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The Effects of Casimir Interactions in Experiments on Gravitationally-induced Entanglement

Bachelor Thesis

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Abstract

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1 Introduction

1.1 Feynman's Gedankenexperiment

2 A first look

Testing the quantum nature of gravity is no easy task and many proposals seek to detect gravitationally induced entanglement between two masses [1–4] as a form of proof. For all these proposals, gravity is assumed to be mediated by a gravitational field. During a time evolution, this field (like any other external field) can only perform local operations (LO) on the states of the test masses. If gravity is now assumed to behave classically, the propagation between the masses can be described by a classical communication (CC) channel [4, 5]. These LOCC operations however cannot turn an initially unentangled state into an entangled one [6, 7]. It immediately follows, that if one measures the involved masses to be entangled after a mutual gravitational interaction, gravity necessarily has to be quantum in some way. It is important to note, that the opposite of this statement is not true. Measuring unentangled masses does not directly imply the classicality of the gravitational field. This can be seen by considering operations that are non-LOCC and also produce unentangled states like for example the swap operation $|\psi\rangle_A |\phi\rangle_B \rightarrow |\phi\rangle_A |\psi\rangle_B$. This operations obviously can't induce entanglement to initially unentangled states, but requires the perfect exchange of quantum information between the states - which is not possible using classical communication alone. In other words: If one prepares masses initially in a pure product state and measures *any* state which cannot be obtained by LOCC-operations after some final time evolution, it is impossible for gravity to behave classical. One can even go so far and define the term ***quantum gravity*** as any interaction mediated by gravity that cannot be described by LOCC operations alone [5].

A plausible and logical idea for an experiment to test for gravitational induced entanglement is described in this chapter - which is, as a reminder, enough to prove a quantum nature of gravity. It requires the generation of coherent delocalized quantum superpositions of massive objects either as so-called Schrödinger-cat states or squeezed gaussian states [4, 8]. Theses masses are brought close enough together for gravity to have a measurable effect. The distances between different parts of the spatial superpositions must have different distances to the delocalized second mass. As a result - and of course *if gravity behaves quantum* - the states should get entangled. To see this, consider the ideal simplification of a real experimental setup where two bodies with mass m are trapped in an harmonic potential wall (like for example in an optical trap) with frequency ω separated by a distance d . The local Hamiltonian of the system is given by

$$\hat{H}_0 = \sum_{i=1,2} \frac{\hat{p}_i^2}{2m} + \frac{1}{2}m\omega^2 \hat{x}_i^2 \quad (2.1)$$

where \hat{x} and \hat{p} are the position and momentum operators satisfying the canonical com-

mutation relation $[\hat{x}_i, \hat{p}_j] = i\hbar\delta_{ij}$. For now, all non-gravitational interactions between the masses have been ignored. In the low energy regime, where the energy transfer during a process is far below the Planck scale $m_p c^2 \sim 10^{19}$ GeV, gravity can be traded as an effective field theory with tools available similar to those for the electromagnetic field and QED [9]. In the non-relativistic limit $v \ll c$, the gravitational interaction can be described by a Newtonian $1/r$ potential acting on the center-of-mass positions, with all classical quantities are replaced by quantum operators [8–10]. Spatial superpositions lead to superpositions of the metric and consequently (in the non-relativistic limit) to a superposed Newtonian potential. The interaction Hamiltonian \hat{H}_G should therefore be describable by

$$\hat{H}_G = -\frac{Gm^2}{|d - \hat{x}_1 + \hat{x}_2|}, \quad (2.2)$$

where $G = 6.6743 \times 10^{-11} \text{ m}^3\text{kg}^{-1}\text{s}^{-2}$ is the gravitational constant. The separation of the masses d is chosen much larger than the extension of the delocalization (in this setup comparable to the position variance of the harmonic oscillator). This condition is realistic given that the biggest spatial delocalization is in the order of $\sqrt{\hbar/m\omega}$. Expanding the Hamiltonian \hat{H}_G for small \hat{x}_i , only the second order term proportional to $(\hat{x}_1 - \hat{x}_2)^2$ can induce entanglement [1]. The zeroth order term is just a overall energy offset, the first order term $\propto (\hat{x}_1 - \hat{x}_2)$ as well as the terms \hat{x}_i^2 result only in a local interaction for each mass separately. The coupling term $-(\hat{x}_1\hat{x}_2 + \hat{x}_2\hat{x}_1) = -2\hat{x}_1\hat{x}_2$ however is very interesting as it couples both oscillators and can thus mediate entanglement. Introducing the ladder operators, the Hamiltonian $\hat{H} = \hat{H}_0 + \hat{H}_G$ can be expressed as [9]:

$$\hat{H} = \sum_{i=1,2} \hbar\omega\hat{a}_i^\dagger\hat{a}_i - \frac{Gm^2}{d^3} \left(\sqrt{\frac{\hbar}{2m\omega}} \right)^2 (\hat{a}_1\hat{a}_2 + \hat{a}_1\hat{a}_2^\dagger + \hat{a}_1^\dagger\hat{a}_2 + \hat{a}_1^\dagger\hat{a}_2^\dagger) \quad (2.3)$$

Applying the *rotating-wave approximation*¹, the terms $\hat{a}_1\hat{a}_2 + \hat{a}_1^\dagger\hat{a}_2^\dagger$ can be dropped. Defining the coupling strength g of the interaction as $g = Gm/\omega d^3$, eq (2.3) can be rewritten as

$$\hat{H} = \sum_{i=1,2} \hbar\omega\hat{a}_i^\dagger\hat{a}_i - \hbar g (\hat{a}_1\hat{a}_2^\dagger + \hat{a}_1^\dagger\hat{a}_2). \quad (2.4)$$

Now, for simplicity and as a simple example, the evolution of the initial Fock state $|\psi(0)\rangle = |10\rangle$ is considered. The gravitational interaction H_G can be treated as a time dependent perturbation and the state evolution is given as (for calculation see appendix A.1.1) [9]

$$|\psi(t=0)\rangle = |10\rangle \xrightarrow{\text{time } t} |\psi(t)\rangle = \mathcal{N} (|10\rangle - i g t |01\rangle + \mathcal{O}(g^2)) \quad (2.5)$$

where \mathcal{N} is an appropriate normalization constant. The evolved state (2.5) is entangled and cannot be reduced into a product of two oscillator Fock states. The entanglement

¹This approximation is known from quantum optics, where all fast oscillating terms in the Hamiltonian can be dropped [5, 9]. In the interaction picture, the ladder operators evolve as $\hat{a}(t) = \hat{a}e^{-i\omega t}$. The terms like $\hat{a}_1(t)\hat{a}_2(t)$ oscillate with frequency 2ω whereas $\hat{a}_1^\dagger(t)\hat{a}_2(t)$ does not oscillate at all. Due to the small coupling, this approximation works very well here.

is very small since it is proportional to the gravitational coupling constant gt ² but can still be measured by ????. !!!SOURCES!!!. Another interesting result, which underlines the false inference of a classical gravity from observed non-entanglement discussed above can be seen by considering the time evolution of a coherent product state $|\alpha\rangle \otimes |\beta\rangle$ where $\hat{a}|\alpha\rangle = \alpha|\alpha\rangle$. The time evolution is derived in appendix A.1.2 and results in

$$e^{-i\hat{H}t/\hbar}(|\alpha\rangle \otimes |\beta\rangle) = |e^{-i\omega t}(\alpha \cos gt - \beta \sin gt)\rangle \otimes |e^{-i\omega t}(-\alpha \sin gt + \beta \cos gt)\rangle. \quad (2.6)$$

This state is clearly a product state and thus not entangled. But for a time $t_0 = \pi/2g$ the state is effectively the swapped initial state $|\beta\rangle \otimes |\alpha\rangle$ up to a local phase. This swap operation is however, as established earlier, not possible under a LOCC protocol. Thus, even if the resulting state after time evolution under a gravitational interaction is unentangled, we can rule out the classicality of gravity [5, 9]. Gravity must therefore be capable of transmitting quantum information and must be described by a quantum channel.

Experimentally, one requires the ability to generate spatial superpositions of two massive objects with large enough coherence times. Usually the weak gravitational interaction requires coherence times in the order of 100 ms – 10 s for any meaningful and measurable entanglement to build up. The masses should additionally be massive enough for their gravitational effects to be measurable. These requirements impose huge experimental and engineering challenges. To contextualize: The most massive object ever put into a spatial superposition is in the order of 4×10^{-23} kg, whereas the smallest object whose gravitational field has been measured was just below 100 mg [11] - a difference of 19 orders of magnitude. One way to experimentally create such spatial superpositions is giving the masses a spin-1/2 degree of freedom. For example, a nitrogen-vacancy diamonds can be used [4], where the NV site provides the required spin of 1/2. An applied magnetic gradient $\partial_x B$ functions like a “beam splitter” and creates a delocalized state. The extent of this superposition can be calculated and separations in the order of 100 μ m are theoretically achievable [4]. Levitated, trapped particles isolated and shielded in a vacuum can increase environmental isolation by avoiding contact with surrounding noise. The additional forces due to the trapping potential or the gravitational acceleration can be studied in advance. In this thesis, I assume that all required states and superpositions can be prepared experimentally.

The general and idealized problem considered is illustrated in fig. 2.1. Two massive bodies with masses M_A and M_B are initially separated by a center-to-center distance $2L$. The masses are prepared in a coherent delocalized quantum superposition Schrödinger-cat-like state in, for now, a parallel orientation as depicted in fig. 2.1. The extension of the superposition is denoted by Δx and is the same for both masses. It is important to choose the positions of the masses such that the distances between each part of the delocalized mass A and B are not always identical. Otherwise, all built up phases are the same and no entanglement is observable. With the notation introduced in fig. 2.1,

²The amount of entanglement can for example be measured with the later introduced *logarithmic negativity* E_N . For this state, this quantity is given as $E_N(|\psi(t)\rangle\langle\psi(t)|) \simeq 2tg/\log 2 + \mathcal{O}(g^2) \geq 0$.

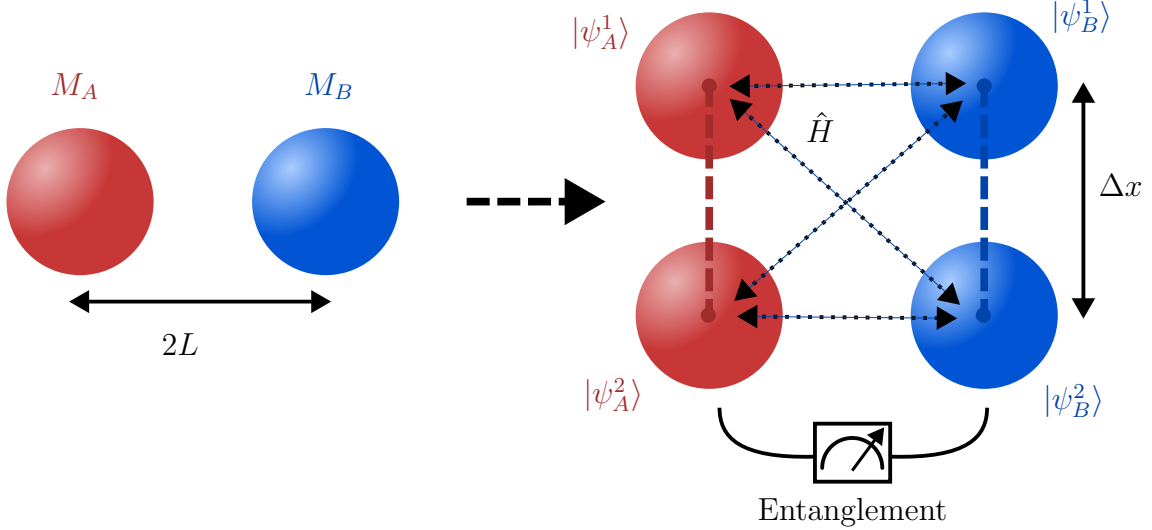


Figure 2.1: Schematic figure of the proposed experiment with two masses prepared in a spatial superposition state. The gravitational interaction \hat{H} induces different phases to each of the superpositions due to the different distances between all masses. This results in measurable entanglement after some time evolution.

the initial state at $t = 0$ is given by

$$|\psi(t=0)\rangle = \frac{1}{2} \left(|\psi_A^1\rangle + |\psi_A^2\rangle \right) \otimes \left(|\psi_B^1\rangle + |\psi_B^2\rangle \right). \quad (2.7)$$

The state evolves under a Hamiltonian \hat{H} and after some time the position of each mass is measured and checked for entanglement. For now I assume that all interactions except gravity can be neglected. In reality, electromagnetic forces and Casimir-Polder interactions [12, 13] need to be considered.

As established earlier in this chapter, with some assumptions made, gravitational interaction can generate entanglement. In the time scales of the experiment, the acceleration of the masses due to the mutual gravitational interaction can be neglected³. The Hamiltonian therefore only needs to include the gravitational potential

$$\hat{V} = -\frac{GM_A M_B}{|\hat{D}|} \quad (2.8)$$

where \hat{D} is the distance operator between the masses. It depends on the individual positions \hat{x}_A and \hat{x}_B . During time evolution, the different parts of the superpositions built up different local phases. I am interested in calculating, how much entanglement one can expect from this kind of interactions.

³Take for example a silica sphere ($\rho = 2648 \text{ kg/m}^3$) with $R = 10^{-5} \text{ m}$ separated by $2L = 4R$. The mutual gravitational acceleration for each sphere is around $a = GM/(2L)^2 = 5 \times 10^{-13} \text{ m/s}^2$ which results for $t \sim 1 \text{ s}$ in a distance traveled of $\sim 10^{-13} \text{ m}$.

2.1 Time evolution under a gravitational potential

Proposition 2.1. *The time evolution under a static and constant Hamiltonian $\hat{H} = \hat{V}(\hat{x}_i) = \text{const.}$ is given by the eigenenergies of the system $\hat{V} |n\rangle = V_n |n\rangle$ proportional to $e^{-iV_n t/\hbar}$.*

Proof. This is a trivial statement. The time evolution is governed by the Schrödinger equation

$$i\hbar \frac{\partial}{\partial t} |\psi(t)\rangle = \hat{H} |\psi(t)\rangle. \quad (2.9)$$

The formal solution of this first order PDE is given by

$$|\psi(t)\rangle = e^{-i\hat{V}t/\hbar} |\psi(t=0)\rangle. \quad (2.10)$$

The constant (hermitian) potential operator can be expressed in the energy-eigenbasis $\{|n\rangle\}$ as $\hat{V} |n\rangle = V_n |n\rangle$. The initial state can be expressed as a superposition in the same eigenstates like $|\psi\rangle = \sum_n c_n |\psi_n\rangle$. Putting both together and using the Taylor expansion of the exponential function, one arrives at the simple form

$$|\psi(t)\rangle = \sum_n e^{-i\hat{V}t/\hbar} |n\rangle \langle n|\psi\rangle = \sum_{n,k} \frac{(-i\hat{V}t/\hbar)^k}{k!} |n\rangle c_n |\psi_n\rangle \quad (2.11)$$

$$= \sum_{n,k} \frac{(-iV_n t/\hbar)^k}{k!} c_n |\psi_n\rangle = \sum_n e^{-iV_n t/\hbar} c_n |\psi_n\rangle \quad (2.12)$$

where in the second to last step $\hat{V}^k |n\rangle = \hat{V}^{k-1} \hat{V} |n\rangle = \hat{V}^{k-1} |n\rangle V_n = \dots = V_n^k |n\rangle$ was used. \square

Using the preceding proposition, the initial state eq. (2.7) can be evolved in time. The potential operator eq. (2.8) acts on every state in the $\{|\psi_A^1\rangle, |\psi_A^2\rangle\} \otimes \{|\psi_B^1\rangle, |\psi_B^2\rangle\}$ basis differently. This is because of the different distances between the states $|\psi_A^i\rangle$ and $|\psi_B^j\rangle$ for different $i, j \in \{1, 2\}$. This results in phases ϕ_{ij} to be built up during time evolution according to proposition 2.1. The state $|\psi(t)\rangle$ after some time evolution is therefore given as

$$|\psi(t)\rangle = \frac{1}{2} \left(e^{i\phi_{11}} |\psi_A^1\rangle |\psi_B^1\rangle + e^{i\phi_{12}} |\psi_A^1\rangle |\psi_B^2\rangle + e^{i\phi_{21}} |\psi_A^2\rangle |\psi_B^1\rangle + e^{i\phi_{22}} |\psi_A^2\rangle |\psi_B^2\rangle \right), \quad (2.13)$$

where the \otimes symbol was omitted. The phases are

$$\phi \equiv \phi_{11} = \phi_{22} = \frac{GM_A M_B}{2\hbar L} t \quad \text{and} \quad \phi_{12} = \phi_{21} = \frac{GM_A M_B}{\hbar \sqrt{4L^2 + (\Delta x)^2}} t. \quad (2.14)$$

Assuming again that the superposition size Δx is much smaller than the distance L between the masses - like before in eq. (2.2) - the phases $\phi_{12} = \phi_{21}$ can be expanded and a global phase ϕ can be factored:

$$\phi_{12} = \phi_{21} \approx \frac{GM_A M_B}{\hbar} \left[\frac{1}{2L} - \frac{(\Delta x)^2}{16L^3} \right] t \equiv \phi - \Delta\phi. \quad (2.15)$$

The state eq. (2.13) can now be expressed in the form

$$|\psi(t)\rangle = e^{i\phi} \frac{1}{\sqrt{2}} \left[|\psi_A^1\rangle \otimes \frac{|\psi_B^1\rangle + e^{-i\Delta\phi} |\psi_B^2\rangle}{\sqrt{2}} + |\psi_A^2\rangle \otimes \frac{e^{-i\Delta\phi} |\psi_B^1\rangle + |\psi_B^2\rangle}{\sqrt{2}} \right], \quad (2.16)$$

where the entanglement dynamics can be directly seen. This state is entangled, if it is not representable as a product state $|\psi\rangle \neq |\psi_A\rangle \otimes |\psi_B\rangle$. That is the case, if the states containing $|\psi_B^i\rangle$ are not both equal to each other (i.e. differ only by a phase) and thus cannot be factored. The system is therefore entangled, if and only if $\Delta\phi \neq k\pi$ with integer $k \in \mathbb{Z}$.

In order to assess in a more quantitative way how entangled the state $|\psi\rangle$ is, a more sophisticated entanglement measure is needed. In the next chapter, the **logarithmic negativity** is motivated and introduced. In the rest of this thesis, I will repeatedly opt for this measure.

2.2 Entanglement measures

Checking whether an arbitrary state ρ is entangled or not is no easy task. In fact, this problem is known to be NP-hard [14]. A state $\rho_{AB} \in \mathcal{H}_A \otimes \mathcal{H}_B$ is called entangled, if it is **non-separable**, that is, it cannot be expressed as a tensor product of two subsystems $\rho_A \in \mathcal{H}_A$ and $\rho_B \in \mathcal{H}_B$. Only for specific cases - like the case of two qubits or qubit-qutrit - a simple sufficient criterion for determining the separability of a general mixed state is known: The positive partial transpose (PPT) criterion states, that if the partial transpose of the density matrix is positive ($\rho^{\Gamma_A} > 0$ ⁴), the state ρ is separable [6, 7]. In other words, if ρ^{Γ_A} has negative eigenvalues, ρ is guaranteed to describe an entangled state. The inverse is true, if and only if the dimension of $\rho_A \otimes \rho_B$ is 2×2 or 3×2 [6] - otherwise, only having non-negative eigenvalues doesn't necessarily result in an unentangled system (such states are called "bound states"). The partial transpose with respect to a subsystem i can be understood in the same way as the partial trace, where the operation (in this case the transform) is performed only on indices corresponding the subsystem ρ_i . To see the necessity of the PPT criterion, consider a separable mixed state ρ , which can be generally expressed as

$$\rho = \sum p_i \rho_A^i \otimes \rho_B^i. \quad (2.17)$$

The partial transpose is in this case trivial:

$$\rho^{\Gamma_A} = \sum p_i (\rho_A^i)^T \otimes \rho_B^i. \quad (2.18)$$

Since the transpose preserves eigenvalues, the transposed subsystem A is still positive $(\rho_A^i)^T > 0$ and describes again a valid quantum state. It follows, that ρ^{Γ_A} is positive as well. If somehow ρ^{Γ_A} has any negative eigenvalues, this can only mean that the initial

⁴A matrix is defined as positive ("positive definite"), if all eigenvalues are positive.

state ρ is not separable and cannot be expressed in the form of eq. (2.17) and the necessity of the criterion is shown.

For quantifying entanglement in a more precise way, a mathematical quantity called **entanglement measure** can be used. A good measure should be able to capture the essential features of entanglement. One can axiomatically state what properties such a measure $E(\rho)$ should have [6, 7]:

Normalization An entanglement measure should be a mapping from densities to real positive values between 0 and 1:

$$\rho \rightarrow E(\rho) \in \mathbb{R}^+ \quad (2.19)$$

where usually the maximally entangled state has $E = 1$.

Monotonicity under LOCC E should not increase under local operations and classical communications. This is the most important postulate for an entanglement measure and often cited as the *only* required postulate.

Vanishing on separable states $E(\rho) = 0$ if ρ is separable

Often one finds additional properties useful like *convexity* $E(\sum p_i \rho_i) \leq \sum p_i E(\rho_i)$ or (full) *additivity* $E(\rho \otimes \sigma) = E(\rho) + E(\sigma)$.

A function that satisfies the most important of these conditions is often called an *entanglement monotone*.

The **negativity** \mathcal{N} is such an entanglement monotone [7, 15] that used the PPT criterion to determine if a state is entangled or not. It is defined as

$$\mathcal{N} = \frac{\|\rho^{\Gamma_A}\|_1 - 1}{2} \quad (2.20)$$

where $\|A\|_1 = \text{tr} |A| = \text{tr} \sqrt{A^\dagger A}$ is the trace norm. The negativity however is not additive and a more suitable and widely used entanglement measure is the **logarithmic negativity** [6, 7, 16]

$$E_N(\rho) = \log_2 \|\rho^{\Gamma_A}\|_1. \quad (2.21)$$

The monotonicity of the logarithm implies, that E_N is an entanglement monotone as well. Furthermore, for the calculations it does not matter which subsystem is transposed.

Proposition 2.2. a) The partial transpose w.r.t. subsystem A is equal to the transposed partial transpose w.r.t. subsystem B : $\rho^{\Gamma_A} = (\rho^{\Gamma_B})^T$. b) The trace norms of partially transposed density operators w.r.t. any subsystem are equal: $\|\rho^{\Gamma_A}\|_1 = \|\rho^{\Gamma_B}\|_1$.

Proof. a) A general density matrix ρ can be expressed as

$$\rho = \sum_{i,j,k,l} \rho_{ij,kl} |i\rangle\langle j|_A \otimes |k\rangle\langle l|_B$$

The partial transpose with respect to subsystem B is then defined as

$$\rho^{\Gamma_B} \equiv \sum_{i,j,k,l} \rho_{ij,kl} |i\rangle\langle j|_A \otimes (|k\rangle\langle l|_B)^T = \sum_{i,j,k,l} c_{ij,kl} |i\rangle\langle j|_A \otimes |l\rangle\langle k|_B$$

The complete transpose of this is

$$(\rho^{\Gamma_B})^T = \sum_{i,j,k,l} \rho_{ij,kl} (|i\rangle\langle j|_A)^T \otimes (|l\rangle\langle k|_B)^T = \sum_{i,j,k,l} c_{ij,kl} |j\rangle\langle i|_A \otimes |k\rangle\langle l|_B \equiv \rho^{\Gamma_A}$$

b) Clear by a) and by using lemma 2.1 and the fact that the eigenvalues of a square matrix A and A^T are equal. \square

The logarithmic negativity is very easy to calculate compared to other entanglement measures. It is enough to compute the square root of the eigenvalues of $(\rho^\Gamma)^\dagger \rho^\Gamma$ or the absolute sum of the eigenvalues of ρ^Γ . For practical and numeric calculations it is often more easy and stable to take a single eigenvalue than the need to compute the sum of multiple. For all numerical calculations in this thesis, I therefore opt for an alternative way to compute the logarithmic negativity.

Lemma 2.1. *The trace norm $\|A\|_1 \equiv \text{tr} \sqrt{A^\dagger A}$ of a hermitian matrix A is equal to the sum of the absolute eigenvalues of A .*

Proof. This can be immediately seen by the spectral theorem:

$$\text{tr} \sqrt{A^\dagger A} = \text{tr} \sqrt{A^2} = \text{tr} \left\{ U \sqrt{\text{diag}(\lambda_1, \dots)^2} U^\dagger \right\} = \sum_i \sqrt{\lambda_i^2} = \sum_i |\lambda_i|.$$

\square

Proposition 2.3. *The negativity eq. (2.20) is given as the absolute sum of all negative eigenvalues of ρ^Γ :*

$$\mathcal{N}(\rho) \equiv \frac{\|\rho^\Gamma\|_1 - 1}{2} = \left| \sum_{\lambda_i < 0} \lambda_i \right|. \quad (2.22)$$

Proof. The proof is in parts given by Vidal and Werner [15]. It is known that the density matrix is hermitian: $\rho = \rho^\dagger$. Using lemma 2.1, the trace norm of the density matrix is given as $\|\rho\|_1 = \sum \lambda_i = \text{tr} \rho = 1$. The partial transpose ρ^Γ obviously also satisfies $\text{tr} \rho^\Gamma = 1$ but might have negative eigenvalues. Since ρ^Γ is still hermitian, the trace norm is given by

$$\|\rho^\Gamma\|_1 = \sum_i |\lambda_i| = \sum_{\lambda_i \geq 0} \lambda_i + \sum_{\lambda_i < 0} |\lambda_i| = \sum_i \lambda_i + 2 \sum_{\lambda_i < 0} |\lambda_i| = 1 + 2 \sum_{\lambda_i < 0} |\lambda_i|,$$

where in the last step $\sum \lambda_i = \text{tr} \rho^\Gamma = 1$ was used. The negativity can be defined as $\mathcal{N}(\rho) = \left| \sum_{\lambda_i < 0} \lambda_i \right|$ and the statement is shown. \square

Remark. The PPT criterion states, that if ρ^Γ has negative eigenvalues, the state ρ is entangled. The negativity uses this criterion for a quantification of entanglement. This proposition makes sense of the name *negativity*.

Calculating the logarithmic negativity of the evolved state eq. (2.13), it is possible to quantify how the entanglement behaves in time. A straight forward computation following the calculation methods established above yields (for detailed calculations see appendix A.2)

$$E_N(|\psi(t)\rangle\langle\psi(t)|) = \log_2(1 + |\sin \Delta\phi|). \quad (2.23)$$

It is interesting to see, that the maximum entanglement $E_N = 1$ is reached for $\Delta\phi = 2\pi k \pm \pi/2$, $k \in \mathbb{Z}$ and no entanglement ($E_N = 0$) is measurable for $\Delta\phi = k\pi$. This result aligns with the previous observations by demanding that the evolved state eq. (2.13) is separable. The complete entanglement dynamics are shown in fig. 2.2. The time

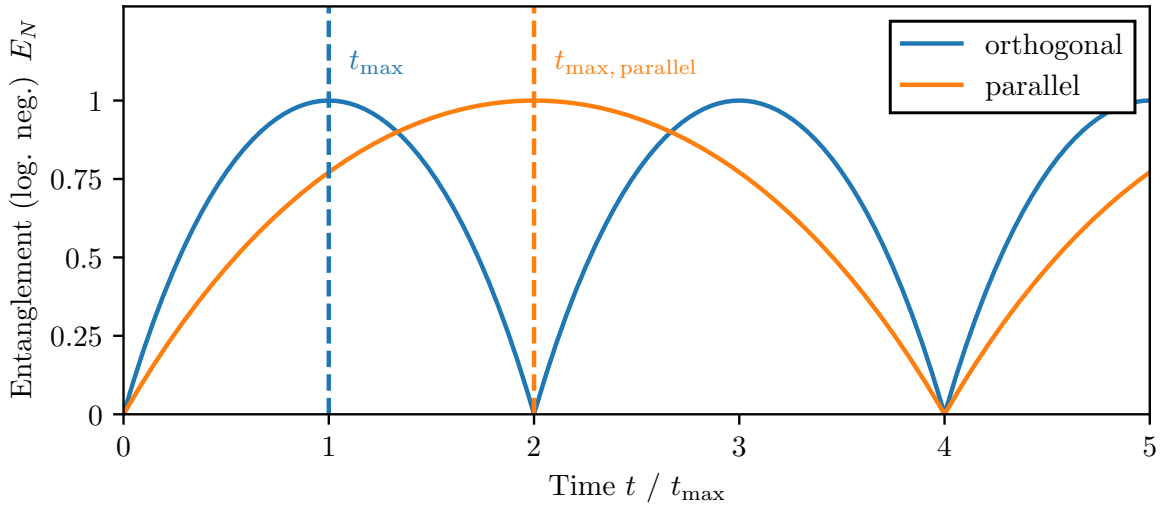


Figure 2.2: Entanglement dynamics quantified by the logarithmic negativity for two different orientations of the spatial superpositions. The parallel orientation was considered in this chapter (see eq. (2.23)), the “orthogonal” one in Ref. [8]. The time of maximum entanglement t_{\max} for the orthogonal configuration is reached after $t_{\max} = 4\pi\hbar L^3/(GM_A M_B \Delta x^2) \simeq 129$ ms.

$t_{\max, \text{parallel}}$ at which the entanglement is maximal (for the first time) can be calculated by using the definition of $\Delta\phi$ from eq. (2.15) as

$$t_{\max, \text{parallel}} = \frac{8\pi L^3 \hbar}{GM_A M_B (\Delta x)^2}. \quad (2.24)$$

In fig. 2.2 the entanglement dynamics for a different orientation considered in Ref. [8] is also shown. There, the superpositions are aligned in the same line as the direct connection between the masses (“orthogonal” to the parallel configuration before), maximizing the differences in distances between them and thus creating entanglement faster. This

expected behavior can be well seen in fig. 2.2: The time t_{\max} until the maximum entanglement is reached, is precisely by a factor of 2 faster than in the here considered parallel configuration [8]. For a practical experiment, this suggests that using the orthogonal orientation could be beneficial and would require shorter coherence times for the superpositions. To give an estimate, consider two identical silica spheres with a density of $\rho = 2648 \text{ kg/m}^3$ with a radius of $R = 10^{-5} \text{ m}$, a separation of $2L = 4R$ and a superposition size $\Delta x = 100 \text{ nm}$ (which is realistic considering theoretical sizes of up to micrometers [4]), the maximum entanglement is reached after about $t_{\max} \approx 129 \text{ ms}$ which is a quite long coherence time and challenging experimentally.

2.3 Issues with the experimental procedure

Additional effects: - Casimir forces entangle as well - Coulomb forces

- solution: conducting faraday shield, UV discharge [17]

How to measure? Many measurements are necessary - Entanglement witness? Measure density directly - Small variations for each measurement in angle of the superposition and in distance to the newly introduced shield

3 Casimir effect

Casimir forces can be viewed in a very similar way to the *van der Waals forces*. In fact, both phenomena describe just two different sides of the same coin. They define the so-called ***dispersion forces*** between neutral atoms or bodies. The quantum theory of van der Waals forces between two neutral atoms was developed by London in 1930 who found the attractive potential $\propto 1/r^6$ for small separations [18]. Casimir and Polder showed in 1948, that for separations larger than the resonance wavelength of the atoms, retardation effects need to be taken into account and the potential decays by a power law of $1/r^7$ [13]. Additionally, they calculated the interaction with a atom or molecule and a perfectly conducting plate, showing that macroscopic objects could experience these ***Casimir-Polder interactions*** as well. It becomes evident, that a full description of dispersion forces cannot be given by classical electrodynamics alone. Additional considerations regarding relativistic effects and quantum electrodynamics have to be made [19–21]. Casimir, following a suggestion by Bohr [22], found a simple derivation using the zero-point energy of the vacuum to calculate the attraction between two conducting plates. In quantum electrodynamics each point in the electromagnetic field can be described by an quantized harmonic oscillator with ground state energy $E_0 = \hbar\omega/2$. The total *zero-point energy* of the ground state of the field (the vacuum) is therefore given by summing over the energies E_0 for each possible mode n

$$E_{\text{vacuum}} = \frac{\hbar}{2} \sum_n \omega_n. \quad (3.1)$$

These sums are clearly divergent since there are infinitely many possible excitations. Electrostatic boundary conditions require the field to be zero at the surface of conductors restricting the possible modes between two parallel plates. Precisely the finite difference between the infinite vacuum energy with and without the macroscopic plates give rise to the ***Casimir forces***. Often, this divergence is simply dripped, motivated by the fact that energy is normally defined only up to a constant [19]. Casimir was able to use regularization techniques to deal with the infinite quantities and arrived at his famous formula [12]

$$E_{\text{Casimir}} = -\frac{\hbar c \pi^2}{720 L^3} A. \quad (3.2)$$

for the attractive Casimir-potential between two plates with area A and separation L . The attractive force $F = -\nabla E$ between the plates can now be simply expressed as

$$F_{\text{Casimir}} = -\frac{\hbar c \pi^2}{240 L^4} A. \quad (3.3)$$

3 Casimir effect

It is remarkable, that such a simple relation arises out of the infinities of the vacuum. Up until now, these Casimir forces are topic of modern scientific research. They are generally very difficult to calculate for other geometries than two infinitely large plates or for real materials with dielectric properties. For simple geometries, even the sign of the force is not always intuitively clear: As an example, the Casimir force can be repulsive for an ideal metal spherical shell [20]. For other simple and important bodies like the sphere-plane or sphere-sphere geometry, no universally valid formula for any separation between the bodies exists. This is discussed in more detail in section 3.2.

Almost ten years after the discovery of Casimir and Polder, Lifshitz was the first to find an expression for the Casimir force between two dielectric plates with arbitrary relative permittivity $\varepsilon_{r,1,2}$ for separations larger than the resonant wavelength⁵ [24]. The expression he found facilitates the general complexity of the Casimir interactions and is only expressible as a complicated integral [24]

$$F/A = -\frac{\hbar c}{32\pi^2 L^4} \int_0^\infty dx \int_1^\infty dp \frac{x^3}{p^2} \left\{ \left[\frac{(s_1 + p)(s_2 + p)}{(s_1 - p)(s_2 - p)} e^x - 1 \right]^{-1} + \left[\frac{(s_1 + \varepsilon_{r,1}p)(s_2 + \varepsilon_{r,2}p)}{(s_1 - \varepsilon_{r,1}p)(s_2 - \varepsilon_{r,2}p)} e^x - 1 \right]^{-1} \right\} \quad (3.4)$$

with

$$s_{1,2} = \sqrt{\varepsilon_{r,1,2} - 1 + p^2}.$$

In the limit of two perfectly conducting plates ($\varepsilon_{r,1} = \varepsilon_{r,2} \rightarrow \infty$), the integral can be solved analytically and one gets the same expression already obtained by Casimir

$$F_{\text{cond.}}/A = -\frac{\hbar c}{16\pi^2 L^4} \int_0^\infty dx \int_1^\infty dp \frac{x^3}{p^2(e^x - 1)} = -\frac{\hbar c \pi^2}{240 L^4}. \quad (3.5)$$

Lifshitz determined the Casimir force between a conducting metal plate and a dielectric plate (denoted DM) as well as the force between two dielectric plates with the same dielectric constant ε_r (DD) as

$$F_{\text{DM}} = -\frac{\hbar c \pi^2}{240 L^4} \frac{\varepsilon_r - 1}{\varepsilon_r + 1} \varphi(\varepsilon_r) \quad (3.6)$$

$$F_{\text{DD}} = -\frac{\hbar c \pi^2}{240 L^4} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 1} \right)^2 \varphi(\varepsilon_r) \quad (3.7)$$

where $\varphi(\varepsilon_r)$ is a numerical function obtained by solving eq. (3.4), which approaches 1 for a perfect conductor. I calculated the function numerically and the result is shown in fig. 3.1. For a dielectric and metal plate, the function φ approaches the finite value $\varphi(\varepsilon_r \rightarrow 1) \approx 0.46$ for small dielectric constants. However, this limit is practically already reached at $\varepsilon_r \approx 4$ and φ stays approximately constant for smaller ε_r .

⁵The “resonance wavelength” for a macroscopic body in this case can be understood as e.g. the plasma frequency in the Drude model [23]. Different models for light-matter interaction result in slightly different resonant wavelength. The Lifshitz formula however holds true for the cases of separations in the micro-meter regime for all practical materials [17].

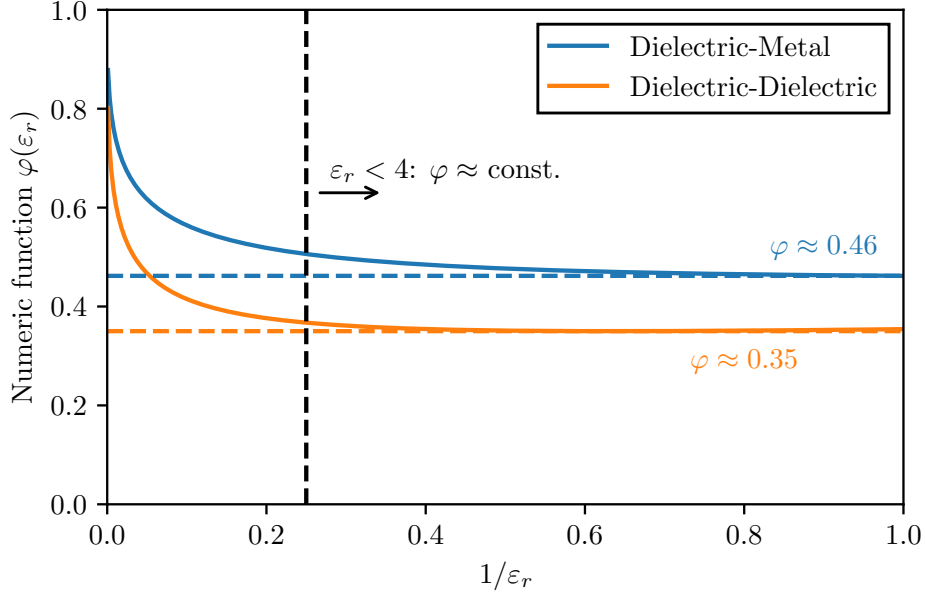


Figure 3.1: Numeric calculations of the function $\varphi(\varepsilon)$ used in the Lifshitz formula eq. (3.6) and (3.7). The function was calculated for **(blue)** a dielectric and a metal plates and **(orange)** two dielectric plates. The function approaches unity for $\varepsilon_r \rightarrow \infty$ and a finite value for $\varepsilon_r \rightarrow 1$.

3.1 Proximity force approximation

The Casimir-Polder force cannot be computed easily for different shapes. There even exists no analytic expression for the simple (and for this thesis relevant) plate-sphere geometry for all ratios L/R and plate-sphere separations. For a general shape, even the sign of the force, i.e. whether it is attractive or repulsive, is often unknown. Fortunately, approximation methods exist and in particular the **proximity-force-approximation (PFA)** can be calculated very easily [25–27]. The PFA is only valid for small separations ($L/R \approx 1$) between the considered smooth bodies. The idea of this approximation is to divide the surfaces of the two bodies into infinitesimal small parallel plates with area dA and summing over the forces dF (or the Casimir-energy dE) between them (see fig. 3.2):

$$E_{\text{PFA}} = \iint_A dA \frac{E_{\text{plate-plate}}}{A} \quad (3.8)$$

where for the casimir energy per unit area $E_{\text{plate-plate}}/A$ either eq. (3.2) or any of the Lifshitz equations (3.7), (3.6) can be chosen. For the following calculations, it is important to distinguish between the distance between the plates center and the spheres center L (like used before) and the edge-to-edge distance $\mathcal{L} = L - R$.

The problem with this approximation is, that it is ambiguous, what surface the area element dA represents. For the plate-sphere geometry, the element can be either chosen tangential to the sphere or parallel to the plate (or in theory any other fictitious surface

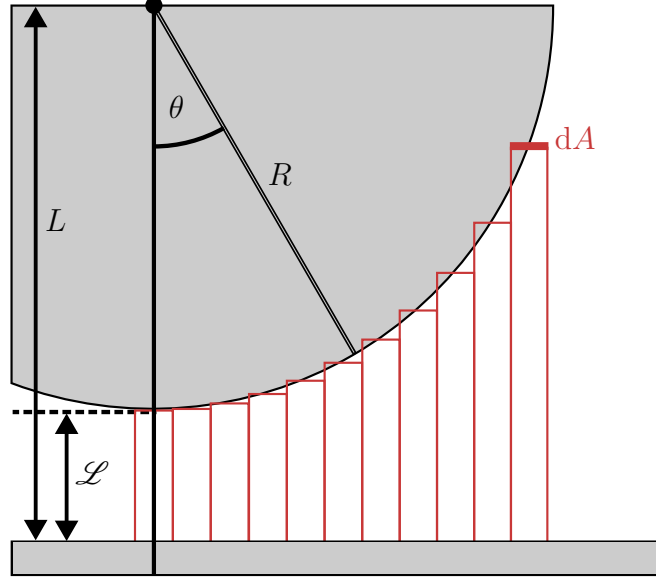


Figure 3.2: In the proximity force approximation the sphere is divided into infinitesimal plane areas dA which all exert a force dF according to eq. (3.3). All the contributions are added up together.

somewhere in between) [27]. For the plate-sphere geometry, in the limit of the validity of the PFA $\mathcal{L} \ll R$ both methods yield the same result. For the following calculations, I choose dA parallel to the plate and the area can be parameterized with $r \in [0, R]$ and $\varphi \in [0, 2\pi]$ resulting in a distance z between the infinitesimal area elements $z(r) = \mathcal{L} + R - \sqrt{R^2 - r^2}$ ⁶. The PFA eq. (3.8) then yields for a dielectric sphere against a perfectly conducting plate

$$E_{\text{plate-sphere}} = -\frac{\hbar c \pi^2}{720} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 1} \right) \varphi(\varepsilon_r) \int_0^R dr \int_0^{2\pi} r d\varphi \frac{1}{z(r)^3} \quad (3.9)$$

$$= -\frac{\hbar c \pi^3}{360} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 1} \right) \varphi(\varepsilon_r) \frac{R^2}{2\mathcal{L}^2(R + \mathcal{L})} \quad (3.10)$$

$$\approx -\frac{\hbar c \pi^3}{720} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 1} \right) \varphi(\varepsilon_r) \frac{R}{\mathcal{L}^2} \quad (3.11)$$

⁶Taking dA tangential to the sphere, it can be parameterized with $\theta \in [0, \pi/2]$ and $\varphi \in [0, 2\pi]$ resulting in $z(\theta) = \mathcal{L} + R - R \cos \theta$. The PFA eq. (3.8) yields with $dA = R^2 \sin \theta d\theta d\varphi$ the result $\propto \frac{\pi R^2 (R + 2\mathcal{L})}{\mathcal{L}^2 (R + \mathcal{L})^2}$ which in the limit of $\mathcal{L} \ll R$ results in the same expression as eq. (3.11).

3.2 Casimir forces between a conducting plate and a dielectric sphere

An empirical derivation for power law of the casimir energy between a sphere and a conducting plate can be made directly from the energy between two atoms with static polarizability α_i given by Casimir and Polder [13]. They derived an expression for the Casimir-Polder potential of ⁷

$$E = -\frac{23\hbar c\alpha_1\alpha_2}{4\pi L^7}. \quad (3.12)$$

The polarizability of a sphere with radius R is derived in appendix B.1 and is given for an dielectric with ε_r as

$$\alpha_{\text{sphere}} = \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right) R^3. \quad (3.13)$$

If one atom is now replaced by a conducting sphere ($\varepsilon_r \rightarrow 1$) of radius $\sim L$ (much larger than the atom) with a polarizability of L^3 , it get obvious that between this big sphere and the atom, the energy is given by a power law of $1/L^4$. It is therefore natural to assume, that for a macroscopic sphere and a macroscopic plate, the Casimir energy behaves similar to a $1/L^4$ law - at least for the *large separation limit* (LSL). The exact calculation for this problem is very hard. In fact, no analytic solution is known.

Ford was able to determine an integral expression using a macroscopic approach in 1998 [23]:

$$F = -\frac{\hbar c}{4\pi L^4} \int_0^\infty d\omega \alpha(\omega) [3 \sin 2\omega L - 6L\omega \cos 2\omega L - 6L^2\omega^2 \sin 2\omega L + 4L^3\omega^3 \cos 2\omega L]. \quad (3.14)$$

The expression depends on the polarizability, which is generally not constant for a dielectric. Especially for small separations between the sphere and the plate, this dependence and the non-constant polarizability make this integral nearly unsolvable. For large separations, much larger than the absorption wavelength of the dielectric or much larger than the wavelength corresponding to the plasma frequency in the Drude-Model, the polarizability can be assumed to be static $\alpha = \text{const}$ [17, 23]. In this simplifying case, the integral can be solved using an exponential convergence factor and results in

$$F = -\frac{6\hbar c}{4\pi L^5} \alpha \quad (3.15)$$

and thus

$$E = -\frac{3}{8} \frac{\hbar c}{\pi L^4} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right) R^3. \quad (3.16)$$

⁷For two macroscopic spheres, the casimir potential looks very similar to eq. (3.12). The polarizability of a sphere is given by eq. (3.13). Using this result, the Casimir energy between two identical dielectric spheres in the large separation limit is given as $-\frac{23\hbar c}{4\pi L^7} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right)^2 R^6$ [28].

For the large separation limit, the Casimir interaction between a sphere and a plate behaves like expected considering the motivation of the $1/L^4$ -law above in this section.

A second method to calculate the Casimir energy for arbitrary compact objects and a conducting wall was developed by Emig *et. al.* [28]. For the sphere-plate geometry, this results in the large separation limit (LSL) in a infinite series, where the first-order term ($\propto 1/L^4$) is precisely given by eq. (3.16) [26, 29]. The first 8 terms of this series are shown in fig. 3.3 as well as a few specific numerical points for higher terms. It becomes

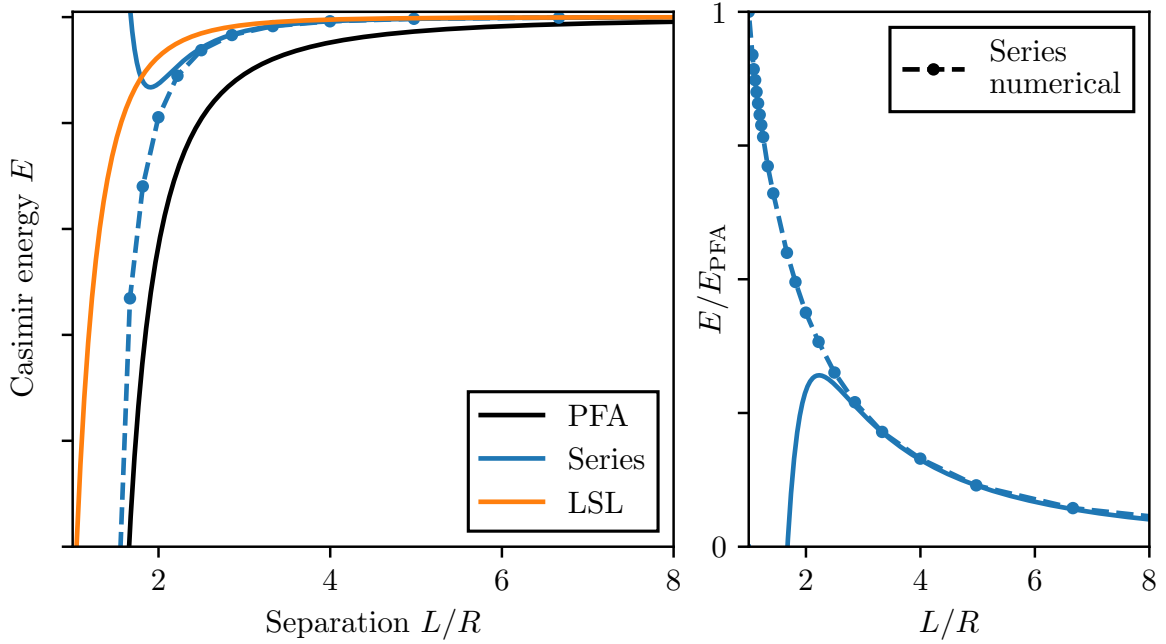


Figure 3.3: Behavior of different approximations of the casimir interaction between a conducting sphere and a perfectly conducting plate. Additionally, a comparison between the PFA and an exact numerical series expansion from Ref. [26] is shown.

evident, that the series expansion converges to the LSL eq. (3.16) for large separations and to the PFA eq. (3.11) for small separations. However, quantitatively the numerics show, that $E/E_{\text{PFA}} \leq 1$ and thus the PFA predicts a stronger Casimir energy for all separations.

Theorem 3.1. *The PFA model for a **conducting sphere** and a conducting plate predicts a stronger casimir energy for all separations L/R than the LSL or the exact series expansion.*

Proof. a) The numerical series expansion in fig. 3.3 from Ref. [26] shows that $|E_{\text{series}}| \leq |E_{\text{PFA}}|$ for all L/R .

b) Comparing the PFA eq. (3.11) for a conducting sphere with a dielectric one, it becomes evident, that $|E_{\text{cond.}}| \geq |E_{\text{diel.}}|$. This can be seen by the fact that for $\varepsilon_r \in [1, \infty)$ $\varphi(\varepsilon_r) \leq 1$ and $(\varepsilon_r - 1)/(\varepsilon_r + 1) \leq 1$. Thus the PFA for conductors is always a worse approximation.

c) Analytically, it can be shown that the energy predicted by the LSL eq. (3.16) is always smaller than the PFA:

$$|E_{\text{PFA}}| > |E_{\text{LSL}}| \quad (3.17)$$

$$\frac{\hbar c \pi^3}{720} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 1} \right) \varphi(\varepsilon_r) \frac{R}{\mathcal{L}^2} > \frac{3\hbar c}{8\pi L^4} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right) R^3 \quad (3.18)$$

$$\frac{8\pi^4}{3 \cdot 720} \frac{\varepsilon_r + 2}{\varepsilon_r + 1} \varphi(\varepsilon_r) > \frac{(L - R)^2 R^2}{L^4} = \left(\frac{R}{L} \right)^2 - 2 \left(\frac{R}{L} \right)^3 + \left(\frac{R}{L} \right)^4 \quad (3.19)$$

Maximizing the RHS, and using the fact that $R/L \leq 1$, the maximum is $1/16$ at $R/L = 1/2$. The minimum of the LHS is a little more complicated. The minimum of $\varphi(\varepsilon_r)$ is reached for $\varepsilon_r = 1$ and is at a minimum 0.46. The minimum of $(\varepsilon_r + 2)/(\varepsilon_r + 1)$ is reached at $\varepsilon_r \rightarrow \infty$ and is 1. Thus the LHS is at the absolute (unreachable) minimum $8\pi^4/2160 \cdot 0.46 \approx 0.166$ which is still larger than $1/16$. \square

Remark. This is rather unintuitive, because one would expect it to be the other way around. The $1/L^4$ dependence in the LSL should - for small separations - tend much faster to zero than the $1/\mathcal{L}^2$ dependence in the PFA. However considering the differences between L and $\mathcal{L} = L - R$, this is no longer surprising.

Admittedly, for separations $L/R \gtrsim 2$, the difference is not very large. Nevertheless will I use the PFA eq. (3.11) for the rest of the thesis as a worst-case estimation. Keep in mind, that the some quantities might be overestimated.

3.3 Imperfect plate and spheres

Python numerical approach, gaussian modes (vibration modes of a spherical plane), perlin noise

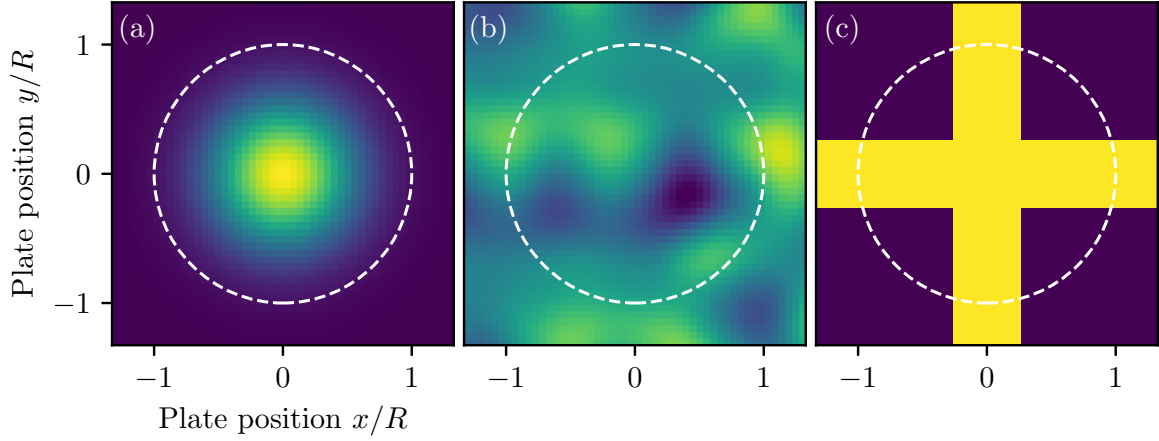


Figure 3.4: A selection of imperfect plates. (a) A simple gaussian displacement in the same size as the sphere. (b) Random excitations in the shape of *perlin noise*. (c) a cross-shape in the center of the plate. All displacements are in the size of $0.1R$.

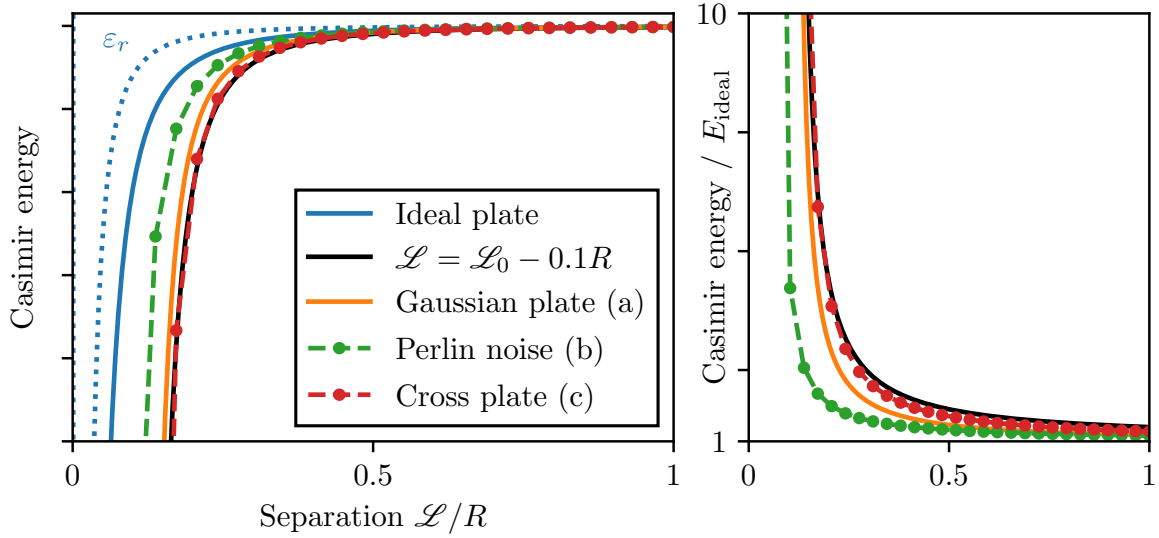


Figure 3.5: Casimir energy for the different noisy plates displayed in fig. 3.4. It becomes evident, that local deviations in the distance between the plate and the sphere in the order of $\Delta\mathcal{L} \sim 0.1R$ increase the Casimir force exponentially for very small separations. Larger separations however are seemingly unaffected.

4 Entanglement generation in front of a static shield

The generalized form of the system described in chapter 2 with the addition of a conducting Faraday shield is shown in fig. 4.1. As before, the masses are delocalized with superposition separations Δx_A for mass A and Δx_B for mass B respectively. The su-

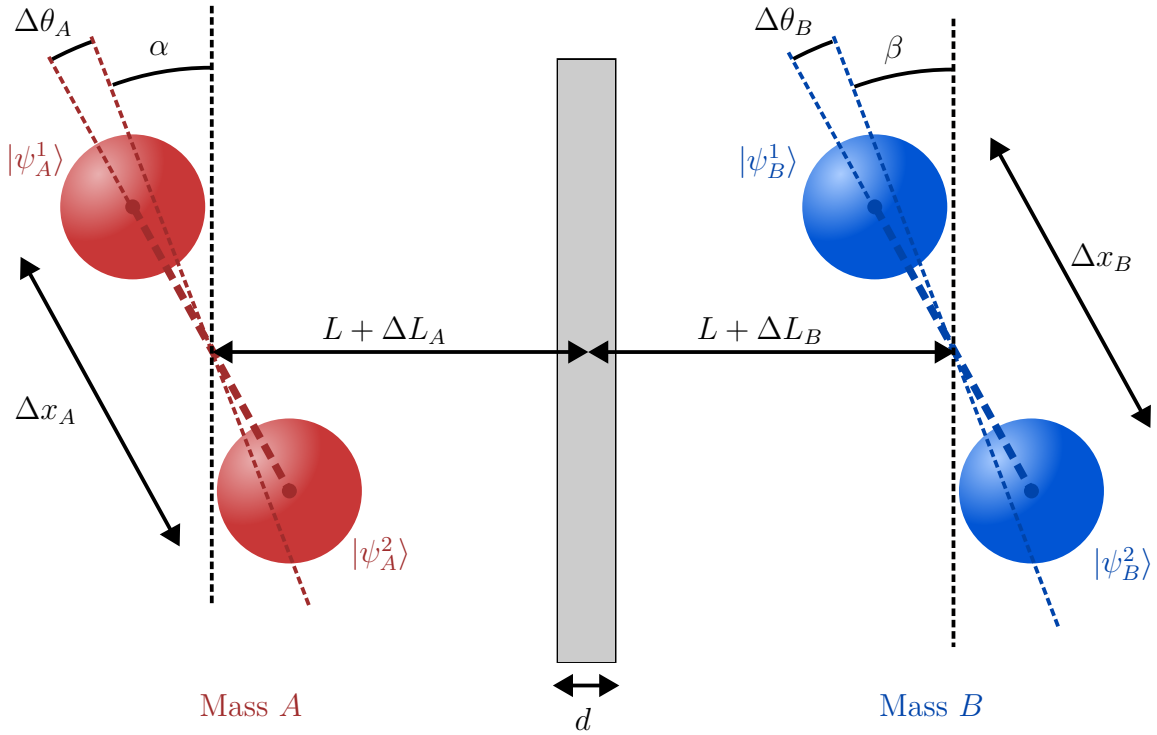


Figure 4.1: Schematic depiction of a experimental setup for the detection of gravitationally induced entanglement between two masses A and B . They are separated by a distance of $2L + \Delta L_A + \Delta L_B$ in arbitrary orientations given by the angles α and β with small variations $\Delta\theta_{A(B)}$. All variations are assumed to be gaussian distributed. The masses are delocalized in a cat state with a separation $\Delta x_{A(B)}$ between the states $\psi^1_{A(B)}$ and $\psi^2_{A(B)}$. A conducting Faraday shield with thickness d is placed in the center between the masses.

perpositions are extended in arbitrary orientations α and β ($\alpha, \beta \in [0, \pi)$) because of

symmetry). Most notably, the orientation of $\alpha = \beta = 0$ represents the same “parallel configuration” as discussed earlier and I will refer to $\alpha = \beta = \pi/2$ as the “orthogonal configuration”. As demonstrated earlier, if gravity is assumed to be able to mediate entanglement, the above system can generate entanglement between states A and B due to their gravitational interaction. Placing a shield in the center to shield potential interactions between the masses because of various electromagnetic interactions such as Coulomb- or Casimir-forces should in theory not influence the entanglement generation. However, Casimir interactions between the shield and the spheres are still present at small variations as discussed in chapter 3. It is straight forward to convince yourself, that these interactions can only induce local phases to each of the delocalized states depending only on the shield-mass distance of set state. These interactions can therefore - assuming a static shield e.g. at zero temperature - not induce any entanglement at all between different masses.

For the complete picture however, one has to consider, how the entanglement is going to be measured in a real experiment. Most likely, the density matrix of the evolved system after some time t is measured by **full state tomography** !!!CITATIONS!!!. Other proposals aim to measure the entanglement using a suitable witness [2, 4], but the creation of such a witness requires knowledge of the specifics on the experimental realization of massive delocalized states. In this thesis, I will focus on the most general and universally applicable case of measuring the complete density matrix of the system and checking for entanglement. The density matrix of a 2 qubit system however consists of 16 different entries (only 9 independent ⁸) and thus at least 9 measurements of the system are required for a full state tomography. However because of the statistical nature of quantum mechanics and the fact, that each measurement destroys the previously evolved state, in practice a lot more measurements are required. One cannot be sure that the masses are placed in the exact same position each run or if the superposition is extended in precisely the same orientation over and over again. Even if it was somehow possible, to place the particles at the *exact* same position each measurement, thermal vibrations of the shield, the masses or the setup still induce small deviations in the placement. Thus, each run the entanglement of a slightly different system is measured creating effectively a result which looks like a mixed state

$$\rho = \int_{-\infty}^{\infty} d\theta \frac{1}{\sqrt{2\pi}\Delta\theta} e^{-\theta^2/2(\Delta\theta)^2} |\psi_\theta\rangle\langle\psi_\theta|. \quad (4.1)$$

Here, $|\psi_\theta\rangle$ is the pure state of a single measurement dependent on the angular variations θ . These variations are assumed to be gaussian distributed around mean 0 and standard deviation $\Delta\theta_{A,B}$. The same argument can be made for variations in the distance, which are distributed with $\Delta L_{A,B}$ (compare with fig. 4.1). In some cases, like for example if the plate position moves slightly, the variations are correlated and $\Delta L_A = \Delta L_B$. But in the most general case, they are assumed to be independent and follow their own probability distribution.

⁸Using the known characteristics of the density matrix like hermiticity $\rho^\dagger = \rho$ and $\text{tr} \rho = 1$, it is possible to reconstruct ρ by measuring only 9 specific entries.

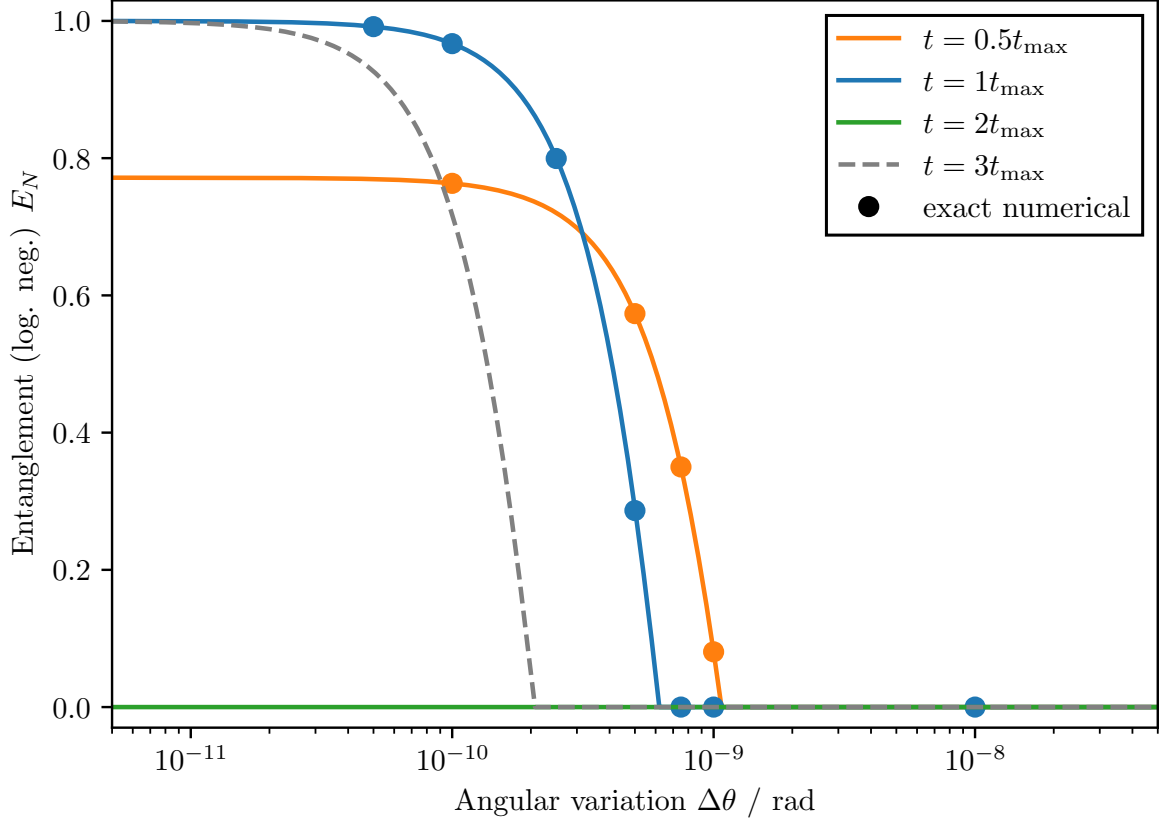


Figure 4.2: Entanglement quantified by the logarithmic negativity (eq. (2.21)) dependent on the angular variation $\Delta\theta$. The entanglement is shown at different times, where t_{\max} is the time of maximal entanglement calculable by eq. (4.4). Additionally, a few selected exact numerical results are shown to align precisely with the approximated version.

4.1 Stability in different orientations

As could be already seen in fig. 2.2 in chapter 2, the entanglement dynamics and especially the time t_{\max} of maximum entanglement highly depend on the orientation. There, it was shown already, that for the parallel configuration the maximum entanglement is reached twice as slow as in the orthogonal orientation. This result can be more generalized for an arbitrary orientation quantified by α and β from fig. 4.1.

$$E_N = \log_2 (1 + |\sin \Delta\phi|) \quad (4.2)$$

where $\Delta\phi$ is now dependent on the orientation and is given by (for $\Delta x \ll L$)

$$\Delta\phi = \frac{GM_A M_B t}{\hbar} \frac{\Delta x_A \Delta x_B}{8L^3} \left[\sin \alpha \sin \beta - \frac{1}{2} \cos \alpha \cos \beta \right]. \quad (4.3)$$

The maximum entanglement is again reached after $\Delta\phi = \pi/2$ and thus is also dependent on the orientation

$$t_{\max} = \frac{4\pi L^3 \hbar}{GM_A M_B \Delta x_A \Delta x_B} \left| \sin \alpha \sin \beta - \frac{1}{2} \cos \alpha \cos \beta \right|^{-1}. \quad (4.4)$$

The resulting times are shown in fig. 4.3 and align with earlier results for the special cases of the orthogonal and parallel configuration. The global minima of t_{\max} is given in the orthogonal configuration meaning that of all possible configurations, this one induces entanglement the fastest. Considering the discussions in the end of chapter 2, this is not very surprising. Much more interesting are the apparent singularities which arise for

$$\sin \alpha \sin \beta = \frac{1}{2} \cos \alpha \cos \beta. \quad (4.5)$$

For $\beta = 0$, the singularity in t_{\max} at $\alpha = \pi/2$ is not surprising. At this configuration, the distances $|\psi_A^1\rangle \leftrightarrow |\psi_B^{1,2}\rangle$ are identical as well as $|\psi_A^2\rangle \leftrightarrow |\psi_B^{1,2}\rangle$. For the case $\alpha = \beta$, the two singularities are precisely given at

$$\alpha = \beta = 2 \arctan(\sqrt{3} \pm \sqrt{2}) \approx 90^\circ \pm 54.74^\circ. \quad (4.6)$$

There does not exist a clear geometric explanation why no entanglement is generated in this configuration, however, all distances between the superpositions for a harmonic mean as visualized in fig. 4.4.

4.2 Discussion on the optimal setup

- No global maximum
- dependent on experimental stuff
- Masses and distances and so on must be bounded by $10x E_{\text{Casimir}} ; E_{\text{Gravity}}$

How can one find the best parameters depending on the experimental requirements?

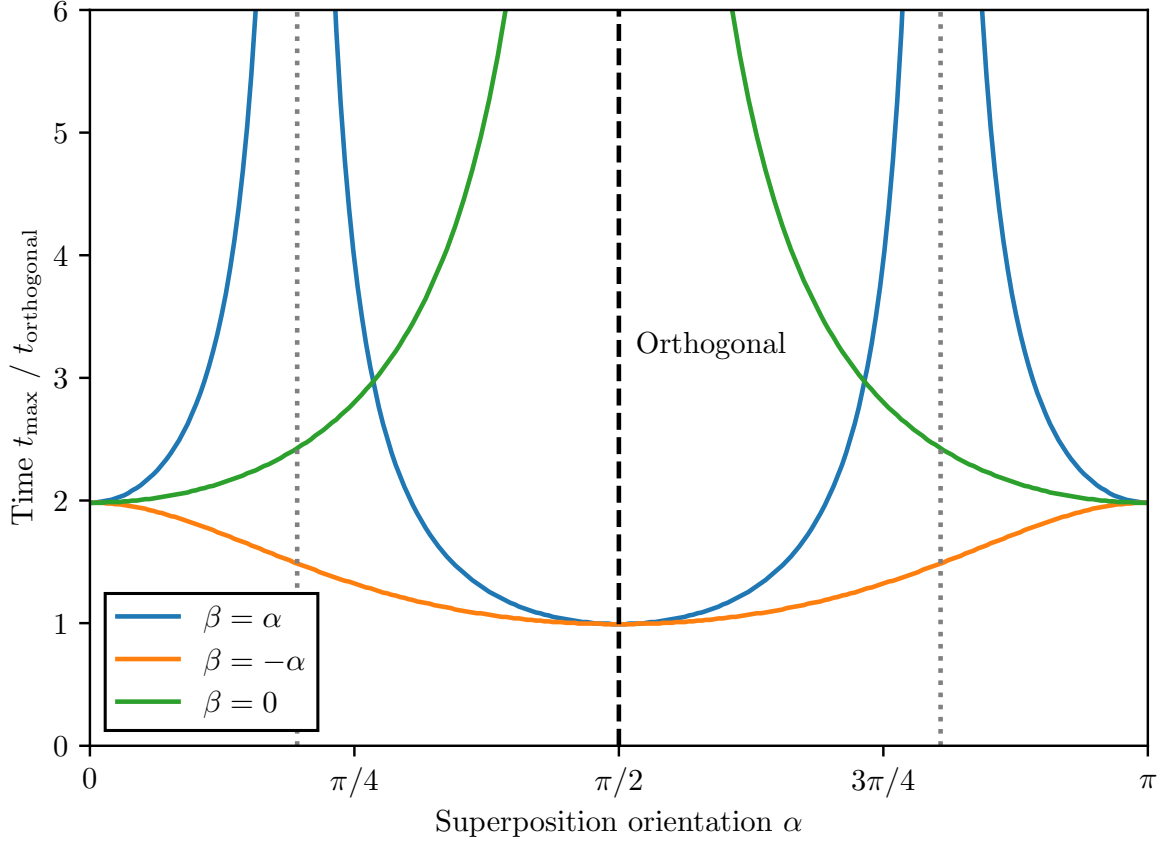


Figure 4.3: Time of maximum entanglement t_{\max} relative to $t_{\max, \text{orthogonal}}$ of the orthogonal configuration for different orientations. Some orientations like $\alpha = \pi/2$ and $\beta = 0$ cannot induce any entanglement at all because of symmetry considerations.

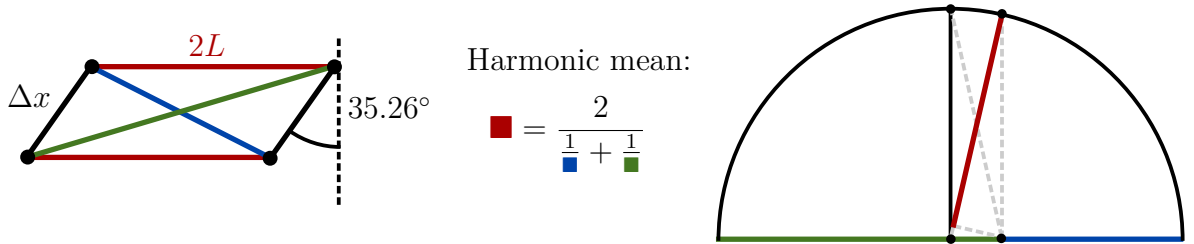


Figure 4.4: **left:** Arrangement in the orientation $\alpha = \beta = 2 \arctan(\sqrt{3} - \sqrt{2})$. All distances exactly lie in the *harmonic mean* (for $\Delta x \ll L$). **right:** Geometric visualization of the harmonic mean.

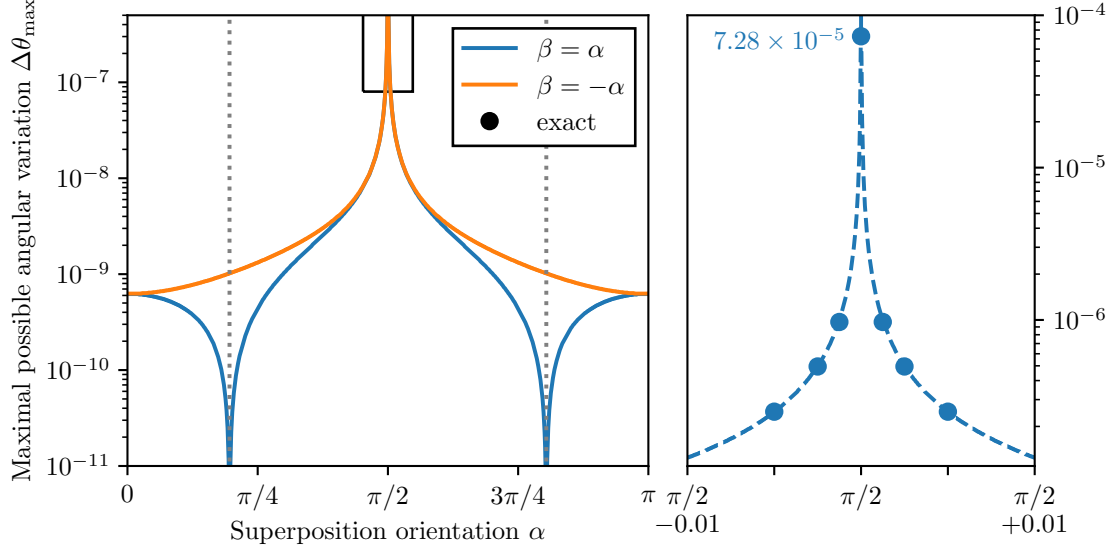


Figure 4.5: Maximum possible allowed angular variation $\Delta\theta_{\max}$ for different orientations. All data-points were calculated at the time of maximum entanglement shown in fig. 4.3. The *orthogonal configuration* is very stable against angular disturbances. At $\alpha = \beta = \pi/2$, only exact numerical results show a finite value. The singularities on the left figure arise from the fact, that these configurations need infinite time to entangle as already seen in fig. 4.3.

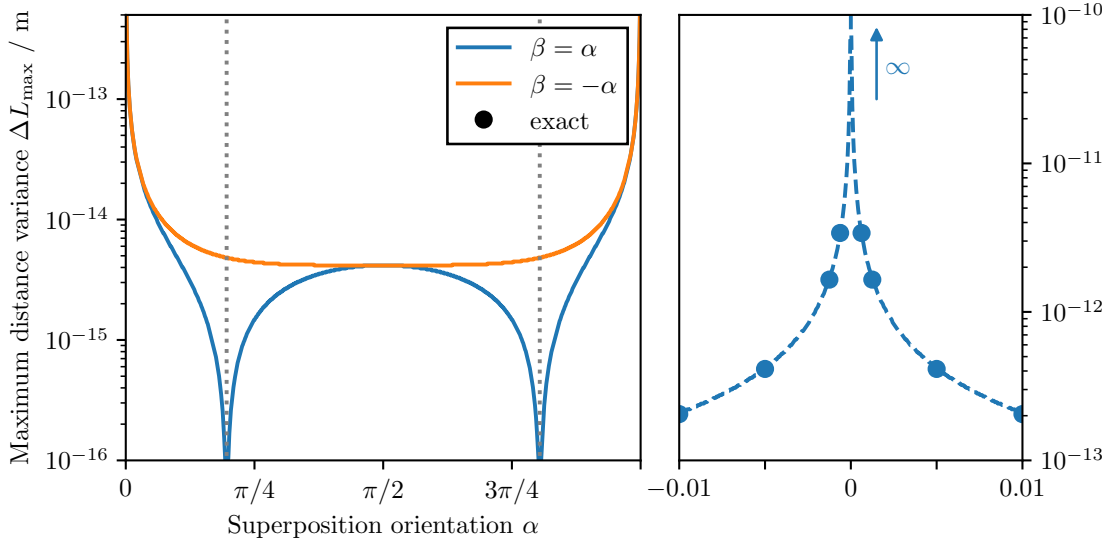


Figure 4.6: Maximum possible allowed distance variation ΔL_{\max} for different orientations. This figure is similar to fig. 4.5, only for distance variations. On the contrary to angular variations, the *parallel configuration* is infinitely stable against changes in the distance between the shield and the cat-state.

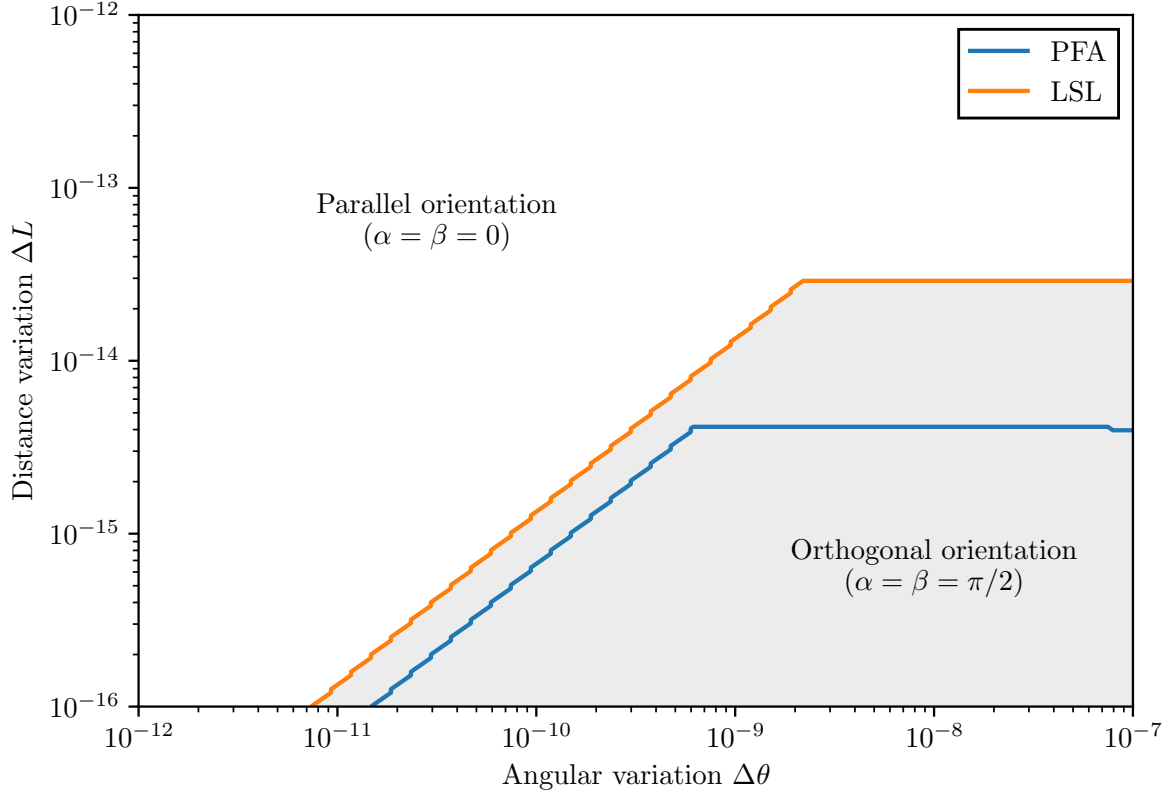


Figure 4.7: Optimal orientation for arbitrary variations in the angle $\Delta\theta$ and the distance ΔL . The optimum was calculated for different models of the Casimir-interaction (PFA eq. (3.11) and LSL eq. (3.16)). If angular variations dominate, the orthogonal configuration is best, whereas for large distance variations, a parallel orientation is advised.

1. Specify maximum coherence time (how long the experiment can take)
2. This fixes $L^3/(M_A M_B \Delta x_A \Delta x_B)$
3. Look up, what maximum $\Delta\theta$ and ΔL you can get at this time (Dependent on the amount of entanglement you want to have at the end)
4. Take the minimum possible L/R in account (dependent on trap stiffness and so on) Ideally, large enough to reduce casimir interactions

1. Given a target measuring time t_{target} , this fixes the ratio

$$\frac{L^3}{M_A M_B \Delta x_A \Delta x_B} = 5.036 \times 10^{22} \text{ m/kg}^2\text{s} \cdot t_{\text{target}} = \text{const.} \quad (4.7)$$

Most likely, the superposition size $\Delta x_{A,B}$ is also not changeable and one has to be satisfied by what can be achieved experimentally. The delocalized states of mass M has to interact gravitationally with each other in a laboratory setting for at least the duration of the measuring process t_{target} . These considerations limits the coherence time and usually, low times in the order of milliseconds to seconds are favorable.

2. In principle, one does not require to measure at the time of maximum entanglement where $E_N = 1$. If a lower quantity of entanglement is enough for a set experimental goal, I recommend, measuring at a lower time than $t_{\text{target}} = \tau t_{\text{max}}$ ($\tau < 1$) !!!!FIGURE!!!!. This on the other hand increases the fixed ratio eq. (4.7) by a factor of $1/\tau$. Therefore, it is possible to use smaller superposition sizes Δx , lighter masses or increase the distance L which decreases Casimir interactions.
3. Most likely, the minium angular

5 The consequences of a thermal shield

Why a shield, what does the shield do, maybe short introduction

5.1 Thickness and size of the shield

The thickness and size of the shield can be estimated by considering a real material. The shield material should have a large electric conductivity σ and should therefore behave like an conductor. Even a superconducting shield could be considered. For a real metal like for example copper, the transmission T of electromagnetic waves is given by [30]

$$T = \left| \frac{\mathbf{E}_{\text{after}}}{\mathbf{E}_{\text{before}}} \right| = \frac{2}{Z_0 \sigma d} \quad (5.1)$$

where σ is the electric conductivity, $Z_0 = 377 \Omega$ the impedance of free space (provided the shield is placed in a vacuum or air) and d the thickness of the shield. The electric conductivity has a strong dependence on temperature and decreases with $1/T^5$ ⁹ with increasing temperatures [32, p. 284-286]. The electric conductivity for copper at room temperature ($\sigma = 59.6 \times 10^6 \text{ S/m}$) should therefore be a very valid worst-case approximation. Measured data suggest a conductivity of $\sigma(T = 10) \approx 1.5 \times 10^{10} \text{ S/m}$ [31].

In general should the thickness of the shield be chosen as thin as possible. Otherwise, the distance between the masses has to increase accordingly and the gravitational potential between them gets smaller. This would require longer coherence times till a sufficient amount of entanglement could be measured. To estimate the thickness of the shield, I am using the condition, that entanglement between the masses should be built up mainly due to gravity. All other possible interactions like Coulomb or Casimir forces, should be suppressed sufficiently by the shield. To quantify the amount of entanglement built-up over time, I obt for a measure called ***entanglement rate***

$$\Gamma = \left. \frac{d}{dt} E_N(\rho) \right|_{t=0} \quad (5.2)$$

⁹This behavior is only true for temperatures below the Debye temperature. For copper, this limit is around $\Theta_D = 343 \text{ K}$. In the experiment, the shield is cooled down, so the low-temperature limit of the electric conductivity for metals is very valid [31].

where E_N is an appropriate entanglement measure - in this case the logarithmic negativity from section 2.2. For the gravitational entanglement, this can be calculated by eq. (2.23) and for the parallel configuration this is given by

$$\Gamma_{\text{Gravity}} = \frac{GM_A M_B (\Delta x)^2}{16\hbar L^3 \log 2}. \quad (5.3)$$

The entanglement rate for other interactions should be smaller (and ideally a lot smaller) than the entanglement rate due to gravity. In the following, this is considered for Coulomb- and Casimir interactions.

5.1.1 Shielding Coulomb-Interactions

The main focus of the shield is to block electromagnetic interactions between the spheres. Experimentally, it might be beneficial for the spheres to carry a small amount of charge to help with the trapping. !!!CITATIONS!!! The Coulomb interaction potential energy

$$V = \frac{1}{4\pi\epsilon_0} \frac{q_A q_B}{2L} \quad (5.4)$$

between the spheres has the same structure as the gravitational potential and can therefore induce entanglement by the same logic. Using a shield, this coupling is suppressed by a factor of T . The entanglement rate Γ_{Coulomb} is therefore easily calculable in a similar form as eq. (5.3) to

$$\Gamma_{\text{Coulomb}} = \frac{T |q_A q_B| (\Delta x)^2}{64\pi\epsilon_0 \hbar L^3 \log 2} \quad (5.5)$$

Requiring that $\Gamma_{\text{Gravity}} > \Gamma_{\text{Coulomb}}$, this inequality yields a maximum transmission and thus a minimum thickness of

$$T \frac{|q_A q_B|}{4\pi\epsilon_0} < GM_A M_B \quad (5.6)$$

$$\iff d > \frac{1}{2\pi\epsilon_0 Z_0 \sigma G} \frac{|q_A q_B|}{M_A M_B} = \frac{9}{32} \frac{1}{Z_0 \sigma} \frac{1}{\pi^3 \epsilon_0 G \rho^2} \frac{e^2}{R^6} \quad (5.7)$$

where in the last step $M_A = M_B$ and $q_A = q_B = e$ has been assumed. The shield thickness heavily depends on the radius of the spheres where small and light masses make the shield unworkably thick. A silica sphere with a radius of $R = 10^{-5}$ m would require a shield around the thickness of 10 nm at 4 K up to $2.5 \mu\text{m}$ at room temperature. A realistic thickness for the shield at low temperatures could therefore be $d = 100$ nm.

Electric fields however can still propagate around a Faraday shield of finite size and induce entanglement. It is however possible, to estimate the required radius r_s of the shield to block a specific amount η of the electric field. The resulting radius corresponding to η is derived in appendix C.1 and the result is shown in fig. 5.1. The transmission T should be replaced by a modified transmission $\tilde{T} = T\eta + (1 - \eta)$ where the shield

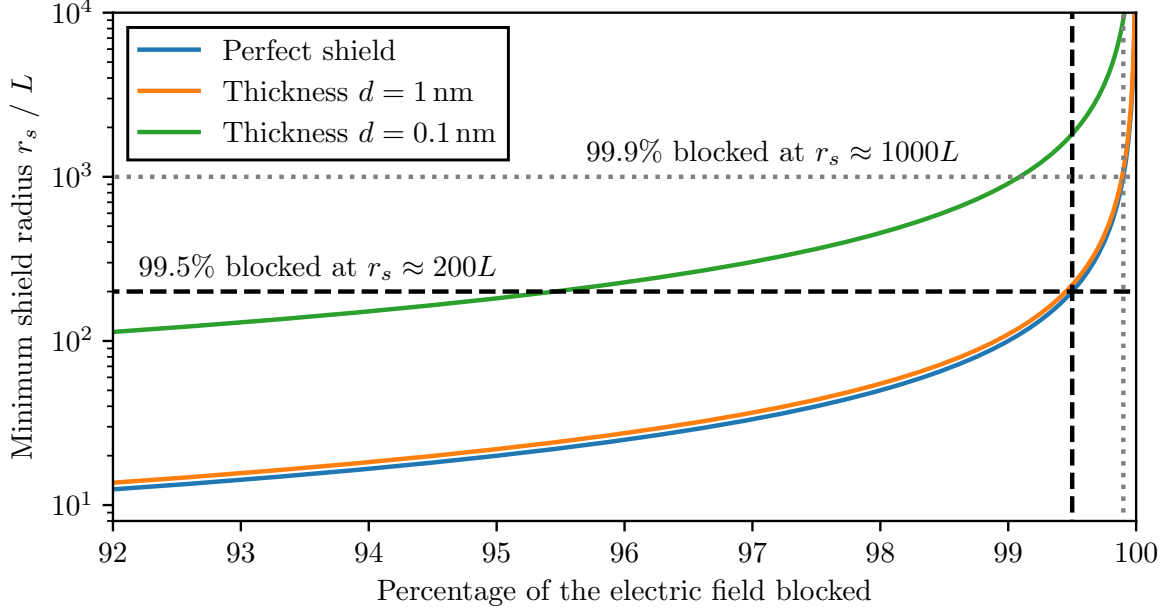


Figure 5.1: Radius of the shield depending on the shield effectiveness η . Additionally a real shield with different thicknesses d at room temperature is considered. For a shielding between 99.5...99.9% ($\eta = 0.995...0.999$), a radius of at least $r_s = 200...1000L$ should be used. For a sphere-plate distance of $L = 2 \times 10^{-5}$ m, the shield radius should be in the order of millimeters and centimeters.

effectiveness η depends on r_s . Now, the condition eq. (5.6) introduces a limit for the minimum effectiveness and thus a limit on the minimum radius of

$$\eta_{\min} = 1 - \frac{4\pi\epsilon_0 G M_A M_B}{|q_1 q_2|}. \quad (5.8)$$

Thus using the setup with the parameters from earlier, a minimum effectiveness of $\eta_{\min} \gtrsim 0.99997$ and thus a radius of $\gtrsim 66$ cm is required. This shield is too large for all practical purposes and it might be beneficial to choose slightly heavier masses to reduce the shield size to the orders of ~ 1 cm. This would require both spheres to have approximately double the radius than before. Using neutral masses without any charge would be also beneficial and the shield size could be reduced to only the size of the spheres themselves, however it might be an engineering challenge, to trap and levitate uncharged massive particles.

5.1.2 Shielding Casimir-Interactions

Similarly to Coulomb interactions, it is possible to estimate the required thickness of the shield necessary to sufficiently suppress Casimir interactions. Between two spheres

with radius R and separation $2L$, the Casimir potential reads [28]

$$V = -\frac{23\hbar c}{4\pi \cdot 128L^7} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right)^2 R^6. \quad (5.9)$$

The entanglement rate can be obtained similarly as before by expanding the casimir potential between the spheres in small Δx and computing the logarithmic negativity as before:

$$\Gamma_{\text{Casimir}} = T^2 \frac{161}{4096} \frac{cR^6(\Delta x)^2}{\pi L^9 \log 2} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right)^2. \quad (5.10)$$

The dependence on T^2 is only a systematic guess but should be sufficient for a basic estimation. Casimir and van der Waals forces are second order effects in the dipole-dipole interaction [19]. Demanding again, that the entanglement due to gravity should be mediated faster than due to Casimir interactions $\Gamma_{\text{Gravity}} > \Gamma_{\text{Casimir}}$, one arrives at the expression

$$T^2 \frac{161cR^6}{256\pi L^6} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right)^2 < \frac{GM_A M_B}{\hbar} \quad (5.11)$$

$$\iff d^2 > \frac{4 \cdot 161c\hbar R^6}{256Z_0^2\sigma^2\pi L^6 GM_A M_B} \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right)^2 \quad (5.12)$$

$$\iff d > \sqrt{\frac{1449}{4096} \frac{c\hbar}{G\pi^3} \frac{2}{Z_0\sigma\rho L^3} \frac{\varepsilon_r - 1}{\varepsilon_r + 2}} \quad (5.13)$$

where again in the last step I assume $M_A = M_B$. For large separations, the shield thickness can go arbitrary low because of the weakness of the casimir interactions at larger distances. Between two silica spheres separated in the order of magnitude as the radius ($L = 2 \times 10^{-5}$ m), the required minimum thickness is between 4×10^{-11} m at 4 K and 10 nm at room temperature. Either way, it is much thinner than a Faraday shield required to shield electrostatic Coulomb interactions. The effect of ε_r was neglected in these numbers. The minimum thickness only changes by a factor between 0...1 due to dielectric materials. In fact, these thicknesses are much thinner than recommended, because the shield loses rigidity. It turns out that the vibrational frequency depends linearly on the thickness and thus more vibrations are possible. A more detailed discussion of this matter is given in the next section.

5.1.3 Gravitational effects of the shield

For simplicity, in the following estimations of the attractive gravitational force between an infinitely large plate with thickness d and a sphere positioned a distance L in front of the plate is calculated. This should slightly overestimate the actual force due to a finite plate¹⁰. The gravitational force between the sphere and a infinitesimal mass segment

¹⁰In fact, using a finite shield with radius $r_s = 0.01$ m, the total force decreases only by 0.2% compared to an infinitely sized one.

$dm = r\rho d\varphi dr$ of the plate with density ρ at a distance r from the center is given by

$$d\mathbf{F} = \frac{GMdm}{\ell^2} \hat{\ell} \Rightarrow dF_z = \frac{GM\rho d}{\ell^2} dr d\varphi \cos \theta, \quad (5.14)$$

where $\ell^2 = r^2 + L^2$ denotes the distance between the sphere and the mass segment and $\theta = \arccos L/\ell$ is the angle between them. The total force between the sphere and the infinite plate is therefore given by

$$F_z = GM\rho dL \int_0^\infty dr \int_0^{2\pi} d\varphi \frac{r}{(r^2 + L^2)^{3/2}} = 2\pi GM\rho d. \quad (5.15)$$

This result is independent of the distance L and a numerical value for a copper shield ($\rho = 8960 \text{ kg/m}^3$) with thickness of $d = 100 \text{ nm}$ and the silica sphere already used in chapter 2 is given by $F \approx 4.2 \times 10^{-24} \text{ N}$. Compared with the attractive force between two silica spheres separated by a distance of $2L = 4 \times 10^{-5} \text{ m}$, the force due to the plates is smaller by a factor of ≈ 0.8 . Both forces are therefore comparable and the thickness of the shield should be chosen as thin as possible to not influence the gravitational sensing of the masses. If the shield is not perfectly placed in the center between the two masses or if the shield is uneven due to e.g. thermal vibrations, these gravitational effects might be able to destroy entanglement by inducing non-global phases to the cat-states.

5.2 Thermal shield vibrations

A spherical plate that is fixed at the edge can vibrate in different vibrational modes labeled by the indices (k, l) where $k \in [1, \infty)$ and $l \in [0, \infty)$. The exact vibrational frequency and the mode shape can only be given in terms of the Bessel functions. In fact, one of the first occurrences of these functions can be traced back to Euler trying to solve the very similar problem of a vibrating perfectly flexible membrane [33]. In general, the vibrations of a plate with thickness d can be described by the differential equation [34, p. 490]

$$D\nabla^2 \nabla^2 u = -\rho d \ddot{u} \quad (5.16)$$

where D is given by material properties like Youngs module E and the poisson number ν as

$$D = \frac{d^3 E}{12(1 - \nu^2)}. \quad (5.17)$$

The general solution of this differential equation can be written in terms of the Bessel functions as (derived in Ref. [34, p. 490-495])

$$u_{kl}(r, \theta, t) = z \left[J_l(\beta_k r) - \frac{J_l(\beta_k r_s)}{I_l(\beta_k r_s)} I_l(\beta_k r) \right] \cos(l\theta + \phi_1) \sin(\omega_{kl} t + \phi_2) \quad (5.18)$$

with

$$\beta_k = \frac{\tilde{r}_k}{r_s} \quad \text{and} \quad \omega_{kl} = \frac{\tilde{r}_k^2}{r_s^2} \sqrt{\frac{D}{\rho d}} = \tilde{r}_k^2 \frac{d}{r_s^2} \sqrt{\frac{E}{12\rho(1 - \nu^2)}}, \quad (5.19)$$

where \tilde{r}_k is the k -th solution of the equation

$$J_l(\tilde{r}_k)I_{l+1}(\tilde{r}_k) + I_l(\tilde{r}_k)J_{l+1}(\tilde{r}_k) = 0. \quad (5.20)$$

The phases ϕ_1 and ϕ_2 can be determined by initial conditions and refer to the rotation of the plate as well as temporal offsets. The first few modes are shown in fig. 5.2. In general,

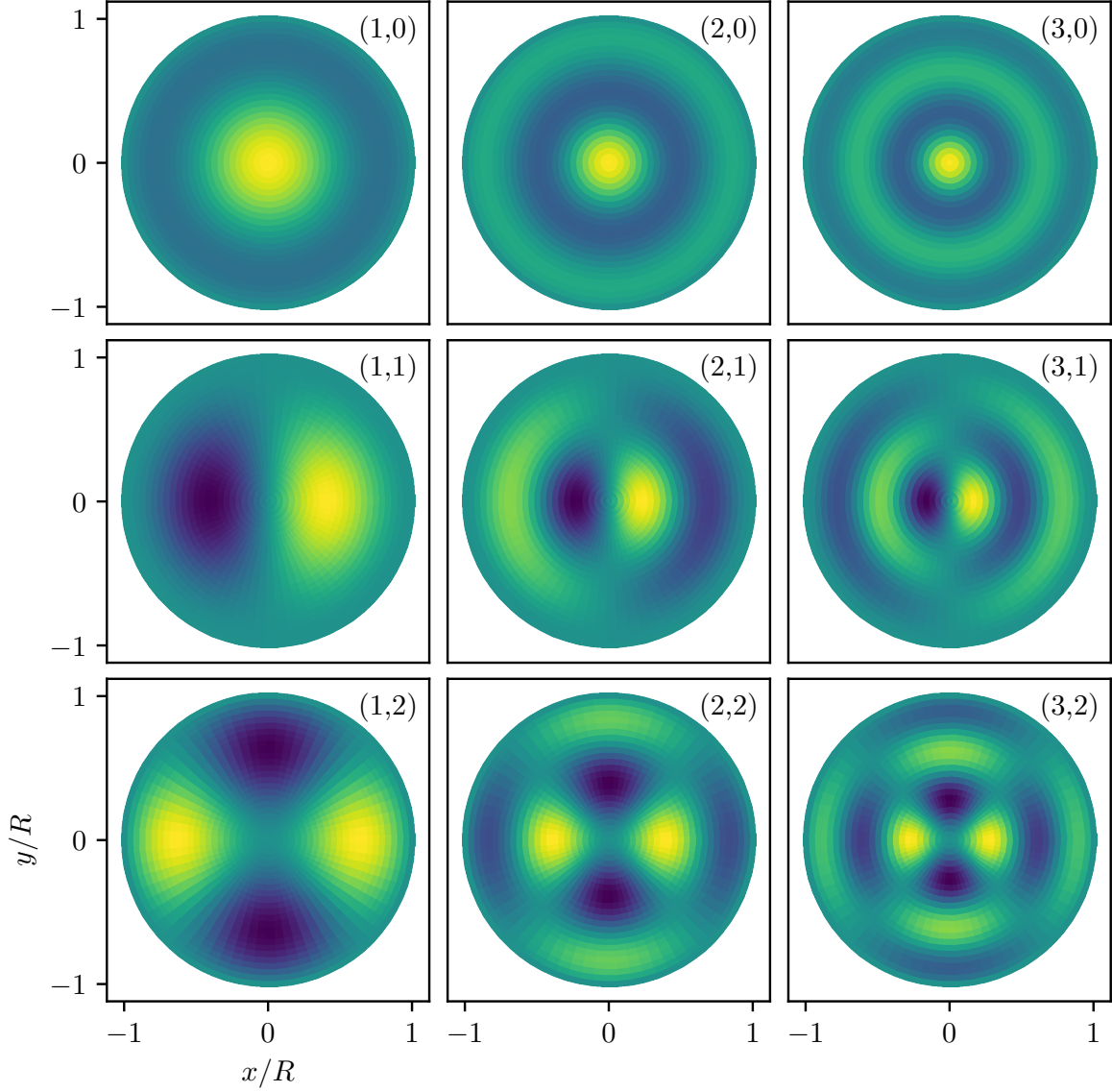


Figure 5.2: Normalized shape of the vibrational modes (k, l) of a vibrating spherical plate fixed at the edge with $R/d = 1000$.

every possible vibration of the plates can be expressed as a sum of these default modes u_{kl} . The amplitude z of the vibrations are determined by the thermal temperature and can be estimated by treating the amplitude of each vibration as a quantum harmonic

5 The consequences of a thermal shield

oscillator with frequency ω_{kl} . The expectation value of the amplitude $\langle \hat{z} \rangle$ is obviously zero and the variance $(\Delta \hat{z})^2 = \langle \hat{z}^2 \rangle - \langle \hat{z} \rangle^2$ at temperature T is given by

$$(\Delta \hat{z}_{kl})_T^2 = \frac{\hbar}{2\tilde{m}\omega_{kl}} \coth\left(\frac{\hbar\omega_{kl}}{2k_B T}\right) \approx \frac{k_B T}{\tilde{m}\omega_{kl}^2} \quad (5.21)$$

A derivation of this is given in appendix C.2. In the last step $\hbar\omega \ll k_B T$ was used. \tilde{m} is the *effective mass* of the mode in which the precise shape is considered. A intuitive estimation for this mass can be given by the average of the mode

$$\tilde{m} = m \frac{1}{\pi r_s^2} \int_0^{r_s} dr \int_0^{2\pi} r d\theta u_{kl}(r, \theta, t). \quad (5.22)$$

with $m = \rho \pi r_s^2 d$ being the total mass of the shield.

5.2.1 Occupation of the vibrational modes

$$\hat{H} = \hbar\omega \left(\hat{a}^\dagger \hat{a} + \frac{1}{2} \right) + \tilde{g}_A(\hat{z}) |\psi_A\rangle\langle\psi_A| + \tilde{g}_B(\hat{z}) |\psi_B\rangle\langle\psi_B| \quad (5.23)$$

Solution:

$$\rho(t) = \frac{1}{2} \begin{pmatrix} 1 & e^{-i\varphi(t)} e^{-\gamma(t)} \\ e^{i\varphi(t)} e^{-\gamma(t)} & 1 \end{pmatrix} \quad (5.24)$$

with the phase

$$\varphi(t) = \frac{1}{\hbar^2 \omega^2} (\sin(\omega t) - \omega t) (g_A^2 - g_B^2) \quad (5.25)$$

and the decoherence term

$$\gamma(t) = \frac{4(g_A - g_B)^2}{\hbar^2 \omega^2} \sin^2\left(\frac{\omega t}{2}\right) \left[\bar{n} + \frac{1}{2} \right] \quad (5.26)$$

5.3 Discussions on the shield

- ****Charged or uncharged**** Compare requirements in size and thickness as well as thermal vibrations, which are less a problem for smaller shields
- Small shield = local casimir effect. Compare with sec. 3.3
- How large are the effects of thermal vibration and how bad is everything (Maybe in the next section, otherwise a little bit chaotic)
- Maybe a cross-structure or a different geometry of the shield could reduce vibrations and a larger shield might be okay. But local structures again
- Rectangular plate (frequencies only one order of magnitude ($\sqrt{\pi^4}$) better -i no real improvements)
- Possible to measure casimir forces with a similar technique of dephasing (very sensitive)

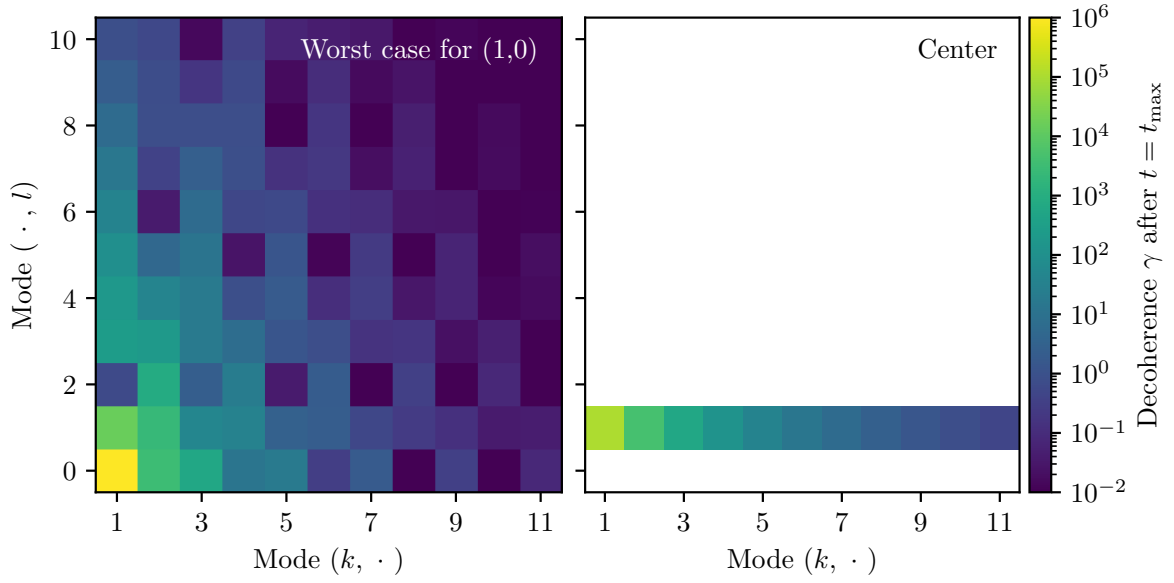


Figure 5.3: Maximum decoherence γ at 4 K for all modes if the cat-state is placed **left:** at the point with maximum gradient of the mode (1,0) and **right:** in the center of the plate. It becomes evident, that only a few low modes play an actual role for the total dephasing.

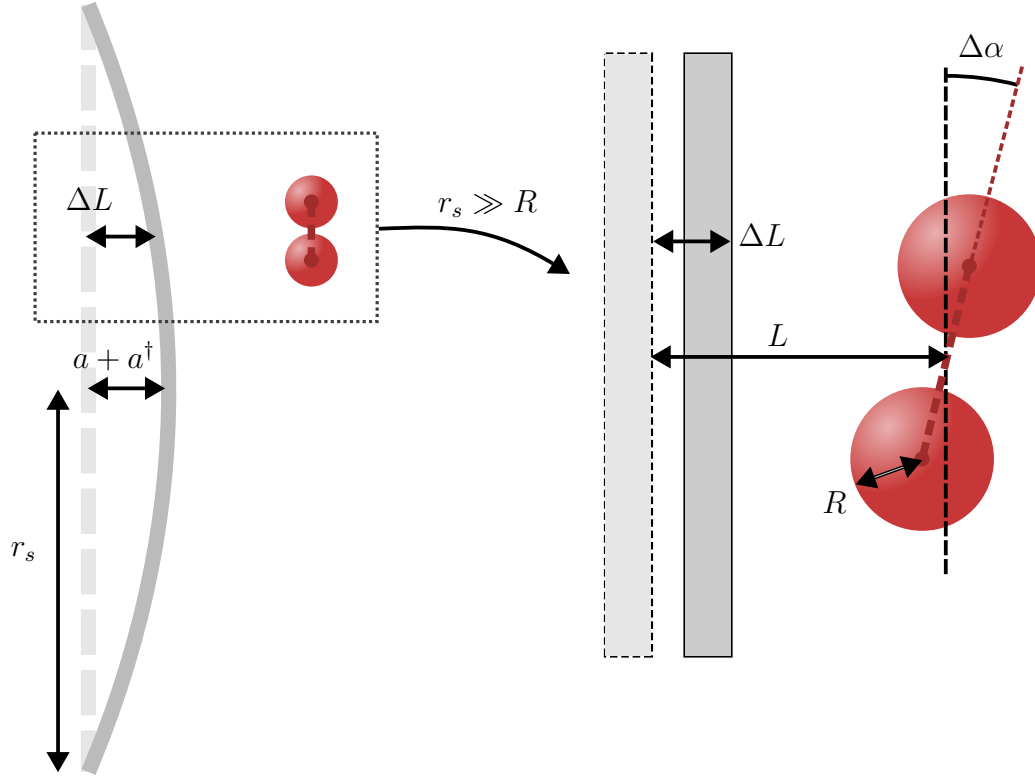


Figure 5.4: For a large shield, vibrations can be interpreted locally as if one mass was a distance ΔL further away from a shield and the other one ΔL closer and as if both masses would be rotated by an angle $\Delta\theta$.

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A TITLE TO BE DONE

A.1 Evolution under a gravitational Hamiltonian

In this section the time evolution of a system under Hamiltonian eq. (2.3) is calculated a) using the gravitational interaction \hat{H}_G as a perturbation b) using an exact time evolution of coherent states.

A.1.1 Using time dependent perturbation theory

A general biparty Fock state $|\psi_0\rangle = |kl\rangle$ with $k, l \in \mathbb{N}_0$ can be evolved in time under a Hamiltonian eq. (2.3) treating the gravitational interaction $H_G = -\hbar g(\hat{a}_1\hat{a}_2^\dagger + \hat{a}_1^\dagger\hat{a}_2)$ as a perturbation. The resulting state $|\psi(t)\rangle$ after some time t is in the most general form given as

$$|\psi(t)\rangle = \sum_{i,j \geq 0} c_{i,j}(t) |i, j\rangle \quad (\text{A.1})$$

where the coefficients $c_{i,j}(t)$ are given by first order perturbation theory as

$$c_{i,j}(t) = c_{i,j}(t=0) - \frac{i}{\hbar} \int_0^t dt' \langle ij | \hat{H}_G | kl \rangle e^{-i(E_{kl} - E_{ij})t'/\hbar}. \quad (\text{A.2})$$

The exponent is given by the energy of the appropriate Fock states $E_{kl} - E_{ij} = \hbar\omega(k + l - (i + j))$ and the matrix element in the integrand can be calculated to

$$\langle ij | \hat{H}_G | kl \rangle = \begin{cases} -\hbar g & \text{if } i = k \pm 1 \text{ and } j = l \mp 1 \\ 0 & \text{otherwise} \end{cases}. \quad (\text{A.3})$$

The coefficients for $t = 0$ are trivially given from the initial state as

$$c_{i,j}(t=0) = \begin{cases} 1 & \text{for } i, j = k, l \\ 0 & \text{otherwise} \end{cases}. \quad (\text{A.4})$$

For the non-zero states the energies in the exponent equate to zero and the evolved state is given by (up to a normalization)

$$|\psi(t)\rangle = |kl\rangle - igt |k-1, l+1\rangle - igt |k+1, l-1\rangle + \mathcal{O}(g^2). \quad (\text{A.5})$$

The result eq. (2.5) is represented by eq. (A.5) for the case of $k = 1$ and $l = 0$.

A.1.2 Using an exact time evolution

The Hamiltonian eq. (2.3) can be rewritten using symmetric and antisymmetric normal modes

$$\hat{a}_{\pm} = \frac{1}{\sqrt{2}}(\hat{a}_1 \pm \hat{a}_2) \quad (\text{A.6})$$

in the form of

$$\hat{H} = \hbar\omega_+ \hat{a}_+^\dagger \hat{a}_+ + \hbar\omega_- \hat{a}_-^\dagger \hat{a}_-, \quad \omega_{\pm} = \omega \pm (-g) \quad (\text{A.7})$$

The initial state consisting of two coherent oscillator states is in the new modes given by

$$|\psi(t)\rangle = |\alpha\rangle_1 |\beta\rangle_2 = \left| \frac{1}{\sqrt{2}}(\alpha + \beta) \right\rangle_+ \left| \frac{1}{\sqrt{2}}(\alpha - \beta) \right\rangle_- \quad (\text{A.8})$$

A general coherent state $|\gamma\rangle$ evolves in time under an Hamiltonian $\hat{H} = \hbar\omega\hat{a}^\dagger\hat{a}$ like $|\gamma(t)\rangle = |e^{-i\omega t}\gamma\rangle$ which can be used to evolve the state in eq. (A.8):

$$|\psi(t)\rangle = \left| \frac{1}{\sqrt{2}}e^{-i\omega_+ t}(\alpha + \beta) \right\rangle_+ \left| \frac{1}{\sqrt{2}}e^{-i\omega_- t}(\alpha - \beta) \right\rangle_- \quad (\text{A.9})$$

$$= \left| e^{-i\omega t}(\alpha \cos gt - \beta \sin gt) \right\rangle_1 \left| e^{-i\omega t}(-\alpha \sin gt + \beta \cos gt) \right\rangle_2, \quad (\text{A.10})$$

where in the last line the back-transformation from the \pm -modes (A.8) was used.

A.2 Exemplary calculation of E_N

In this section, the logarithmic negativity E_N eq. (2.21) is exemplary calculated for the state eq. (2.13). The density matrix of this system is given by

$$\rho(t) = |\psi(t)\rangle\langle\psi(t)| = \frac{1}{4} \begin{pmatrix} 1 & e^{i\Delta\phi} & e^{i\Delta\phi} & 1 \\ e^{-i\Delta\phi} & 1 & 1 & e^{-i\Delta\phi} \\ e^{-i\Delta\phi} & 1 & 1 & e^{-i\Delta\phi} \\ 1 & e^{i\Delta\phi} & e^{i\Delta\phi} & 1 \end{pmatrix}. \quad (\text{A.11})$$

Consequently, the partially transposed density ρ^{Γ_B} is given by

$$\rho^{\Gamma_B}(t) = \frac{1}{4} \begin{pmatrix} 1 & e^{-i\Delta\phi} & e^{i\Delta\phi} & 1 \\ e^{i\Delta\phi} & 1 & 1 & e^{-i\Delta\phi} \\ e^{-i\Delta\phi} & 1 & 1 & e^{i\Delta\phi} \\ 1 & e^{i\Delta\phi} & e^{-i\Delta\phi} & 1 \end{pmatrix}. \quad (\text{A.12})$$

The eigenvalues were calculated using `Mathematica` and equate to

$$\left\{ \sin^2 \left(\frac{\Delta\phi}{2} \right), \cos^2 \left(\frac{\Delta\phi}{2} \right), \frac{\sin \Delta\phi}{2}, -\frac{\sin \Delta\phi}{2} \right\}$$

According to lemma 2.1, $\|\rho^{\Gamma_B}\|_1$ is given by the sum of the absolute eigenvalues, which is equal to $1 + |\sin \Delta\phi|$. The negativity as the absolute sum of all negative eigenvalues (demonstrated in proposition 2.3) equates to $\mathcal{N} = |\sin \Delta\phi|/2$. Both methods result in a logarithmic negativity of $E_N = \log_2(1 + |\sin \Delta\phi|)$.

B Casimir interactions

B.1 Polarizability of a dielectric sphere

The polarizability α is defined via

$$\mathbf{E}_\infty \alpha = \mathbf{p}, \quad (\text{B.1})$$

where \mathbf{p} is the induced dipole moment and \mathbf{E}_∞ is the external electric field that induces the dipole moment. For a linear and uniform dielectric, it is given as $\mathbf{p} = \mathcal{V} \varepsilon_0 (\varepsilon_r - 1) \mathbf{E}_\text{in}$ [36, p. 220-226]. Here, \mathcal{V} is the volume of the object and \mathbf{E}_in is the electric field inside the dielectric. The electrostatic boundary conditions for the problem are given by

$$V_\text{in}|_{r=R} = V_\text{out}|_{r=R} \quad \text{and} \quad \varepsilon_r \varepsilon_0 \frac{\partial V_\text{in}}{\partial r} \Big|_{r=R} = \varepsilon_0 \frac{\partial V_\text{out}}{\partial r} \Big|_{r=R} \quad (\text{B.2})$$

and the electric potential outside of the sphere at $r \rightarrow \infty$ should be equal to the external dipole-inducing field $V_\text{out}|_{r \rightarrow \infty} = -\mathbf{E}_\infty \cdot \mathbf{r} = -E_\infty r \cos \theta$. The electric potential inside and outside the sphere can be calculated using the spherical decomposition of the general electric potential $V \propto 1/|\mathbf{r} - \mathbf{r}'|$ into Legendre Polynomials P_l [36, p. 188-190]:

$$V_\text{in}(r, \theta) = -E_\infty r \cos \theta + \sum_{l=0}^{\infty} A_l r^l P_l(\cos \theta), \quad (\text{B.3})$$

$$V_\text{out}(r, \theta) = -E_\infty r \cos \theta + \sum_{l=0}^{\infty} \frac{B_l}{r^{l+1}} P_l(\cos \theta). \quad (\text{B.4})$$

Applying both boundary conditions, it follows that [36, p. 249-251]

$$\begin{cases} A_l = B_l = 0 & \text{for } l \neq 1, \\ A_1 = -\frac{3}{\varepsilon_r + 2} E_\infty, \quad B_1 = \frac{\varepsilon_r - 1}{\varepsilon_r + 2} R^3 E_\infty \end{cases} \quad (\text{B.5})$$

and the resulting homogenous electric field $\mathbf{E}_\text{in} = -\nabla V_\text{in}$ inside the sphere is given as

$$\mathbf{E}_\text{in} = \frac{3}{\varepsilon_r + 2} \mathbf{E}_\infty. \quad (\text{B.6})$$

The field is shown on the right in fig. B.1. The polarizability α of the sphere can be now be determined to

$$\alpha_\text{sphere} = 4\pi \varepsilon_0 R^3 \left(\frac{\varepsilon_r - 1}{\varepsilon_r + 2} \right). \quad (\text{B.7})$$

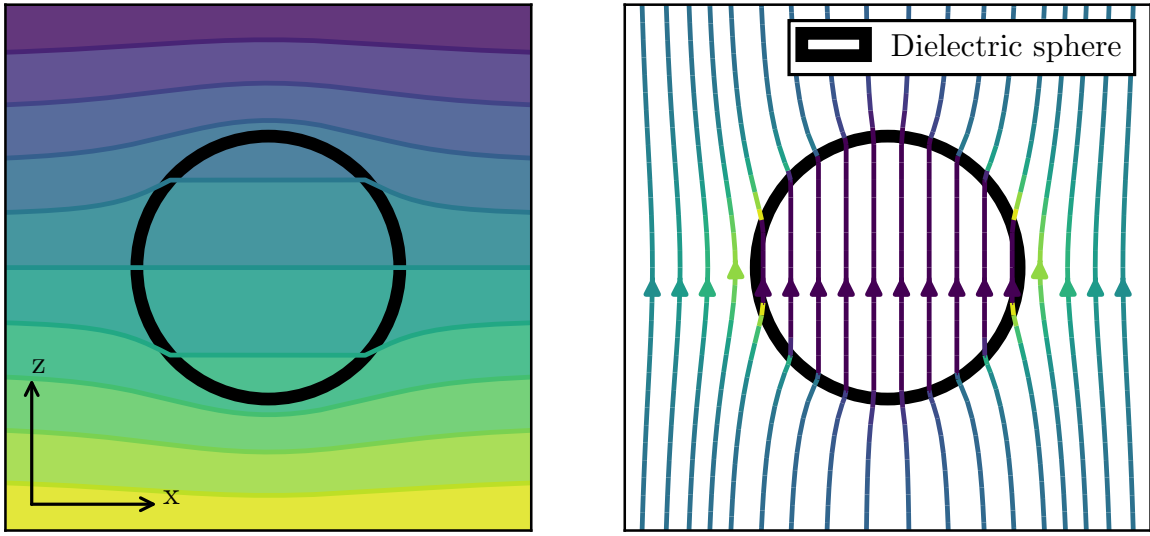


Figure B.1: **left:** Electric potential V of a dielectric sphere in a external electric field $\mathbf{E}_\infty \parallel \mathbf{e}_z$. **right:** The corresponding electric field lines inside and outside the dielectric sphere.

C The shield and its consequences

C.1 Blocking of the shield

Assume both spheres with charge q separated by a distance $2L$ and a shield is placed perfectly in the center of the spheres orthogonal to the direct connection between them. The magnitude field at a distance z from this connection line is given by

$$E(z) = \frac{2Lq}{4\pi\epsilon_0(L^2 + z^2)^{3/2}} \quad (\text{C.1})$$

The total flux a spherical region with radius r_s is given by

$$\Phi = \int_0^{r_s} dz \int_0^{2\pi} z d\varphi E(z) = \frac{q}{\epsilon_0} \left[1 - \frac{L}{\sqrt{L^2 + r_s^2}} \right]. \quad (\text{C.2})$$

Comparing the total flux Φ_{tot} for $r_s \rightarrow \infty$ with the flux through the shield, one can arrive at the effectiveness η of the shield as

$$\eta = \frac{\Phi}{\Phi_{\text{tot}}} = 1 - \frac{L}{\sqrt{L^2 + r_s^2}} \quad (\text{C.3})$$

and thus

$$r_s = L \sqrt{\frac{1 - (1 - \eta)^2}{(1 - \eta)^2}} \quad (\text{C.4})$$

C.2 Thermal harmonic oscillator

The amplitude z of a single vibrational shield-mode (k, l) with frequency $\omega_{kl} \equiv \omega$ behaves like a quantum harmonic oscillator. The average amplitude $\langle z \rangle_n = 0$. The variance $(\Delta z)^2 = \langle z^2 \rangle - \langle z \rangle^2$ however is given by

$$(\Delta z)_n^2 = \langle z^2 \rangle_n = \frac{\hbar}{2m\omega} (1 + 2n). \quad (\text{C.5})$$

At a temperature T , the occupation of the modes is described by the boltzmann distribution:

$$\langle z^2 \rangle_T = \sum_{n=0}^{\infty} \frac{1}{Z} e^{-\beta E_n} \langle z^2 \rangle_n, \quad (\text{C.6})$$

where $\beta = 1/k_B T$, $E_n = \hbar\omega(n + 1/2)$ is the energy of mode n and

$$Z = \sum_{n=0}^{\infty} e^{-\beta E_n} = \frac{e^{-\beta \frac{\hbar\omega}{2}}}{1 - e^{-\beta \hbar\omega}} \quad (\text{C.7})$$

is the partition function. Using known series, the expression eq. (C.6) can be evaluated to

$$(\Delta z)_T^2 = \langle z^2 \rangle_T = \frac{\hbar}{2m\omega} \sum_{n=0}^{\infty} \frac{1}{Z} [e^{-\beta E_n} + 2ne^{-\beta E_n}] \quad (\text{C.8})$$

$$= \frac{\hbar}{2m\omega} \left[1 + \frac{2}{Z} \sum_{n=0}^{\infty} ne^{-\beta E_n} \right] \quad (\text{C.9})$$

$$= \frac{\hbar}{2m\omega} \left[1 + \frac{2e^{-\beta \hbar\omega}}{1 - e^{-\beta \hbar\omega}} \right] = \frac{\hbar}{2m\omega} \coth\left(\frac{\hbar\omega}{2k_B T}\right) \quad (\text{C.10})$$