \quad (1.17)$$

Here, the form function $\varphi : \mathbb{R}^3 \rightarrow \mathbb{R}_{\geq 0}$ is chosen to be smooth, compactly supported, with total mass $\|\varphi\|_1 = 1$, and $\varphi(0) > 0$.

Altogether, this yields the Burgers vector density as a function of $I$:

$$b_{ijk}(I) = \sum_{l=1}^3 \varepsilon_{ijl} \varphi * J_{lk}(I) = \sum_{l=1}^3 \varepsilon_{ijl} \varphi * \sum_{e \in E} (n_e)_i(I_e)_k \lambda_e. \quad (1.18)$$

For a graphical illustration of $I$ and $b(I)$ see Fig. 1.

It is shown in Appendix A.2 that the Kirchhoff node law (1.14) implies that $\tilde{b}(I)$ is sourceless, i.e., Eq. (1.12) holds for it, or equivalently, that the integrability condition $d_2 b = 0$ is valid.
We now impose an additional discreteness condition on \( I \), which encodes the restriction that Burgers vectors should take values in a \textit{microscopic} lattice reflecting the atomic structure of the solid. Let \( \Gamma \subset \mathbb{R}^3 \) be a lattice, interpreted as the microscopic lattice (scaled to length scale 1). We set

\[
\mathcal{I} = \mathcal{I}(E) = \{ I \in \Gamma^E : (1.14) \text{ holds for } I \}. 
\] (1.19)

Note that the current \( I \) is indexed by the edges in the mesoscopic graph \((V,E)\) but takes values in the microscopic lattice \( \Gamma \). One should not confuse the mesoscopic graph \((V,E)\) nor the mesoscopic lattice \( \Lambda \) with the microscopic lattice \( \Gamma \); they have nothing to do with each other. A motivation for the introduction of two different lattices \( \Gamma \) and \( \Lambda \) on two different length scales is described in the discussion of the model at the end of this section.

From now on, we abbreviate

\[
H_{\text{disl}}(I) := H_{\text{disl}}(b(I)) \quad \text{and supp } I := \{ e \in E : I_e \neq 0 \}. 
\]

We require the following general assumptions:

\textbf{Assumption 1.1.}  

- \textbf{Symmetry:} \( H_{\text{disl}}(I) = H_{\text{disl}}(-I) \) for all \( I \in \mathcal{I} \).
- \textbf{Locality:} For \( I = I_1 + I_2 \) with \( I_1, I_2 \in \mathcal{I} \) such that no edge in supp \( I_1 \) has a common vertex with another edge in supp \( I_2 \) we have \( H_{\text{disl}}(I) = H_{\text{disl}}(I_1) + H_{\text{disl}}(I_2) \). Moreover, \( H_{\text{disl}}(0) = 0 \).
- \textbf{Lower bound:} For some constant \( c > 0 \) and all \( I \in \mathcal{I} \),

\[
H_{\text{disl}}(I) \geq c \| I \|_1 := c \sum_{e \in E} |I_e|. 
\] (1.20)

Condition (1.20) reflects the local energetic costs of dislocations in addition to the elastic energy costs reflected by \( H_{\text{el}}(w) \).

One obtains typical examples for \( H_{\text{disl}}(I) \) by requiring a number of assumptions on the form function \( \varphi \) and \( H_{\text{disl}}(b) \): First, for all \( \{u, v\}, \{x, y\} \in \Lambda \) with \( \{u, v\} \cap \{x, y\} = \emptyset \), we assume

\[
([u, v] + \text{supp } \varphi) \cap ([x, y] + \text{supp } \varphi) = \emptyset. 
\]

Furthermore,

\[
\inf_{e=\{x,y\} \in \Lambda} \lambda_e \left( \left\{ z \in [x, y] : (z + \text{supp } \varphi) \cap (w + \text{supp } \varphi) = \emptyset \right. \right. 
\]

for all \( w \in [u, v] \) with \( \{u, v\} \in \Lambda \backslash \{e\} \) > 0. 

(1.21)
Roughly speaking, the last condition means that different edges \(e \in \Lambda_d\) on \(\otimes\) overlap too much after broadening with \(\text{supp} \varphi\). Finally, for all \(b\),

\[
\mathcal{H}_{\text{disl}}(b) = \mathcal{H}_{\text{disl}}(-b) \geq c_1 \int_{\mathbb{R}^3} |b(x)|_1 \, dx = c_1 \sum_{i,j,k} \int_{\mathbb{R}^3} |b_{ijk}(x)| \, dx
\]

(1.22)

for some constant \(c_1 > 0\), and for \(b = b_1 + b_2\) with \(\text{supp} b_1 \cap \text{supp} b_2 = \emptyset\), it is true that \(\mathcal{H}_{\text{disl}}(b) = \mathcal{H}_{\text{disl}}(b_1) + \mathcal{H}_{\text{disl}}(b_2)\). One particular example is obtained by taking equality in (1.22).

1.3. Model and Main Result

One summand in the Hamiltonian of our model is defined by

\[
H^*_\text{el}(I) = \inf_{w \in C^\infty_c(\mathbb{R}^3, \mathbb{R}^3 \times \mathbb{R}^3)} d_1 w = b(I)
\]

(1.23)

for all \(I \in (\mathbb{R}^3)^E\) satisfying (1.14). The condition that \(w\) is compactly supported reflects the boundary condition, in the sense that close to infinity the solid must not be moved away from its reference location. The symmetry with respect to linearized global rotations is reflected by the fact \(H_{\text{el}}(w) = H_{\text{el}}(w + w_{\text{const}})\) for every constant antisymmetric matrix \(w_{\text{const}} \in \mathbb{R}^{3 \times 3}\). Only the boundary condition, i.e., only the restriction that \(w\) should have compact support, breaks this global symmetry. This paper is about the question whether this symmetry breaking persists in the thermodynamic limit \(E \uparrow \Lambda\).

Because \(H_{\text{el}}\) is positive semidefinite, \(H^*_\text{el}\) is positive semidefinite as well, cf. (2.36). This gives us the following linearized model for the dislocation lines at inverse temperature \(\beta < \infty\):

\[
Z_{\beta,E} := \sum_{I \in \mathcal{I}(E)} e^{-\beta(H^*_\text{el}(I) + H_{\text{disl}}(I))},
\]

(1.24)

\[
P_{\beta,E} := \frac{1}{Z_{\beta,E}} \sum_{I \in \mathcal{I}(E)} e^{-\beta(H^*_\text{el}(I) + H_{\text{disl}}(I))} \delta_I,
\]

(1.25)

where \(\delta_I\) denotes the Dirac measure in \(I \in \mathcal{I}\) and we use the convention \(e^{-\infty} = 0\) throughout the paper. Whenever the \(E\)-dependence is kept fixed, we use the abbreviations \(Z_{\beta} = Z_{\beta,E}\) and \(P_{\beta} = P_{\beta,E}\).

The following preliminary result shows that any sequence of smooth configurations satisfying the boundary conditions (i.e., being compactly supported) with prescribed Burgers vectors has a limit \(w^*\) in \(L^2\) provided that the energy is approaching the infimum of all energies within the class. We show later that this limit is a unique minimizer of \(H_{\text{el}}\) in a suitable Sobolev space. An explicit description of \(w^*\) is provided in Lemma 2.2.

**Proposition 1.2** (Compactly supported approximations of the minimizer). For any \(I \in \mathcal{I}\), there is a bounded smooth function \(w^*(\cdot, I) \in L^2(\mathbb{R}^3, \mathbb{R}^{3 \times 3})\) such that for any sequence \((w^n)_{n \in \mathbb{N}}\) in \(C_c^\infty(\mathbb{R}^3, \mathbb{R}^{3 \times 3})\) with \(d_1 w^n = b(I)\) for all \(n \in \mathbb{N}\) and \(\lim_{n \to \infty} H_{\text{el}}(w^n) = H_{\text{el}}(I)\) we have \(\lim_{n \to \infty} \|w^n - w^*(\cdot, I)\|_2 = 0\).
In the whole paper, constants are denoted by \( c_1, c_2, \) etc. They may depend on the fixed model ingredients: the microscopic lattice \( \Gamma \), the mesoscopic lattice \( \Lambda \), the constant \( c \) from formula (1.20), and the form function \( \varphi \). All constants keep their meaning throughout the paper. Similarly, the expression “\( \beta \) large enough” means “\( \beta > \beta_0 \) with some constant \( \beta_0 \) depending also only on \( \Gamma, \Lambda, c, \) and \( \varphi \).”

The following theorem shows that the breaking of linearized rotational symmetry \( w \rightarrow w + w_{\text{const}} \) induced by the boundary conditions persists in the thermodynamic limit \( E \uparrow \Lambda \), provided that \( \beta \) is large enough.

**Theorem 1.3** (Spontaneous breaking of linearized rotational symmetry). There is a constant \( c_2 > 0 \) such that for all \( \beta \) large enough and for all \( t \in \mathbb{R} \),

\[
\inf_{E \in \Lambda} \inf_{x, y \in \mathbb{R}^3} \min_{i, j \in [3]} E_{P_{\beta, E}} \left[ e^{it (w_{ij}^*(x, I) - w_{ij}^*(y, I))} \right] \geq \exp \left\{ - \frac{t^2}{2} e^{-c_2 \beta} \right\},
\]

and consequently

\[
\sup_{E \in \Lambda} \sup_{x, y \in \mathbb{R}^3} \max_{i, j \in [3]} \text{var}_{P_{\beta, E}} \left( w_{ij}^*(x, I) - w_{ij}^*(y, I) \right) \leq e^{-c_2 \beta}.
\]

We remark that the symmetry \( I \leftrightarrow -I \) implies that \( w^*(x, I) \) is a centered random matrix. Since \( w^*(x, I) \) encodes in particular the orientation of the crystal at location \( x \), this result may be interpreted as the presence of long-range orientational order in the thermodynamic limit.

**Discussion of the model** If we compare our model to the rotator model, purely elastic deformations correspond to “spin wave” contributions, while deformations induced by the Burgers vectors correspond to “vortex” contributions. In our model, the purely elastic deformations are orthogonal to the deformations induced by the Burgers vectors in a suitable inner product \( \langle \cdot, \cdot \rangle_F \); this is made precise in Eq. (2.40). Therefore, we do not model the purely elastic part stochastically. It would not be relevant for our purposes, because in a linearized model, it is expected to be independent of the Burgers vectors anyway.

The mixed continuum/lattice structure of the model has the following motivation: A realistic microscopic description of a crystal at positive temperature would be very complicated, including, e.g., vacancies and interstitial atoms. This makes a global indexing of all atoms by a lattice intrinsically hard. On a mesoscopic scale, we expect that these difficulties can be neglected when the elastic parameters of the model are renormalized. Our model should be understood as a two-scale description of the crystal in which the microscopic Burgers vectors are represented on a scale such that their discreteness is still visible, but the physical space is smoothed out; recall that the factor \( \varepsilon \) used in approximation (1.4) encodes the ratio between the two scales. Hence, we use continuous derivatives in physical space, but discrete calculus for the Burgers vectors. The mesoscopic lattice \( V_\Lambda \) serves only as a convenient spatial regularization. Unlike the microscopic lattice \( \Gamma \), it has no intrinsic physical meaning.

**Organization of the paper.** In Sect. 2, we identify the minimal energy configuration \( w^* \) in the sense of Proposition 1.2 in the appropriate Sobolev space.
Section 3 deals with the statistical mechanics of Burgers vector configurations by means of a Sine–Gordon transformation and a cluster expansion in the spirit of the Fröhlich-Spencer treatment of the Villain model [12,13]. Section 4 provides the bounds for the observable, which manifests the spontaneous breaking of linearized rotational symmetry. It uses variants of a dipole expansion, which we provide in the Appendix.

2. Minimizing the Elastic Energy

In this section, we collect various properties of \(H^*_e(I)\) defined in (1.23). In particular, we prove Proposition 1.2.

2.1. Sobolev Spaces

Let \(V\) be a finite-dimensional \(\mathbb{C}\)-vector space with a norm \(| \cdot |\) coming from a scalar product \(\langle \cdot, \cdot \rangle_V\). For integrable \(f : \mathbb{R}^3 \to V\), let

\[
\hat{f}(k) = (2\pi)^{-3/2} \int_{\mathbb{R}^3} e^{-i(k \cdot x)} f(x) \, dx
\]

denote its Fourier transform, normalized such that the transformation becomes unitary. For any \(\alpha \in \mathbb{R}\) and \(f \in C^\infty_c(\mathbb{R}^3, V)\), we define

\[
\|f\|_\alpha^V := \|\hat{f}\|_{2,\alpha},\text{ where } \|g\|_{2,\alpha} := \int_{\mathbb{R}^3} |k|^{2\alpha} |g(k)|^2 \, dk.
\]

We set

\[
C_\alpha(V) := \{ f \in C^\infty_c(\mathbb{R}^3, V) : \|f\|_\alpha^V < \infty \}.
\]

Then, \(\|f\|_\alpha^V\) is a norm on the \(\mathbb{C}\)-vector space \(C_\alpha(V)\). For \(\alpha > -3/2\), we know that \(C_\alpha(V) = C^\infty_c(\mathbb{R}^3, V)\) because \(|k|^{2\alpha}\) is integrable near 0 and \(\hat{f}\) decays fast at infinity. Let \((L^2_\alpha(V), \| \cdot \|_\alpha)\) denote the completion of \((C_\alpha(V), \| \cdot \|_\alpha^V)\) and \(L^2_\alpha(V) := \{ g : \mathbb{R}^3 \to V \text{ measurable mod changes on null sets : } \|g\|_{2,\alpha} < \infty \}\).

The Fourier transform \(f \mapsto \hat{f}\) gives rise to a natural isometric isomorphism \(L^2_\alpha(V) \to L^2_\alpha(V)\). For any \(\alpha \in \mathbb{R}\), the sesquilinear form

\[
\langle \cdot, \cdot \rangle : C_{-\alpha}(V) \times C_\alpha(V) \to \mathbb{C}, \quad \langle f, g \rangle = \int_{\mathbb{R}^3} \langle f(x), g(x) \rangle_V \, dx
\]

extends to a continuous sesquilinear form

\[
\langle \cdot, \cdot \rangle : L^2_{-\alpha}(V) \times L^2_\alpha(V) \to \mathbb{C}.
\]

Partial derivatives \(\partial_j : C^\infty_c(\mathbb{R}^3, V) \to C^\infty_c(\mathbb{R}^3, V)\) (acting component-wise) extend to bounded operators \(\partial_j : L^2_{\alpha+1}(V) \to L^2_\alpha(V)\). Consequently, the (component-wise) Laplace operator \(\Delta = \sum_{j \in [3]} \partial_j^2\) extends to an isometric isomorphism \(\Delta : L^2_{\alpha+2}(V) \to L^2_\alpha(V)\).

We will mainly have \(V = V_j\) for \(j \in \mathbb{N}_0\), where

\[
V_j = \{ (a_{i_1 \ldots i_j})_{i_1, \ldots, i_j, k} \in \mathbb{C}^{3 \times \ldots \times 3} : a_{i_1 \ldots i_j k} \text{ is antisymmetric in } i_1, \ldots, i_j \},
\]
endowed with the norm
\[
|a| = \left( \frac{1}{j!} \sum_{i_1 \ldots i_j, k \in [3]} |a_{i_1 \ldots i_j k}|^2 \right)^{\frac{1}{2}}.
\] (2.7)

Functions with values in \( \mathbb{V}_j \) are just \( C^3 \)-valued \( j \)-forms. Note that the last index \( k \) has a special role since there is no antisymmetry condition for it. The real part of the space \( \mathbb{V}_0 = C^3 \) may be interpreted as a vector space containing Burgers vectors. For any \( \mathbb{L} \)

\[
\Delta = \sum_{i=1}^{3} \partial_i \partial_i.
\]

They are adjoint to each other in the sense that for any \( \alpha \in \mathbb{R} \),

\[
\langle d_j^* a, b \rangle = \langle a, d_j b \rangle \quad \text{for } a \in L^2_{-\alpha+1}(\mathbb{V}_{j+1}), \ b \in L^2_{\alpha}(\mathbb{V}_j).
\] (2.10)

Since in the following we are mostly interested in the cases \( j = 0, 1, 2 \), we spell out the definition of \( d_j \) explicitly:

\[
(d_0 f)_{ij} = \partial_i f_j,
\] (2.11)

\[
(d_1 w)_{ijk} = \partial_i w_{jk} - \partial_j w_{ik},
\] (2.12)

\[
(d_2 b)_{ijkl} = \partial_i b_{jkl} + \partial_j b_{ikl} + \partial_k b_{ijl} = \partial_l b_{jkl} - \partial_j b_{ikl} + \partial_k b_{ijl}.
\] (2.13)

The Laplace operator \( \Delta : L^2_{\alpha+2}(\mathbb{V}_j) \rightarrow L^2_{\alpha}(\mathbb{V}_j) \) then fulfills

\[
\Delta = -(d_j^* d_j + d_{j-1}^* d_{j-1}^*), \quad (j \in \mathbb{N}_0); \quad (2.14)
\]

here, \( d_{-1} \) and \( d_{-1}^* \) should be interpreted as 0. Equation (2.14) is only important for \( j = 0, 1, 2, 3 \) because \( \mathbb{V}_j = \{0\} \) holds for \( j \geq 4 \) in three dimensions. To see (2.14), we calculate

\[
(-d_j^* d_j)_{i_1 \ldots i_j k} = \sum_{i_0=1}^{3} \partial_{i_0} (d_j a)_{i_0 i_1 \ldots i_j k}
\]

\[
= \sum_{i_0=1}^{3} \partial_{i_0} \left[ \partial_{i_0} a_{i_1 \ldots i_j k} + \sum_{l=1}^{j} (-1)^{l+2} \partial_{i_l} a_{i_0 i_1 \ldots i_l \ldots i_j k} \right]
\]

\[
= \Delta a_{i_1 \ldots i_j k} - \sum_{i_0=1}^{3} \sum_{l=1}^{j} (-1)^{l+1} \partial_{i_0} \partial_{i_l} a_{i_0 i_1 \ldots i_l \ldots i_j k} \quad (2.15)
\]
and
\[
(-d_j d^{*}_{j-1} a)_{i_1...i_j} = - \sum_{l=1}^{j} (-1)^{l+1} \partial_{i_l} (d^{*}_{j-1} a)_{i_1...\hat{i}_l...i_j}
= \sum_{i_0=1}^{3} \sum_{l=1}^{j} (-1)^{l+1} \partial_{i_0} a_{i_0 i_1...\hat{i}_l...i_j}.
\]
(2.16)

### 2.2. Elastic Hamiltonian

Given \( I \in \mathcal{I} \) and \( b = b(I) \), we calculate now \( H^*_\text{el} (I) \). To begin with, we observe that \( H^*_\text{el} \), introduced in (1.7), is a quadratic form, and therefore, it can be written as
\[
H^*_\text{el} (w) = \langle w, w \rangle_F,
\]
with a sesquilinear form \( \langle \cdot, \cdot \rangle_F \) depending on \( F \) defined in (1.5). More precisely, using \( \langle A, B \rangle = \text{Tr}(AB^*) \) for the Euclidean scalar product for matrices \( A, B \), we introduce \( \langle \cdot, \cdot \rangle_F : L^2(\mathbb{R}^3, \mathbb{C}^{3 \times 3}) \times L^2(\mathbb{R}^3, \mathbb{C}^{3 \times 3}) \to \mathbb{C} \) through
\[
\langle w, \tilde{w} \rangle_F := \int_{\mathbb{R}^3} \left[ \frac{\lambda}{2} (\text{Tr}(w(x) + w^t(x)) \text{Tr}(\tilde{w}(x) + \tilde{w}^t(x)) + \mu \text{Tr}([w(x) + w^t(x)](\tilde{w}(x) + \tilde{w}^t(x))) \right] dx
= 2 \sum_{i,j=1}^{3} \int_{\mathbb{R}^3} \left[ \lambda (w_{i\alpha}(x) \tilde{w}_{j\alpha}(x)) + \mu (w_{i\beta}(x) + w_{j\beta}(x)) \tilde{w}_{i\beta}(x) \right] dx.
\]
(2.18)

Because of the stability condition (1.5), the inner product \( \langle \cdot, \cdot \rangle_F \) is positive semidefinite. For any \( \alpha \in \mathbb{R} \), we consider the restriction of \( \langle \cdot, \cdot \rangle_F \) to \( C_{-\alpha} (V_1) \times C_{\alpha} (V_1) \), and then extend it to a sesquilinear form \( \langle \cdot, \cdot \rangle_F : L^2_{-\alpha} (V_1) \times L^2_{\alpha} (V_1) \to \mathbb{C} \) being continuous w.r.t. \( \| \cdot \|_{\alpha} \) and \( \| \cdot \|_{\alpha}^* \).

In (1.23), we may take the infimum over \( w^b + \ker (d_1) \) with a suitable \( w^b \in L^2_{0\alpha} (V_1) = L^2(\mathbb{R}^3, \mathbb{C}^{3 \times 3}) \) satisfying \( d_1 w^b = b(I) \). We claim that a convenient choice is
\[
w^b := -\Delta^{-1} d_1^* b \in L^2_{0\alpha} (V_1).
\]
(2.19)

To see this, we observe that \( \Delta^{-1} \) commutes with \( d_1 \) and \( d_1^* \) because \( \Delta^{-1} \) corresponds to multiplication with the scalar \( |k|^{-2} \) in Fourier space, and \( d_1, d_1^* \) correspond to (multi-component) multiplication operators in Fourier space, as well. Therefore, since \( b = b(I) \in \ker (d_2 : L^2_{-1}(V_2) \to L^2_{-\alpha}(V_3)) \), we obtain
\[
d_1 w^b = -d_1 d_1^* \Delta^{-1} b = -\Delta^{-1} d_1^* b = -\Delta^{-1} (d_1 d_1^* + d_2 d_2^*) b = b.
\]
(2.20)

Using that \( \ker (d_1 : C_0 (V_1) \to C_1 (V_2)) \) is dense in \( \ker (d_1 : L^2_{0\alpha} (V_1) \to L^2_{0\alpha} (V_2)) = \text{range} (d_0 : L^2_{1\alpha} (V_0) \to L^2_{0\alpha} (V_1)) \), we obtain
\[
H^*_\text{el} (I) = \inf_{w \in L^2_{0\alpha} (V_1) : d_1 w = 0} \langle w^b + w, w^b + w \rangle_F = \inf_{\psi \in L^2_{1\alpha} (V_0)} \langle w^b + d_0 \psi, w^b + d_0 \psi \rangle_F.
\]
(2.21)
2.3. Minimizer

Differential operators In order to analyze the $d_0\psi$-dependence in (2.21), we derive an adjoint $\nabla^F$ for $d_0$ with respect to $\langle \cdot , \cdot \rangle_F$ and $\langle \cdot , \cdot \rangle$. Let $\nabla^F : L^{2\gamma}_{-\alpha+1}(\mathbb{V}_1) \to L^{2\gamma}_{-\alpha}(\mathbb{V}_0)$,

$$
(\nabla^F g)_j := -2 \sum_{i=1}^{3} [\lambda \partial_j g_{ii} + \mu \partial_i (g_{ij} + g_{ji})], \quad (j \in [3]).
$$

Indeed, it satisfies the following adjointness relation for any $g \in L^{2\gamma}_{-\alpha}(\mathbb{V}_1)$ and $f \in L^{2\gamma}_{-\alpha+1}(\mathbb{V}_0)$ (using that $-\partial_j$ is adjoint to $\partial_j$ w.r.t. $\langle \cdot , \cdot \rangle$):

$$
\langle g , d_0 f \rangle_F = 2 \sum_{i,j=1}^{3} [\lambda \langle g_{ii} , \partial_j f_j \rangle + \mu \langle g_{ij} + g_{ji} , \partial_i f_j \rangle] = -2 \sum_{i,j=1}^{3} (\lambda \partial_j g_{ii} + \mu \partial_i (g_{ij} + g_{ji}) , f_j) = \langle \nabla^F g , f \rangle. \quad (2.23)
$$

The identity $\langle d_0 \psi , d_0 \psi \rangle_F = \langle \nabla^F d_0 \psi , \psi \rangle$ motivates us to introduce the following differential operator for any $\alpha \in \mathbb{R}$:

$$
D := \frac{1}{2} \nabla^F d_0 : L^{2\gamma}_{\alpha+2}(\mathbb{V}_0) \to L^{2\gamma}_{\alpha}(\mathbb{V}_0),
$$

$$
D\psi = -\mu \Delta \psi - (\mu + \lambda) \text{grad div } \psi = \left(-\mu \Delta \psi_j - (\mu + \lambda) \partial_j \left( \sum_{i=1}^{3} \partial_i \psi_i \right) \right)_{j \in [3]}. \quad (2.25)
$$

At this moment, we are most interested in the case $\alpha = -1$; the case of general values for $\alpha$ is needed for regularity considerations in the proof of Proposition 1.2 later on.

**Lemma 2.1** (Properties of $D$). For any $\alpha \in \mathbb{R}$, the map $D : L^{2\gamma}_{\alpha+2}(\mathbb{V}_0) \to L^{2\gamma}_{\alpha}(\mathbb{V}_0)$ is invertible with the inverse $D^{-1} : L^{2\gamma}_{\alpha}(\mathbb{V}_0) \to L^{2\gamma}_{\alpha+2}(\mathbb{V}_0)$,

$$
D^{-1}\psi = -\Delta^{-1} \left( \frac{1}{\mu} \psi + \left( \frac{1}{2\mu + \lambda} - \frac{1}{\mu} \right) \Delta^{-1} \text{grad div } \psi \right) . \quad (2.26)
$$

In coordinate notation,

$$
(D^{-1}\psi)_j = -\Delta^{-1} \left( \frac{1}{\mu} \psi_j + \left( \frac{1}{2\mu + \lambda} - \frac{1}{\mu} \right) \Delta^{-1} \partial_j \left( \sum_{i=1}^{3} \partial_i \psi_i \right) \right) . \quad (2.27)
$$

The map $D$ is symmetric for $\alpha = -1$, i.e., $\langle \tilde{\psi} , D\psi \rangle = \langle D\tilde{\psi} , \psi \rangle$ for $\psi, \tilde{\psi} \in L^{2\gamma}_{1}(\mathbb{V}_0)$, and bounded from above and from below as follows:

$$
0 \leq \mu \langle \psi , -\Delta \psi \rangle \leq \langle \psi , D\psi \rangle \leq (2\mu + \lambda) \langle \psi , -\Delta \psi \rangle . \quad (2.28)
$$

**Proof.** Using $\text{div grad} = \Delta$ and abbreviating

$$
\gamma := \frac{1}{2\mu + \lambda} - \frac{1}{\mu} ,
$$

(2.29)
we calculate

\[ D^{-1}D\psi = -\Delta^{-1}\left(\frac{1}{\mu} \text{Id} + \gamma \Delta^{-1} \text{grad div}\right) (-\mu \Delta \psi - (\mu + \lambda) \text{grad div} \psi) \]

\[ = \psi + \gamma \mu \Delta^{-1} \text{grad div} \psi + \frac{\mu + \lambda}{\mu} \Delta^{-1} \text{grad div} \psi \]

\[ + \gamma(\mu + \lambda) \Delta^{-1} \text{grad div} \psi = \psi \]  

(2.30)

and similarly \( DD^{-1} = \text{id} \).

By the adjointness property (2.23), the symmetry of \( D \) is obvious by its definition:

\[ \langle D\psi', \psi \rangle = \langle \nabla^F_d d_0 \psi', \psi \rangle = \langle d_0 \psi', d_0 \psi \rangle_F \]  

for \( \psi, \psi' \in L^2_1(V_0) \).

Furthermore, one has

\[ \langle \psi, D\psi \rangle = \mu \langle \psi, -\Delta \psi \rangle + (\mu + \lambda) \langle \text{div} \psi, \text{div} \psi \rangle. \]  

(2.31)

We claim that

\[ \langle \text{div} \psi, \text{div} \psi \rangle \leq \langle \psi, -\Delta \psi \rangle. \]  

(2.32)

This is best seen using a Fourier transform and the Cauchy–Schwarz inequality in \( \mathbb{C}^3 \):

\[ \langle \text{div} \psi, \text{div} \psi \rangle = \| \text{div} \psi \|^2_2 = \| ik \cdot \tilde{\psi}(k) \|^2_2 \]

\[ \leq \| k \| \cdot \| \tilde{\psi}(k) \|^2_2 = \| d_0 \psi \|^2_2 = \| d_0 \psi \|^2_2 = \langle \psi, -\Delta \psi \rangle \]  

(2.33)

where \( k \cdot \tilde{\psi}(k) \) denotes the Euclidean scalar product in \( \mathbb{C}^3 \). Using fact (2.32) and the stability condition for \( \mu \) and \( \lambda \) given in (1.5), which implies \( \mu + \lambda > 0 \), we obtain also claim (2.28).

**Definition of the minimizer.** In the next lemma, it is shown that the minimizer of the elastic energy has the following form:

\[ w^* := w^b + d_0 \psi^* \]  

(2.34)

with \( w^b \) defined in (2.19),

\[ \psi^* := D^{-1}v^b, \quad \text{and} \quad v^b := -\frac{1}{2} \nabla^F w^b. \]  

(2.35)

**Lemma 2.2** (Minimizer of the elastic energy). The infimum in (2.21) is a minimum:

\[ H_{\text{el}}^*(I) = \langle w^*, w^* \rangle_F. \]  

(2.36)

It is unique in the following sense: For all \( w \in L^2_0(V_1) \) with \( d_1 w = b(I) \) and \( \langle w, w \rangle_F = \langle w^*, w^* \rangle_F \), we have \( w = w^* \). The summands of the minimizer \( w^* \) given in (2.34) have the following components:
\[w^b_{ij} = -(d^*_1 \Delta^{-1} b)_{ij} = \sum_{l=1}^{3} \Delta^{-1} \partial_l b_{lij},\]  

(2.37)

\[d_0 \psi^*_i j = \Delta^{-1} \partial_i \sum_{k=1}^{3} \left( \partial_k (d^*_1 \Delta^{-1} b)_{jk} + \frac{\lambda}{2\mu + \lambda} \partial_j (d^*_1 \Delta^{-1} b)_{kk} \right)\]

\[= -\partial_i \Delta^{-2} \sum_{k,l=1}^{3} \left( \partial_k \partial_l b_{ljk} + \frac{\lambda}{2\mu + \lambda} \partial_j \partial_l b_{lkk} \right), \quad i, j \in [3].\]  

(2.38)

**Proof.** The calculation \(\nabla F(w^b + d_0 \psi^*) = \nabla F(w^b - \frac{1}{2} d_0 D^{-1} \nabla F w^b) = \nabla F w^b - DD^{-1} \nabla F w^b = 0\) shows that the function \(\psi^*\) solves the system of equations

\[\langle \nabla F(w^b + d_0 \psi^*) , f \rangle = 0, \quad (f \in L^{2^\vee}_1(V_0)),\]  

(2.39)

or equivalently, using (2.23) and (2.34),

\[\langle w^* , d_0 f \rangle_F = 0, \quad (f \in L^{2^\vee}_1(V_0)).\]  

(2.40)

By the above, the following calculation shows that \(w^*\) is a minimizer in (2.21) as claimed: For all \(f \in L^{2^\vee}_1(V_0)\):

\[\langle w^* + d_0 f , w^* + d_0 f \rangle_F = \langle w^* , w^* \rangle_F + 2 \text{Re} \langle w^* , d_0 f \rangle_F + \langle d_0 f , d_0 f \rangle_F\]

\[= \langle w^* , w^* \rangle_F + \langle d_0 f , d_0 f \rangle_F \geq \langle w^* , w^* \rangle_F.\]  

(2.41)

Furthermore, using (2.28) we obtain:

\[\langle d_0 f , d_0 f \rangle_F = \langle \nabla F d_0 f , f \rangle = 2 \langle D f , f \rangle\]

\[\geq 2\mu \langle f , -\Delta f \rangle = 2\mu \langle d_0 f , d_0 f \rangle = 2\mu \|d_0 f\|_2^2.\]  

(2.42)

In particular, \(d_0 f \neq 0\) implies \(\langle d_0 f , d_0 f \rangle_F > 0\), which yields the claimed uniqueness of the minimizer. Let \(i, j \in [3]\). Identity (2.37) follows from definition (2.19) of \(w^b\). Using it, we express \(v^b \in L^{2^\vee}_1(V_0)\) as follows:

\[v^b_j = \sum_{k,l=1}^{3} \left[ \lambda \partial_j \partial_l (\Delta^{-1} b)_{lkk} + \mu \partial_k \partial_l ((\Delta^{-1} b)_{lkj} + (\Delta^{-1} b)_{ljk}) \right]\]

\[= \Delta^{-1} \sum_{k,l=1}^{3} \left[ \lambda \partial_j \partial_l b_{lkk} + \mu \partial_k \partial_l b_{ljk} \right].\]  

(2.43)
Because of the antisymmetry $\partial_k \partial_l (\Delta^{-1} b)_{lkj} = -\partial_l \partial_k (\Delta^{-1} b)_{klj}$, one term dropped out in the last step. It follows that
\[
\psi^*_j = (D^{-1} v^b)_j = -\Delta^{-1} \left( \frac{1}{\mu} v^b_j + \left( \frac{1}{2\mu + \lambda} - \frac{1}{\mu} \right) \Delta^{-1} \partial_j \sum_{m=1}^3 \partial_m v^b_m \right)
\]
\[
= -\Delta^{-2} \sum_{k,l=1}^3 \left( \frac{1}{\mu} [\lambda \partial_j \partial_l b_{llk} + \mu \partial_k \partial_l b_{ljk}] + \left( \frac{1}{2\mu + \lambda} - \frac{1}{\mu} \right) \Delta^{-1} \partial_j \sum_{m=1}^3 \partial_m [\lambda \partial_m \partial_l b_{llk} + \mu \partial_k \partial_m b_{lmk}] \right). \quad (2.44)
\]
Using $\sum_{l,m} \partial_m \partial_l b_{lmk} = 0$ from the antisymmetry $b_{lmk} = -b_{mlk}$, this equals
\[
\psi^*_j = -\Delta^{-2} \sum_{k,l=1}^3 \left( \partial_k \partial_l b_{ljk} + \frac{\lambda}{2\mu + \lambda} \partial_j \partial_l b_{llk} \right). \quad (2.45)
\]
This shows that $d_0 \psi^*$ has the form given in (2.38).

\section*{Regularity of the minimizer.}

\textbf{Proof of Proposition 1.2.} We set $L^{2\gamma}_{>\alpha}(V) := \cap_{\alpha'>\alpha} L^{2\gamma}_{\alpha'}(V)$. From $b(I) \in C_c^\infty(\mathbb{R}^3, V_2) = \cap_{\alpha>3/2} C_\alpha(V_2) \subseteq L^{2\gamma}_{>-3/2}(V_2)$ it follows from (2.19) that $w^b \in L^{2\gamma}_{>-1/2}(V_1)$. Hence, by (2.35), $\psi^b \in L^{2\gamma}_{>-3/2}(V_0)$, and then $\psi^* = D^{-1} \psi^b \in L^{2\gamma}_{>-1/2}(V_0)$. We conclude $w^* = w^b + d_0 \psi^* \in L^{2\gamma}_{>-1/2}(V_1)$. By Sobolev’s embedding theorem, $w^*$ is a bounded smooth function with all derivatives being bounded. In particular, pointwise evaluation $w^*(x)$ of $w^*$ makes sense for every $x \in \mathbb{R}^3$.

For the remaining claim, take a sequence $f^n \in L^{2\gamma}_1(V_0)$, $n \in \mathbb{N}$, with $H_{el}(w^* + d_0 f^n) \to H_{el}(w^*) = H_{el}(I)$ as $n \to \infty$. Using $\ker(d_1 : L^{2\gamma}_{\alpha}(V_1) \to L^{2\gamma}_{\alpha}(V_2)) = \text{range}(d_0 : L^{2\gamma}_{\alpha}(V_0) \to L^{2\gamma}_{\alpha}(V_1))$, it suffices to show that $\|d_0 f^n\|_2$ converges to 0 as $n \to \infty$. In view of system (2.40) of equations, we know
\[
\begin{align*}
2 \langle f^n, D f^n \rangle &= \langle d_0 f^n, d_0 f^n \rangle_F + 2 \Re \langle w^*, d_0 f^n \rangle_F = \langle w^* + d_0 f^n, w^* + d_0 f^n \rangle_F - \langle w^*, w^* \rangle_F \\
&= H_{el}(w^* + d_0 f^n) - H_{el}(w^*) \xrightarrow{n \to \infty} 0. \quad (2.46)
\end{align*}
\]
Using comparison (2.28) between $D$ and $-\Delta$, we conclude
\[
\|d_0 f^n\|_2^2 = \langle d_0 f^n, d_0 f^n \rangle = \langle f^n, -\Delta f^n \rangle \xrightarrow{n \to \infty} 0. \quad (2.47)
\]

We remark that the facts $d_2 b(I) = 0$ and $b(I) \in C_c^\infty(\mathbb{R}^3, V_2)$ imply $\int_{\mathbb{R}^3} b(I)(x) \, dx = 0$ and hence $b(I) \in C_\alpha(V_2)$ for all $\alpha > -5/2$, not only for all $\alpha > -3/2$. As a consequence, $w^* \in L^{2\gamma}_{>-3/2}(V_1)$.\[\square\]
3. Cluster Expansion

We now develop a cluster expansion (polymer expansion) of the measures \( P_{\beta,E} \) defined in (1.25), using the strategy of Fröhlich and Spencer [13]. In the following, \( E \subseteq \Lambda \) is a given finite set of edges in the mesoscopic lattice. We take the thermodynamic limit \( E \uparrow \Lambda \) only in the end.

3.1. Sine–Gordon Transformation

The elastic energy \( H^*_{el}(I) \) defined in (1.23) is a quadratic form. If \( I = I_1 + \cdots + I_n \) is the decomposition of \( I \) into its connected components, the mixed terms in \( H^*_{el}(I) \) induce non-local interactions between different \( I_i \) and \( I_j \). The Sine–Gordon transformation introduced now is a tool to avoid these non-localities.

Because the quadratic form \( H^*_{el} \) is positive semidefinite, the function \( \exp\{ -\beta H^*_{el}(I) \} \) is the Fourier transform of a centered Gaussian random vector \( \phi = (\phi_e)_{e \in E} \) on some auxiliary probability space with corresponding expectation operator denoted by \( E \):

\[
E[e^{i\langle \phi, I \rangle}] = e^{-\beta H^*_{el}(I)}. \tag{3.1}
\]

For any observable of the form \( I \ni I \mapsto \langle \sigma, I \rangle \) with \( \sigma \in \mathbb{R}^E \), we define

\[
Z_{\beta,\phi} := \sum_{I \in I} e^{i\langle \phi, I \rangle} e^{-\beta H_{disl}(I)}, \tag{3.2}
\]

\[
Z_\beta(\sigma) := \sum_{I \in I} e^{i\langle \sigma, I \rangle} e^{-\beta (H^*_{el}(I)+H_{disl}(I))} = E[Z_{\beta,\sigma+\phi}]. \tag{3.3}
\]

In order to exchange expectation and summation, we used that \( e^{-\beta H_{disl}(I)} \) is summable over the set \( I \) by (1.20). Note that \( Z_\beta = Z_\beta(0) \) implies

\[
\frac{Z_\beta(\sigma)}{Z_\beta(0)} = E_{P_\beta}[e^{i\langle \sigma, I \rangle}]. \tag{3.4}
\]

3.2. Preliminaries on Cluster Expansions

In this section, we collect some background on cluster expansions (polymer expansions). For recent treatments of cluster expansions, see in particular Poghosyan and Ueltschi [20] or Bovier and Zahradník [5] and references. To make our presentation most accessible, we use the textbook version given in [10].

Let \( B \) denote the set of all non-empty connected subsets of \( E \). We call \( X, Y \in B \) compatible, \( X \not\sim Y \), if no edge in \( X \) has a common vertex with an edge in \( Y \). Otherwise \( X, Y \) are called incompatible, \( X \sim Y \). In particular, \( X \sim X \). Recall \( \text{supp} I = \{ e \in E : I_e \neq 0 \} \) for \( I \in I \), where \( I \) is defined in (1.19). Let

\[
J = \{ I \in I : \text{supp} I \in B \}. \tag{3.5}
\]

The incompatibility relation \( \sim \) on \( B \) is inherited to an incompatibility relation, also denoted by \( \sim \), on \( J \) via

\[
I \sim I' \iff \text{supp} I \sim \text{supp} I'. \tag{3.6}
\]
Every subset of $E$ can be uniquely decomposed in a set of pairwise compatible connected components, which is a subset of $B$. For $n \in \mathbb{N}$, let
\[ \mathcal{J}_n = \{(I_1, \ldots, I_n) \in \mathcal{J}^n : I_i \not\sim I_j \text{ for all } i \neq j\}. \tag{3.7} \]
Consider $I \in \mathcal{I}$ and the connected components $X_1, \ldots, X_n$ ($n \in \mathbb{N}_0$) of $\text{supp} \, I$. We set $I_j := I_1 X_j \in \mathcal{J}$. Here, it is crucial that Kirchhoff rule (1.14) holds for $I$ if and only if it holds for all $I_j$. Then, using the locality of $H_{\text{disl}}$ given in Assumption 1.1, we obtain
\[ \langle \phi, I \rangle = \sum_{j=1}^{n} \langle \phi, I_j \rangle, \quad H_{\text{disl}}(I) = \sum_{j=1}^{n} H_{\text{disl}}(I_j). \tag{3.8} \]
For $I \in \mathcal{I}$ and some $\beta > 0$, we abbreviate
\[ K(I, \phi) := e^{i \langle \phi, I \rangle} e^{-\beta H_{\text{disl}}(I)}. \tag{3.9} \]
The function $K$ fulfills the following important factorization property: For $I \in \mathcal{I}$ with connected components $I_1, \ldots, I_n$ as above, one has
\[ K(I, \phi) = \prod_{j=1}^{n} K(I_j, \phi). \tag{3.10} \]
This fact relies on the dimension being at least 3. In $d = 2$, the Burgers vector density would not be locally neutral, resulting in a significant complication of the argument (as in [12] compared to [13]). In view of definition (3.2) of $Z_{\beta, \phi}$, equation (3.10) yields
\[ Z_{\beta, \phi} = \sum_{I \in \mathcal{I}} K(I, \phi) = 1 + \sum_{n=1}^{\infty} \frac{1}{n!} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}_n^\sim} \prod_{j=1}^{n} K(I_j, \phi). \tag{3.11} \]
The summand 1 comes from the contribution of $I = 0$, using $H_{\text{disl}}(0) = 0$. Recall that by (1.20) $|K(I, \phi)| = e^{-\beta H_{\text{disl}}(I)} \leq e^{-\beta c \|I\|_1}$ is summable over $I \in \mathcal{I}$, which shows that all expressions in (3.11) are absolutely summable. To control $Z_{\beta, \phi}$, we use a cluster expansion. Next we cite the relevant theorems.

Let $G_n$ denote the set of all connected subgraphs $G_n = ([n], E_n)$ of the complete graph with vertex set $[n] = \{1, \ldots, n\}$. Let $\mathcal{E}_n = \{E_n : G_n = ([n], E_n) \in \mathcal{G}_n\}$ denote the set of all corresponding edge sets. Consider the Ursell functions
\[ U(I_1, \ldots, I_n) = \frac{1}{n!} \sum_{E_n \in \mathcal{E}_n} \prod_{\{i,j\} \in E_n} (-1_{\{I_i \sim I_j\}}). \tag{3.12} \]
Let $\tilde{\mathcal{J}}$ be any finite set endowed with a reflexive and symmetric incompatibility relation $\sim$. We define $\tilde{\mathcal{J}}_n^\sim$ by (3.7) with $\mathcal{J}$ replaced by $\tilde{\mathcal{J}}$.

**Fact 3.1** (Formal cluster expansion, [10, Proposition 5.3]). For every $I \in \tilde{\mathcal{J}}$, let $K(I)$ be a variable. Consider the polynomial in these variables
\[ Z := 1 + \sum_{n=1}^{\infty} \frac{1}{n!} \sum_{(I_1, \ldots, I_n) \in \tilde{\mathcal{J}}_n^\sim} \prod_{j=1}^{n} K(I_j). \tag{3.13} \]
As a formal power series

\[
\log Z = \sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \tilde{J}^n} U(I_1, \ldots, I_n) \prod_{j=1}^{n} K(I_j). \tag{3.14}
\]

Moreover, if the right-hand side in (3.14) is absolutely summable, then equation (3.14) holds also in the classical sense as follows: \( \exp(\text{rhs}(3.14)) = Z \).

A criterion for convergence of the cluster expansion is cited in the following fact:

**Fact 3.2** (Convergence of cluster expansions, [10, Theorem 5.4]). Assume that there are “sizes” \((a(I))_{I \in \tilde{J}} \in \mathbb{R}_{\geq 0}^J\) and “weights” \((K(I))_{I \in \tilde{J}} \in \mathbb{C}^\tilde{J}\) such that for all \(I \in \tilde{J}\), the following bound holds:

\[
\sum_{J \in \tilde{J}} |K(J)| 1_{\{I \sim J\}} e^{a(J)} \leq a(I). \tag{3.15}
\]

Then, we have for all \(J \in \tilde{J}\):

\[
1 + \sum_{n=2}^{\infty} \sum_{(I_1, \ldots, I_{n-1}) \in \tilde{J}^{n-1}} |U(J, I_1, \ldots, I_{n-1})| \prod_{j=1}^{n-1} |K(I_j)| \leq e^{a(J)}. \tag{3.16}
\]

Moreover, in this case, series (3.14) is absolutely convergent.

### 3.3. Partial Partition Sums

We take a sequence \((\tilde{J}_m)_{m \in \mathbb{N}}\) of finite subsets \(\tilde{J}_m \subseteq \tilde{J}\) with \(\tilde{J}_m \uparrow \tilde{J}\) and set \(\tilde{J}_\infty := \tilde{J}\), with \(\tilde{J}\) being defined in (3.5). For \(m \in \mathbb{N} \cup \{\infty\}\) and \(I \in \tilde{I}\), let

\[
z_m(\beta, I) := \sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \tilde{J}_m^n} U(I_1, \ldots, I_n) 1_{\{I_1+\cdots+I_n=I\}} \prod_{j=1}^{n} e^{-\beta H_{\text{disl}}(I_j)} \in \mathbb{R} \tag{3.17}
\]

whenever this double series is absolutely convergent. Note that \(z_m(\beta, I) = z_m(\beta, -I)\) because \(H_{\text{disl}}(I) = H_{\text{disl}}(-I)\) by Assumption 1.1. We abbreviate also \(z(\beta, I) := z_\infty(\beta, I)\). Uniformly in \(m\), the summands in series (3.17) are dominated by the corresponding ones in \(z^+(\beta, I) := z^+_\infty(\beta, I)\), where

\[
z^+_m(\beta, I) := \sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \tilde{J}_m^n} |U(I_1, \ldots, I_n)| 1_{\{I_1+\cdots+I_n=I\}} \prod_{j=1}^{n} e^{-\beta H_{\text{disl}}(I_j)} \in [0, \infty]. \tag{3.18}
\]

By monotone convergence for series,

\[
z^+_m(\beta, I) \uparrow z^+_\infty(\beta, I) \text{ as } m \to \infty. \tag{3.19}
\]

For \(I \in \tilde{I}\), we define its size

\[
\text{size } I := \|I\|_1 + \text{diam supp } I. \tag{3.20}
\]
Here, diam denotes the diameter in the graph distance in the mesoscopic lattice $G = (V, E)$. The size has the following properties. For $I_1, I_2 \in \mathcal{I}$ with $I_1 \sim I_2$, one has

$$\text{size}(I_1 + I_2) \leq \text{size} I_1 + \text{size} I_2. \quad (3.21)$$

Recall that $I$ takes values in the microscopic lattice $\Gamma$. We set

$$\eta := \min\{|\gamma| : \gamma \in \Gamma \setminus \{0\}\} \quad (3.22)$$

and observe for all $I \in \mathcal{I}$:

$$\eta |\text{supp} I| \leq \|I\|_1. \quad (3.23)$$

If in addition $\text{supp} I$ is connected, we have $\text{diam} \text{supp} I \leq |\text{supp} I|$ and hence

$$\|I\|_1 \leq \text{size} I \leq c_3 \|I\|_1 \quad (3.24)$$

with the constant $c_3 := 1 + \eta^{-1}$. Using the constant $c$ from (1.20), let $c_4 = c_4(c, \eta) := c/(2c_3)$. Then, for $I \in \mathcal{I}$, it follows

$$H_{\text{disl}}(I) \geq c\|I\|_1 \geq \frac{c}{c_3} \text{size} I \geq c_4 \text{size} I. \quad (3.25)$$

We choose now a constant $c_5 = c_5(c, \eta)$ with $0 < c_5 < c_4$ and set $c_6 = c_6(c, \eta) := c_4 - c_5 > 0$. Fact 3.2 is applied twice, later with the weight $K(I, \phi)$ introduced in (3.9), but first with the weight

$$K(J) := e^{-\beta c_6 \text{size} J}, \quad J \in \mathcal{J}, \quad (3.26)$$

and the size function $a : \mathcal{J} \to \mathbb{R}_{>0}$,

$$a(J) := \beta c_5 \eta |\text{supp} J|, \quad J \in \mathcal{J}. \quad (3.27)$$

The following lemma serves to verify hypothesis (3.15) of the cluster expansion.

**Lemma 3.3** (Peierls argument). There is $c_7 > 0$ such that for all $\beta$ large enough,

$$\sup_{E \in \Lambda} \sup_{o \in E} \sum_{J \in \mathcal{J}: \ o \in \text{supp} J} e^{-\beta c_6 \text{size} J} \leq e^{-\beta c_7}. \quad (3.28)$$

Furthermore, one has

$$\sup_{m \in \mathbb{N}} \sup_{E \in \Lambda} \sup_{o \in E} \sum_{J \in \mathcal{J}_m: \ o \in \text{supp} J} |K(J)| e^{a(J)} \leq e^{-\beta c_7}, \quad (3.29)$$

and hypothesis (3.15) holds for $\mathcal{J} = \mathcal{J}_m$ for all $m \in \mathbb{N}$ and all $\beta$ large enough.

**Proof.** Claim (3.28) is verified as follows: Take $o \in E \subseteq \Lambda$. We estimate

$$\sum_{J \in \mathcal{J}: \ o \in \text{supp} J} e^{-\beta c_6 \text{size} J} \leq \sum_{J \in \mathcal{J}: \ o \in \text{supp} J} e^{-\beta c_6 \|J\|_1} = \sum_{X \in \mathcal{B}: \ o \in X} \sum_{J \in \mathcal{J}: \ \text{supp} J = X} e^{-\beta c_6 \|J\|_1}. \quad (3.30)$$
Dropping the condition that $J$ should fulfill the Kirchhoff rules, we obtain the following bound for any given $X \in \mathcal{B}$:

$$
\sum_{J \in \mathcal{J} : \text{supp } J = X} e^{-\beta c_6 \|J\|_1} \leq \sum_{J \in (\Gamma \setminus \{0\})^X} e^{-\beta c_6 \|J\|_1} = c_8(\beta)|X| \tag{3.31}
$$

with the abbreviation

$$
c_8(\beta) := \sum_{\iota \in \Gamma \setminus \{0\}} e^{-\beta c_6 |\iota|}. \tag{3.32}
$$

Because $\Gamma$ is a three-dimensional lattice, for any $k \in \mathbb{N}$ there are at most $c_9 k^2$ lattice points within distance $[\eta k, \eta (k + 1)]$ from 0, where $c_9 > 0$ is a constant only depending on $\Gamma$. Thus,

$$
c_8(\beta) \leq \sum_{k=1}^{\infty} c_9 k^2 e^{-\beta c_6 \eta k} \leq e^{-\beta c_{10}} \tag{3.33}
$$

for all large $\beta$ and a positive constant $c_{10} = c_{10}(\eta, c_6, c_9)$. Substituting (3.31) and (3.33) into (3.30), we obtain

$$
\sum_{J \in \mathcal{J} : o \in \text{supp } J} e^{-\beta c_6 \text{size } J} \leq \sum_{X \in \mathcal{B} : o \in X} e^{-\beta c_{10} |X|}. \tag{3.34}
$$

The last sum is estimated with the following Peierls argument: Let $M < \infty$ be the maximal vertex degree in the mesoscopic lattice with edge set $\Lambda$. Let $n \in \mathbb{N}$. For every set $X \in \mathcal{B}$ with $o \in X$ and $|X| = n$, there is a closed path of length $2n$ steps that starts in $o$ and visits every edge in $X$. There are at most $M^{2n}$ choices of closed paths of length $2n$ starting in $o$, and therefore at most $M^{2n}$ choices of $X$. We conclude for all large $\beta$:

$$
\text{lhs}(3.28) \leq \sup_{E \Subset \Lambda} \sup_{o \in E} \sum_{X \in \mathcal{B} : o \in X} e^{-\beta c_{10} |X|} \leq \sum_{n=1}^{\infty} M^{2n} e^{-\beta c_{10} n} = \frac{M^2 e^{-\beta c_{10}}}{1 - M^2 e^{-\beta c_{10}}} \tag{3.35}
$$

with $c_7 > 0$ only depending on $c_{10}$ and $M$. This proves claim (3.28).

Next, we prove claim (3.29). We observe that (3.23) and (3.24) imply $\eta |\text{supp } J| \leq \|J\|_1 \leq \text{size } J$. Using this, $c_4 - c_5 = c_6$, and (3.28), claim (3.29) follows from the estimate

$$
\sum_{J \in \mathcal{J}_m : o \in \text{supp } J} |K(J)| e^{a(J)} = \sum_{J \in \mathcal{J}_m : o \in \text{supp } J} e^{-\beta (c_4 \text{ size } J - c_5 \eta |\text{supp } J|)} \\
\leq \sum_{J \in \mathcal{J}_m : o \in \text{supp } J} e^{-\beta (c_4 - c_5) \text{ size } J} = \sum_{J \in \mathcal{J}_m : o \in \text{supp } J} e^{-\beta c_6 \text{ size } J} \leq e^{-\beta c_7}, \quad (m \in \mathbb{N}). \tag{3.36}
$$

Note that we have dropped the index $m$ in the last two sums.
To verify (3.15) for \( \tilde{J} = J_m \), we define the closure of any edge set \( F \subseteq E \) by
\[
F := \{ f \in E | f \text{ has a common vertex with some } e \in F \}.
\]
(3.37)
Let \( m \in \mathbb{N} \) and \( I \in J_m \). Summing (3.36) over \( o \in \text{supp}I \), we conclude
\[
\sum_{J \in J_m} |K(J)| |I(J,J)\rangle e^{a(J)} \leq \sum_{o \in \text{supp}I} \sum_{J \in J_m; \ o \in \text{supp}J} |K(J)| e^{a(J)}
\]
\[
\leq e^{-\beta \epsilon_7} |\text{supp}I| \leq e^{-\beta \epsilon_7} M |\text{supp}I| \leq \beta c_5 \eta |\text{supp}I| = o(I)
\]
(3.38)
for all large \( \beta \), uniformly in \( I \in J_m \). Here, we have used that \( e^{-\beta \epsilon_7} M \leq \beta c_5 \eta \) for large \( \beta \).

**Lemma 3.4** (Exponential decay of partial partition sums). For all sufficiently large \( \beta > 0 \) and \( m \in \mathbb{N} \cup \{\infty\} \), the following holds with the constants \( c_4 = c/(2c_3) \) and \( c_7 \) as in Lemma 3.3:
\[
\sup_{m \in \mathbb{N} \cup \{\infty\}} \sup_{E \in \Lambda} \sup_{o \in E} \sum_{I \in \mathcal{I}} \sum_{o \in \text{supp}I} e^{\beta c_4 \text{size} I} z_m^+(\beta, I) \leq e^{-\beta \epsilon_7}.
\]
(3.39)
In particular, in this case, \( z_m(\beta, I) \) is well defined for all \( I \in \mathcal{I} \) and fulfills the same bound
\[
\sup_{m \in \mathbb{N} \cup \{\infty\}} \sup_{E \in \Lambda} \sup_{o \in E} \sum_{I \in \mathcal{I}} \sum_{o \in \text{supp}I} e^{\beta c_4 \text{size} I} |z_m(\beta, I)| \leq e^{-\beta \epsilon_7}.
\]
(3.40)

**Proof.** Using (3.19) and monotone convergence for series, it suffices to consider only finite \( m \in \mathbb{N} \) to prove (3.39). Let \( \beta > 0 \), \( I \in \mathcal{I} \setminus \{0\} \), and \( m \in \mathbb{N} \). Inserting (3.25) into definition (3.18) of \( z_m^+ \) yields
\[
z_m^+(\beta, I) \leq \sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}_m^n; \ I_1 + \cdots + I_n = I} |U(I_1, \ldots, I_n)| \prod_{j=1}^{n} e^{-\beta \frac{2}{c_3} \text{size} I_j}.
\]
(3.41)
For \( (I_1, \ldots, I_n) \in \mathcal{J}_m^n \) with \( U(I_1, \ldots, I_n) \neq 0 \) and \( I = I_1 + \cdots + I_n \) as in the above summation, we have
\[
\sum_{j=1}^{n} \text{size} I_j \geq \text{size} I
\]
(3.42)
from (3.21), and hence, taking again the constant \( c_4 = c/(2c_3) \)
\[
\prod_{j=1}^{n} e^{-\beta \frac{2}{c_3} \text{size} I_j} \leq e^{-\beta c_4 \text{size} I} \prod_{j=1}^{n} e^{-\beta c_4 \text{size} I_j}.
\]
(3.43)
We choose a reference edge \( o \in \text{supp}I \). Substituting (3.43) and (3.26) in (3.41) yields
\[
z_m^+(\beta, I) \leq e^{-\beta c_4 \text{size} I} \sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}_m^n; \ I_1 + \cdots + I_n = I} |U(I_1, \ldots, I_n)| \prod_{j=1}^{n} K(I_j).
\]
(3.44)
The inner sum on the right-hand side can be extended to run over all \( n \)-tuples \((I_1, \ldots, I_n)\) in \( \mathcal{J}^n_m \) with \( o \in \text{supp } I_1 \cup \ldots \cup \text{supp } I_n \), since by definition, for any \( I \) which cannot be written as a sum of such \( I_1, \ldots, I_n \), for some \( n \in \mathbb{N} \), \( z^+_m(\beta, I) = 0 \) or \( o \notin \text{supp } I \). It follows
\[
\sum_{I \in \mathcal{I}; \ o \in \text{supp } I} e^{\beta c_4 \text{size } I} z^+_m(\beta, I) \leq \sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}^n_m; \ o \in \text{supp } I_1 \cup \ldots \cup \text{supp } I_n} |U(I_1, \ldots, I_n)| \prod_{j=1}^{n} K(I_j)
=: C_{m,o,\beta}. \tag{3.45}
\]

As we observed above, it suffices to consider only finite \( m \) in claim (3.39). It remains to show that for \( \beta \) large enough it is true that
\[
\sup_{m \in \mathbb{N}} \sup_{E \subset \Lambda} \sup_{o \in E} C_{m,o,\beta} \leq e^{-\beta c_7}. \tag{3.46}
\]

Note that this condition does not involve \( I \). Because \( |U(I_1, \ldots, I_n)| \) is invariant under permutation of its arguments, we can bound \( C_{m,o,\beta} \) by
\[
C_{m,o,\beta} \leq \sum_{n=1}^{\infty} n \sum_{(I_1, \ldots, I_n) \in \mathcal{J}^n_m; \ o \in \text{supp } I_1} |U(I_1, \ldots, I_n)| \prod_{j=1}^{n} K(I_j)
= \sum_{I_1 \in \mathcal{J}^1_m; \ o \in \text{supp } I_1} K(I_1) \left( 1 + \sum_{n=2}^{\infty} n \sum_{(I_2, \ldots, I_n) \in \mathcal{J}^{n-1}_m} |U(I_1, \ldots, I_n)| \prod_{j=2}^{n} K(I_j) \right); \tag{3.47}
\]
for the summand indexed by \( n = 1 \) we have used \( U(I_1) = 1 \).

By Lemma 3.3, we may apply Fact 3.2, yielding
\[
1 + \sum_{n=2}^{\infty} n \sum_{(I_2, \ldots, I_n) \in \mathcal{J}^{n-1}_m} |U(I_1, \ldots, I_n)| \prod_{j=2}^{n} K(I_j) \leq e^{a(I_1)}. \tag{3.48}
\]

Combining (3.47), (3.48), and (3.29) from Lemma 3.3 gives
\[
\sup_{m \in \mathbb{N}} \sup_{E \subset \Lambda} \sup_{o \in E} C_{m,o,\beta} \leq \sup_{m \in \mathbb{N}} \sup_{E \subset \Lambda} \sup_{o \in E} \sum_{I_1 \in \mathcal{J}^1_m; \ o \in \text{supp } I_1} K(I_1) e^{a(I_1)} \leq e^{-\beta c_7}, \tag{3.49}
\]
yielding claim (3.46). Since \( |z_m(\beta, I)| \leq z^+_m(\beta, I) \), claim (3.40) is an immediate consequence of (3.39).

3.4. Gaussian Lower Bound for Fourier Transforms

Next, we apply a cluster expansion with \( K(I, \phi) \) defined in (3.9) to obtain a representation of \( Z_{\beta,\phi} \) and finally a bound for the Fourier transform of the observable.
Lemma 3.5 (Partition sums in the presence of $\phi$). For all $\beta$ large enough, the following identity holds for any $\phi \in \mathbb{R}^E$:

$$0 < Z_{\beta, \phi} = \exp \left( \sum_{I \in \mathcal{I}} z(\beta, I) e^{i \langle \phi, I \rangle} \right) = \exp \left( \sum_{I \in \mathcal{I}} z(\beta, I) \cos \langle \phi, I \rangle \right)$$

$$\leq \exp \left( \sum_{I \in \mathcal{I}} z^+(\beta, I) \right) < \infty. \quad (3.50)$$

Proof. Recall definitions (3.17) and (3.18) of $z_m$ and $z^+_m$. Take any $\phi \in \mathbb{R}^E$. Rearranging a multiple series with positive summands and using (3.39) for $\beta$ large enough, we obtain

$$\sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}_n} |U(I_1, \ldots, I_n)| \prod_{j=1}^{n} e^{-\beta H_{ \text{disl}}(I_j)} = \sum_{I \in \mathcal{I}} z^+(\beta, I) \leq |E| e^{-\beta c_7} < \infty. \quad (3.51)$$

Using this as a dominating series and the fact $|K(I, \phi)| = e^{-\beta H_{ \text{disl}}(I)}$, the following rearrangement of the series is valid for all $m \in \mathbb{N}$:

$$\sum_{n=1}^{\infty} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}_n} U(I_1, \ldots, I_n) \prod_{j=1}^{n} K(I_j, \phi) = \sum_{I \in \mathcal{I}} z_m(\beta, I) e^{i \langle \phi, I \rangle}. \quad (3.52)$$

By (3.25),

$$|K(I, \phi)| = e^{-\beta H_{ \text{disl}}(I)} \leq e^{-\beta c_4 \text{ size } I} = K(I). \quad (3.53)$$

According to Lemma 3.3 and Facts 3.1 and 3.2, one has for all $m \in \mathbb{N}$:

$$\exp(\text{lhs}(3.52)) = 1 + \sum_{n=1}^{\infty} \frac{1}{n!} \sum_{(I_1, \ldots, I_n) \in (\mathcal{J}_m)_n} \prod_{j=1}^{n} K(I_j, \phi). \quad (3.54)$$

From monotone convergence, we know

$$1 + \sum_{n=1}^{\infty} \frac{1}{n!} \sum_{(I_1, \ldots, I_n) \in (\mathcal{J}_m)_n} \prod_{j=1}^{n} |K(I_j, \phi)| \xrightarrow{m \to \infty} 1 + \sum_{n=1}^{\infty} \frac{1}{n!} \sum_{(I_1, \ldots, I_n) \in (\mathcal{J}_m)_n} \prod_{j=1}^{n} |K(I_j, \phi)| < \infty; \quad (3.55)$$

the finiteness follows as in the argument described below (3.11). Consequently, applying dominated convergence in (3.54) and using $\mathcal{J}_m \uparrow \mathcal{J}$ and (3.11) yields

$$\exp(\text{lhs}(3.52)) \xrightarrow{m \to \infty} 1 + \sum_{n=1}^{\infty} \frac{1}{n!} \sum_{(I_1, \ldots, I_n) \in \mathcal{J}_n} \prod_{j=1}^{n} K(I_j, \phi) = Z_{\beta, \phi}. \quad (3.56)$$

On the other hand, from (3.51) and dominated convergence for series,

$$\sum_{I \in \mathcal{I}} z_m(\beta, I) e^{i \langle \phi, I \rangle} \xrightarrow{m \to \infty} \sum_{I \in \mathcal{I}} z(\beta, I) e^{i \langle \phi, I \rangle}. \quad (3.57)$$
Taking the limit $m \to \infty$ in equation (3.52) yields the first equality in claim (3.50).

The second equality of claim (3.50) follows from the following symmetry consideration. One has $I \in \mathcal{I}$ if and only if $-I \in \mathcal{I}$ and $z(\beta, I) = z(\beta, -I)$ by definition, and hence

$$\sum_{I \in \mathcal{I}} z(\beta, I) e^{i \langle \phi, I \rangle} = \frac{1}{2} \left[ \sum_{I \in \mathcal{I}} z(\beta, I) e^{i \langle \phi, I \rangle} + \sum_{I \in \mathcal{I}} z(\beta, -I) e^{i \langle \phi, -I \rangle} \right] = \sum_{I \in \mathcal{I}} z(\beta, I) \cos \langle \phi, I \rangle \, .$$

(3.58)

The last series converges absolutely, and its absolute value is bounded by $\sum_{I \in \mathcal{I}} z^+(\beta, I) < \infty$.

□

**Lemma 3.6** (Gaussian lower bound for Fourier transforms). For all $\beta$ large enough, the following holds for any $\sigma \in \mathbb{R}^E$:

$$\mathbb{E}_{P_\beta} \left[ e^{i \langle \sigma, I \rangle} \right] \geq \exp \left( -\frac{1}{2} \sum_{I \in \mathcal{I}} |z(\beta, I)| \langle \sigma, I \rangle^2 \right) \, .$$

(3.59)

**Proof.** By (3.4), (3.3), and Lemma 3.5, we have

$$\mathbb{E}_{P_\beta} \left[ e^{i \langle \sigma, I \rangle} \right] = \frac{Z_\beta(\sigma)}{Z_\beta(0)} = \frac{\mathbb{E}[Z_{\beta, \sigma+\phi}]}{\mathbb{E}[Z_{\beta, \phi}]} \, ,$$

(3.60)

$$Z_{\beta, \sigma+\phi} = \exp \left( \sum_{I \in \mathcal{I}} z(\beta, I) \cos \langle \sigma + \phi, I \rangle \right) \, .$$

(3.61)

Using

$$\cos \langle \sigma + \phi, I \rangle = \cos \langle \phi, I \rangle (\cos \langle \sigma, I \rangle - 1)+\cos \langle \phi, I \rangle - \sin \langle \phi, I \rangle \sin \langle \sigma, I \rangle$$

(3.62)

and the bound

$$\cos \langle \phi, I \rangle (\cos \langle \sigma, I \rangle - 1) \geq -| \cos \langle \sigma, I \rangle - 1| \geq -\frac{1}{2} \langle \sigma, I \rangle^2 \, ,$$

(3.63)

we obtain

$$\sum_{I \in \mathcal{I}} z(\beta, I) \cos \langle \sigma + \phi, I \rangle$$

$$\geq -\frac{1}{2} \sum_{I \in \mathcal{I}} |z(\beta, I)| \langle \sigma, I \rangle^2 + \sum_{I \in \mathcal{I}} z(\beta, I) \left[ \cos \langle \phi, I \rangle - \sin \langle \phi, I \rangle \sin \langle \sigma, I \rangle \right] \, .$$

(3.64)

We take the average over an auxiliary sign $\Sigma$ taking values $\pm 1$. We substitute $\phi$ by $\Sigma \phi$ in (3.64). Then, using the facts $\cos \langle \Sigma \phi, I \rangle = \cos \langle \phi, I \rangle$ and $\sin \langle \Sigma \phi, I \rangle = \Sigma \sin \langle \phi, I \rangle$, it follows
\[
\frac{1}{2} \sum_{\Sigma \in \{\pm 1\}} Z_{\beta, \sigma + \Sigma \phi} = \frac{1}{2} \sum_{\Sigma \in \{\pm 1\}} \exp \left( \sum_{I \in \mathcal{I}} z(\beta, I) \cos \langle \sigma + \Sigma \phi, I \rangle \right) \\
\geq \exp \left( -\frac{1}{2} \sum_{I \in \mathcal{I}} |z(\beta, I)| \langle \sigma, I \rangle^2 \right) \\
\cdot \frac{1}{2} \sum_{\Sigma \in \{\pm 1\}} \exp \left( \sum_{I \in \mathcal{I}} z(\beta, I) \left[ \cos \langle \phi, I \rangle - \sin \langle \Sigma \phi, I \rangle \sin \langle \sigma, I \rangle \right] \right) \\
= \exp \left( -\frac{1}{2} \sum_{I \in \mathcal{I}} |z(\beta, I)| \langle \sigma, I \rangle^2 \right) Z_{\beta, \phi} \\
\cdot \frac{1}{2} \sum_{\Sigma \in \{\pm 1\}} \exp \left( -\sum_{I \in \mathcal{I}} z(\beta, I) \sin \langle \phi, I \rangle \sin \langle \sigma, I \rangle \right). \\
\] (3.65)

Note that \( \sum_{I \in \mathcal{I}} z(\beta, I) \sin \langle \phi, I \rangle \sin \langle \sigma, I \rangle \) converges absolutely for \( \beta \) large enough by Lemma 3.5. Since \( \frac{1}{2} (e^x + e^{-x}) \geq 1 \) for all \( x \in \mathbb{R} \), we obtain

\[
\frac{1}{2} \sum_{\Sigma \in \{\pm 1\}} Z_{\beta, \sigma + \Sigma \phi} \geq \exp \left( -\frac{1}{2} \sum_{I \in \mathcal{I}} |z(\beta, I)| \langle \sigma, I \rangle^2 \right) Z_{\beta, \phi}. \\
\] (3.66)

Using that \( \phi \) is centered Gaussian and \( Z_{\beta, \phi} \) is bounded and positive, we conclude

\[
\mathbb{E}[Z_{\beta, \sigma + \phi}] = \mathbb{E} \left[ \frac{1}{2} \sum_{\Sigma \in \{\pm 1\}} Z_{\beta, \sigma + \Sigma \phi} \right] \\
\geq \exp \left( -\frac{1}{2} \sum_{I \in \mathcal{I}} |z(\beta, I)| \langle \sigma, I \rangle^2 \right) \mathbb{E}[Z_{\beta, \phi}]. \\
\] (3.67)

In view of (3.60) and \( \mathbb{E}[Z_{\beta, \phi}] > 0 \), this proves the claim. \( \square \)

4. Proof of the Main Result

4.1. Bounding the Observable

For \( I \in \mathcal{I} \) and \( b = b(I) \) as in (1.18), we take the minimizer \( w^* : \mathbb{R}^3 \rightarrow \mathbb{R}^{3 \times 3} \) defined in (2.34) and set \( w^*(x, I) := w^*(x) \). For arbitrary \( x, y \in \mathbb{R}^3 \), we choose \( \sigma(x, y) = (\sigma_{ij}(x, y))_{i,j \in [3]} \in (\mathbb{R}^E)^{[3] \times [3]} \) satisfying the equation

\[
\langle \sigma_{ij}(x, y), I \rangle = w_{ij}^*(x, I) - w_{ij}^*(y, I) \\
\] (4.1)

for all \( I \in \mathcal{I} \). Such a choice is possible because \( w^*(x, I) \) is a linear function of \( I \).
Lemma 4.1 (Bounding the observable). There are a function $W : \Lambda \times \mathbb{R}^3 \times \bigcup_{E \in \Lambda} (I(E) \setminus \{0\}) \to \mathbb{R}_{\geq 0}$ with
\[
c_{11} := \sup_{x \in \mathbb{R}^3} \sup_{E \in \Lambda} \sup_{I \in I(E) \setminus \{0\}} \sum_{o \in \Lambda} W(o, x, I) < \infty
\] (4.2)
and $\beta_1 > 0$ such that for all $E \in \Lambda$, $I \in I(E) \setminus \{0\}$, $o \in \text{supp } I$, $x \in \mathbb{R}^3$, $\beta \geq \beta_1$ and $i, j \in [3]$ we have
\[
w_{ij}^*(x, I)^2 \leq W(o, x, I)e^{\beta c_4 \text{size } I}.
\] (4.3)

Proof. Take $E \in \Lambda$, $I \in I(E) \setminus \{0\}$, $o \in \text{supp } I$, $x \in \mathbb{R}^3$, and $i, j \in [3]$. We choose a vertex $v(o) \in o$. We set
\[
M_1(I, o) := \max_{i, j \in [3]} \sum_{l=1}^3 \int_{\mathbb{R}^3} |u - v(o)||b_{lij}(u)| \, du,
\] (4.4)
\[
R(I, o) := \max\{|x| : x \in \text{supp } \varphi\}
+ \max\{|v' - v(o)| : v' \in e \text{ for some } e \in \text{supp } I\}.
\] (4.5)

Since size $I$ is bounded away from 0, there is a constant $c_{12} > 0$ such that $R(I, o) \leq c_{12}\text{size } I$. By definition (1.18) of $b(I)$, one has the bound $\|b_{lij}(I)\|_1 \leq c_{13}\|I\|_1 \leq c_{13}\text{size } I$ for all its components, with some constant $c_{13} > 0$. Hence, we obtain
\[
M_1(I, o) \leq R(I, o) \max_{i, j \in [3]} \sum_{l=1}^3 \|b_{lij}(I)\|_1 \leq 3c_{12}c_{13}(\text{size } I)^2.
\] (4.6)

Because $b$ is compactly supported and divergence-free in the sense of equation (1.12),
\[
Q(I) := \int_{\mathbb{R}^3} b(I)(u) \, du = 0
\] (4.7)
by the fundamental theorem of calculus. Recall the representation $w^* = w^b + d_0 \psi^*$ with $w^b$, $d_0 \psi^*$ as in (2.37) and (2.38). We now establish bound (4.3) in two steps, first for $x$ far from $o$, then close to $o$. A key estimate is provided by bounds on integral kernels proven in Appendix A.3.

Case 1: First we consider the case $|v(o) - x| \geq 2R(I, o)$. It follows from (2.37) and the second inequality in (A.15) from Lemma A.1 (see Appendix A.3) in the case that $v(o) = 0$ is the origin that
\[
|w_{ij}^b(x)| \leq \sum_{l=1}^3 |\partial_l \Delta^{-1} b_{lij}(I)(x)| \leq \frac{24M_1(I, o)}{\pi} \frac{1}{|v(o) - x|^3} \leq c_{14}(\text{size } I)^2
\] (4.8)
with the constant $c_{14} = 72c_{12}c_{13}/\pi$.

The stability condition (1.5) implies $|\lambda|/|2\mu + \lambda| \leq 1$. In the same way as in (4.8), (2.38) and the second inequality in (A.37) from Lemma A.3 (see again Appendix A.3) in the case that $v(o) = 0$ is the origin give
\[ |d_0 \psi_{ij}^*(x)| \leq \sum_{k,l=1}^{3} \left[ |\partial_i \Delta^{-2} \partial_k \partial b_{ijk}(I)(x)| + |\partial_i \Delta^{-2} \partial_j \partial b_{lkk}(I)(x)| \right] \]
\[ \leq \frac{2 \cdot 9 \cdot 36 M_1(I,o)}{\pi} \frac{1}{|v(o) - x|^3} \leq \frac{c_{15}(\text{size } I)^2}{|v(o) - x|^3} \tag{4.9} \]
with the constant \( c_{15} = 1944c_{12}c_{13}/\pi \). It follows still in the case \( v(o) = 0 \)
\[ |w_{ij}^*(x,I)| \leq |w_{ij}^b(x)| + |d_0 \psi_{ij}^*(x)| \leq \frac{(c_{14} + c_{15})(\text{size } I)^2}{|v(o) - x|^3}. \tag{4.10} \]

The next step involves translation invariance: Shifting both \( x \) and \( I \) by a mesoscopic lattice vector \( v \in V_\Lambda \) does not change \( w_{ij}^*(x,I) \) because \( (x,I) \mapsto b(I)(x) \) has the same translation invariance. Because inequality \( (4.10) \) is written in a translation-invariant form, it holds also if we drop the assumption \( v(o) = 0 \).

This yields
\[ w_{ij}^*(x,I)^2 \leq \frac{(c_{14} + c_{15})^2(\text{size } I)^4}{|v(o) - x|^6} \leq |v(o) - x|^{-6} e^{\beta c_4 \text{ size } I} \tag{4.11} \]
for all \( \beta \geq \beta_1 \) for sufficiently large \( \beta_1 \), neither depending on \( o, x, \) nor \( I \).

Case 2: Next we consider the case \( |v(o) - x| < 2R(I,o) \). We recall the definition of \( J_{jk}(I) \) from \( (1.16) \). We now use the symbol \( \|\cdot\|_1 \) in two different ways. On the one hand, \( \|I\| = \sum_{e \in E} |I_e| \) for \( I \). On the other hand, \( \|J_{jk}(I)\|_1 \) denotes the total unsigned mass of the signed measure \( J_{jk}(I) \) given by the following definition: For any signed measure \( \tilde{J} \) on \( \mathbb{R}^3 \) with Hahn decomposition \( \tilde{J} = \tilde{J}_+ - \tilde{J}_- \), we define \( \|\tilde{J}\|_1 := \tilde{J}_+(\mathbb{R}^3) + \tilde{J}_-(\mathbb{R}^3) \). With this interpretation, we have
\[ \|J_{jk}(I)\|_1 \leq \sup_{e \in \Lambda} \lambda_e(\mathbb{R}^3) \|I\|_1. \tag{4.12} \]
Combining this with \( (A.50) \) and \( (A.51) \) from Lemma A.4 in Appendix A.3.3 yields the bound
\[ |w_{ij}^*(x,I)| \leq c_{16} \|I\|_1 \leq c_{16} \text{ size } I \tag{4.13} \]
for all \( x \in \mathbb{R}^3 \) and all \( I \in \mathcal{T}\setminus\{0\} \) with a constant \( c_{16} > 0 \). Note that
\[ |\{o \in \Lambda : |v(o) - x| < 2R(I,o)\}| \leq c_{17}(\text{size } I)^3 \tag{4.14} \]
with a constant \( c_{17} > 0 \) depending on the lattice spacing in \( \Lambda \). Squaring \( (4.13) \), we obtain
\[ |w_{ij}^*(x,I)|^2 \leq (\text{size } I)^{-3} c_{16}^2 (\text{size } I)^5 \leq (\text{size } I)^{-3} e^{\beta c_4 \text{ size } I} \tag{4.15} \]
again for all \( \beta \geq \beta_1 \) for sufficiently large \( \beta_1 \), neither depending on \( o, x, \) nor \( I \).

Combining the two cases, claim \( (4.3) \) holds for
\[ W(o,x,I) := 1_{|v(o) - x| \geq 2R(I,o)} |v(o) - x|^{-6} + 1_{|v(o) - x| < 2R(I,o)} (\text{size } I)^{-3}. \tag{4.16} \]

To bound \( \sum_{o : |v(o) - x| \geq 2R(I,o)} |v(o) - x|^{-6} \), observe that \( R(I,o) \) is bounded away from 0 uniformly in \( I \) and \( o \), and that \( |v(o) - x|^{-6} \) is summable away from \( x \).
in three dimensions. We use (4.14) to bound \( \sum_{o: |v(o) - x| < 2R(I, o)} (\text{size } I)^{-3} \). We conclude

\[
\sum_{o \in \Lambda} W(o, x, I) \leq c_{11} \tag{4.17}
\]

uniformly in \( x, E \in \Lambda \), and \( I \in \mathcal{I}(E) \), with a constant \( c_{11} \) depending on \( \Lambda \).

\[\square\]

### 4.2. Identifying Long-Range Order

We finally prove now our main result.

**Proof of Theorem 1.3.** Applying Lemma 3.6 yields for \( t \in \mathbb{R}, E \in \Lambda, x, y \in \mathbb{R}^3 \), and \( i, j \in [3] \)

\[
E_{\beta, E} \left[ e^{it(\sigma_{ij}(x,y), I)} \right] \geq \exp \left( -\frac{t^2}{2} \sum_{I \in \mathcal{I}(E)} |z(\beta, I)| \langle \sigma_{ij}(x,y), I \rangle^2 \right). \tag{4.18}
\]

We may drop the summand indexed by \( I = 0 \) because \( \langle \sigma_{ij}(x,y), 0 \rangle = 0 \). Inserting (4.3) and employing Lemma 3.4 in the last line in (4.19), we obtain for sufficiently large \( \beta \) that

\[
\frac{1}{2} \sum_{I \in \mathcal{I}(E) \setminus \{0\}} |z(\beta, I)| \langle \sigma_{ij}(x,y), I \rangle^2
\]

\[
\leq \sum_{I \in \mathcal{I}(E) \setminus \{0\}} |z(\beta, I)| (w_{ij}^*(x, I)^2 + w_{ij}^*(y, I)^2)
\]

\[
\leq \sum_{o \in E} \sum_{I \in \mathcal{I}(E): o \in \text{supp } I} |z(\beta, I)| (w_{ij}^*(x, I)^2 + w_{ij}^*(y, I)^2)
\]

\[
\leq \sup_{I \in \mathcal{I}(E) \setminus \{0\}} \sum_{o \in E} \left( W(o, x, \tilde{I}) + W(o, y, \tilde{I}) \right) \sum_{I \in \mathcal{I}(E): o \in \text{supp } I} e^{\beta c_4 (\text{size } I)} |z(\beta, I)|
\]

\[
\leq 2c_{11} \sup_{E \in \Lambda} \sup_{E \in \mathcal{I}(E): o \in \text{supp } I} \sum_{o \in E} e^{\beta c_4 (\text{size } I)} |z(\beta, I)| \leq 2c_{11} e^{-\beta c_7} \leq e^{-\beta c_2}, \tag{4.19}
\]

with a constant \( c_2 = c_2(c_{11}, c_7) > 0 \), where \( c_{11} \) was defined in (4.2). Mind that \( c_2 \) does not depend on \( x, y, i, j, E, \beta \). This proves the first claim.

By Theorem 3.3.9 in [7], for \( \beta \) large enough, the variance of \( \langle \sigma_{ij}(x,y), I \rangle \) exists and fulfills

\[
\text{var}_{\beta, E}(\langle \sigma_{ij}(x,y), I \rangle) \leq E_{\beta, E}(\langle \sigma_{ij}(x,y), I \rangle^2)
\]

\[
\leq -\lim_{t \downarrow 0} t^{-2} \left( E_{\beta, E}[e^{it\langle \sigma_{ij}(x,y), I \rangle}] - 2 + E_{\beta, E}[e^{-it\langle \sigma_{ij}(x,y), I \rangle}] \right) \leq e^{-\beta c_2}, \tag{4.20}
\]

where the last inequality is a consequence of the lower bound (1.26).

\[\square\]

We remark that the two reflection symmetries \( H_{el}(-w) = H_{el}(w) \) and \( H_{diss}(-I) = H_{diss}(I) \) imply that \( w^*(x, -I) = -w^*(x, I) \) and \( w^*(x, I) \) are equal.
in distribution with respect to $P_{\beta,E}$, jointly in $x \in \mathbb{R}^3$. In particular, the first inequality in (4.20) is actually an equality.

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**Appendix A. Appendix**

**A.1. Elasticity Theory**

**A.1.1. Description of an Elastically Deformed Solid in Continuum Approximation.** From Sect. 1.2.1, we recall that the elastic deformation energy $E_{\text{el}}(f)$ is modeled to be an integral over a smooth elastic energy density $\rho_{\text{el}} : \mathbb{R}^{3 \times 3} \to \mathbb{R}$:

$$E_{\text{el}}(f) = \int_{\mathbb{R}^3} \rho_{\text{el}}(\nabla f(x)) \, dx. \quad (A.1)$$

We assume furthermore:

- $\rho_{\text{el}}$ takes its minimum value 0 at the identity matrix $\text{Id}$. This means that the non-deformed solid has minimal energy.
- The elastic energy density is insensitive to rotations of the solid:

$$\rho_{\text{el}}(RM) = \rho_{\text{el}}(M), \quad (R \in \text{SO}(3), M \in \text{GL}_+ (3)), \quad \text{(rot inv)}$$

where $\text{GL}_+ (3) = \{ M \in \mathbb{R}^{3 \times 3} : \det M > 0 \}$. In other words, deforming the solid and then rotating it costs the same elastic energy as only deforming it with the same deformation.

Note that reflections $R \in \text{O}(3) \backslash \text{SO}(3)$ or singular or orientation reversing linearized deformations $M \in \mathbb{R}^{3 \times 3}$ with $\det M \leq 0$ do not make sense physically in this context.

For $M, N \in \text{GL}_+ (3)$, it is equivalent that $M^t M = N^t N$ and that there exists $R \in \text{SO}(3)$ such that $N = RM$. As a consequence, $\rho_{\text{el}}(M)$ is a function of $M^t M$. We set $\tilde{\rho}_{\text{el}}(M^t M) := \rho_{\text{el}}(M)$.

Note that assumption (rot inv) of rotational invariance does not imply isotropy of the solid, which is defined by

$$\rho_{\text{el}}(MR) = \rho_{\text{el}}(M), \quad (R \in \text{SO}(3), M \in \text{GL}_+ (3)). \quad \text{(isotropy)}$$
Thus, anisotropy means that first rotating the solid and then deforming it with a given deformation might cost a different elastic energy than deforming it with the same deformation without rotating it first. Although the isotropy assumption is an oversimplification for any real monocrystal, we assume it to keep the presentation simple.

A.1.2. Linearization. We now consider only small perturbations $f = \text{id} + \varepsilon u : \mathbb{R}^3 \to \mathbb{R}^3$ of the identity map as deformation maps. For $\nabla f = \text{Id} + \varepsilon \nabla u$, we obtain

\[
(\nabla f)^t \nabla f = \text{Id} + \varepsilon (\nabla u + (\nabla u)^t) + \varepsilon^2 (\nabla u)^t \nabla u. \tag{A.2}
\]

Substituting this into (1.4) gives

\[
\tilde{\rho}_\text{el}((\nabla f)^t \nabla f) = \varepsilon^2 F(\nabla u + (\nabla u)^t) + O(\varepsilon^3), \quad (\varepsilon \to 0). \tag{A.3}
\]

If $\mathbb{R} \ni \varepsilon \mapsto R_\varepsilon \in \text{SO}(3)$ is a path of rotations with $R_0 = \text{Id}$, then $\frac{d}{d\varepsilon} R_\varepsilon|_{\varepsilon=0}$ is antisymmetric. Hence, the Taylor-approximated energy density $\varepsilon^2 F(\nabla u + (\nabla u)^t)$ is not influenced by linearized rotations.

If assumption (isotropy) holds, then

\[
\tilde{\rho}_\text{el}(A) = \tilde{\rho}_\text{el}(R^t A R) \tag{A.4}
\]

holds for all positive definite matrices $A = A^t$ and $R \in \text{SO}(3)$. Taylor-expanded this means

\[
F(U) = F(R^t U R) \tag{A.5}
\]

for all symmetric matrices $U = U^t$ and $R \in \text{SO}(3)$. Thus, $F(U)$ depends only on the list $a, b, c$ of eigenvalues of $U$ (with multiplicities). For diagonal matrices $U = \text{diag}(a, b, c)$, the only quadratic forms which are symmetric in $a, b, c$ are linear combinations of $(\text{Tr} U)^2 = (a + b + c)^2$ and $|U|^2 = a^2 + b^2 + c^2$. Thus, under an isotropy assumption, we have

\[
F(U) = \frac{\lambda}{2} (\text{Tr} U)^2 + \mu |U|^2 = (a, b, c) \left( \mu \text{Id} + \frac{\lambda}{2} e e^t \right) \begin{pmatrix} a \\ b \\ c \end{pmatrix}, \tag{A.6}
\]

with real constants $\lambda$ and $\mu$ and $e^t = (1, 1, 1)$. The matrix $\mu \text{Id} + \frac{\lambda}{2} e e^t$ has the double eigenvalue $\mu$ with eigenspace $e^\perp$ and a single eigenvalue $\mu + 3\lambda/2$ with eigenspace $\mathbb{R} e$. Hence, the quadratic form $F$ is positive definite if and only if $\mu$ and $\lambda$ satisfy the conditions given in (1.5). Summarizing, we have the model for the linearized elastic deformation energy given in (1.6)–(1.7).

A.2. From Kirchhoff’s Node Rule to Continuum Sourceless Currents

We show that Kirchhoff’s node rule for the discrete current $I$ implies absence of sources for its smoothed variant $\tilde{b}(I)$. For $e \in E$ from $x \in V$ to $y \in V$ in its
counting direction, we write \( x = v_-(e) \) and \( y = v_+(e) \). We rewrite (1.17) as

\[
\tilde{b}_{jk}(I)(x) = \int_{\mathbb{R}^3} \varphi(x - y) J_{jk}(I)(dy)
\]

\[
= \sum_{e \in E} (n_e)_{j} (I_e)_{k} |v_+(e) - v_-(e)| \int_{0}^{1} \varphi(x - v_-(e) - t(v_+(e) - v_-(e))) \, dt
\]

\[
= \sum_{e \in E} (I_e)_{k} (v_+(e) - v_-(e))_{j} \int_{0}^{1} \varphi(x - v_-(e) - t(v_+(e) - v_-(e))) \, dt,
\]

\((j, k \in \{3\}, I \in (\mathbb{R}^3)^E \text{ with (1.14), } x \in \mathbb{R}^3).\)  

(A.7)

As a consequence of Kirchhoff’s rule (1.14), \( \tilde{b}(I) \) is indeed divergence-free:

\[
(\text{div} \tilde{b}(I))_{k}(x)
\]

\[
= \sum_{j=1}^{3} \sum_{e \in E} (I_e)_{k} (v_+(e) - v_-(e))_{j} \int_{0}^{1} \partial_j \varphi(x - v_-(e) - t(v_+(e) - v_-(e))) \, dt
\]

\[
= - \sum_{e \in E} (I_e)_{k} \int_{0}^{1} \frac{d}{dt} \varphi(x - v_-(e) - t(v_+(e) - v_-(e))) \, dt
\]

\[
= - \sum_{e \in E} (I_e)_{k} \left[ \varphi(x - v_+(e)) - \varphi(x - v_-(e)) \right]
\]

\[
= - \sum_{e \in E} \sum_{v \in V} s_{ve} (I_e)_{k} \varphi(x - v) = 0.
\]  

(A.8)

A.3. Integral Kernels

A.3.1. Bounds for the Dipole Expansion. In this Appendix, we derive a simplified version of the dipole expansion for the Coulomb potential, which we need as ingredient to identify long-distance bounds for the observable \( u^* \) in the proof of Lemma 4.1.

Let \( \rho \in C_0^\infty(\mathbb{R}^3, \mathbb{R}) \). We define its total charge

\[
Q := \int_{\mathbb{R}^3} \rho(x) \, dx.
\]  

(A.9)

We take a radius \( R > 0 \) such that

\[
R \geq \sup\{|x| : x \in \mathbb{R}^3, \rho(x) \neq 0\}
\]  

(A.10)

and the first unsigned moment

\[
M_{1} := \int_{\mathbb{R}^3} |y| |\rho(y)| \, dy.
\]  

(A.11)

The inverse Laplacian \(-\Delta^{-1}\) is described by convolution with the Coulomb potential

\[
G(y) = \frac{1}{4\pi |y|}.
\]  

(A.12)
Lemma A.1 (Simplified dipole expansion). For all \( x \in \mathbb{R}^3 \) with \( |x| \geq 2R \) and \( i \in [3] \), we know that

\[
-\Delta^{-1} \rho(x) = \frac{1}{4\pi} \frac{Q}{|x|} + r_1(x), \tag{A.13}
\]

\[
-\partial_i \Delta^{-1} \rho(x) = -\frac{Q}{4\pi |x|^3} x_i + \partial_i r_1(x) \tag{A.14}
\]

with the error bounds

\[
|r_1(x)| \leq \frac{M_1}{\pi} \frac{1}{|x|^2}, \quad |\partial \partial_1 (x)| \leq \frac{8M_1}{\pi} \frac{1}{|x|^3}. \tag{A.15}
\]

Proof. Let \( x \in \mathbb{R}^3 \) with \( |x| \geq 2R \) and \( y \in \text{supp} \rho \). Then for all \( t \in [0,1] \), we have

\[
|x - ty| = |x| - R \geq \frac{|x|}{2} \geq R. \tag{A.16}
\]

We apply

\[
f_{x,y}(1) = f_{x,y}(0) + \int_0^1 f'_{x,y}(t) \, dt \tag{A.17}
\]

to

\[
f_{x,y}(t) := \frac{1}{|x - ty|}. \tag{A.18}
\]

In this proof, \( \partial_i = \partial/\partial x_i \) always refers to the variable \( x_i \). Using \( \partial_i |x|^\alpha = \alpha x_i |x|^\alpha - 2 \), we calculate:

\[
f'_{x,y}(t) = \frac{\langle x - ty, y \rangle}{|x - ty|^3}, \quad \partial_i f_{x,y}(t) = -\frac{x_i - ty_i}{|x - ty|^3}, \tag{A.19}
\]

\[
\partial_i f'_{x,y}(t) = \frac{y_i}{|x - ty|^3} - 3 \frac{\langle x - ty, y \rangle (x_i - ty_i)}{|x - ty|^5}. \tag{A.20}
\]

Consequently, we can bound the integrand in (A.17) and its \( \partial_i \) derivative as follows:

\[
|f'_{x,y}(t)| = \frac{|\langle x - ty, y \rangle|}{|x - ty|^3} \leq \frac{|y|}{|x - ty|^2} \leq 4\frac{|y|}{|x|^2}, \tag{A.21}
\]

\[
|\partial_i f'_{x,y}(t)| \leq 4\frac{|y|}{|x - ty|^3} \leq \frac{32|y|}{|x|^3}. \tag{A.22}
\]

We obtain

\[
-\Delta^{-1} \rho(x) = \frac{1}{4\pi} \int_{\mathbb{R}^3} \frac{1}{|x - y|} \rho(y) \, dy \]

\[
= \frac{1}{4\pi} \int_{\mathbb{R}^3} \left( f_{x,y}(0) + \int_0^1 f'_{x,y}(t) \, dt \right) \rho(y) \, dy, \tag{A.23}
\]

and hence claim (A.13) holds with the error term

\[
r_1(x) = \frac{1}{4\pi} \int_{\mathbb{R}^3} \int_0^1 f'_{x,y}(t) \, dt \rho(y) \, dy. \tag{A.24}
\]
It is bounded as follows:

$$|r_1(x)| \leq \frac{1}{4\pi} \int_{\mathbb{R}^3} \int_0^1 \frac{4|y|}{|x|^2} \, dt \, |\rho(y)| \, dy = \frac{M_1}{\pi} \frac{1}{|x|^2}. \quad (A.25)$$

This shows the first claimed bound in (A.15).

Substituting (A.22) in the $\partial_i$ derivative of equation (A.23), i.e., in

$$- \partial_i \Delta^{-1} \rho(x) = \frac{1}{4\pi} \int_{\mathbb{R}^3} \left( \partial_i f_{x,y}(0) + \int_0^1 \partial_i f'_{x,y}(t) \, dt \right) \rho(y) \, dy, \quad (A.26)$$

the second claimed bound in (A.15) follows, too. \hfill \Box

A.3.2. Properties of $\partial_i \partial_j \Delta^{-2}$. Next, we describe $\partial_i \partial_j \Delta^{-2}$ explicitly by an integral kernel. We derive it from its Fourier transform, cf. (2.1).

**Lemma A.2** (Integral kernel of $\partial_i \partial_j \Delta^{-2}$). For any Schwartz functions $f, g : \mathbb{R}^3 \to \mathbb{C}$, we have

$$- \int_{\mathbb{R}^3} \hat{g}(k) \frac{k_i k_j}{|k|^4} \hat{f}(k) \, dk$$

$$= \frac{1}{8\pi} \int_{\mathbb{R}^3} \int_{\mathbb{R}^3} \frac{g(x)}{|x-y|} \frac{1}{|x-y|^2} \left( \frac{(x_i - y_i)(x_j - y_j)}{|x-y|^2} - \delta_{ij} \right) f(y) \, dx \, dy. \quad (A.27)$$

**Proof.** Using

$$\frac{1}{|k|^4} = \int_0^\infty e^{-\frac{1}{2} z |k|^2} \frac{z}{4} \, dz \quad (A.28)$$

for $k \in \mathbb{R}^3 \setminus \{0\}$, we rewrite the left hand side in claim (A.27) as follows:

$$\text{lhs}(A.27) = - \int_{\mathbb{R}^3} \int_0^\infty \hat{g}(k) k_i k_j e^{-\frac{1}{2} z |k|^2} \frac{z}{4} \hat{f}(k) \, dz \, dk$$

$$= - \int_0^\infty \frac{z}{4} \int_{\mathbb{R}^3} \hat{g}(k) k_i k_j e^{-\frac{1}{2} z |k|^2} \hat{f}(k) \, dk \, dz \quad \text{(by Fubini)}$$

$$= \frac{1}{4(2\pi)\frac{3}{2}} \int_0^\infty z^{-\frac{1}{2}} \int_{\mathbb{R}^3} \hat{g}(k) \left[ \partial_i \partial_j \left( e^{-\frac{|k|^2}{2z}} \ast f \right) \right] \hat{f}(k) \, dz \, dk$$

$$= \frac{1}{4(2\pi)\frac{3}{2}} \int_0^\infty z^{-\frac{1}{2}} \int_{\mathbb{R}^3} \hat{g}(x) \partial_i \partial_j \left( e^{-\frac{|k|^2}{2z}} \ast f \right) (x) \, dx \, dz \quad (A.29)$$

(by Plancherel). Here, the application of Fubini’s theorem is justified because of

$$\int_{\mathbb{R}^3} \int_0^\infty \left| \hat{g}(k) k_i k_j e^{-\frac{1}{2} z |k|^2} \frac{z}{4} \hat{f}(k) \right| \, dz \, dk = \int_{\mathbb{R}^3} \left| \hat{g}(k) \frac{k_i k_j}{|k|^4} \hat{f}(k) \right| \, dk \leq \int_{\mathbb{R}^3} \left| \frac{1}{|k|^2} \hat{f}(k) \right| \, dk < \infty, \quad (A.30)$$

since $1/|k|^2$ is integrable near 0 in 3 dimensions. We transform $t = \frac{1}{2}, z^{-\frac{1}{2}} \, dz = -t^{-\frac{3}{2}} \, dt$ and use

$$\partial_i \partial_j e^{-\frac{t|x|^2}{2}} = (t^2 x_i x_j - t \delta_{ij}) e^{-\frac{t|x|^2}{2}} \quad (A.31)$$
to obtain
\[
\text{lhs}(A.27) = \frac{1}{4(2\pi)^{\frac{3}{2}}} \int_0^\infty \int_{\mathbb{R}^3} g(x) \partial_i \partial_j \left( e^{-\frac{t|x|^2}{2}} * f \right)(x) \, dx \, \frac{dt}{t^{\frac{3}{2}}}
\]
\[
= \frac{1}{4(2\pi)^{\frac{3}{2}}} \int_0^\infty \int_{\mathbb{R}^3} g(x) (t^2(x_i - y_i)(x_j - y_j) - t\delta_{ij})e^{-\frac{t|x-y|^2}{2}} f(y) \, dy \, dx \, \frac{dt}{t^{\frac{3}{2}}}
\]
\[
= \frac{1}{4(2\pi)^{\frac{3}{2}}} \int_{\mathbb{R}^3} g(x) \int_0^\infty (t^2(x_i - y_i)(x_j - y_j) - t\delta_{ij})e^{-\frac{t|x-y|^2}{2}} \, dt \, f(y) \, dx \, dy.
\]  
(A.32)

The application of Fubini’s theorem in the last step is justified below. Now, substituting \( s = \frac{t}{2}|x|^2 \), \( t = \frac{2s}{|x|^2} \), \( \sqrt{t} \, dt = 2\sqrt{2} \sqrt{s} \, ds \), \( t^{-\frac{1}{2}} \, dt = \sqrt{\frac{2}{|x|^2}} s^{-\frac{1}{2}} \, ds \) and using \( \Gamma(\frac{1}{2}) = \sqrt{\pi} \), \( \Gamma(\frac{3}{2}) = \frac{\sqrt{\pi}}{2} \), we calculate
\[
\frac{1}{4(2\pi)^{\frac{3}{2}}} \int_0^\infty (t^2 x_i x_j - t\delta_{ij})e^{-\frac{t|x|^2}{2}} \, dt
\]
\[
= \frac{2\sqrt{2}}{4(2\pi)^{\frac{3}{2}}} \int_0^\infty \sqrt{s} e^{-s} \, ds \frac{x_i x_j}{|x|^3} - \frac{\sqrt{2}}{4(2\pi)^{\frac{3}{2}}} \int_0^\infty \frac{1}{\sqrt{s}} e^{-s} \, ds \frac{\delta_{ij}}{|x|}
\]
\[
= \frac{\Gamma(\frac{1}{2})}{\sqrt{2}(2\pi)^{\frac{3}{2}}} \frac{x_i x_j}{|x|^3} - \frac{\Gamma(\frac{1}{2})}{2\sqrt{2}(2\pi)^{\frac{3}{2}}} \frac{\delta_{ij}}{|x|}
\]
\[
= \frac{1}{8\pi} \left( \frac{x_i x_j}{|x|^3} - \frac{\delta_{ij}}{|x|} \right)
\]  
(A.33)

and similarly
\[
\frac{1}{4(2\pi)^{\frac{3}{2}}} \int_0^\infty |t^2 x_i x_j - t\delta_{ij}| e^{-\frac{t|x|^2}{2}} \, dt \leq \frac{1}{8\pi} \left( \frac{|x_i x_j|}{|x|^3} + \frac{\delta_{ij}}{|x|} \right) \leq \frac{1}{4\pi|x|},
\]  
(A.34)

which is integrable near 0. Together with the fast decay of \( f \) and \( g \), this bound justifies the application of Fubini’s theorem in the last step of (A.32). Substitution of calculation (A.33) in (A.32) proves claim (A.27). □

Next, we derive a variant of the simplified dipole expansion presented in Sect. A.3.1 (using the notation from there), but now for the operator \( \partial_i \partial_j \Delta^{-2} \) rather than \( \Delta^{-1} \). This is used in the proof of Lemma 4.1.

**Lemma A.3** (Long-range asymptotics of \( \partial_i \partial_j \Delta^{-2} \)). For all \( x \in \mathbb{R}^3 \) with \( |x| \geq 2R \) and \( i, j, k \in [3] \), we have
\[
\Delta^{-2} \partial_i \partial_j \rho(x) = \frac{Q}{8\pi} \left( \frac{x_i x_j}{|x|^3} - \frac{\delta_{ij}}{|x|} \right) + r'_1(x),
\]  
(A.35)
\[
\partial_k \Delta^{-2} \partial_i \partial_j \rho(x) = \frac{Q}{8\pi} \left( \frac{\delta_{ik} x_j + \delta_{jk} x_i + \delta_{ij} x_k}{|x|^3} - \frac{3 x_i x_j x_k}{|x|^5} \right) + \partial_k r'_1(x)
\]  
(A.36)

with the error bounds
\[
|r'_1(x)| \leq \frac{3M_1}{\pi} \frac{1}{|x|^2}, \quad |\partial_k r'_1(x)| \leq \frac{36M_1}{\pi} \frac{1}{|x|^3}.
\]  
(A.37)
Proof. Recall that for $x \in \mathbb{R}^3$ with $|x| \geq 2R$, $y \in \text{supp } \rho$, and all $t \in [0,1]$, we have $|x-ty| \geq |x|/2 \geq R$. We apply (A.17) to
\[
 f_{x,y}(t) := \frac{(x_i - ty_i)(x_j - ty_j)}{|x-ty|^3} - \frac{\delta_{ij}}{|x-ty|}, \tag{A.38}
\]
\[
 f_{x,y}(0) = \frac{x_i x_j}{|x|^3} - \frac{\delta_{ij}}{|x|}, \tag{A.39}
\]
\[
 \partial_k f_{x,y}(0) = \frac{\delta_{ik} x_j + \delta_{jk} x_i + \delta_{ij} x_k}{|x|^3} - 3 \frac{x_i x_j x_k}{|x|^5}. \tag{A.40}
\]

We calculate:
\[
 f'_{x,y}(t) = -\frac{y_i (x_j - ty_j)}{|x-ty|^3} - \frac{(x_i - ty_i) y_j}{|x-ty|^3} + 3 \frac{\langle y, x-ty \rangle (x_i - ty_i)(x_j - ty_j)}{|x-ty|^5}, \tag{A.41}
\]
\[
 \partial_k f'_{x,y}(t) = -\frac{y_i \delta_{jk}}{|x-ty|^3} + 3 \frac{y_i (x_j - ty_j)(x_k - ty_k)}{|x-ty|^5} - \frac{\delta_{ik} y_j}{|x-ty|^3}
 + 3 \frac{(x_i - ty_i) y_j (x_k - ty_k)}{|x-ty|^5} + 3 \frac{y_k (x_i - ty_i)(x_j - ty_j)}{|x-ty|^5}
 + 3 \frac{\langle y, x-ty \rangle \delta_{ik} (x_j - ty_j)}{|x-ty|^5} + 3 \frac{\langle y, x-ty \rangle (x_i - ty_i) \delta_{jk}}{|x-ty|^5}
 - 15 \frac{\langle y, x-ty \rangle (x_i - ty_i)(x_j - ty_j)(x_k - ty_k)}{|x-ty|^7}
 - \delta_{ij} \left( \frac{y_k}{|x-ty|^3} - 3 \frac{(x_k - ty_k) \langle x-ty, y \rangle}{|x-ty|^5} \right). \tag{A.42}
\]

Using the Cauchy–Schwarz inequality, it follows
\[
 \left| f'_{x,y}(t) \right| \leq \frac{6|y|}{|x-ty|^2} \leq \frac{24|y|}{|x|^2}, \quad \left| \partial_k f'_{x,y}(t) \right| \leq \frac{36|y|}{|x-ty|^3} \leq \frac{288|y|}{|x|^3}. \tag{A.43}
\]

We obtain
\[
 \Delta^{-2} \partial_i \partial_j \rho(x) = \frac{1}{8\pi} \int_{\mathbb{R}^3} \frac{1}{|x-y|} \left( \frac{(x_i - y_i)(x_j - y_j)}{|x-y|^2} - \delta_{ij} \right) \rho(y) \, dy
 = \frac{1}{8\pi} \int_{\mathbb{R}^3} \left( f_{x,y}(0) + \int_0^1 f'_{x,y}(t) \, dt \right) \rho(y) \, dy \tag{A.44}
\]
and hence claim (A.35) with an error term $r'_1$ bounded by
\[
 |r'_1(x)| \leq \frac{1}{8\pi} \int_{\mathbb{R}^3} \int_0^1 \frac{24|y|}{|x|^2} \, dt \, |\rho(y)| \, dy = \frac{3M_1}{\pi} \frac{1}{|x|^2}. \tag{A.45}
\]

This shows the first claimed bound in (A.37).

For the partial derivatives $\partial_k$ with respect to $x_k$, we obtain:
\[
 \partial_k \Delta^{-2} \partial_i \partial_j \rho(x) = \frac{1}{8\pi} \int_{\mathbb{R}^3} \left( \partial_k f_{x,y}(0) + \int_0^1 \partial_k f'_{x,y}(t) \, dt \right) \rho(y) \, dy. \tag{A.46}
\]
Consequently,
\[
|\partial_k r'_1(x)| \leq \frac{1}{8\pi} \int_{\mathbb{R}^3} \int_0^1 \frac{288|y|}{|x|^3} \, dt \, |\rho(y)| \, dy = \frac{36M_1}{\pi} \frac{1}{|x|^3}.
\]  
(A.47)

This proves the second bound in (A.37).

A.3.3. Integral Kernels Close to the Diagonal. Recall that \(\|\tilde{J}\|_1\) denotes the total unsigned mass of a signed measure \(\tilde{J}\).

**Lemma A.4** (Uniform bounds). For the form function \(\varphi\) introduced in Sect. 1.2.3, one has

\[
c_{18} := \max_{l \in [3]} \|\partial_l \Delta^{-1} \varphi\|_\infty < \infty,
\]
(A.48)

\[
c_{19} := \max_{l, i, j \in [3]} \|\partial_l \partial_i \partial_j \Delta^{-2} \varphi\|_\infty < \infty.
\]
(A.49)

Additionally, for any signed measure \(J\) with \(\|J\|_1 < \infty\) and for all \(l, i, j \in [3]\), one has

\[
\|\partial_l \Delta^{-1} \varphi \ast J\|_\infty \leq c_{18} \|J\|_1,
\]
(A.50)

\[
\|\partial_l \partial_i \partial_j \Delta^{-2} \varphi \ast J\|_\infty \leq c_{19} \|J\|_1.
\]
(A.51)

**Proof.** The operators \(\partial_l \Delta^{-1}\) and \(\partial_l \partial_i \partial_j \Delta^{-2}\) have the integral kernels \(k_1(x-y)\) and \(k_2(x-y)\), respectively, where

\[
k_1(z) := \frac{1}{4\pi} \frac{z_l}{|z|^3} \quad \text{and}
\]

\[
k_2(z) := \frac{1}{8\pi} \left( \frac{\delta_{il} z_j + \delta_{jl} z_i + \delta_{ij} z_l}{|z|^3} - 3 \frac{z_i z_j z_l}{|z|^5} \right),
\]
(A.52)

cf. (A.27), (A.38), and (A.40). Since both kernels are bounded by \(O(|z|^{-2})\), they are locally integrable and decay at infinity. Because \(\varphi\) is compactly supported and bounded, claims (A.48) and (A.49) follow. The remaining claims follow from \(\|k \ast J\|_\infty \leq \|k\|_\infty \|J\|_1\) with \(k = k_1\) and \(k = k_2\), respectively. 

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